

# **Hydrogeochemical Investigations of Historic Mining Districts, Central Western Slope of Colorado, Including Influence on Surface-Water Quality**

*By* J. Thomas Nash



**USGS Digital Data Series  
DDS-73  
Version 1.0 2002**

U.S. Department of the Interior  
U.S. Geological Survey

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**U.S. Department of the Interior**

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February 2002, version 1.0

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## Data Files

[Data files, in Excel spreadsheet format (.xls), are stored in the “Data” folder at the root directory of this CD-ROM. Files with .csv extensions are “comma separated value” files. Filenames and brief descriptions of the files are provided below]

1.	NSAMDL.xls	Descriptions and locations of sample sites
2.	NSAMDL.csv	
3.	NGMR.xls	Chemical analyses of mineralized rock samples
4.	NGMR.csv	
5.	NGWC.xls	Analytical results for surface-water samples 1997–98
6.	NGWC.csv	
7.	NGWC9.xls	Analytical results for surface-water samples 1999
8.	NGWC9.csv	
9.	NGLC.xls	Analytical results for leachates of mineralized samples

- 10. NGLC.csv
- 11. NGMT.xls           Inventory of mills and tailings on the CWS
- 12. NGMT.csv
- 13. NGCZI.xls        Copper-zinc indices for surface waters
- 14. NGCZI.csv

# Hydrogeochemical Investigations of Historic Mining Districts, Central Western Slope of Colorado, Including Influence on Surface-Water Quality

By J. Thomas Nash

## Abstract

Reconnaissance studies of mining districts in the central part of the Western Slope (CWS) of Colorado were undertaken to characterize mined materials, mine drainage, and the influence of historic mining on nearby streams. The geochemical studies included sampling and chemical analysis of mine dumps, mill tailings, mine drainage, and surface waters. Descriptions and interpretations are based on more than 400 new geochemical analyses. Area geology, mining history, and geochemical methods are reviewed in relatively nontechnical style to provide a background for the descriptions of mining districts.

These geochemical investigations show that contamination from historic mines is generally less than that asserted by some or suggested by geochemical models. Most of the studied mining districts have at least a few sources of contaminants, but, in the majority of cases, the metals and acidity of concern are naturally mitigated to acceptable levels within about 2 or 3 mi of the sources. For these self-mitigating areas, the major concern is zinc and cadmium concentrations that are at or somewhat higher than (1× to 2×) the chronic exposure standard for aquatic life. A few districts having a combination of stronger sources and limited buffering capacity have greater impacts on water quality, and presumably on aquatic health, beyond the area of mining and downstream for distances of about 3 to 10 mi. In these headwater streams, metals such as Al, Cu, Fe, Mn, and Pb, in addition to Zn and Cd, tend to be above standards for aquatic life. In such situations, land managers and owners need to consider remediation methods that will reduce the release of metals and acid in mining-related sources. Several areas have unmined sources of acid and metals that complicate mitigation planning.

In this study of 22 mining districts, only one poses a significant threat to watershed health—the Red Mountain district in the headwaters of the Uncompahgre River watershed. Red Mountain Creek in the Red Mountain district appears to be the most contaminated of those studied here; there are indications of numerous sources in mines, dumps, mill tailings, and in the highly altered unmined rocks of Red Mountain. Several other

watersheds have mining-impacted waters of less magnitude or extent. Short reaches of streams show relatively low-level effects of historic mining: the Howard Fork of the San Miguel River, Elk Creek above Coal Creek (west of Crested Butte), Canyon Creek above Ouray, uppermost Henson Creek, and uppermost Uncompahgre River. The problems in these reaches involve acidic mine drainage before it can be buffered by natural processes and elevated Zn, Cd, and Cu concentrations. More study of these waters is needed, and ways to reduce acidity and metals concentrations to levels that will sustain or improve aquatic life communities must be found.

Tailings are identified here as important contributors of contamination in many mining areas. Also, many tailings impoundments appear to be at risk of catastrophic failure in an extreme storm. Most of the damage from a tailings failure cannot be reversed once the tailings are in a stream because cleanup is costly and likely to damage riparian habitat. Many tailings impoundments are collapsing after 50 or more years of neglect, and some are located in flood plains where they are at risk in a major storm. Locally focused flash floods are fairly common in Colorado, especially in the summer, and are capable of moving millions of tons of earth materials. A case can be made that failure of a tailings impoundment in a storm event can cause far more intensive and extensive damage to a watershed than the more predictable effects of acid mine drainage. Further study of tailings stability in extreme storms is needed to make an appropriate risk analysis.

## Introduction

More than 130 years of mining on the Western Slope of Colorado has left a legacy of trails, ruins, and waste piles that are viewed with concern or affection by residents, visitors, and Federal land managing agencies (FLMA's). Newspapers and journals contain divergent descriptions of the effects of historic mining that range from prophesy of doom to naive assertions that there are no problems. The truth, if it can be found, is somewhere in between: the effects of historic mining on surface-water quality and biologic habitat are enor-

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mously variable from site to site and area to area. Predictions of mining impacts based on scientific measures of geology, ore character, mining style, climate, and other parameters are improving but tend to lack the specificity required by FLMA's to plan restoration activities. This investigation was undertaken to provide, most importantly, new descriptive data and, secondly, some interim interpretations and suggestions. Data and observations in this study are not perfect but should hold up under scrutiny; the interpretations are subject to state-of-the-art understanding and should be viewed as concepts, not facts.

Mining on the central Western Slope (CWS) of Colorado played a major role in the settlement and growth of the region, starting unofficially a few years before treaties with the Native Americans in the 1860's. Mining and related service industries brought millions of dollars of new wealth to dozens of communities. Mining started with few Federal regulations, and those that existed in the first 50 years tended to have the intent of encouraging new mining activity. Handling of mine wastes does not appear to have been regulated significantly until 1935, when an Executive Order stipulated that mill tailings could not be dumped into streams. Concerns today about contamination from historic mines are valid but should be tempered with an understanding of the fact that few or no regulations existed for the first 75 years of mining on the CWS.

Environmental assessment of watersheds and mine sites can be viewed as rigorous science or evolving art. I will not debate this issue here, but the reader needs to understand (1) that some aspects of geochemistry may not be quite as they appear and (2) that highly precise measurements by a skilled scientist may not be reproducible or appropriate to the perceived problem. The author brings more than 30 years of geologic and geochemical experience to this study but just a few years of work on some techniques such as water chemistry. The field observations, sampling, and interpretations in this study are based much more on geology than on protocols used by regulators and health scientists. Readers and FLMA's need to understand that my observations at a site are likely to be quite different from those of a skilled person with Colorado Department of Health; it is possible that while our data or observations might differ, both could be correct in their context. The results and interpretations reported here are reconnaissance in nature, designed to identify and rank the major problems on the CWS, and are given with the caveat that further detailed studies by more precise methods will be required to confirm and quantify chemical conditions.

Scale is a major operational problem for geo-environmental studies. Focus and detail are required to evaluate point sources and sites. In some situations a scientist may need to spend days characterizing water from one pipe or the waste on an acre site. At the other extreme is the need to understand numerous kinds of mines and geologic settings across more than 10,000,000 acres of land on the CWS, with many kinds of ownership, to begin to comprehend trends in geochemical data. In an attempt to deal with issues of scale, discussions in this report will start with watershed scale of the four major rivers and will shift to site scale for limited discussion of problem

sites. Because my fieldwork was reconnaissance in nature, I generally did not have detailed maps or knowledge of land status, thus cannot stipulate for most sites whether the status is private, public, or mixed. The study area on the CWS and the districts that were investigated and sampled for geochemical analysis are shown on figure 1.

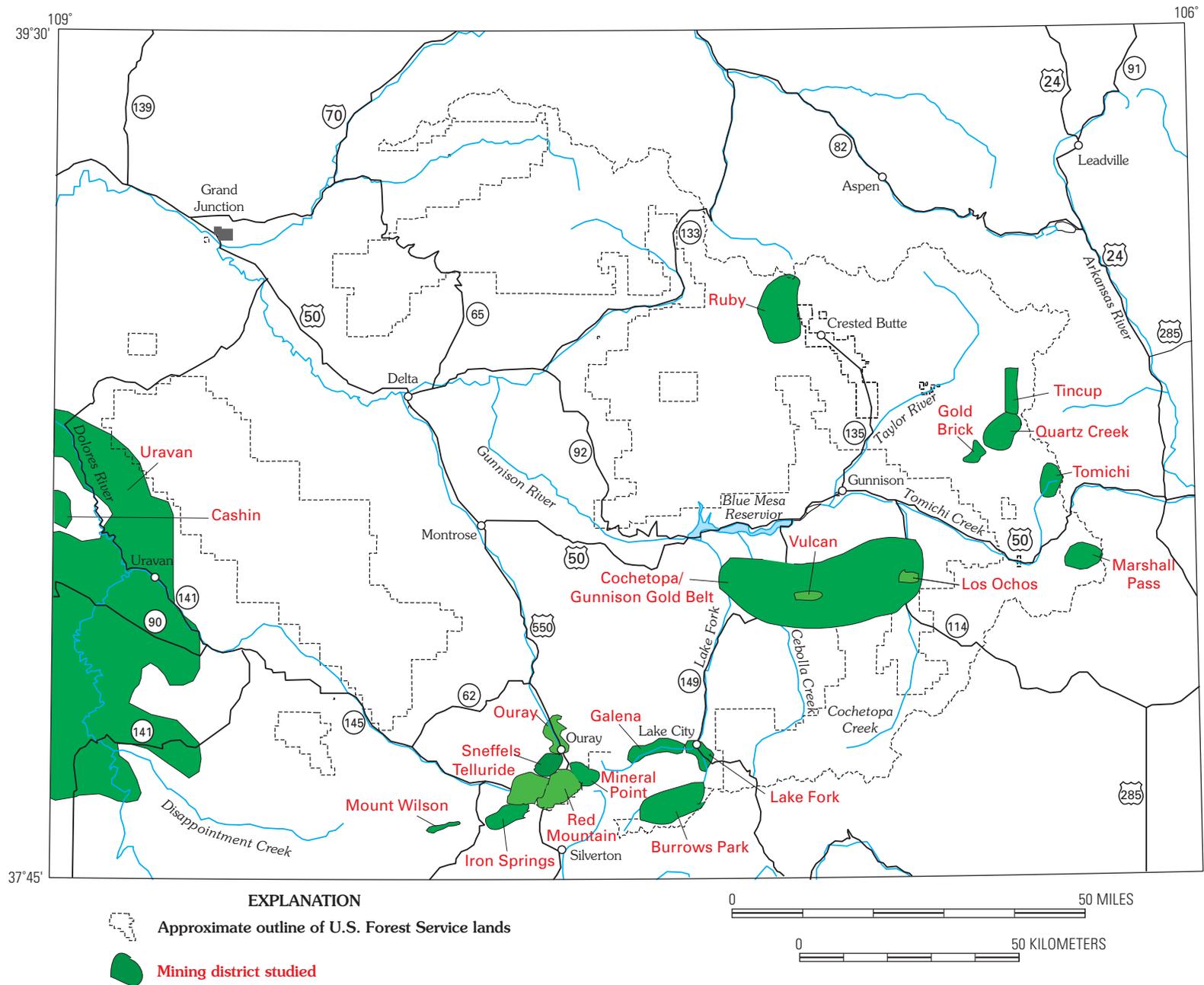
Work in the field was guided by information in the literature, in digital databases, and on published 1:24,000-scale U.S. Geological Survey (USGS) topographic maps that show major mining operations. The goal in most districts was to visit representative and substantial mines and mill tailings to characterize the district<sup>1</sup> and observe and sample the most likely sources of contamination. While in the field, the next criterion was accessibility—both to save time and with an eye on potential for reclamation (experience elsewhere suggests that reclamation is less likely to happen if new roads must be constructed). And, finally, hands-on work and sampling was influenced by visible indications of land status. If mine sites were posted as private, or fenced and gated, these sites were not examined. If there was no diligence (signage or fences), and if there were indications of substantial traffic by the public, then I would go on the property and collect samples for chemical analysis. For water characterization, an attempt was made to sample streams above, immediately below, and a mile or two below the major mines or mills.

This study was made in parallel with geologic and mineral resources studies of the same area by colleagues in the USGS (Bankey, in press). Studies of area geology (Day and Bove, in press), mineral deposits (Wilson and Spanski, in press), and geochemistry (Smith, in press) were accomplished in that program.

*Acknowledgments.*—This study has benefited from the support and suggestions of many people in the USGS, U.S. Forest Service (USFS), and U.S. Bureau of Land Management (BLM). The broad goals of geo-environmental investigations and the need for more information on the CWS were defined by Geoff Plumlee, USGS. I appreciate the support given by my supervisors for experiments with reconnaissance methods and new ways to present results. Colleagues on the CWS mineral assessment project have shared their knowledge and have provided preliminary maps. Staff scientists with the USFS and BLM in district offices have suggested sites and areas in need of study. I thank in particular Barbara Hite, BLM, and Daryl Gusey, USFS, for numerous suggestions and feedback as this work evolved. Preliminary drafts of manuscripts were reviewed by Bob Eppinger, Harley King, Bill Miller, and Brad Van Gosen, who offered numerous helpful suggestions. Errors of fact or interpretation that remain are the responsibility of the author.

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<sup>1</sup> District is used informally in this report, chiefly for geography and history rather than for legal implications. Names applied to districts can vary, and I have attempted to use the most common one. In a companion study of mineral resources of this study area, my colleagues Wilson and Spanski (in press) use the term "mining area," which is appropriate but not in general usage.



**Figure 1.** Location of mining districts studied, central Western Slope, Colorado.

## Methods

Fieldwork for this study was conducted in June, July, and September of 1997, July and August of 1998, and in September of 1999. Many of the districts were visited more than once as the goals of the project evolved or to seek better water samples earlier in the year. Sites visited were determined by published information, USGS records in the Mineral Resources Data System (MRDS), information on USGS 1:24,000-scale topographic maps, by suggestions from BLM and USFS colleagues, and by access. Sample localities were recorded on standard 1:24,000-scale topographic maps and measured onsite with a conventional GPS (global positioning system) instrument having an accuracy of about  $\pm 200$  ft based on tests at known sites; the GPS unit was run under standard WGS84 for datum and spheroid. Sample sites and analyzed samples are briefly described, and locations given, in table 1<sup>2</sup>.

## Rock and Mine-Waste Geochemical Studies

Mineralized samples were collected from mine dumps, outcrops, mill tailings, and smelter sites. In most cases the intent was to collect a representative sample, but for some sites a special sample was collected to determine a special property. The most commonly employed sampling method for dumps and tailings involved the collection of numerous small portions at 10 to 30 sites, then processing those subsamples through a 2-mm stainless steel sieve to derive a composite sample weighing about 2 to 4 pounds.

Rock samples were crushed and pulverized for analysis by the USGS and analyzed by a commercial laboratory. In one method, the rocks are dissolved in a mixture of four acids, then the concentration of 35 major and trace elements are determined using ICP-AES (inductively coupled plasma-atomic emission spectroscopy); this is considered a total analysis because the strong acids dissolve all but the very refractory minerals such as zircon. Mineralized samples were also analyzed by a method that employs weaker acids to dissolve most minerals, then uses an organic reagent to collect 11 metals of interest (Ag, As, Au, Bi, Cd, Co, Cu, Mo, Pb, Sb, and Zn); analysis is by ICP-AES. This second analytical method has lower levels of determination and works well for some metallic elements. Quality-assurance monitoring showed that the deviation of the reported values from correct ones is less than 5 percent for most elements.

Analytical results for mineralized rock samples (dumps, tailings, altered rocks) are in table 2<sup>2</sup>; these results are also available in the Data folder at the root directory of this CD-ROM as the original spreadsheet file (NGMR.xls and NGMR.csv).

## Water Geochemical Studies

Water compositions were investigated by a reconnais-

sance sampling method when water was observed at mine portals, seeping from mine dumps or tailings, or where streams were found near mining operations. The volume of flow was estimated at each site but is very approximate. Notes were taken on the nature of any mineral coatings related to the water, if any, because these often are an indicator of degraded water quality. However, in the case of highly acidic waters, there is no immediate precipitate; thus, the waters have the appearance of good quality, neutral waters.

Methods used to measure pH, conductivity, and sample waters in the field are described in Appendix 1. The routine method for sample collection in plastic bottles is not effective for mercury because it becomes adsorbed on the container and is not available for analysis (J. Gray, oral commun., 1998). In 1998, some representative samples were collected in special glass bottles containing a chromic acid preservative for Hg, courtesy of Jim Crock, USGS. The filtered and acidified water samples were analyzed by a commercial laboratory in Denver, Colo., using ICP-MS. The elements determined by this method and quality control are discussed in Appendix 1.

Analytical results for water samples collected and analyzed for this study are in table 3<sup>2</sup>; the results are also included in the Data folder at the root directory of this CD-ROM in spreadsheet form as NGWC.xls and NGWC.csv. Results for 30 samples collected in 1999 are shown at the end of table 3 but appear in a separate file in the Data folder at the root directory of this CD-ROM (NGWC9.xls, NGWC9.csv).

## Laboratory Leach Tests of Mineralized Samples

Solid materials from dumps, mill tailings, outcrops, and slag piles have been processed in the laboratory by a passive leach method (Fey and others, 2000a) that provides a measure of reactions in nature, such as during storm events. Quite simply, 100 g of rock sample was placed in a beaker with 2,000 mL of deionized water, stirred slightly, and an initial pH measured. After about 20 hours, the upper part of the liquid was stirred slightly to mix the leachate solution. At 24 hours, the pH and conductivity of the leachate were measured and a 60-mL aliquot was taken with a syringe and passed through a 0.45- $\mu$ m filter, just as was done for water samples in the field. The leachate samples were acidified with 5 drops of 1:1 ultrapure HNO<sub>3</sub> to stabilize metals<sup>3</sup> in the solution—chiefly to minimize adsorption on the plastic bottle or formation of precipitates. Results from this method closely resemble those from EPA method 1312 that tumbles the sample (Fey and others, 2000a). For this project, the leachate was sent to a

<sup>2</sup> Tables 1–3, because of their length, appear in Appendix 2 at the end of this report. Appendix 2 begins on page 152.

<sup>3</sup> The term “metal” will be used in a broad sense in many places in this report to include metalloids and nonmetals, such as As and Se; this broad context will not consider speciation such as As<sup>III</sup> and As<sup>V</sup>, or oxidation state, as in the case of oxyanions (MoO<sub>4</sub>).

commercial laboratory and analyzed by ICP-MS in the same manner as water samples. This analytical method is well suited for this task because the leachates are highly varied in composition and ICP-MS provides results for elemental concentrations that span the range from less than 0.1 to 1,000,000 ppb (parts per billion, the same as microgram/liter except at very high concentration); the more concentrated solutions are diluted 1:10 to 1:50 to stay within the calibration range. Anions are not determined by ICP-MS; no supplemental analyses for them were made because the dominant anion is expected to be sulfate. Quality-control tests using replicate samples and standard waters show that the results have a precision and accuracy of about  $\pm 10$  percent for most elements, which is well within the sampling error and probable seasonal variability of waters in this study. These analyses are effective for the description and ranking of samples or sites.

For most rock samples, no sample preparation was done for these tests. The majority of dump samples were sieved in the field to pass through a 2-mm sieve. Mill tailings are, by their nature, ground materials of fine sand size, thus they did not require grinding or sieving. A special type of dump or tailing sample, efflorescent crusts, was picked in the field and cleaned somewhat in the laboratory to provide material for leaching that was more than 50 percent crust. The samples were dry and friable when weighed and placed in the reaction vessel.

The behavior of samples during the 24-hour tests was variable. For many samples, especially those with abundant iron oxides or jarosite (rusty-appearing minerals), the pH dropped to below 3.5 in just a few minutes. In some samples, the pH evolved to lower values during the 24 hours, and in a smaller number of samples (20 percent), the pH rose a few tenths of a unit, reflecting acid consumption in reactions with minerals in those samples. Most of the materials created orange- to pale-yellow-colored water; a few were light gray and some introduced little color. Suspended material in the leachates tended to settle in about 10 hours, and these were easy to filter. A small percentage of samples created milky solutions that did not clear overnight, and some of these were difficult to filter because clay- or colloid-sized material clogged the filter. Elements in the solution for analysis, which passed through the 0.45- $\mu\text{m}$  filter, are commonly termed “dissolved” in the literature (as opposed to “colloidal” or “total”).

Analytical results for the leachate solutions and the leachate pH are in table 4; these results are also in the Data folder at the root directory of this CD-ROM in spreadsheet format (NGLC.xls, NGLC.csv).

## Geographic Setting and Climate

The study area spans diverse ecoregions from the alpine zone at the Continental Divide to pinyon-juniper or sagebrush-vegetated semidesert of the Colorado Plateau at the Utah border (Bailey, 1998). The climate of mining areas varies

both across the study area and locally. Because water is the key issue in this study, and because water drives most of the chemical reactions that create acid or mobilize metals, annual precipitation and style of precipitation are important factors in the geo-environmental setting of ore deposits (Plumlee, 1999). Climate records (Benci and McKee, 1977; Colorado Climate Center, 1984) provide helpful information, but one must remember that local conditions on a mountain can be much more severe than at the nearest weather station.

Precipitation in the CWS ranges from less than 10 inches per year at low elevations on the western side to more than 50 inches above 11,000 ft elevation. Many of the major mining districts receive snow for 10 months of the year, and avalanches were major threats to the early miners who lived near the early mines. The high mountains also tend to get major rainfall in the late summer from storms originating to the southwest; these storms can dump several inches of rain in less than 24 hours and create flash floods (Follansbee and Sawyer, 1948). Precipitation and evaporation vary locally on sunny southern slopes and dark north-facing slopes; this appears to be a significant factor in detail, such as observed in mine drainages of the Silverton area (Nash, 1999a) but will not be considered further here. Just as changes in vegetation are easily seen as a function of climate, one can see the higher volume of mine runoff and adit drainage in snowy or wet mountain settings that contrast with relatively dry settings, as in the Cochetopa district (fig. 1), where adits are dry and crusts of sulfate minerals form when waters evaporate on mine dumps. The kinds and magnitudes of mining-related geochemical problems are predictably related to climate: mine drainage is most voluminous and often most degraded at wet, high elevation sites; seasonal or episodic flash floods or flushes tend to be more significant at dryer, lower elevations. These associations will be discussed in later sections.

Many books chronicle the history of mining camps in Colorado and provide useful information on the mining districts described here. Some of the books, for example by Benham (1980), Smith (1994), and Southworth (1997), also provide prints of historic photographs that clearly show the rugged conditions and simple structures of the early mines and associated communities.

## Introduction to Ore-Deposit Types and Their Environmental Signatures

Many types of ore deposits have been located or mined in the CWS over the past 130 years. Economic geologists utilize numerous classification systems to compare and summarize geologic and economic attributes of deposits (Lindgren, 1933; Cox and Singer, 1986; Guilbert and Park, 1986). In the resource assessment of the CWS, more than 25 mineral-deposit types were considered (Spanski and Bankey, in press; Wilson and Spanski, in press). These classifications subdivide

**Table 4.** Analytical results by ICP-MS for selected elements in leachate solutions, CWS, Colorado.

[Samples passively leached for 24 hours in deionized water; leachate filtered (<0.45 µm) and acidified; solution analyzed by ICP-MS in commercial laboratory. Sample type: third character in sample ID indicates type: D, dump; T, tailing; R, rock (unmined); C, crust on dump or tailing sample. Values are reported in parts per billion; <, not detected at value shown; number of figures reported by spreadsheet may not be significant; nd, not determined. Districts: CO, Cochetopa; GB, Gold Brick; GN, Galena; IS, Iron Spring; LO, Los Ochos; MP, Mineral Point; OU, Ouray; QC, Quartz, Creek; RB, Ruby; RM, Red Mountain; SN, Sneffels; TE, Telluride; TM, Tomichi; TC, Tincup; UV, Uruvan. Deposit type: BP, breccia pipe; FE, fericrete; PMR, polymetallic replacement; PMV, polymetallic vein; SC, sediment-hosted copper; SK, skarn; SU, sandstone uranium; TV, volcanic-hosted vein; VMS, volcanogenic massive sulfide]

Sample ID	District	Deposit type	pH	Conductivity (µS/cm)	Al (ppb)	As (ppb)	Cd (ppb)	Cu (ppb)	Fe (ppb)	Mg (ppb)	Mn (ppb)	Mo (ppb)	Pb (ppb)	Sb (ppb)	Se (ppb)	U (ppb)	Zn (ppb)
NGCL223																	
NGCL225	CO	VMS	3.3	nd	1,537,171.9	21.4	1,494.5	70,581.8	60,976	3,064,039	275,626.5	4.4	8.0	0.2	360.5	29.7	322,773.8
NGCL227	CO	VMS	2.1	nd	299,391.8	12,507.1	1,212.4	9,584.0	1,908,102	191,911	1,745.2	950.3	1,781.3	2.9	27,422.3	13.8	350,125.8
NGCL228C	CO	VMS	1.9	>2,000	308,410.4	233.6	3,777.1	222,516.2	1,878,930	315,452	4,723.1	1,277.2	22.3	0.9	905.6	47.0	934,449.8
NGCL504	CO	VMS	2.3	>2,000	429,021.9	346.4	3,734.2	446,646.8	2,577,782	356,969	4,352.8	55.7	201.6	<0.1	602.4	40.3	962,023.5
NGDL101	UV	SU	7.1	nd	4.9	19.3	0.1	1.3	1,344	1,018	0.7	11.3	0.3	0.4	32.3	0.2	2.8
NGDL102	UV	SU	7.2	nd	6.4	23.5	0.1	0.6	1,469	2,153	0.8	18.8	0.1	0.4	81.0	0.1	2.0
NGDL108	UV	SU	8.1	nd	69.2	8.9	0.1	6.7	48	763	1.7	4.6	3.4	0.2	4.7	1.1	1.8
NGDL109	UV	SU	8.2	nd	343.5	18.1	0.2	3.7	109	434	0.6	32.4	0.6	0.2	228.7	2.4	1.2
NGDL110	UV	SU	7.4	200	30.8	24.6	9.9	1.5	167	822	0.7	1,977.5	0.5	1.8	79.5	0.7	2.7
NGDL111	UV	SU	7.2	330	19.2	13.6	2.4	1.2	232	1,878	7.8	797.4	0.5	1.7	706.6	1.0	5.1
NGDL112	UV	SU	6.0	nd	44.9	22.3	1.0	1.6	28	534	0.2	198.0	0.4	0.7	4.8	7.9	0.3
NGDL113	UV	SU	7.6	40	70.3	64.5	0.5	1.4	97	773	1.2	144.1	1.2	0.9	4.2	3.7	4.5
NGDL114	UV	SU	5.0	nd	11.9	2.1	0.1	1.1	1,366	432	1.0	11.6	0.1	0.1	16.0	0.7	2.5
NGDL116	UV	SU	6.2	nd	67.5	10.2	0.1	1.2	24	591	0.2	18.9	0.3	0.3	8.5	3.0	0.3
NGDL118	UV	SU	6.6	nd	104.2	32.4	0.1	0.7	49	1,339	0.6	10.8	0.7	0.4	18.8	3.7	2.9
NGDL201	TM	PMR	7.1	nd	80.6	1.6	0.2	9.8	1,843	599	75.8	4.8	595.6	0.2	-0.2	0.3	486.2
NGDL204	TM	PMR	4.5	nd	6.7	1.3	1,305.8	13.0	3,649	703,424	121,467.9	5.6	485.8	0.2	40.8	11.5	130,908.6
NGDL206	TM	PMR	5.7	nd	2.4	0.2	33.4	4.8	1,355	4,218	759.0	0.8	8.7	0.1	4.6	0.1	2,468.9
NGDL207	TM	PMR	7.9	nd	6.2	2.7	8.3	2.9	24	541	2.7	1.6	18.1	1.3	0.2	0.0	37.5
NGDL209	TM	PMR	7.4	nd	6.7	0.5	1.5	1.4	59	2,073	1.5	2.8	16.4	0.2	6.7	0.0	17.9
NGDL213	TM	PMR	5.3	nd	2.6	0.1	102.0	4.3	1,434	17,712	2,367.0	0.6	777.5	0.3	2.5	0.1	6,259.7
NGDL217	CO	VMS	3.5	nd	15,228.3	1.4	435.8	50,143.1	1,210	34,359	2,375.3	3.3	1.7	0.1	80.8	21.2	83,820.1
NGDL223D	CO	VMS	3.8	nd	9,986.4	0.2	33.3	141,013.7	572	267,471	7,781.5	0.8	6.6	0.0	53.3	4.9	4,588.7
NGDL224	CO	VMS	3.6	nd	20,043.8	1.8	171.8	6,410.7	11,214	52,791	858.3	1.5	0.8	0.0	66.7	2.3	35,534.4
NGDL226	CO	VMS	2.5	nd	9,726.7	53.7	35.1	1,962.7	100,278	8,518	111.1	6.2	17.5	0.1	81.4	1.3	8,873.9
NGDL229	CO	VMS	3.4	nd	595.9	0.6	9.5	920.2	4,358	504	15.9	0.5	4.3	0.0	5.0	0.4	2,602.3
NGDL231	CO	VMS	3.3	>2,000	68,449.0	10.6	166.7	9,541.5	10,971	122,412	4,387.1	61.6	0.5	0.1	101.3	11.2	30,477.7
NGDL235	GB	PMV	5.9	nd	45.5	2.4	0.4	7.4	47	344	6.9	0.4	2.1	0.2	0.3	0.0	17.9
NGDL240	TC	PMR	6.2	nd	34.8	1.6	0.0	2.8	49	808	4.7	0.8	3.0	0.8	0.2	0.0	0.8
NGDL241	TC	PMR	5.6	nd	15.8	4.6	0.2	24.4	54	502	10.8	1.6	17.5	1.7	0.2	0.0	20.1
NGDL242	TC	PMR	4.3	60	312.5	0.5	0.2	16.4	2,006	86	264.9	0.8	239.8	1.7	0.2	0.2	64.0
NGDL250	LO	VU	3.5	nd	4,432.5	12.9	3.7	94.0	178	1,141	109.6	0.6	2.6	0.6	0.4	253.3	286.8
NGDL251	LO	VU	3.5	nd	9,006.0	4.1	1.5	138.9	3,384	912	122.4	1.5	3.3	0.2	9.8	405.9	218.8
NGDL252	LO	VU	3.1	nd	5,670.2	11.3	3.5	23.9	1,607	564	21.3	1.5	0.7	0.2	3.2	901.1	80.9
NGDL253	LO	VU	5.4	nd	57.1	16.4	0.1	3.9	21	88	1.4	10.0	0.8	1.6	0.2	7.4	0.9
NGDL255	LO	VU	7.7	nd	10.9	2.0	0.1	41.5	16	65	5.9	1.6	0.6	0.7	0.6	9.4	11.7
NGDL255	LO	VU	4.2	nd	13.9	2.0	0.1	40.3	16	63	6.2	1.0	0.3	0.7	0.9	10.4	10.4
NGDL256	LO	VU	3.1	nd	4,390.0	8.7	32.2	136.3	971	1,536	352.0	9.2	1.7	3.0	2.6	3,402.7	1,120.1
NGDL261	CO	VMS	4.8	nd	7.5	9.3	544.9	659.8	611	17,194	2,959.7	1.5	0.7	0.2	12.1	1.1	148,329.6
NGDL262	CO	VMS	4.2	nd	813.2	112.0	487.8	33,287.7	489	62,698	10,074.0	0.9	7.3	1.0	54.8	7.2	139,927.7
NGDL265	QC	PMV	3.7	nd	48.2	0.4	0.3	50.8	62	73	9.5	20.0	0.4	0.1	0.7	6.8	34.8
NGDL270	QC	PMV	3.5	nd	1,553.7	17.6	25.5	4,671.2	1,959	512	51.9	13.1	8.0	17.2	1.8	1.9	1,919.2
NGDL274	TC	PMR	5.7	nd	8.3	0.6	2.1	3.0	109	2,117	79.5	4.9	0.9	0.4	1.9	1.0	39.0

**Table 4.** Analytical results by ICP-MS for selected elements in leachate solutions, CWS, Colorado—*Continued.*

Sample ID	District	Deposit type	pH	Conductivity (µS/cm)	Al (ppb)	As (ppb)	Cd (ppb)	Cu (ppb)	Fe (ppb)	Mg (ppb)	Mn (ppb)	Mo (ppb)	Pb (ppb)	Sb (ppb)	Se (ppb)	U (ppb)	Zn (ppb)
NGDL280	TC	PMR	4.1	nd	112.9	0.3	6.7	681.5	61	499	38.2	12.5	120.4	1.9	0.3	3.1	574.0
NGDL288	GN	TV	2.9	nd	10,606.7	6.1	134.8	2,050.8	16,930	9,902	3,315.2	22.0	12.7	0.3	2.5	4.0	19,465.1
NGDL303	OU	PMR	4.1	nd	13.4	0.2	1.3	37.9	32	303	296.2	1.8	35.2	0.4	0.4	0.5	176.3
NGDL321	IS	TV	3.1	850	582.7	0.6	3.8	377.2	1,520	716	1,447.9	112.9	11.1	0.2	1.0	2.4	624.0
NGDL322	IS	TV	3.1	590	775.0	0.3	6.5	333.2	2,042	811	411.6	1.3	44.2	1.8	0.3	0.5	1,098.3
NGDL327	RM	BP	2.7	nd	2,334.8	22.2	14.1	2,534.4	16,796	1,281	112.2	0.8	5,654.0	8.3	2.8	0.5	1,479.5
NGDL329	RM	BP	2.9	760	1,023.7	0.8	0.4	475.5	5,253	784	51.4	33.0	4.6	0.2	0.4	0.5	83.4
NGDL330	RM	BP	2.7	1,550	337.1	110.5	1.1	994.0	14,229	322	7.7	1.2	2,650.0	102.5	5.3	2.4	299.9
NGDL331	RM	BP	3.1	nd	474.9	1.6	1.4	250.4	688	415	73.8	0.7	3,867.4	2.6	0.5	0.4	286.3
NGDL332	RM	BP	3.4	160	76.5	2.2	1.1	396.5	69	470	40.7	0.8	1,708.3	1.0	0.2	0.1	361.0
NGDL339	RM	BP	3.2	390	111.8	0.4	0.4	74.4	1,279	280	51.1	15.9	297.8	0.4	0.3	0.1	187.5
NGDL341	RM	BP	2.5	nd	3,512.9	34.0	14.6	618.2	37,143	1,487	58.8	2.7	4,684.8	3.9	4.6	2.4	3,394.5
NGDL350	RM	TV	2.7	nd	2,337.1	0.2	31.8	119.8	6,501	1,278	633.9	0.7	2.6	0.0	1.8	0.4	6,399.7
NGDL351	RM	TV	3.0	1,420	442.6	0.4	95.0	150.9	3,919	500	564.1	1.7	731.8	0.1	3.1	0.2	12,012.9
NGDL352	RM	TV	3.7	90	49.5	<0.1	3.7	7.7	23	194	139.4	1.3	2.8	0.1	0.2	0.1	900.9
NGDL355	RM	BP	3.1	nd	705.9	0.3	0.7	66.6	2,604	313	46.0	0.4	5,015.0	0.7	0.7	0.7	145.6
NGDL358	RM	BP	3.7	nd	9.9	1.0	0.0	4.0	64	8	1.1	0.3	46.2	0.6	0.2	0.0	9.5
NGDL361	RM	BP	2.6	nd	739.1	37.3	1.1	1,608.0	22,392	226	18.6	0.6	1,411.9	23.3	3.8	1.3	290.4
NGDL366	RM	TV	3.4	nd	10.6	2.7	0.4	59.0	230	25	3.2	0.3	2,153.4	5.3	0.4	0.0	64.6
NGDL372	SN	TV	5.5	70	8.1	0.1	0.2	0.6	84	993	20.4	10.5	2.6	0.2	0.5	0.0	19.6
NGDL379	MP	TV	3.4	420	1,276.9	0.1	38.8	278.1	240	1,317	2,134.7	1.6	242.3	0.2	0.6	0.1	3,542.4
NGDL381	MP	TV	3.4	nd	3,395.8	0.5	16.8	1,044.4	795	860	1,011.6	0.3	559.0	0.9	0.8	1.5	2,077.0
NGDL389	TE	TV	4.0	nd	260.7	0.2	123.3	648.9	284	604	4305.3	1.0	3769.1	0.3	<2	0.5	27115.1
NGDL390	TE	TV	3.6	nd	106.6	2.2	7.2	61.6	165	291	714.3	0.6	122.7	0.5	<2	0.2	1658.9
NGDL391	TE	TV	2.8	940	757.9	33.5	3.2	127.8	8,626	180	87.8	3.2	205.3	1.0	0.2	0.1	643.2
NGDL395	GN	TV	2.7	nd	10,469.8	79.4	176.2	1,288.7	12,924	4,312	5,525.8	5.0	3.4	1.4	4.4	3.3	21,480.7
NGDL507	RB	TV	3.3	334	865.2	30.8	215.1	1,553.0	9,405	1,994	10,973.0	2.1	2,607.5	0.1	27.4	0.5	28,313.8
NGDL524	RB	TV	3.2	441	771.6	5.1	283.5	1,035.5	879	1,117	17,614.2	<0.05	3,863.3	0.2	12.3	3.4	39,459.7
NGDL525	RB	TV	2.9	652	905.5	2.0	33.7	241.7	4,559	404	566.7	<0.05	1,182.8	0.1	1.6	0.3	2,443.0
NGDL528	RB	TV	3.2	227	19.1	0.3	4.2	32.6	116	115	235.9	<0.05	1,086.7	0.0	0.7	0.0	366.9
NGDL556	RB	TV	3.0	485	2,139.3	1.5	39.4	379.7	5,452	723	4,154.2	<0.05	5,305.8	0.4	3.9	0.8	5,988.2
NGDL585	RM	TV	2.7	903	1,643.5	143.0	2.2	1,392.5	35,098	551	53.5	5.6	624.7	4.6	5.1	0.2	381.0
NGDL591	RM	TV	2.9	525	1,807.4	9.1	10.3	1,973.1	23,966	2,845	133.4	<0.05	2,579.6	4.3	0.2	0.9	2,667.3
NGDL598	RM	TV	3.1	209	501.5	2.0	13.6	638.5	531	625	174.5	0.3	2,164.8	3.8	0.3	1.4	1,748.6
NGDL606	RM	TV	3.0	325	1,144.0	0.4	1.4	111.5	4,072	212	93.1	3.0	112.5	0.7	0.3	0.1	211.9
NGDL661	TE	TV	3.1	390	539.3	0.4	53.0	801.3	861	222	1,307.0	0.4	47.4	0.3	0.9	1.3	7,168.7
NGDL679	SN	TV	3.1	762	14,474.6	1.8	29.0	113.6	4,615	11,714	15,488.4	<0.05	16.3	0.1	2.5	0.9	4,818.6
NGGL201	TM	PMR	3.1	nd	80.6	1.6	0.2	9.8	1,843	599	75.8	4.8	595.6	0.2	0.2	0.3	486.2
NGRL311	IS	FC	3.8	60	252.1	<0.1	1.8	95.4	4	64	22.9	1.3	2.8	0.1	0.2	0.4	78.6
NGRL326	RM	BP	4.6	30	8.9	2.2	0.1	3.7	98	116	7.8	0.4	4.9	0.7	0.2	0.0	23.6
NGRL329	RM	BP	4.3	60	362.8	0.4	0.2	12.1	2,112	79	247.4	0.7	256.9	1.2	0.2	0.2	74.6
NGRL617	IS	FC	3.5	49	188.6	<0.1	1.6	65.9	20	22	4.3	0.1	12.5	0.0	0.2	0.4	50.3
NGRL682	SN	TV	4.9	29	37.5	0.6	0.2	3.8	56	127	33.1	4.7	1.2	0.2	0.2	0.0	16.2
NGTL203	TM	PMR	5.4	nd	1.7	0.2	90.4	1.0	1,939	271,529	16,299.3	1.1	46.8	0.1	1.7	0.1	2,908.5
NGTL212	TM	PMR	5.9	nd	30.5	0.1	43.5	2.7	953	67,029	367.9	1.1	21.1	0.1	2.7	0.1	515.0
NGTL237	GB	PMV	4.2	nd	23.3	32.6	36.0	174.1	1,324	930	528.5	3.6	3,386.4	0.1	73.5	0.1	5,807.4
NGTL238	GB	PMV	3.4	nd	365.5	13.4	79.2	152.8	601	1,391	480.3	1.6	2,247.5	0.4	22.2	18.2	4,352.2
NGTL239	GB	PMV	5.1	nd	23.5	0.3	1.1	9.1	30	313	27.2	0.4	35.8	0.1	0.2	0.0	39.1
NGTL272	TC	PMR	4.3	nd	159.7	0.3	213.8	2,010.2	97	348	49.0	1.4	61,122.1	0.2	0.3	0.1	1,907.0

Table 4. Analytical results by ICP-MS for selected elements in leachate solutions, CWS, Colorado—Continued.

Sample ID	District	Deposit type	pH	Conductivity (µS/cm)	Al (ppb)	As (ppb)	Cd (ppb)	Cu (ppb)	Fe (ppb)	Mg (ppb)	Mn (ppb)	Mo (ppb)	Pb (ppb)	Sb (ppb)	Se (ppb)	U (ppb)	Zn (ppb)
NGTL283	GN	TV	4.2	nd	169.4	0.2	6.9	99.2	16	95	1,563.3	1.3	484.0	0.1	0.3	0.2	981.6
NGTL284	GN	TV	4.4	nd	201.5	0.1	2.3	42.6	19	169	943.6	1.3	839.6	0.2	0.2	0.1	299.9
NGTL302	OU	PMR	3.7	1,220	2,711.9	2.7	1,096.0	2,316.6	6,419	6,936	60,229.0	2.2	1,481.3	28.8	20.9	2.1	164,969.4
NGTL307	OU	PMR	3.2	nd	789.5	0.7	3.8	83.4	470	522	962.4	1.3	201.2	0.6	0.6	0.7	605.7
NGTL308	OU	PMR	4.0	nd	1,301.4	0.7	60.7	1,421.6	424	1,346	7,240.8	1.0	67.8	0.3	2.5	1.1	8,501.0
NGTL313	IS	TV	2.7	nd	1,330.7	0.8	38.5	82.8	12,846	293	108.2	1.0	1,825.0	0.6	2.6	0.6	5,585.8
NGTL316	IS	TV	3.3	350	508.8	1.0	1.8	454.8	491	310	305.5	0.5	1.0	2.3	0.2	0.2	375.0
NGTL325	IS	TV	3.5	nd	33.8	0.2	6.1	112.6	201	574	827.5	0.6	2,464.4	0.7	0.2	0.2	410.6
NGTL345	RM	TV	3.5	nd	34.1	0.3	0.5	13.4	185	141	75.8	0.6	5.2	0.1	0.2	0.0	96.5
NGTL356	RM	FC	2.8	830	297.2	0.1	0.0	16.1	4,492	207	20.7	0.1	29.2	0.1	0.2	0.2	23.4
NGTL375	SN	TV	4.3	50	63.4	0.4	1.1	4.2	13	122	65.1	0.3	14.4	1.0	0.2	0.0	294.5
NGTL376	SN	TV	5.2	30	0.2	0.1	21.9	2.2	20	33	101.3	2.5	101.2	9.5	0.2	0.0	911.1
NGTL377	SN	TV	5.6	40	1.4	<0.1	0.5	0.4	106	430	18.2	13.1	1.8	0.1	0.2	0.0	6.4
NGTL386	TE	TV	7.7	nd	281.7	0.0	33.2	77.5	2,761	2,356	1304.9	8.4	20.0	0.5	<10	1.6	1355.2
NGTL530	GB	PMV	5.3	3	21.0	1.6	0.1	0.7	9	46	3.2	0.9	11.3	0.2	0.2	0.0	8.5
NGTL534	TC	PMR	5.8	24	20.9	4.2	0.0	0.6	61	62	3.5	0.2	5.6	0.7	0.2	0.0	3.3
NGTL625	IS	TV	2.8	1,250	5,758.6	4.0	8.9	332.2	50,235	4,067	1,726.9	0.7	3.9	0.3	0.2	0.7	2,005.7
NGTL626	IS	TV	2.6	>2,000	4,363.9	48.3	5.0	1,132.7	122,801	3,920	145.7	36.3	322.0	0.4	6.1	2.2	526.2
NGTL631	IS	TV	4.1	37	42.7	0.7	0.2	13.9	993	210	26.3	1.1	5.6	1.7	0.2	0.0	26.7
NGTL659	SN	TV	4.9	48	1.4	<0.1	1.9	0.8	145	229	45.8	12.4	6.7	0.1	0.5	<0.1	69.5
NGTL660	SN	TV	2.8	621	309.0	0.2	51.2	712.2	2,935	181	304.2	1.8	7,453.2	0.2	0.7	0.4	5,391.2
NGTL661	TE	TV	4.2	88	0.2	<0.1	49.2	766.3	171	88	4,655.2	<0.05	7,527.6	0.1	2.5	0.0	11,663.6
NGTL699	SN	TV	2.8	620	589.3	0.6	51.4	1,090.4	4,474	352	514.5	13.2	7,045.4	0.7	0.2	0.3	8,326.6

deposits according to finer points of geologic age, ore genesis, mineralogy, and additional attributes necessary to properly evaluate the economic potential (likelihood of occurrence, ore grade, ore tonnage, and ore metal-lurgy) of a geologic terrain or prospect. These special-ized classifications include much detail that is not neces-sary for a general understanding of mine-related geo-chemical impact on the environment. A much simpler system is used here (table 5) that emphasizes fundamen-tal geology (host-rock lithology) and ore composition, both of which play major roles in the generation and dispersion of acidic waters and metal contamination. Compositions of mined and mineralized rock samples are summarized in figures 2 and 3.

### Polymetallic Deposits

Many of the mines and prospects in the CWS are of the polymetallic type in which silver was the com-modity of prime interest in the past. These deposits also carry substantial amounts of base-metal sulfide minerals, allowing some of the mines to recover copper, lead, or zinc. Gold was a byproduct in some. Many of the classic original mines of the region worked ores of this type. Although the polymetallic character, produced by abun-dant base-metal sulfide minerals with important traces of silver and gold, is generally similar among this group, economic geologists will recognize that there are many differences within this group. Some examples of this type are small, simple veins filled by quartz and sulfide minerals, formerly amenable to mining by small under-ground operations, as in the Ruby and Galena districts. Some were larger and deeper vein systems, as at the Idarado or Camp Bird mines of the Red Mountain and Sneffels districts. Some are vein and replacement zones in limestone, as at the Tincup or Tomichi districts. Some of the systems formed close to intrusive stocks and are rich in molybdenum and tungsten, as in the Quartz Creek or Eureka districts.

Polymetallic deposits have the potential to create acidic mine or rock drainage from the high amounts of sulfide minerals, and they have the potential to release many toxic heavy metals in those acidic waters (Plumlee and others, 1995a, 1995b; Plumlee and others, 1999). However, in this study and in ongoing research for the Animas watershed near Silverton (Nash, 1999a, 1999b), it is clear that geologic setting can be even more impor-tant than deposit mineralogy in the source and fate of metals and acid waters: host rocks can effectively miti-gate locally acidic mine drainage at the mine site or close to it. Thus, the polymetallic group of deposits in the



**Photograph 1.** The Dolores River at Slickrock creates an oasis for plants and wildlife and is in sharp contrast to the dry hills nearby. Uranium mine dumps are close to the riparian zone in many places. There is no mine drainage in this area, but rare extreme storm events create runoff and can erode mine dumps.



**Photograph 2.** Cedar-pinyon forests are characteristic of mesas in the Uravan district of western Colorado. This vegetation indicates that somewhat higher precipitation falls on the mesas relative to the lower valleys.



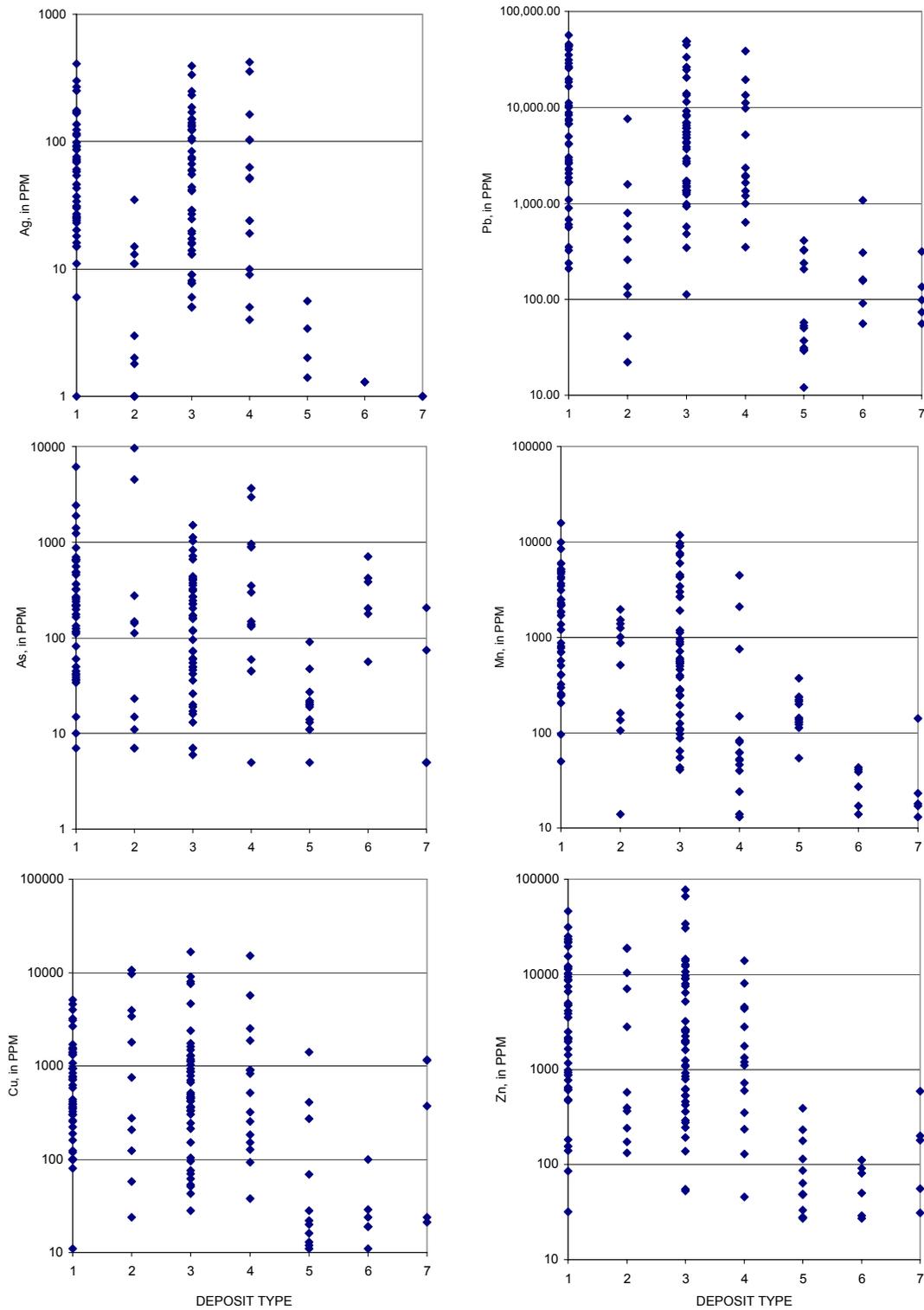
**Photograph 3.** The Cochetopa district is typical of sagebrush semiarid climate of the hills south of Gunnison at an elevation of about 9,000 ft.



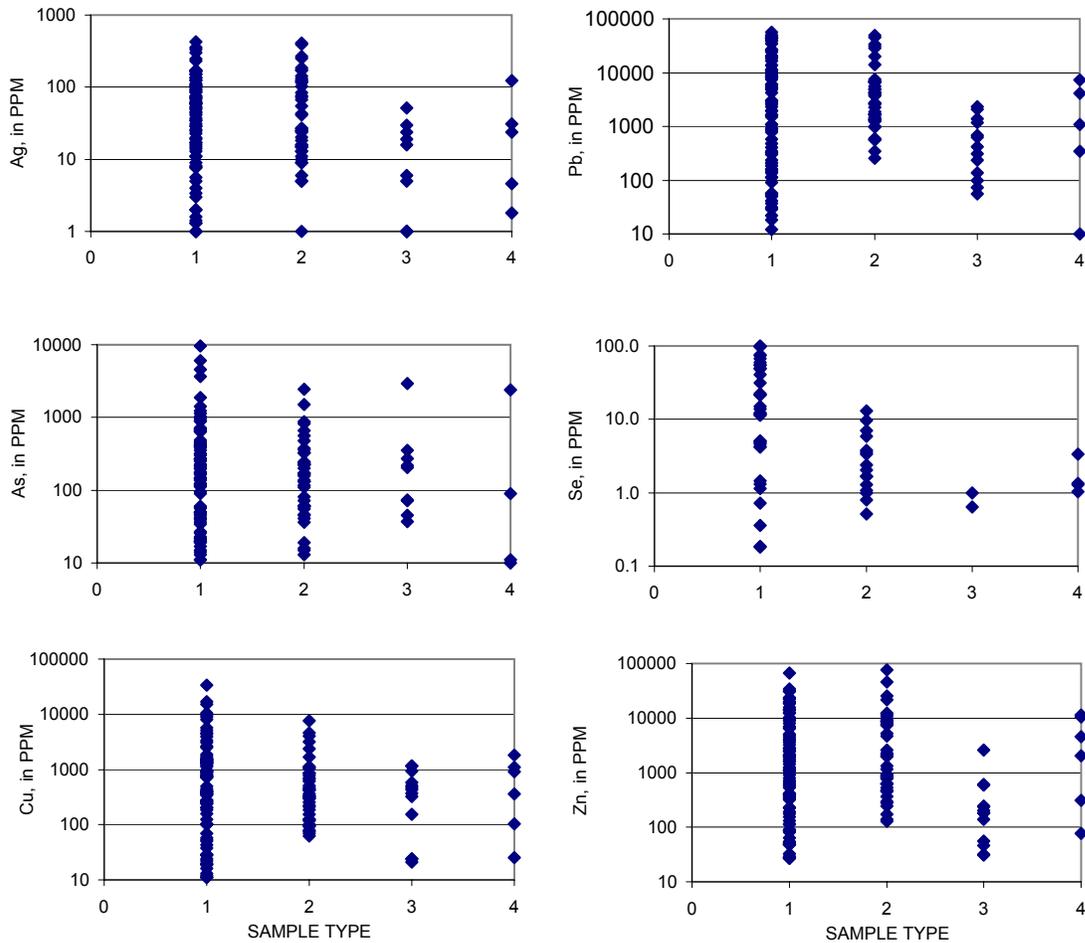
**Photograph 4.** The Marshall Pass uranium mining area at about 11,000 ft elevation receives more than 200 inches of snow. Reclamation work has been done on waste dumps and the open pit of the Pitch mine, in the distance. The high amount of precipitation here, relative to other mining areas on the CWS, helps revegetation but creates problems in the pit lake and mine drainage that do not occur in other, drier uranium mining areas.

**Table 5.** Simple classification of mineral deposits for environmental assessment, CWS, Colorado.

Geologic setting	Ore controls	Mineralogy	Geochemistry	Examples
<b>1. Polymetallic replacement, veins, or breccia pipe</b>				
<b>A.</b> In altered igneous, sedimentary rocks, limestone	Faults, veins, replacements	Multiple sulfides: pyrite, chalcopyrite, galena, sphalerite, Ag-minerals, arsenopyrite.	High Ag, As, Cd, Cu, Mo, Pb, Zn, with Au, Se, Te	Whitepine; Tincup
<b>B.</b> In volcanic rocks	Faults, veins, caldera margin	As above	As above	Idarado, Camp Bird
<b>C.</b> In breccia pipe	Pipes, small intrusions	As above, silica	As above, high As	National Belle, Guston
<b>2. Gold-quartz vein</b>				
Tertiary volcanic rocks, some in pre-Tertiary rocks	faults, breccias	Low sulfide content, low galena, sphalerite, high adularia	High As, Sb, Hg, Ag, Au, Se, Te, $\pm$ Cu, Pb, Zn	Late-stage veins Idarado, Camp Bird, few mined alone
<b>3. Massive sulfide</b>				
Metavolcanic terrane	Bedded layers, shears	Very high pyrite, Cu-sulfides, sphalerite,	Cu, Cd, Fe, Zn, $\pm$ Pb, As, Sb, Ag	Vulcan, Cochetopa
<b>4. Sedimentary-hosted copper</b>				
Sandstones, esp. redbed sequences	Faults, bedding	Cu-sulfide, Cu-oxides low sulfide content	High Cu, Ag? Local U-V?	Cashin
<b>5. Vein uranium</b>				
Igneous, sedimentary; or volcanic rocks	Faults, veins igneous contacts	High pyrite, uraninite, autunite, iron oxides,	High U, radium, $\pm$ As, Cu, Pb, Se, Zn $\pm$ Cu-Pb-Zn sulfides	Pitch mine, Los Ochos
<b>6. Sandstone uranium</b>				
Sandstone, shale	Bedding faults, organics	Uraninite, carnotite autunite, pyrite	High U, V, Se, Mo, radium	Uravan district
<b>7. Porphyry Cu-Mo</b>				
Igneous stocks, intruded sediments	Contacts, veins stockworks	High pyrite, Cu-sulfides molybdenite, Cu-oxides $\pm$ fluorite	As, Cu, F, Mo, S, Pb, U, Zn	Mount Emmons



**Figure 2.** Composition of mined materials by deposit type, central Western Slope, Colorado. Deposit-type codes: 1, polymetallic vein and replacement; 2, volcanogenic massive sulfide; 3, polymetallic veins in volcanic rocks; 4, breccia pipe; 5, sandstone uranium; 6, vein uranium; 7, ferricrete.



**Figure 3.** Composition of mineralized rocks by sample type, central Western Slope, Colorado. Sample types: 1, mine dump; 2, mill tailing; 3, unmined altered rock; 4, smelter slag.

CWS can be subdivided according to host rocks and wallrock alteration into five classes: (1) volcanic-hosted, highly altered red rocks; (2) volcanic-hosted breccia pipes, silicified rocks; (3) volcanic-hosted, weakly altered green rocks; (4) carbonate-hosted, in sedimentary rocks with high acid-neutralizing capacity (ANC); and (5) other settings with low ANC, such as granite and quartzite. Deposits of types 2 and 3 have natural capacities to mitigate harmful effects by neutralizing acid and causing reactions that reduce metal concentrations (Smith and others, 1994; Plumlee and others, 1999). Deposits of type 4 tend to neither enhance nor mitigate acid generation and metal transport, whereas the setting of type 1 tends to enhance problems because they contribute acid and metal sources in unmined altered rocks and the unmined wallrocks have little or no ANC (alteration processes previously destroyed minerals capable of consuming acid). Because the polymetallic deposits are so numerous and important in the CWS, discussion of the four types is expanded below.

*Volcanic-hosted deposits in highly altered red rocks.—*

The prototype area is Red Mountain Pass where several mountains of bright red volcanic rocks help create some of the

worst metal contamination in Colorado from both mined and unmined sources (Plumlee and others, 1995c; Plumlee and others, 1999). Mine drainage from portals, and natural springs in unmined rocks in this large area, commonly have pH values below 3, and metal concentrations are commonly 10 to 1,000 times greater than water-quality standards. The Summitville mine in the eastern San Juan Mountains is in this class (Plumlee and others, 1999) and is now famous for creating terrible water contamination because of inadequate mining and milling methods (King, 1995). Some examples of this class contain relatively rare minerals such as alunite (aluminum sulfate) that can help contribute acid, but the chief problems are widespread pyrite in altered rocks and the low ANC of altered host rocks (Plumlee and others, 1999). Economic geologists term this alteration “acid sulfate type” for the sulfate minerals and the acid hydrothermal conditions of formation, not the acid mine drainage. This alteration, shown by the bright red iron oxide minerals that form during weathering, and the associated polymetallic deposits tend to form in volcanic centers and along caldera margins, but they can form in other settings, as in the small intrusion at Calico Peak, west of Rico, Colo. For

the CWS, examples are the Red Mountains of Red Mountain district and Red Mountain of the Galena district near Lake City.

*Volcanic-hosted breccia pipe deposits in silicified rocks.*—Breccia pipe deposits rich in silver were the first bonanza ores on the CWS and were aggressively mined in the Red Mountain district through the mid-1890's. These pipe deposits are associated with the red, acid-sulfate alteration zones but are distinguished here because these deposits are mined in many places, are characterized by intense silicification of brecciated and leached volcanic rocks, and are known to be rich in arsenic-bearing ore minerals (Burbank and Luedke, 1968; Burbank and others, 1972; Fisher and Leidy, 1972). The silica alteration zones have very low ANC, thus the pyrite-rich mine workings and waste materials create very acidic drainage, commonly below pH 2.5.

*Volcanic-hosted deposits in weakly altered green rocks.*—Large areas of San Juan volcanic rocks are green from pervasive alteration called "propylitic" in which minerals such as chlorite, epidote, and calcite are added by hydrothermal fluids (heated waters, such as steam) (Burbank, 1960; Burbank and Luedke, 1961). Calcite and other carbonate minerals are well known for their high ANC, but even the chlorite in these green rocks has similar ANC according to recent research (Desborough and others, 1998). The beneficial properties of green alteration has been recognized (Smith and others, 1994), and this study demonstrates the wide distribution and importance of the green alteration in volcanic rocks. In some detail, pH's near 7 rather than 8 suggest that the buffering may be produced by chlorite rather than calcite; water in equilibrium with calcite should have a pH of about 8.1. The importance of these rocks is well shown in the Red Mountain district, where the eastern side, in red rocks of the Red Mountains, has numerous sources of highly acidic, metal-rich waters, whereas the western side, in green rocks, is characterized by circum-neutral waters. Equally dramatic, the extensive mining and milling in the Sneffels district creates relatively minor contamination in Canyon Creek, whereas similar practices in the Red Mountain Creek area have contributed highly polluted waters that overwhelm natural attenuation processes. Fortunately for the CWS, type-3 green alteration is present in more mining districts than the red type 1.

*Carbonate-hosted deposits.*—Polymetallic sulfide deposits were important as veins and replacements in pre-Tertiary sedimentary rocks in the Ouray, Tomichi, and other districts on the CWS, and these are characterized by relatively minor water contamination. The pH of surface waters is neutral to slightly alkaline, and the metals of concern are Cd and Zn. The beneficial effects of carbonate rocks are well known (Smith and others, 1994; Plumlee and others, 1999) and more easily predicted from geologic maps and mine descriptions than the similar effects of green volcanic rocks. However, the near-neutral pH environment of carbonate rocks can pose a distinct set of problems relative to the better known acidic mine waters, including the transport of Mn, Mo, Se, As, and U in addition to Cd-Zn (Smith and others, 1994). Further, neutralization

processes on acidic waters that precipitate many toxic metals do not occur in these drainages.

*Other settings.*—Polymetallic veins also occur in siliceous sedimentary, igneous, and metamorphic rocks that lack significant ANC. Veins in Precambrian granite gneiss near Cumberland Pass create local, small volumes of acidic mine drainage. If the deposits were large, there would be potential for major impact because silicate minerals in host rocks and alluvium are not capable of neutralizing the acidic waters. Deposits in these settings were mined in the Gold Brick and Quartz Creek districts, but the deposits of this type are neither large nor common on the CWS and do not appear to pose significant problems.

## Gold-Quartz Vein Deposits

Veins in which gold is the chief or only commodity of value have been mined in many parts of the Western United States, and some have been mined on the CWS. Because of the high unit value of gold in recent years, deposits containing only trace concentrations of gold (about 2 parts per million) have been mined and continue to be mined in Nevada. In many of these Nevada deposits, the concentrations of sulfide minerals and base metals is very low, which means they pose few environmental hazards (Nash and Trudel, 1996); most large mines in Nevada are in sedimentary rocks with high ANC, and acidic drainage is rarely a problem (Hofstra and others, 1995). On the CWS, gold-quartz-adularia veins are well known and have contributed important Au values to many mines. These veins occur along polymetallic veins, but they formed at a slightly different stage, and, generally, both vein types are mined together. In the 1980's the possibility of selectively mining gold veins in the upper part of the Camp Bird mine was investigated. The waste products from gold-bearing veins are nearly always commingled with waste from polymetallic veins. In this study, it was not possible to collect samples of mine waste or mine drainage to characterize this type of deposit, thus gold veins will not be considered further as a separate ore type in this investigation.

## Massive Sulfide Deposits

These deposits are extremely rich in pyrite and can be mined for their content of sulfur, copper, zinc, or byproduct silver and gold. The deposits occur as layers or lenses of pyritic rock that contain high to very high concentrations of As, Cd, Cu, Fe, Pb, Zn, and other metals (Day and others, in press). Their chemical and mineralogical composition is similar to polymetallic deposits. The rock units containing these sulfide deposits on CWS have been metamorphosed, but their bulk composition has not changed greatly and the newly formed minerals contribute to a generally high ANC that is similar to the green-altered volcanic rocks described above.

The high concentrations of pyrite and base-metal sulfide

minerals in massive sulfide deposits can produce extremely acidic drainage, as in the Shasta district, California (Alpers and Nordstrom, 1991; Taylor and others, 1995; Plumlee and others, 1999). The combination of bad mining practices and wet climate can cause severe environmental degradation, but not all massive sulfides have this impact. The combination of dry climate and deep oxidation in the Basin and Range Province of Arizona and Nevada greatly reduces the tendency for acid generation (Nash and others, 1996). Also, the green propylitic alteration of basaltic host rocks, common in this deposit type, introduces calcite and chlorite that provide high ANC. Water draining from many massive sulfide deposits is neutral to weakly alkaline, caused by the beneficial effects of altered wallrocks. Despite geologic differences, massive sulfide deposits have many geochemical features in common with polymetallic deposits: ore compositions alone would suggest huge problems, but natural mitigation by wallrocks can greatly reduce risks in some areas.

Massive sulfide deposits were mined in the Cochetopa district south of Gunnison (fig. 10). Compared to deposits in Arizona and California, the known deposits are small; the largest mined deposit was the Vulcan–Good Hope (Hartley, 1983). As will be discussed later, the relatively dry climate of this mining area may be a big factor in minimizing contamination from mines in the Cochetopa district.

## Sediment-Hosted Copper Deposits

More than 100 years ago, vein-like deposits rich in copper and silver were located in red-bed sandstones on the Colorado Plateau (Fischer, 1936) in what would later become uranium-vanadium country. The copper-silver deposits were mined intermittently through 1945 by relatively small mining operations that are often forgotten in the shadow of the more famous mines of the San Juans. However, new geologic information indicates that the copper minerals are not confined to fracture zones, but are broadly disseminated in a much larger volume of rock (Hahn and Thorson, 1995). Advances in metallurgical methods suggest that new, environmentally friendly methods can be used to good advantage on these mineralogically simple ores. Consequently, a permit has been granted for mining this variety of ore in the Lisbon Valley, Utah, a short distance west of the Colorado border, and preliminary drilling results suggest a similar mine may be viable at the site of the former Cashin mine in western Montrose County (fig. 1) (Hahn and Thorson, 1995).

These copper deposits, with byproduct silver, are now believed to form in structures related to the salt anticlines of western Colorado–eastern Utah. These structures are fairly well known, but because the copper deposit model is so new, few of the structures in Colorado have been tested for metals. The potential for these deposits is not well known, but several of these relatively small deposits could exist on the Western Slope. Environmentally, the metals of concern are Cu and Ag, with relatively small amounts of sulfur and arsenic. Acidic

waters are not likely to be generated or persist in the calcite-bearing sandstone host rocks.

## Uranium Deposits

Many kinds of uranium deposits in plutonic, volcanic, and sedimentary rocks have been identified in Colorado. The behavior of uranium and associated metals is quite different among the various types of uranium deposits, but many of the problem metals tend to be mobile in either highly acidic or alkaline waters. Mitigation of the metals is technically different for acidic versus alkaline conditions.

On the CWS, three types of uranium deposits have been mined: (1) uranium-vanadium deposits in sandstone that are widespread and famous in the Uravan district, westernmost Colorado-Utah (also called Uravan mineral belt) (Chenoweth, 1981); (2) veins of uranium in sandstone in the Los Ochos subdistrict of the Cochetopa district, southeast of Gunnison (McCulla, 1980); and (3) vein-like and breccia-filling ores in dolomite and adjacent sedimentary rocks at the Pitch mine, Marshall Pass district (Nash, 1988) (fig. 1). Other minor occurrences of uranium have been prospected and mined in small operations at other scattered sites that will not be considered here. Examples of sandstone-type and vein-type uranium deposits were examined briefly for this study.

The environmental geochemistry of uranium mines is more complex than for other metal mines because radiation and decay products of uranium, such as radon and radium, also are involved. Regulatory standards are established for uranium and radionuclides in water and in mill tailings but have not been set for dump waste rocks (CWQCC, 2000). Standards for released water and base metals are the same as those for metal mines. Mine reclamation generally focuses on exposure to radiation rather than chemical concentrations of uranium, but release of uranium and radionuclides to surface and ground waters also is a concern and water-quality standards must be met (CWQCC, 2000).

## Porphyry Cu-Mo Deposits

Deposits of copper and molybdenum, associated with shallow-level igneous stocks having characteristic porphyritic texture (large crystals mixed with very fine ones), are the major source of Cu and Mo in the United States and the world. Mines range in size from large to huge (as at Climax, Colo.), and commonly are open-pit operations. The geologic features with potential environmental implications include (1) large tonnages of rocks with high sulfide and base-metal concentrations; (2) igneous rocks and alteration zones that can have low to moderate acid-neutralizing capacity; and (3) high fracture permeability that promotes percolation of surface waters and encourages acid-generating and metal-liberating reactions. The environmental aspects of porphyry deposits are reviewed elsewhere (Cox and others, 1995; Ludington and others, 1995;



**Photograph 5.** The Hough mine, Palmetto basin, is above treeline at an elevation of about 12,000 ft. High precipitation, mostly as snow, and short growing season complicate the environmental effects of mines in this setting. Mine adits and mine-dump runoff tend to be problems in this environment.

Plumlee and others, 1999). Metals of major concern include As, Cd, Cu, Fe, Mo, Pb, S, Sb, Te, and Zn from the Cu-Mo deposits. The Mo-rich variety (termed “Climax-type”) have the potential to produce acidic waters that are rich in F and U.

No porphyry-type deposits have been mined on the CWS, but several have been identified. The most thoroughly studied is the Mt. Emmons molybdenum deposit near Crested Butte (fig. 10 and fig. 16) (Thomas and Galley, 1982), for which economic feasibility and engineering studies have been made. I did not attempt to sample the molybdenum systems near Crested Butte because AMAX has studied the mineralized rocks and drainage waters for many years and is working closely with the State and USFS on both historic and possible future mine sites. Geologically, the Mt. Emmons deposit differs from mined deposits of the “Climax type” in that sedimentary rocks host the ore. Based on observations on other sulfide deposits in sedimentary rocks (Plumlee and Nash, 1995; Plumlee and others, 1999), we can predict that sedimentary rocks will provide more ANC than granitic host rocks, thus environmental degradation may be less.

### Geochemistry of Mined Mineralized Rocks

The chemical composition of mined rocks in 149 dump and tailings samples collected in this study show a wide range in element concentrations, generally consistent with the introductory descriptions of deposit types above. Some unexpected

trends are evident in the new chemical data, however, that merit discussion. Although the number of samples collected is relatively small, the same methods and philosophy of sampling were used by one person, the author, so these results may be consistent and representative of the spectrum of compositions in the study area. Only a few of the samples were collected as “high grade” (atypical of the mine dump). The unexpected trends, shown in figure 2, is that the four major types of metal deposits (sediment-hosted polymetallic vein and replacement, volcanogenic massive sulfide, volcanic-hosted veins, and breccia pipes) have generally similar ranges in metal concentrations. Only the uranium vein and sandstone deposits are clearly distinct for their lower metal concentrations (which is not surprising). The polymetallic character, with substantial enrichment in many base metals, is evident for the four major deposit types. For this analysis, polymetallic vein and polymetallic replacement deposits in pre-Tertiary sedimentary rocks were lumped together because, in most districts, the two ore types are found within a mine and are mixed on the dumps and tailings piles. The limited data suggest that there are a few distinctions, such as lower concentrations of Ag, Pb, and Mn in massive sulfide deposits relative to the other three, and U is clearly lower in the metallic deposits.

One inference from this chemical information is that the four major metallic deposit types on the CWS can be expected to have generally similar high concentrations of base metals in their mine waste. The important differences between deposit types is the lithology and chemical properties (ANC) of the



**Table 6.** Summary of mills, tailings, and smelters, CWS, Colorado.

[The “Mill sites” column includes several sites that contained tailings but no obvious mill building. The “Tailings sites” column does not include the many sites that have no visible tailings near confirmed structures. At more than half of the tailings sites, it appears that major amounts (50 percent or more?) have been lost to erosion. Approximately 12 of the tailings impoundments have been reclaimed]

District	Mill sites	Tailings sites	Smelter sites
<b>Lake Fork of Gunnison watershed</b>			
Burrows Park	1	0	0
Galena	6	5	3
Lake Fork	2	1	1
<b>Gunnison River watershed</b>			
Cochetopa	1	0	1
Gold Brick	4	3	0
Quartz Creek	2	0	0
Ruby	3	2	0
Tincup	2	2	1
Tomichi	1	2	1
<b>Uncompahgre River watershed</b>			
Ouray	3	3	0
Red Mountain	6	6	1
Sneffels	8	7	0
<b>San Miguel River watershed</b>			
Cashin	1	1	0
Mount Wilson	1	0	0
Iron Springs	7	7	0
Placerville	1	1	0
Telluride	16	8	0
<b>Uranium mills</b>			
	5	5	0
<b>TOTAL</b>	<b>70</b>	<b>53</b>	<b>8</b>

the success or failure of mining ventures. The legacy of mill processing of ores is numerous small to very large piles or “impoundments” of finely ground mineralized rock known as tailings. The technology of mills and smelters, and the location of mills, smelters, and tailings are poorly documented in the literature but were emphasized in this investigation because this class of mine-related waste has not received adequate inventory and study for its role in contamination of waters. Significantly, only a few percent of historic mills and smelters are listed in electronic databases of the Department of Interior, such as MRDS and MAS/MILS (McFaul and others, 2000). Also, electronic search of literature tends to identify chiefly uranium mill tailings because they have been the subject of much litigation and reclamation effort over the past 20 years.

On-site and literature search in this project has identified 8 smelter sites, 70 probable mill sites, and 53 tailings sites (summarized in table 6; descriptions are in table 7); the discrepancy between number of mill sites and tailings sites reflects the total loss of visible tailings from many of the older mills. Locations of most mill tailings and mill sites are shown on figure 4. Presumably there are many more mills and tailings sites hidden in obscure places or beyond access on private property. Most mills are on private property (patented mill claims), but in many mining areas tailings may have been placed on public lands, presenting a mixed-ownership problem that complicates reclamation by Federal lands managers.

What is a mill and why are tailings a concern today? Metals of value were tucked away inside rocks and veins and had to be removed before they were refined (smelted) for use. Depending upon the kind of ore, 80 to more than 99 percent of the ore was actually of no value and had to be separated from the gold, galena, sphalerite, or other ore minerals. The first stage was grinding, generally to a size finer than beach sand. Then a variety of ingenious physical and chemical methods were applied to concentrate the minerals by jigs, tables, amalgamation, and floatation. For some ores, cyanide was used to dissolve the metal of interest; cyanide is highly toxic when fresh, but generally degrades with exposure to air (Smith and Mudder, 1999) and is not a problem in the historic wastes investigated here. After going through several stages, involving tanks and pumps and thickeners, one or more concentrates were created, and a great deal more fine sand (tailings) were pumped out as a slurry (similar to wet concrete) and disposed of. The easiest way to dispose of tailings was to pour or pump them into a nearby stream; in the 1930’s “ponds” were devised to collect the solids while also allowing the water and chemicals to be recycled through the mill.

Mill tailings are a class of mining-related waste that pose special problems for the environment. First, mills and tailings nearly always were located near streams. Early mills used water power, and all mills required large amounts of water for ore processing. Second, gravity was utilized in mills: crushed rock and tailings were processed downward, and tailings were always placed below the mill (which generally was below the mine), and that typically means the tailings were placed in flood plains and riparian zones. Third, the fine grain size of tailings tends to make the constituents more reactive than in coarse mine waste or in unmined rocks. As a rule of thumb, there is much more interaction of tailings with water than there is for other kinds of mine waste. Also, fine tailings are easily eroded during storm events, such as flash floods, and are much more likely to interact with floodwaters than other kinds of mine waste. Mill tailings are not a de facto threat to water and the natural environment, but they should be a high priority for evaluation. The character of mill tailings is currently difficult to assess from the desktop, but there is one good predictor: large mines required large mills that produced large tonnages of tailings (often more than a million tons), and these are more likely to be a significant problem than a 100-ton pile.

Milling and associated waste evolved dramatically from 1875 to the present. Three general stages can be highlighted.

1. Early (1875 to about 1900): Many of the bonanza ores could not be treated by mills close to the mines because milling technology was crude. Thus mining was highly selective and created very rich products that were sent directly to smelters for recovery of valuable metals. The early mines often had very narrow workings for selective mining, and the ore typically was hand sorted, both inside the mine and outside, to create “direct shipping” ore for the smelters. On the CWS, some small smelters were built to serve early mines near Lake City, Red Mountain, Silverton, Tincup, and Whitepine, but these generally were ineffective and failed. Large smelters in Denver, Pueblo, and Durango were capable of efficient recovery of silver and gold, but transportation was difficult and costly, which created the demand for railroads on the Western Slope. The combination

of greatly diminished silver prices between 1891–95 and depletion of rich “bonanza” ores close to the surface forced closure of many of the early mines that created much wealth in a short period of time. The consequences today:

- Relatively small amounts of surface disturbance and limited subsurface workings because the mining was very selective and nearly always underground,
  - Few mills and small volumes of tailings,
  - Small, generally unreactive slag piles created by local smelters.
2. Middle stage (about 1900 to 1935): This was an era of consolidation of small mines into fewer major mines. The mines were served by deep tunnels and long aerial tramways and were supported by mills with new technology capable of separating and concentrating gold, silver-lead, copper, and zinc. The ores mined from



**Photograph 6.** Slag is a black, vitreous rock-like substance produced when ores are melted during smelting. The slag shown here was poured close to Tomichi Creek but probably does not contaminate the water because it is relatively unreactive.

large veins, as at Camp Bird, Idarado, and Sunnyside mines, were described as “low grade” because they contained much lower concentrations of ore metals than the early bonanza ores. The mining was possible chiefly because of the new milling technology (Burbank and Luedke, 1969). During this era, larger tonnages were mined per day, producing larger waste dumps outside the mines and much larger tonnages of mill tailings. The consequences today:

- Large mine dumps that can produce significant amounts of acid and metals in drainage waters,
- Tailings that tend to be either unconfined (released directly into flowing streams) or poorly confined.

Most of the mills dating to this era have only a fraction of their tailings nearby—the majority were never impounded or were lost when impoundments failed in major storm events. The impact of these tailings practices is poorly documented, but there is indirect evidence (described later) and direct measurements (Vincent and others, 1999), that sulfide minerals from tailings now lie in and along streams miles from the mill, slowly releasing base metals to surface waters.

3. Modern stage (post-1935): An Executive order in 1935 placed new regulations on the discharge of tailings into streams. This order enforced new practices developed by major mining companies in Colorado and elsewhere to confine mill tailings in “ponds” and thereby keep the mill sands, slimes, and waters out of nearby streams. Milling technology continued to improve on recovery

methods, but this was not a major change on impact to the environment. Improvements to smelters greatly reduced the output of noxious fumes, but this was not a factor on the Western Slope because the smelters were located elsewhere. After World War II, consolidation tended to create a few very large mines in the major districts, and these mines moved thousands of tons of ore a day. Mass mining by open-pit methods was advanced elsewhere but was tried at only a few small pits on the Western Slope; thus, that technology is not a factor on the CWS as it is in Arizona, Nevada, and other Western States. The consequences today:

- Impoundments of tailings, combined with mills of larger throughput, produced large piles of tailings
- Some tailings impoundments were either inadequately designed or were in the eye of a major flash flood and failed catastrophically.

All tailings impoundments on the CWS lack an impermeable base; thus, water that percolates through the tailings cannot be controlled before it infiltrates ground water. Reclamation of the tailings at the Keystone mill, west of Crested Butte, required a major effort to place a plastic barrier on the downslope edge of the tailings to control drainage. Reclamation of other large tailings impoundments, for example in the Telluride and Red Mountain areas, has not addressed this issue. Risk of catastrophic failure of tailings impoundments has been considered for uranium tailings, but I am not aware of such analysis applied to historic metallic mineral tailings and their management.

*Text continues on page 28*



**Photograph 7.** Mill tailings, produced when mills grind ore, are fine sandy materials that resemble beach sands. In many situations, as shown here at the Camp Bird mine, the tailings are close to streams and are easily eroded during storm events.

**Table 7.** Inventory of mill and tailings sites, CWS, Colorado.

[Name: Informal name from local feature, such as a mine; not reliable or legally correct. Mill: Y, yes, features highly suggestive of mill structure; ?, uncertain. Tailings: amount of tailings observed or reported: S, small (<100 tons); M, medium; L, large; VL, very large (>1 million tons); N, none; Nx, tails eroded. ESRisk: risk of failure in extreme storm; L, low, M, medium, H, high]

TN-ID	Name	District	Mill	Tailings	Lat (N.) (degrees)	Long (W.) (degrees)	ESRisk	Description
<b>Smelter sites</b>								
NGG201	Tomichi	Tomichi	Y	S	38.5169	106.4233	L	Smelter site, with small area of slag pours
NGG454	Lee Smelter Galena?	Henson	Y	Nx	37.9789	107.4975	L	Smelter stack by Henson Creek; no visible slag
NGG501	Vulcan	Cochetopa	Y	S	38.3453	107.0058	L	Some equipment, small mounds of slag
NGG684	Saratoga	Red Mtn	Y	S	37.9461	107.6636	L	Saratoga smelter structure sparse, good piles of slag
NGG992	Lee smelter	Henson	?	Nx	38.0083	107.4500	L	Possible foundation at Smelter Creek jct with Henson Crk; no slag seen
NGG965	Crooke	Galena	Y?	S?	38.0139	107.3111	L	Likely location of smelter on Lake Fork by falls
<b>Mill sites</b>								
NGM102	Cashin	Cashin	Y	S	38.3111	108.9508	M	Wooden mill standing, tails along creek
NGM202	Akron	Tomichi	Y	L	38.5381	106.3961	H	Large concrete foundation, 2 impounds
NGM219	Lot mine	Powderhorn	Y	S	38.3194	107.0917	L	Small mill standing, small tails, dry and ok
NGM232	Carter	Gold Brick	Y	M	38.6128	106.5944	H	Big wooden mill ruins, jigg tables; tails far to east
NGM233	Raymond	Gold Brick	Y	M	38.6208	106.5861	H	Big mill ruins, stone foundation; tails far to east
NGM234	Gold Link	Gold Brick	Y	M?	38.6306	106.5789	H	Big mill ruins ; tails to east
NGM281	Hidden Treasure	Galena	Y	S	38.0189	107.3594	M	Concrete foundation parts;
NGM282	Ute-Ulay	Galena	Y	M	38.0172	107.3486	H	Concrete foundation; several tails impounds near Henson Creek; unstable
NGM286	Pelican?	Galena	Y	NX	38.0200	107.3350	N	Big concrete foundation; no tails
NGM960	Capitol City	Galena	Y	Nx	37.9958	107.4819	L	Mill site reported by B. Hite, BLM (1999)
NGM302	Bachelor	Ouray	Y	S	38.0622	107.6622	L	Wood bldg collapsed, steel machinery; small tails dry above Dexter Crk
NGM306	Banner	Ouray	Y?	Mx	38.0542	107.6875	L	Mill removed, site sold for homes, tails in place beneath homes
NGM308	Silvershield	Ouray	Y	Sx	38.0408	107.6831	L	Tin building stands west of river, tram cable; ocher tails nearby
NGM323	Alta	Iron Springs	Y	L?	37.8850	107.8544	L	Mill burned; foundation clear; tails in two ponds to west T323 and T325
NGM362	Silver Queen	Red Mtn	?	M?	37.9167	107.6864	L	No structure evident at lower level dumps; tails to north must have mill
NGM372	Camp Bird	Sneffels	y	VL	37.9706	107.7272	H	Concrete foundation for mill stands near 14th level portal; large tails to SE
NGM375	Atlas	Sneffels	Y	M	37.9756	107.7578	H	Wooden structure, metal parts, on slope above tailings
NGM386	Society	Telluride	?	M	37.9478	107.8722	M	No structure evident; located near old railroad to south; tails along river

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Table 7. Inventory of mill and tailings sites, CWS, Colorado—Continued.

TN-ID	Name	District	Mill	Tailings	Lat (N.) (degrees)	Long (W.) (degrees)	ESRisk	Description
<b>Mill sites—Continued</b>								
NGM396	Palmetto	Henson	Y	Nx	37.9822	107.5592	N	Small stamp mill below Palmetto Gulch
NGM397	Empire Chief	Henson	Y	Nx	37.9744	107.5222	N	Wooden mill 4 mi west Capitol City; no tails
NGM526	Standard	Ruby	Y	S	38.8786	107.0750	H	Cement foundation next to mine buildings, tails to south
NGM534	Tincup	Tincup	Y	S	38.7278	106.4972	L	Wooden structure collapsed; tram to SE, tails to NW
NGM539	Golden Fleece?	Lake Fork	Y?	Sx?	37.9811	107.2906	N	Concrete structure by lake; tails in lake?; cited by Irving and Bancroft, 1911
NGM542	Sherman	Burrows Park	Y	Nx	37.9022	107.4261	N	Stone foundation; cited by Irving and Bancroft; no tails; waterpowered mill?
NGM606	Mountain Silver	Red Mtn	Y	Sx?	0.9328	107.6611	N	Small mill, metal parts; near adit, no tails found
NGM619	Carbonero	Iron Springs	Y	M	37.8583	107.8194	H	Mill site shown on old topo, nothing visible in trees, some spilled tails
NGM625	Silver Bell	Iron Springs	Y	M	37.8600	107.8689	H	Building intact by old railroad loop, tails to west
NGM628	Caribeau	Iron Springs	Y	M	37.8575	107.8483	H	Some concrete forms, no structures near mine portal, abundant tailings
NGM636	Blistered Horn	Tincup	Y	Sx	38.7028	106.4917	N	Wooden structure collapsed, a few tails at foundation and south
NGM638	Gold King	Iron Springs	Y	S	37.8794	107.8478	L	Collapsed building some steel parts; cited in literature, location on map
NGM649	San Juan	Mineral Point	Y	S	37.9606	107.5925	N	Wooden mill and tanks standing; some red tails in foundation
NGM659	Richmond	Sneffels	Y	M	37.9525	107.7264	H	Richmond basin, mill by falls, medium volume of gray tails in Imogene Creek
NGM660	Camp Bird 3	Sneffels	Y	S	37.9488	107.7312	L	Concrete foundation at upper adit level of Camp Bird; tails in basin/creek
NGM661	Savage	Telluride	Y	S	37.9331	107.7456	M	Small stamp mill remains, jig tables; tails to southwest
NGM663	Tomboy-1	Telluride	Y	S	37.9339	107.7525	M	Wooden structure, tails nearby, SE of main Tomboy mills; Argentine?
NGM665	Tomboy-2	Telluride	Y	S	37.9356	107.7542	M	Foundation and metal, many tails nearby
NGM666	Tomboy-3	Telluride	Y	Nx	37.9375	107.7572	N	Burned remains of mill with 60 or more stamps, largest at Tomboy; no tails
NGM667	Tomboy-4	Telluride	Y	Nx	37.9372	107.7581	N	Burned, just west of large mill, close to creek, used water power?, no tails
NGM668	Tomboy-5	Telluride	Y	Sx	37.9381	107.7583	N	SW corner of cluster, burned; concrete foundation; a few tails in foundation
NGM669	Tomboy-6	Telluride	Y	Nx	37.9372	107.7592	N	Wood and cement piers remain, SW of all; no tails
NGM670	Japan?	Telluride	Y	Nx	37.9378	107.7547	N	Wood remains, with jig tables, no tails
NGM674	Liberty Bell	Telluride	Y	Nx	37.9342	107.7967	L	Stone foundation only, mentioned in literature; tails in river?

**Table 7.** Inventory of mill and tailings sites, CWS, Colorado—*Continued*.

TN-ID	Name	District	Mill	Tailings	Lat (N.) (degrees)	Long (W.) (degrees)	ESRisk	Description
<b>Mill sites—Continued</b>								
NGM675	unknown	Telluride	Y?	Nx	37.9336	107.7944	N	Foundation resembles mill, no tailings evident
NGM679	Eclipse	Sneffels	Y	S	37.9764	107.7472	L	Small wooden mill, small tailings to SW
NGM687	Silver Pick	Mt. Wilson	Y	Nx	37.8733	108.0050	N	Remains of mill at treeline, private cabin site; no tails seen
NGM854	Whitepine	Tomichi	Y	Sx	38.5338	106.3992	H	Stone foundation, not much else; small tails pond to south was breached
NGM870	Hidden Treasure	Sneffels	Y	Sx	37.9499	107.7806	M	Mill burned, some steel, tram cable, foundation, tails
NGM964	Gateway	Uravan	Y	S	38.6750	108.9722	L	Reported in literature; observed by BLM
NGM967	unknown	Ouray	Y	S	38.0589	107.6900	N	Recent custom mill, small capacity
NGM968	unknown	Ouray	Y	Nx	38.0539	107.6906	N	Cement foundation, two ages of building? on railroad ; no tailings visible
NGM969	San Bernardo	Iron Springs	Y	Mx	37.8581	107.8844	L	Wooden mill stands at end of tram to mine; tailings to NW above creek
NGM970	Keystone	Ruby	Y	L	38.8683	107.0361	N	Mill near lowest adit, concrete remains; tails to SE; reclaimed
NGM975	Barstow	Red Mtn	Y	M?	37.9208	107.7111	N	Reported by Smith (1994); location approximated; tails not known
NGM976	Tobasco	Burrows Park	Y	Sx?	37.9344	107.5100	N	Reported by BLM Alpine Loop guide
NGM976	unknown	Quartz Creek	?	Sx?	38.6111	106.4917	N	Reported by Hill (1909), not located, presumed to be small
NGM977	Vancorum	Uravan	Y	R	38.2208	108.6000	N	Older vanadium mill; reclaimed; site from map
NGM978	Slick Rock?	Slick Rock	Y?	R	38.0361	108.9083	N	Slick Rock mill or buying station; reclaimed '90's; site from map
NGM979	Uravan U	Uravan	Y	R	38.3667	108.7361	N	Uravan mill complex; reclaimed '90's; site from map
NGM981	Ohio City	Gold Brick	?	Nx	38.5639	106.5972	N	Small early mill in literature; not visible; no tails?
NGM982	Gunnison U	na	Y	R	38.5250	106.9444	N	Uranium mill for Los Ochos and other U mines; reclaimed in '90's
NGM990	Idarado	Red Mtn	Y	VL	37.9128	107.7014	L	Mill buildings east of Treasury tunnel; tailings to north
NGM993	Pandora	Telluride	Y	VL	37.9306	107.7778	L	Large modern mill from Treasury tunnel (Idarado)
NGM996	Revenue	Sneffels	Y	Mx?	37.9750	107.7494	L	Mill loc from map; described in literature; tails covered by waste rock?
NGM997	Contention	Lake Fork	Y	SX	37.9819	107.2917	M	Mill by Crystobal Lake, tails below along Lake Fork
NGM998	Pitkin	Qtz Creek	Y	NX	38.6072	106.5222	N	Stone foundation, W side of Pitkin village; tails in wetland to SE?
NGM999	Roosevelt	Qtz Creek	Y	NX	38.5772	106.5467	N	Wooden mill with stamps, east side Qtz Creek; no tails; never operated?
<b>Tailings sites</b>								
NGT102	Cashin	Cashin	Y	S	38.3106	108.9508	M?	Light colored sandy tails along creek
NGT202	Akron	Tomichi	Y	M	38.5378	106.3956	H	Gray tailings, southern impound, with pond

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Table 7. Inventory of mill and tailings sites, CWS, Colorado—Continued.

TN-ID	Name	District	Mill	Tailings	Lat (N.) (degrees)	Long (W.) (degrees)	ESRisk	Description
<b>Tailings sites—Continued</b>								
NGT212	Akron	Tomichi	Y	M	38.5383	106.3953	H	Ocher tails, heaped above mill, mixed with waste rocks
NGT216	Whitepine	Tomichi	?	S	38.5247	106.4122	H	Tails in Tomichi Creek; source likely was NMG856 south of Akron mill.
NGT237	Gold Link	Gold Brick	Y	M	38.6294	106.5811	H	Shoestring impound east of mill, ocher
NGT238	Raymond	Gold Brick	Y	M	38.6161	106.5892	H	Shoestring impound east of mill, south and close to creek
NGT239	Carter	Gold Brick	Y	M	38.6111	106.5972	H	Shoestring impound east of mill, north of creek
NGT281	Hidden Treasure	Galena	Y	S	38.0183	107.3578	H	Small tails on S side Henson Crk; mill to N between rd and creek
NGT282	Ute-Ulay	Galena	Y	M	38.0172	107.3819	H	Several tails impounds near Henson Creek; unstable
NGT306	Banner	Ouray	Y?	Mx	38.0542	107.6875	L	Tailings under housing development, on bench above Unc. River
NGT308	Silvershield	Ouray	Y	Sx	38.0408	107.6828	L	Ocher tails on west side of Unc. River
NGT313	Carbonero	Iron Spr	Y?	M-L	37.8558	107.8214	H	Ocher tails in wet area by Howard Fk
NGT315	Carribeau	Iron Spr	Y?	L	37.8575	107.8492	H	Large area ocheer tails on bench near Howard Fk
NGT323	Alta	Iron Springs	Y	S	37.8850	107.8556	L	Smaller of two impounds
NGT325	Alta	Iron Springs	Y	L	37.8856	107.8572	L	Larger area on flat west of mine; reclaimed 1997 but tails still in place
NGT343	Idarado	Red Mtn	Y	S	37.9183	107.8306	M?	One of smaller tails ponds, by Red Mtn creek, not reclaimed
NGT345	Idarado	Red mtn	Y	VL	37.9183	107.6994	L	Large tail impound, mid-north
NGT362	Silver Queen	Red Mtn	?	Sx	37.9178	107.6867	L	Small imound below dump, possibly extends under dump, partly reclaimed?
NGT364	Silver QueenRed Mtn		?	Mx	37.9297	107.6864	L	Larger unconfined slosh of tails
NGT345	Idarado	Red Mtn	Y	M	37.9185	107.6855	H	Unconfined tails in channel of Red Mountain Creek, older stage or spill?
NGT964	Idarado 3	Red Mtn	Y	M	37.9165	107.7010	L	Confined and reclaimed tailings near US550; pond 3
NGT370	Idarado 4	Red Mtn	Y	VL	37.9375	107.6722	L	Huge reclaimed tails, northernmost from Idarado; pond 4
NGT963	Idarado 2	Red Mtn	Y	M	37.9219	107.6994	L	Reclaimed tails, pond 2
NGT962	Idarado 1	Red Mtn	Y	S	37.9189	107.6989	L	Reclaimed tails, pond 1
NGT841	Idarado	Red Mtn	Y	Sx	37.9192	107.6972	H	Fluvial tails as overbank sediments along Red Mtn. Crk
NGT372	Camp Bird	Sneffels	Y	VL	37.9736	107.7236	H	Very large impound along Canyon creek
NGT375	Atlas	Sneffels	Y	M	37.9761	107.7569	H	Gray and ocher tails along creek; mill to west
NGT386	Society	Telluride	?	M	37.9478	107.8722	M?	Surprising area of tails on bank near San Miguel River
NGT526	Standard	Ruby	Y	S	38.8781	107.0750	H	Ocher tails, impound and pond intact, next to Elk Crk.
NGT534	Tincup	Tincup	Y	S	38.7281	106.4978	L	Brown coarse tails, some recently removed; mill to SE

**Table 7.** Inventory of mill and tailings sites, CWS, Colorado—*Continued*.

TN-ID	Name	District	Mill	Tailings	Lat (N.) (degrees)	Long (W.) (degrees)	ESRisk	Description
<b>Tailings sites—Continued</b>								
NGT538	Contention	Lake Fork	Y?	S	37.9867	107.2931	M?	Yellowish tails in draw next to Lake Fork Crk outlet; mill above to west?
NGT619	Carbonero	Iron Springs	Y?	M	37.8561	107.8206	H	Ocher tails in floodplain of Howard fork, side creek flows over tails
NGT625	Silver Bell	Iron Springs	Y	M	37.8617	107.8733	H	Ocher tails piled high above Howard Fork, west of highway
NGT628	Caribeauu	Iron Springs	Y	M	37.8581	107.8492	H	Ocher tails on south side of Howard Fork, wet from mine drainage
NGT638	Gold King	Iron Springs	Y	S	37.8803	107.8494	L	Tan sandy tails west of mill
NGT649	San Juan	Mineral Point	Y	S	37.9608	107.5925	L	Small amount of red tails (roasted) at foundation; low production or in creek?
NGT659	Richmond	Sneffels	Y	M	37.9525	107.7264	H	Richmond basin, mill by falls, medium volum of graytails
NGT660	Hidden Treasure	Sneffels	Y	S	37.9500	107.7303	M	Ocher tails poured down slope into small creek and wetland from NGM870
NGT661	Savage	Telluride	Y	S	37.9328	107.7461	L	Small stamp mill remains; tails to southwest
NGT679	Eclipse	Sneffels	Y	S	37.9761	107.7475	L	Tails about 1 acre, grass growing
NGT868	Camp Bird 3	Sneffels	Y	Sx	37.9482	107.7306	L	Tails east of mill, some probably eroded; covered by 1970's waste dumps
NGT867	X-Richmond	Sneffels	Y	Sx	37.9574	107.7242	H	Fluvial tails as overbank sediments along Imogene Creek, below Richmond
NGT969	San Bernardo	Iron Spr	Y	Mx	37.8589	107.8847	M?	Impounded tailings above creek, not reclaimed
NGT970	Keystone	Ruby	Y	L	38.8678	107.0322	L	Several tailings ponds on topo map; reclaimed in mid-1990's.
NGT985	Pandora	Telluride	Y	VL	37.9333	107.7972	L	Very large impound farther west than T986, S side river
NGT986	Pandora	Telluride	Y	VL	37.9319	107.7833	L	Very large impound W of mill, S side San Miguel River
NGT987	Sutton	RedMtn?	Y	Nx	37.9989	107.6583	N	In Canyon, end of tramway; from map; no tails
NGT988	Idarado	Red Mtn	Y	L	37.9236	107.6931	L	Mid-north in Red Mtn Creek valley, from map
NGT982	Gunnison U	na	Y	R	38.5250	106.9444	N	Tails reclaimed--removed and placed elsewhere
NGT979	Uravan U	Uravan	Y	R	38.3667	108.7361	N	Tailings reclaimed by Superfund action in late 1990's
NGT977	Vancorum	Uravan	Y	R	38.2208	108.6000	N	Tailings and other waste reclaimed
NGT978	Slick Rock?	Slick Rock	Y?	R	38.0361	108.9083	N	Radioactive waste reclaimed in 1990's by UMTRA



**Photograph 8.** The National Belle mine, Red Mountain district, produced rich silver ore from a small breccia pipe within this hill; the mine dump is relatively small (right side) for the value of ore produced prior to 1900. The rusty rocks on the left side are natural exposures.



**Photograph 9.** This picturesque mill on the Crystal River is typical of early mills that utilized water power and discharged mill tailings directly into the river.



**Photograph 10.** Tomboy, east and above Telluride, was the site of at least six mills that served medium-sized mines that converged on this one relatively flat area at 11,000 ft elevation. These mills were active from about 1890 to 1930 and were phased out after difficulties with fires and avalanches in favor of mills at lower elevations that received ores from long tunnels and tramways. Almost no tailings were impounded at Tomboy—because there was no space and no regulations, the tailings were allowed to wash away in Savage and Marshall Creeks.



**Photograph 11.** The Akron mine and mill near Whitepine was active in the 1940's–50's. The large piles of waste rocks and mill tailings were created as ore and waste were hauled out of a tunnel that was driven to the east (right) into several previously mined deposits.



**Photograph 12.** Mining of uranium-vanadium in the UraVan district, from the 1940's through the 1970's, was typically from small- to medium-sized mines (shaft and adits) that brought relatively small volumes of waste to the surface. They left no tailings onsite because ore was hauled to a central mill. The Golden Cycle mine site was reclaimed in 1997.

## Mobility of Metals: Leach Tests of Mineralized Rocks

The soluble or mobile portion of metals in rocks is the part that has the potential to degrade surface waters and harm aquatic life; thus, it is of more interest than the total metal concentrations described earlier. To better understand the mobility of metals, laboratory tests were undertaken to simulate reactions in mined waste materials. Solid materials from dumps, mill tailings, outcrops, and slag piles—all of which can be termed rocks—have been processed in the laboratory by a passive leach method that provides an approximate measure of reactions in nature, such as during storm events. Results of the leach tests, summarized in table 8, span a wide range of concentrations that will be described here and discussed in more detail in later sections on the mining districts.

Leach tests demonstrate great variability in the amount of metal that is soluble in dump and tailings materials; the variability is large from element to element, and from site to site. The results for 116 samples are listed in table 4, summarized in table 9, and shown in figures 5 and 6. The behavior of samples from the CWS are generally similar to trends for more than 150 samples from the Silverton area (Fey and others, 2000a; Fey and others, 2000b) and also from Montana and Nevada (Nash, 2000b). There are many mineralogic and geologic factors that control the water-rock reactions, which are complex

in detail. At the most general level, results for dump, tailings, and altered-rock samples show that leachate metal concentrations generally are highest in dump samples and lowest in altered-rock samples. If one looks more closely, there are large differences even among a group of materials: some tailings samples in a district can release much higher metal concentrations than others (such as the common case of tailings from old, inefficient stamp mills compared to those from modern mills). There also is a wide range in results plotted by deposit type (fig. 5), suggesting that there is almost as much variability within materials from a deposit type as there is between deposit types. This variability is not unexpected to a geologist, but probably is perplexing to a manager who wants to see a clear and predictable pattern to the behavior of metals. From my experience with hundreds of dumps and several hundred tailings impoundments, predictions are a poor substitute for hands-on sampling and testing of materials.

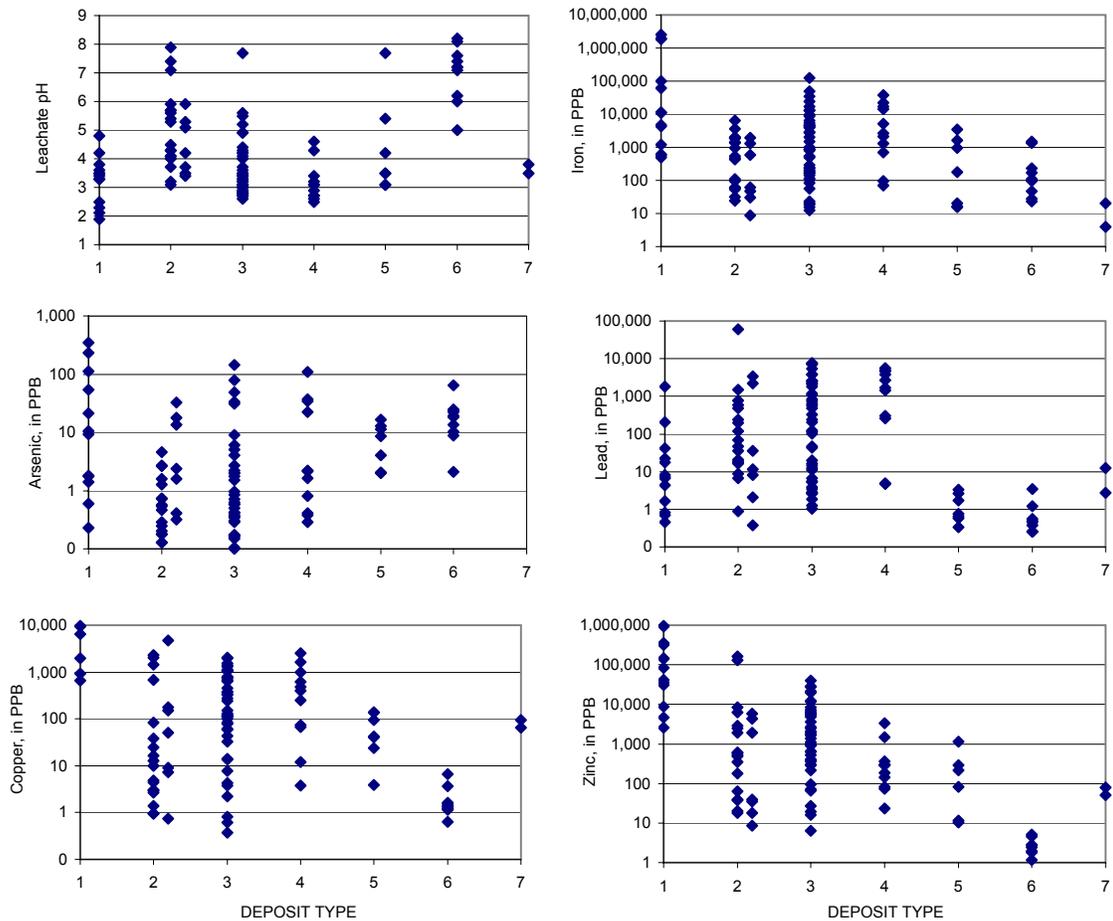
Results of these laboratory leach tests confirm what others are finding (Fey and others, 2000a): release and mobility of metals is strongly correlated to acidity generated by the materials. Nearly all metals of concern have higher concentrations and higher mobility in acidic solutions. The increase is nearly linear from pH 5 to pH 2. The principles underlying this relationship are the same as for mine drainage and stream waters, which are discussed elsewhere in this report and treated in much more chemical detail by others (Drever, 1997; Langmuir, 1997; Nordstrom and Alpers, 1999). The following

**Table 8.** Statistical summary of some selected features of leachate and mine drainage chemistry, central Western Slope, Colorado.

[Chemical data from ICP-MS analyses, reported in parts per billion; pH in standard units]

	Dump leachates (n=76)		Tailings leachates (n=28)		Mine drainage (n=62)	
	Median	Maximum	Median	Maximum	Median	Maximum
pH	3.5	2.5*	4.1	2.6*	4.5	2.0*
Al	344	68,450	43.7	5,760	493	470,000
As	1.8	143	0.4	48.3	3.5	1,000
Cd	3.7	1,310	6.1	1,100	7.0	165
Cu	112	141,000	82.8	2,320	139	28,000
Fe	880	100,300	469	122,800	5,400	1,500,000
Mn	87.8	121,450	146	60,230	1,200	57,800
Mo	1.6	1,980	1.1	36.3	1.7	174
Pb	46.8	61,100	12.7	5,650	28.9	1,374
Sb	0.4	103	0.3	28.8	0.4	13
Se	2.2	707	0.2	73.5	1.2	194
U	0.8	3,400	0.1	18.2	1.1	555
Zn	361	148,300	526	164,900	611	21,500

\* for pH, maximum acidity is minimum pH value.



**Figure 5.** Compositions of leachate solutions by deposit type. Deposit-type codes: 1, massive sulfide; 2, polymetallic vein and replacement; 3, volcanic-hosted polymetallic vein; 4, breccia pipe; 5, sandstone uranium; 6, vein uranium; 7, ferricrete.

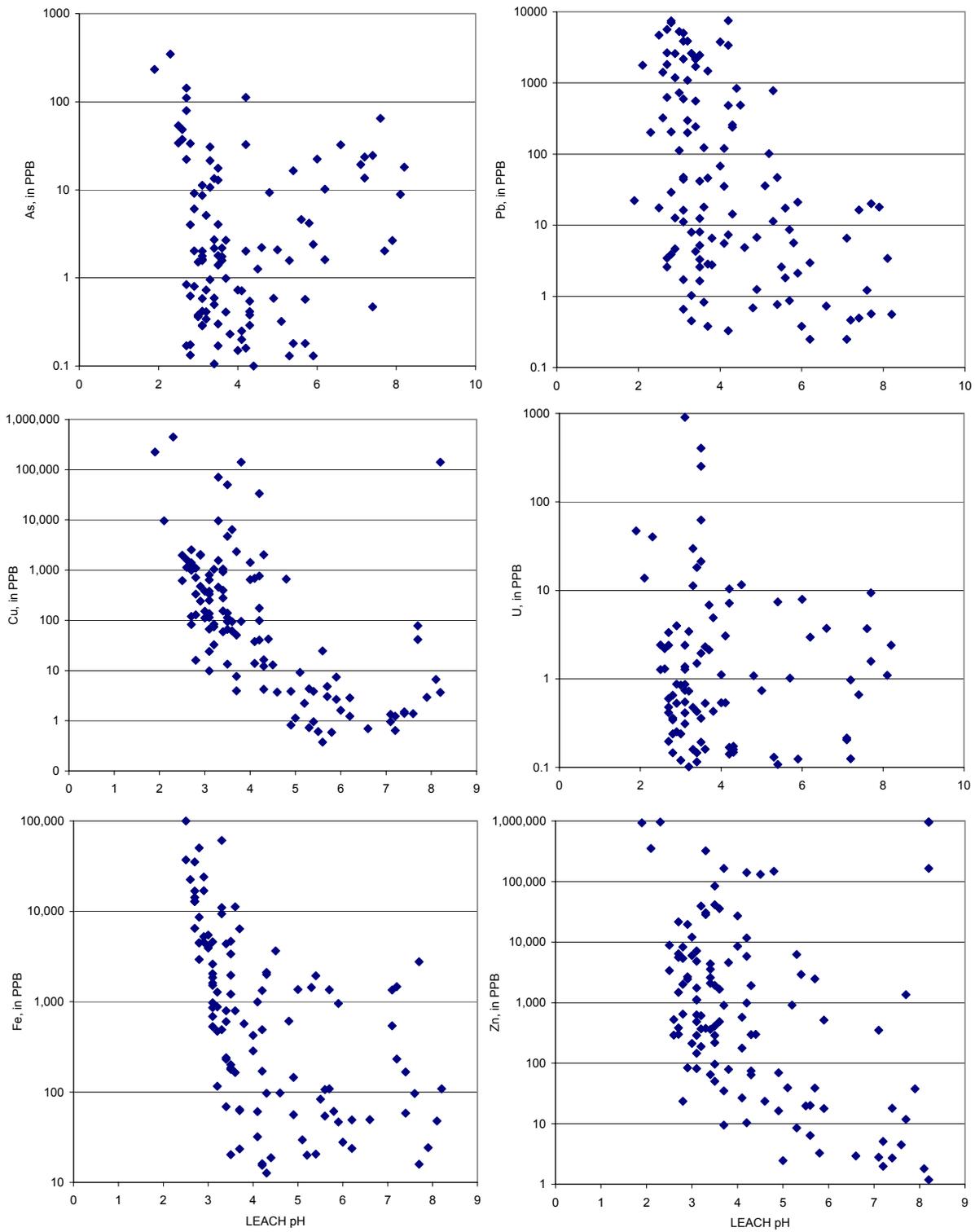


Figure 6. Compositions of leachates from mineralized rocks as a function of pH.

summary comments by element are intended to provide some guidance for the potentially toxic metals expected in metal mining areas of the CWS. However, note that the leach test results are an approximate measure of the potential for the material to release that element, but this may not occur in the real world because of lack of moisture, low permeability, or other factors. Further, the leachate concentrations can only be qualitatively judged relative to regulatory standards for surface waters, a complex topic that will be discussed in the next section of this report. Because the water-quality standard values range from less than 1 ppb to more than 1,000 ppb, for various elements and various beneficial uses, the values mentioned below are provided to suggest an order of magnitude and may not be appropriate for all sites.

**Aluminum.**—Many samples yielded high Al concentrations if the leachate pH was below 4; the median<sup>4</sup> value was 260 ppb. The toxicity standard for aquatic life (chronic exposure) is 100 ppb Al (CWQCC, 2000; discussed later). This tendency for acid mine drainage to carry high amounts of Al is potentially significant to fish habitat.

**Arsenic.**—Mobility of As is not as acid-dependent as true metals, and only 5 percent of the CWS samples released more than 100 ppb (figs. 5, 6). The median was 1.6 ppb. Mined materials on the CWS tend to release relatively low amounts of As, which is good news because higher amounts were expected from predominant polymetallic ores.

**Cadmium.**—This metal is toxic at low concentrations and is commonly released in significant concentrations if the leachate pH is below 5. Nearly half the samples released more than 10 ppb Cd. Mined materials in the study area have the potential to release environmentally significant amounts of Cd.

**Copper.**—This element is toxic to aquatic life (but not humans) at relatively low concentrations (about 5–10 ppb). Mined materials in the study area have a tendency to release significant amounts if the pH is below about 5 (fig. 6). More than half of the samples tested released more than 100 ppb (which is about 20 times higher than the aquatic life toxicity standard). Copper and Zn are potentially the biggest threats to fisheries on the CWS and, for this reason, will be discussed in several subsequent sections.

**Iron.**—This element is mobile in acidic waters (fig. 6) and is commonly released in high concentrations from the materials tested (22 percent yielded >5,000 ppb, which is five times the aquatic life standard). This is consistent with the common occurrence of high Fe in surface waters of the study area and red-ocher precipitates on stream beds.

**Manganese.**—The behavior of Mn is theoretically similar to that of Fe, but in this study it is not as problematic. Although

some samples released high concentrations, only 9 percent were above 5,000 ppb. Manganese appears to be a concern chiefly for drinking-water supplies, for which the standard is 50 ppb, whereas Mn is not a threat to aquatic life until concentrations of about 1,000 ppb are reached (CWQCC, 2000).

**Molybdenum.**—Molybdenum is a common trace metal in the ores studied, but it appears to be less mobile than other metals. Only 6 percent of leached samples released more than 100 ppb (the aquatic life toxicity standard is 19 ppb). This is an element that is relatively mobile under alkaline conditions as the oxyanion molybdate, as shown in leach tests on sandstone-type uranium ores.

**Lead.**—Although lead is abundant in many of the ores, it generally is not as mobile as associated Cd, Cu, or Zn. Leach tests show that a third of the samples released more than 250 ppb at pH <5 (fig. 6); the aquatic life standard is about 25 ppb. This suggests that Pb is potentially a significant problem, but it is less common than some other base metals.

**Antimony.**—Antimony is not listed among the metals that are toxic to aquatic life, but it is of concern to human health. Thus, there is a drinking-water standard of 146 ppb. This element is enriched in many of the ores but does not appear to be very mobile. Only one sample released more than 100 ppb. Because the toxicity threshold is relatively high, and the leach tests suggest that it is not highly mobile, there seems to be low potential for problems from Sb.

**Selenium.**—This metalloid is toxic to aquatic life at relatively low concentrations (5–10 ppb) and is relatively mobile; thus, it may pose problems despite its trace-level concentrations in most ore types. Leach tests show that 16 percent of the CWS samples release more than 25 ppb Se. Selenium is mobile as the oxyanion  $\text{SeO}_4$  at neutral-alkaline pH; thus its behavior is distinct from most of the true metals. Compositions and pH values favorable for Se mobility are most common in sandstone-type uranium ores, but locally it may be significant at massive sulfide deposits such as Vulcan in the Cochetopa district.

**Uranium.**—This element is mobile under both acidic and alkaline conditions, which explains why it can be mobilized from many ore wastes, even ores like those of the massive sulfide type that do not contain much U. Potentially significant levels of uranium were observed in only 4 leachates with pH <5 (fig. 6). The water-quality standard for U is relatively high (44 ppb, converted from the standard of 30 picocuries). The median value for all samples was relatively low at 0.5 ppb.

**Zinc.**—This element is enriched in many ore types and is probably the most mobile and concentrated toxic metal relative to its standard of about 50 ppb for aquatic life. Zinc is not generally a risk for human health. Slightly more than half of the samples tested released more than 500 ppb (10 times the aquatic life standard). Highest values for Zn are in acidic waters below pH 4, but leachates with pH of 5–7 also carried significant Zn (fig. 6). The release of high concentrations of Zn from many CWS samples suggests that Zn is the most important metal to study where there is concern for aquatic life near mining areas on the Western Slope.

<sup>4</sup>Median values are reported here for several reasons: the number of samples is relatively small, the statistical validity of these calculations is not considered to be rigorous, and the data values generally do not have a normal distribution (the presence of extremely high values skews the distribution and creates an average that is much higher than the median).

## Water Quality: Introduction to Standards and Ratings

Water quality is a complex issue that includes both technical (chemical and biological) and legal issues that are far beyond the scope of this report. The framework for water quality utilized in this investigation is the standard set by the Colorado Water Quality Control Commission (CWQCC), which is part of the Colorado Department of Public Health and Environment. This agency sets policies and regulations for protection of water quality as defined by the Colorado Water Quality Control Act, which is very similar to the Federal Clean Water Act, except that the Colorado Act also includes ground waters. The CWQCC works in parallel with the Water Quality Control Division (CWQCD), which “regulates the discharge of pollutants into the State’s surface and ground waters.” The Colorado policies and regulations are complex for the nonspecialist and are most easily accessed through the frequently updated State website ([www.cdph.state.co.us/cdphereg.html](http://www.cdph.state.co.us/cdphereg.html)). Particularly useful from the website is the Colorado Water Quality Management and Drinking Water Protection Handbook with 141 pages of information on policies, methods, and standards. Another source of technical information is the Colorado Nonpoint Assessment Report (CWQCD, 1989).

The Colorado regulations are generally similar to those developed by other States and by the EPA, but, compared to some other Western States, CWQCC is much more detailed in the stream segments that it assigns regulatory standards based on beneficial use (some segments are as short as 1 mi). Beneficial use classes include recreation, agriculture (irrigation), aquatic life, drinking water, and other historic or future uses. In the study area, the designation for most streams near mining areas on the CWS is “cold water aquatic life,” which includes fish and associated biota. Because the standards for metal concentrations that are toxic in various beneficial uses differ in important details that are confusing in an overview such as this report, I will utilize just one standard, for cold water aquatic life, as the reference for metal concentrations (table 9).

The quality of Colorado surface waters is reviewed every 2 years by the CWQCD, in compliance with the U.S. Clean Water Act. Placement on the list of stream segments that are not in compliance with standards (called “303(d) lists”) has significant implications for management of those waters. The Nonpoint Assessment Report (CWQCD, 1989) contains more than 180 pages of technical information and lists of stream segments that were considered impacted by mining, agricultural, or industrial activities. For the study area, many streams were listed as impacted by metal pollutants related to mining; in most cases, the mining source, historic or current, is not specified. In the Gunnison River watershed, seven stream segments were rated low, six medium, and one highly impacted. In the Uncompahgre River watershed, four stream segments were rated medium and two highly impacted. For the San Miguel/Dolores watershed (the part in study area),

**Table 9.** Colorado basic stream standards for metals: class-1 aquatic life.

[Note: These values are for chronic exposure. The selected values shown are for 100-200 mg/L water hardness; criteria for Al, Fe, Mn, and Se do not vary with hardness. Taken from CWQCD (1989, table 9, which cites Colorado Department of Health, 1984, Basic Standards and Methodologies 3.1.0)]

pH or element	Stream standard for class-1 aquatic life (ppb or µg/L, except as noted)
pH	6.5–9.0 pH units
Al	100
As	50
Cd	1
Cu	10
Fe	1,000
Mn	1,000
Pb	25
Se	5
Zn	50

one segment was rated low, seven medium, and six highly impacted. Some of the highly impacted stream segments identified were Oh-Be-Joyful Creek (NW. of Crested Butte), Red Mountain Creek south of Ouray, Marshall Creek and San Miguel River east of Telluride, and San Miguel River below Uravan; these streams will be described and discussed in later sections of this report. The 303(d) report was a significant source of data for the Colorado geo-environmental map constructed by the USGS (Plumlee and others, 1995d). Significantly, the 1998 303(d) list (CWQCD, 1998) has revised many of the ratings from earlier lists: only Slate Creek (near Crested Butte) and three headwater branches of the San Miguel River east of Telluride are listed as impacted by mining-related contamination. The revisions have not been explained by CWQCD but appear to reflect better analytical results for the streams, as well as reclamation in several mining areas. New water sampling for this investigation, described later, is in close agreement with the 1998 list (but is at odds with many of the 1989 ratings). Information in the 1998 303(d) list shows that, where streams are impacted by mining as in the Silverton, Bonanza, and Leadville districts, zinc is the most common contaminant, and copper, iron, and cadmium also are listed as contaminants.

Prompted by the 1989 303(d) list of mining-impacted streams, Plumlee and others (1995d) explained likely sources of acid and metals and provided geologic reasons for the mobility of metals. The Colorado geo-environmental assessment showed graphically the geographic and geologic areas of concern. Because that work was based chiefly on available information, including the stream ratings of CWQCD (1989), this investigation was designed to provide new geochemical information for surface waters in the vicinity of mines—areas that are not normally sampled by agencies monitoring water flow and quality (typically done at gauging stations on main stream segments 5–10 mi from headwaters). Because of time

and budget constraints, and the large area under study, the investigations reported here were necessarily reconnaissance in nature. The goal of this investigation was to identify the major significant geochemical problems related to historic mining, to rank them, but not to define the specific level of contaminant release. Another goal was to determine the beneficial geologic interactions with acidic waters that naturally attenuate local sources of metal contaminants.

Quality of *ground water* is included in CWQCD management of Colorado waters and is an issue around historic mines where surface drainage from various sources infiltrates alluvium or bedrock. There are indications in monitoring wells near mines that subsurface flow can be significant (Nash, 1999b), but wells are rarely available to describe ground water near mines. An exception is the presence of wells in the Uruan district and a sampling program by specialists in the BLM Grand Junction office. Study of historic mining areas would be much more effective if shallow wells could be drilled to sample and monitor ground water.

Water quality in this study will be compared to the standard for aquatic life (class 1, cold water) from CWQCD (1989 and 2000) (table 9); these values are for chronic (long-term) exposure. In this report I will refer to them as aquatic life water standards (ALWS). These reference concentrations are similar to those for domestic drinking-water supplies but are more restrictive for elements such as Cu and Zn that affect aquatic life more than human health, and more stringent than for agricultural use. In detail, biologists prefer that metal concentrations be corrected for the effects of water hardness: increases in hardness reduce the toxic effects of some metals (Cd, Cu, Pb, Zn). Thus, the concentration for the metal to cause chronic or acute toxicity is higher in high-hardness waters (300 ppb Zn at 400 mg/L hardness compared with 50 ppb Zn at 100 mg/L hardness). Hardness values for these mine-related waters are highly variable, in the range about 10 to 1,400 mg/L (as CaCO<sub>3</sub>). Hardness corrections were not made for my results because to do so would imply more precision than appropriate in these reconnaissance hydrogeochemical studies. The values in table 9 provide guidelines for evaluating the severity of degradation, consistent with the nature of my studies. These values are useful for interpretation of processes operating at mine sites, but readers should consult results of other definitive studies (Owens, 1997; Herron and others, 1998; Crock and others, 1999; Smith and Huyck, 1999) for more specific data and regulatory concerns. For more information, excellent material is available from Colorado Department of Public Health and Environment, Water Quality Control Commission under the State website at [www.state.co.us](http://www.state.co.us).

Although the composition of streams and other surface waters is an indicator of impact on the environment, biologic (toxicologic) and health criteria are really the basis for water-quality standards (CWQCD, 1989; Manahan, 1994). In practice, it is much easier to determine water compositions than to measure biological communities such as benthic macroinvertebrates and fish (Besser and others, 1998; Boyle and Bukantis, 1998), which is one reason why water chemistry was sampled

widely in this study. Biological criteria for water quality are based on the biota and food chain that is historically present in the stream, or the desired community at the end of reclamation or stream restoration. These topics and issues are too complex to discuss in any detail here.

The interested reader should seek new biological work done on the upper Animas River watershed near Silverton because the fisheries, or lack of them, are very similar to those in most of the watersheds or sub-basins in this study area. Biological information was collected in 1991 and discussed in the context of water quality by Owens (1997). The Animas Stakeholders group has supported new biological studies of the upper Animas River from 1996 to 2000, and scientists of the USGS Biological Resources Division are completing studies there of topics including bioaccumulation of potentially toxic metals by stream biota; toxicity of stream water, bed sediment, and sediment porewater to fish and invertebrates; toxicity thresholds of zinc and copper to fish and invertebrates that populate the upper Animas basin; and influences of physical habitat on stream trout populations (Besser, 2000). The toxicity tests for the Animas River and its biota (Besser, 2000) show that the concentrations of dissolved copper and zinc that cause mortality to “sensitive” and “tolerant” species differ considerably (for example, 700 vs. 160 ppb Zn and 33 vs. 8 ppb Cu); thus, no single concentration actually defines aquatic health as is implied by the Colorado Basic Stream Standards cited above. Readers should be aware of these complexities, consult appropriate biologic authorities, and not take too literally the chemical results presented and discussed in this report. Because research on fisheries shows that Cu and Zn are of prime importance to the health of many aquatic species, those metals will be emphasized in this report.

## Threats to Aquatic Life by Metals: A Copper-Zinc Index

A copper-zinc index (CZI) has been formulated to provide a simple number that describes the magnitude of copper and zinc concentrations in water samples in relation to aquatic life requirements as determined by toxicologic tests just described (Besser, 2000). The intent of the CZI is to focus on two metals of prime concern to aquatic health on the CWS while minimizing regulatory details of water-quality standards. The CZI is computed as  $[(\text{Cu in ppb}/20) + (\text{Zn in ppb}/200)] / 2$ . The values of 20 and 200, for Cu and Zn respectively, are not precisely defined, but are essentially average values for tolerant and sensitive species in mortality tests and are similar in magnitude to the ALWS in table 9. Because the intent of the CZI is to suggest the magnitude of concern for aquatic health, there seems to be little point in refining the factors to more precise-appearing numbers. The sum is divided by two to conveniently make the index 1 for the break between healthy and unhealthy compositions: CZI values below 1 are “good,”

and values above 1 are “bad.” CZI values for 181 surface-water samples from the CWS (table 10) show a range from 0.01 to 1,405 and a median value of 1.13 (the average, 31.3, is skewed upward by a few very high values). The range of CZI values is shown on figure 7. Many of the very high values (10 to 1,405) are for local water sites such as mine drainages or tailings pond seeps, but some streams that have CZI values >10 seem to be real concerns as they appear to be significant threats to aquatic life.

## Surface-Water Quality Relative to Historic Mining

Water was sampled at many sites during this reconnaissance study in an effort to measure the dispersion of acid and metals away from known mine or mill sites. The water analyses provide much useful information, but three points

must be emphasized: (1) the sampling and analytical methods used here (Appendix 1) are not as rigorous as those used by CWQCC, EPA, and other regulatory agencies for the measurement of trace metals in water; (2) this study does not provide any indication of seasonal variability; and (3) these water analyses are only an indirect measure of the biological habitat and ecological health. The water sampling described here is deemed a cost-effective way of ranking surface waters to identify the most contaminated waters that are in greatest need of further study and, eventually, restoration.

Surface-water compositions are highly variable on the CWS. Measured pH's ranged from less than 2 to more than 8, conductivities (a measure of total dissolved solids) ranges from very low (< 50  $\mu\text{S}/\text{cm}$ , a very good value) to very high (>2,000  $\mu\text{S}/\text{cm}$ , very bad quality). Metal concentrations ranged from less than 1 ppb to more than 1,000,000 ppb (table 3, table 11) in places—far in excess of regulatory standards. Water compositions determined here for many streams are better than implied by some water-quality ratings as “mining impacted” (CWQCD, 1989).

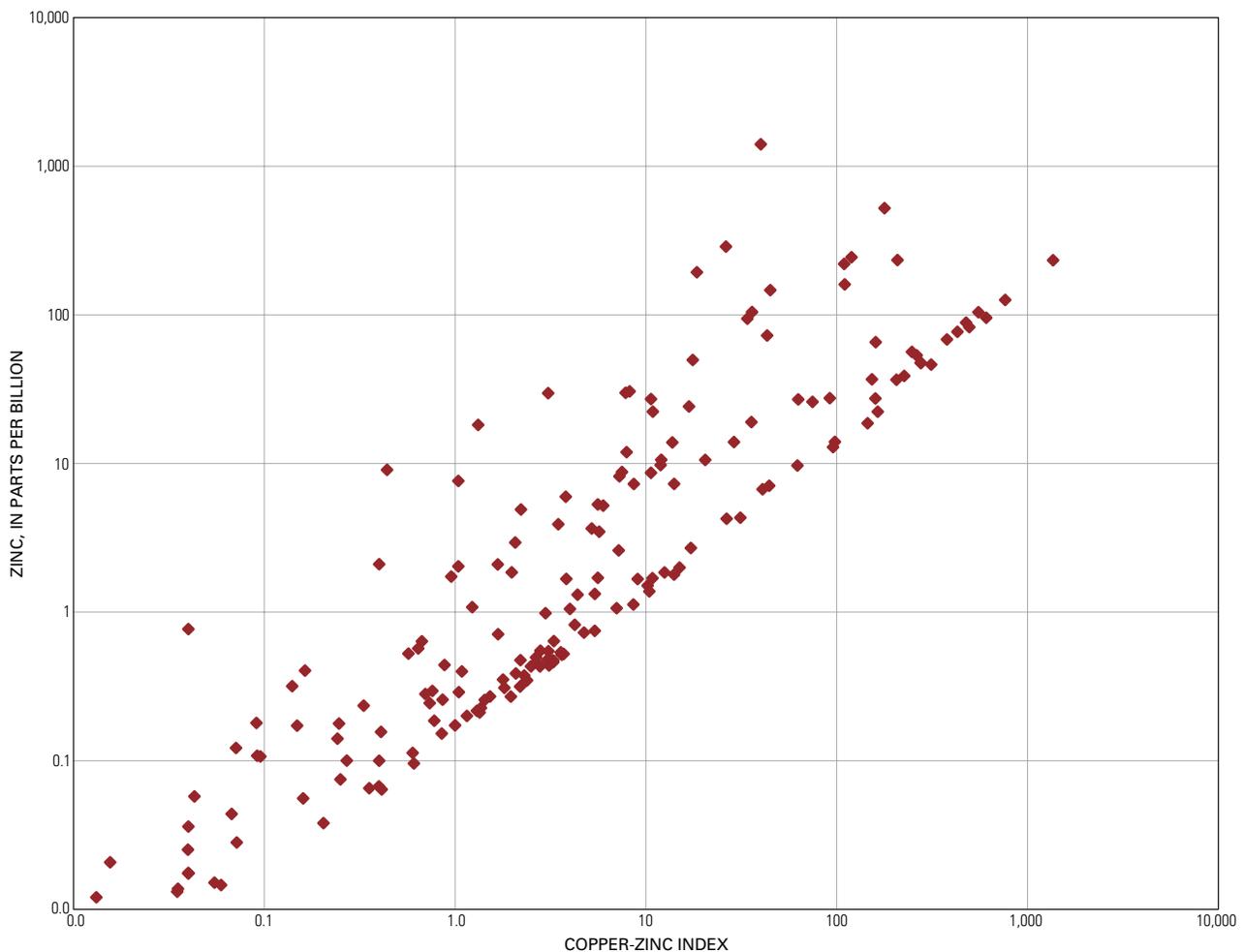


Figure 7. Copper-zinc index (CZI) values for surface waters, central Western Slope, Colorado.

**Table 10.** Copper-zinc index for surface waters, CWS, Colorado.

[Values derived from ICP-MS analyses; units are ppb (parts per billion). Abbreviations of water type, district, and watershed as in other chemical tables. CuI\*: Cu index, Cu (ppb) divided by 10 (ALWS); ZnI\*, Zn index, Zn (ppb) divided by 50 (ALWS); CZI\*: Copper-zinc index, (Cu concentration/20 + Zn concentration/200)/2]

Sample ID	pH	Conductivity ( $\mu$ S/cm)	Water type	District	Watershed	CuI*	ZnI*	CZI*	Site name
NGW105	8.5	440	S	UV	SM	0.4	0.1	0.1	Lion Creek
NGW106	8.3	310	MD	UV	SM	0.5	0.1	0.1	adit drainage
NGW107	8.5	410	MD	UV	SM	0.2	0.0	0.1	adit drainage
NGW115	7.4	1740	SP	UV	SM	0.6	0.1	0.2	no name seep
NGW117	8.1	840	S	UV	SM	0.1	0.0	0.0	Atkinson Creek
NGW205	7.6	410	DP	TM	GU	0.7	7.0	1.1	Akron mill
NGW214	7.9	180	MD	TM	GU	0.4	3.6	0.5	Akron adit
NGW230	7.8	280	S	VU	GU	0.6	0.7	0.2	Camp Creek
NGW231	7.7	310	S	VU	GU	0.6	0.2	0.2	Camp Creek
NGW232	7.9	240	MD	GB	GU	0.4	0.8	0.2	Carter mine
NGW247	7.4	240	MD	TC	GU	0.4	0.2	0.1	New Gold Cup adit
NGW249	7.7	90	SP	MRP	GU	0.4	0.1	0.1	Stock spring, Chester
NGW259	7.9	680	SP	LO	GU	0.3	0.3	0.1	Stock spring
NGW263	3.7	210	MD	GQ	GU	118.1	8.2	30.5	Bon Ton adit drainage
NGW264	5.7	120	MD	GQ	GU	115.6	7.8	29.9	Bon Ton adit-dump drainage
NGW275	7.6	280	MD	TC	GU	2.0	0.6	0.6	Blistered Horn adit
NGW287	7.4	260	MD	BP	LF	0.6	3.1	0.5	American Basin
NGW296	4.7	80	S	MP	UN	1.1	12.5	1.8	Head Mineral Creek
NGW298	3.5	200	MD	MP	UN	7.6	62.3	9.7	London
NGW299	3.6	410	S	RM	UN	31.3	7.5	8.8	Red Mtn Creek
NGW309	3.4	>2000	SP	OY	UN	251.0	1,361.8	233.0	Banner seep
NGW310	3.7	570	SP	IS	SM	17.8	6.0	5.2	Iron Spring mine
NGW315	3.0	1660	TD	IS	SM	122.2	763.7	126.0	Carbonero mill
NGW317	7.3	980	TD	IS	SM	0.3	1.4	0.3	Caribeau mill tailings
NGW318	7.1	790	MD	IS	SM	0.3	3.6	0.5	Caribou mine
NGW320	6.6	290	MD	IS	SM	58.1	35.7	19.0	Alta mine
NGW328	4.8	250	S	RM	UN	7.5	1.7	2.1	National Belle
NGW333	2.9	1340	MD	RM	UN	88.3	16.8	24.2	Champion Gulch
NGW335	2.4	>2000	MD	RM	UN	825.3	109.4	220.0	Guston mine
NGW336	3.1	>2000	MD	RM	UN	76.5	62.9	27.0	Lower Guston adit
NGW338	2.6	>2000	DP	RM	UN	268.9	43.2	72.6	Dump puddle
NGW340	2.7	>2000	MD	RM	UN	916.5	119.6	244.1	Guston mine
NGW342	3.5	640	S	RM	UN	83.8	10.9	22.3	Red Mtn Creek
NGW344	2.9	>2000	TS	RM	UN	83.2	495.1	82.7	Idarado Tailings seep
NGW347	3.6	410	S	RM	UN	29.1	7.3	8.2	Red Mtn Creek
NGW348	2.0	1650	MD	RM	UN	43.6	205.7	36.6	Larson Brothers mine
NGW353	6.3	740	MD	RM	UN	0.3	10.4	1.4	Mountain King
NGW357	2.9	1410	MD	RM	UN	66.5	74.7	26.0	Joker mine
NGW363	5.9	1260	MD	RM	UN	0.3	5.4	0.7	Silver Belle
NGW365	3.7	390	S	RM	UN	33.1	11.9	9.8	Red Mtn Creek
NGW367	3.0	290	S	RM	UN	18.5	2.2	4.9	Corkscrew Gulch
NGW371	6.2	190	S	SN	UN	0.2	2.2	0.3	Canyon Creek
NGW373	6.5	190	S	SN	UN	0.2	3.1	0.4	Canyon Creek
NGW380	6.9	120	S	MP	UN	0.5	2.7	0.5	Upper Uncompahgre River
NGW382	6.5	480	MD	MP	UN	0.3	2.3	0.4	Old Lout mine
NGW385	7.2	360	S	IS	SM	0.2	1.3	0.2	Howard Fork
NGW388	7.3	150	S	TE	SM	0.5	15.0	2.0	Savage Creek
NGW391	7.2	110	S	TE	SM	0.3	2.8	0.4	San Miguel River
NGW393	3.3	1510	MD	GN	LF	3.7	95.7	12.9	Palmetto basin
NGW394	2.9	1500?	MD	GN	LF	7.0	97.8	14.0	Palmetto basin
NGW450	7.1	240	S	GN	LF	32.0	20.4	10.6	Head Henson Creek
NGW455	2.7	nd	MD	MP	UN	42.4	225.8	38.8	San Juan Chief

Table 10. Copper-zinc index for surface waters, CWS, Colorado—Continued.

Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Water type	District	Watershed	CuI*	ZnI*	CZI*	Site name
NGW492	5.9	910	S	SN	UN	0.3	3.3	0.5	Canyon Creek
NGW493	5.9	1540	MD	IS	SM	2.4	3.0	1.0	Caribeau
NGW494	5.9	1540	DH	IS	SM	0.2	1.4	0.2	Drill hole flow
NGW495	2.1	>2000	MD	RM	UN	764.8	18.5	193.5	Head Corkscrew
NGW499	3.0	285	MD	RB	GU	84.2	378.2	68.3	Standard
NGW505	8.2	251	S	VU	GU	0.7	2.7	0.5	Cabin Creek
NGW506	8.1	354	S	VU	GU	0.8	0.3	0.2	Cabin Creek
NGW507	3.1	200	MD	RB	GU	70.8	152.9	36.8	Mid Elk Basin
NGW510	3.1	320	MD	RB	GU	29.8	159.4	27.4	Upper Elk Basin
NGW511	2.9	354	DD	RB	GU	101.5	247.8	56.4	Upper Elk Basin
NGW512	3.0	285	MD	RB	GU	92.8	429.4	76.9	Standard mine
NGW514	3.3	301	S	RB	GU	82.5	262.3	53.4	E Elk Creek
NGW515	5.8	20	S	RB	GU	0.0	0.1	0.0	W Elk Creek
NGW518	3.0	169	DD	RB	GU	140.3	553.0	104.2	Standard mine
NGW519	3.1	448	TP	RB	GU	116.4	476.4	88.6	Standard mine
NGW520	2.9	765	TPS	RB	GU	51.5	276.4	47.4	Standard mine
NGW521	5.4	62	S	RB	GU	6.4	40.9	6.7	Elk Creek
NGW523	5.9	60	S	RB	GU	3.7	26.5	4.2	Elk Creek
NGW529	7.2	97	S	GB	GU	0.2	0.9	0.2	Gold Creek
NGW531	6.9	90	S	GB	GU	0.1	0.4	0.1	Gold Creek
NGW532	6.9	56	S	GB	GU	0.0	0.0	0.0	Gold Creek
NGW533	6.7	40	S	GQ	GU	6.5	1.0	1.7	Tomichi Creek
NGW537	6.8	66	S	LF	LF	0.1	0.1	0.0	Wade Gulch
NGW540	6.9	96	S	LF	LF	0.1	0.2	0.1	Red Mtn Gulch
NGW541	6.6	43	S	LF	LF	0.1	0.4	0.1	Lake Fork Gunnison River
NGW543	6.7	46	S	LF	LF	0.0	0.0	0.0	Cottonwood Creek
NGW544	6.8	34	S	LF	LF	0.0	0.0	0.0	Cattaract Creek
NGW545	6.9	44	S	BP	LF	0.0	0.0	0.0	Grizzly Creek
NGW546	7.1	47	S	BP	LF	0.1	0.6	0.1	Silver Creek
NGW547	7.0	46	S	LF	LF	0.0	0.1	0.0	Bent Creek
NGW548	6.8	76	S	LF	LF	0.1	0.0	0.0	Lake Fk Gunnison River
NGW549	6.9	46	S	GN	LF	0.1	0.1	0.0	W Fk Henson Creek
NGW550	7.2	42	S	GN	LF	0.8	0.7	0.3	Upper Henson Creek
NGW552	7.3	46	S	GN	LF	0.2	0.3	0.1	Lower Henson Creek
NGW555	3.0	779	TS	RB	GU	52.5	275.3	47.5	Standard mine
NGW556	6.6	32	MD	RB	GU	2.1	9.1	1.7	Upper Elk Basin
NGW558	7.3	59	S	RB	GU	1.4	10.8	1.7	Lower Elk Creek
NGW559	7.3	37	S	RB	GU	0.2	1.0	0.2	Slate River
NGW560	7.3	24	S	RB	GU	0.3	1.8	0.3	Oh Be Joyful Creek
NGW561	7.0	62	S	TM	GU	0.7	0.1	0.2	Tomichi Creek
NGW563	7.1	83	S	TM	GU	0.5	2.1	0.4	Tomichi Creek
NGW565	7.2	81	S	TM	GU	0.5	2.5	0.4	Tomichi Creek
NGW570	2.7	1025	S	RM	UN	586.2	110.1	160.3	Red Mtn Creek
NGW571	2.6	1140	MD	RM	UN	563.9	44.9	146.6	Adit into ferricrete
NGW574	3.3	436	S	RM	UN	103.1	10.6	27.1	Red Mtn Creek
NGW575	2.9	385	S	RM	UN	29.3	10.6	8.7	Champion Gulch
NGW576	2.5	>2000	MD	RM	UN	830.7	208.2	233.7	Guston mine
NGW577	4.2	122	MD	RM	UN	13.9	3.5	3.9	Prospect trench
NGW578	2.5	>2000	TS	RM	UN	78.6	607.9	95.6	Idarado Tailings seep
NGW580	2.9	885	DD	RM	UN	41.3	28.9	13.9	Joker mine
NGW581	3.6	319	S	RM	UN	48.5	13.8	13.9	Red Mtn Creek
NGW582	3.0	565	S	RM	UN	43.7	7.9	11.9	Corkscrew Gulch
NGW583	3.3	248	S	RM	UN	1.8	0.6	0.5	Small Creek
NGW584	2.5	1035	S	RM	UN	117.3	3.1	29.7	Corkscrew Gulch
NGW587	2.1	>2000	MD	RM	UN	1137.6	26.3	287.7	Adit, Corkscrew
NGW589	3.9	39	S	RM	UN	3.1	0.0	0.8	Lake at divide

**Table 10.** Copper-zinc index for surface waters, CWS, Colorado—*Continued.*

Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Water type	District	Watershed	CuI*	ZnI*	CZI*	Site name
NGW592	3.4	154	S	RM	UN	1.5	0.2	0.4	Small creek
NGW593	3.1	563	S	RM	UN	6.4	2.0	1.9	Small creek
NGW595	2.9	492	S	RM	UN	3.7	1.2	1.1	Small reek
NGW599	3.2	215	S	RM	UN	10.7	2.1	2.9	Small creek
NGW605	4.9	278	S	RM	UN	2.2	0.7	0.6	Gray Copper Creek
NGW606	6.0	375	MD	RM	UN	0.2	2.4	0.3	Mountain Silver
NGW609	5.9	74	MD	RM	UN	1.3	0.9	0.4	Upper Mt Silver
NGW610	3.4	390	S	RM	UN	36.3	12.0	10.6	Red Mtn Creek
NGW612	6.2	680	S	IS	SM	0.6	1.0	0.3	Chapman Gulch
NGW613	6.4	620	MD	IS	SM	0.2	0.6	0.1	Small adit
NGW614	5.9	1540	DH	IS	SM	0.3	1.5	0.3	Exploration drill hole
NGW616	5.1	95	S	IS	SM	1.1	1.1	0.4	Small creek
NGW617	3.4	955	Sp	IS	SM	24.8	8.6	7.3	Iron Spring
NGW618	5.5	365	S	IS	SM	0.4	0.4	0.2	E Howard Fork
NGW621	5.5	890	S	IS	SM	0.1	14.0	1.8	Side creek
NGW627	6.1	330	S	IS	SM	0.2	1.2	0.2	S Fork San Miguel
NGW632	5.9	910	MD	IS	SM	3.0	4.4	1.3	Caribou mine
NGW633	5.9	820	TD	IS	SM	0.2	1.3	0.2	Flow on Caribou tails
NGW634	5.5	348	S	IS	SM	0.2	2.4	0.3	Howard Fork
NGW635	6.9	135	S	IS	SM	0.1	0.0	0.0	Lake Fk San Migue;
NGW636	4.7	450	MD	IS	SM	22.1	14.1	7.3	Adit W of Campgrnd
NGW637	6.7	348	S	IS	SM	0.5	1.8	0.4	Howard Fork
NGW641	6.2	118	S	MP	UN	0.4	3.0	0.5	Upper Uncompahgre River
NGW642	6.5	472	MD	MP	UN	2.0	145.2	18.6	M Breen adit
NGW644	7.0	83	S	MP	UN	0.5	4.7	0.7	Mineral Creek
NGW645	6.9	128	S	MP	UN	1.2	4.2	0.8	Upper Uncompahgre River
NGW651	3.0	707	S	MP	UN	28.9	312.6	46.3	Mineral Creek
NGW652	3.6	170	S	MP	UN	6.2	44.3	7.1	Upper Unconpahgre River
NGW654	3.1	443	MD	RM	UN	4.8	3.8	1.7	Small mine
NGW656	5.8	202	S	OU	UN	0.9	3.3	0.6	Uncompahgre River
NGW657	5.9	69	S	SN	UN	0.3	3.6	0.5	Sneffels Creek
NGW658	6.9	48	S	SN	UN	0.3	3.1	0.5	Imogene Creek
NGW662	6.5	48	S	TE	SM	0.1	0.4	0.1	Savage Creek
NGW670	6.7	90	S	TE	SM	1.6	31.3	4.3	Savage Creek
NGW671	7.3	48	S	TE	SM	0.4	3.1	0.5	Marshall Creek
NGW673	6.9	53	S	TE	SM	0.9	10.2	1.5	Ingrahm Creek
NGW677	7.2	73	S	TE	SM	0.1	0.0	0.0	Bear Creek
NGW678	5.6	66	S	SN	UN	0.2	3.7	0.5	Sneffels Creek
NGW679	6.0	94	MD	SN	UN	0.1	0.2	0.0	Eclipse mine
NGW680	6.1	44	S	SN	UN	0.1	0.0	0.0	Yankee Boy Creek
NGW681	5.7	121	S	SN	UN	0.2	3.3	0.5	Canyon Creek
NGW683	7.1	12	S	RM	UN	0.1	0.0	0.0	Mineral Creek ditch
NGW685	6.6	150	S	RM	UN	11.0	5.7	3.5	Albany Creek
NGW820	2.9	130	Sp	RM	UN	18.0	0.4	9.1	Spring near mine/vein, red altn
NGW823	4.6	15	S	RM	UN	1.1	17.2	2.7	Headwater Gray Copper Creek
NGW827	2.8	189	MD	RM	UN	36.0	1.3	18.2	Vernon adit, lower
NGW828	3.5	33	S	RM	UN	0.6	0.1	0.3	Side creek from red altn
NGW829	3.0	121	S	RM	UN	15.0	1.0	7.6	Gray Copper Crk W. of Vernon
NGW830	3.0	92	S	RM	UN	4.1	0.4	2.1	Gray Copper Crk E. of Vernon
NGW831	2.0	>2000	MD	RM	UN	2800	40	1405	Corkscrew headwall adit
NGW834	6.8	172	S	IS	SM	0.4	2.2	0.5	Howard Fk at Ames
NGW835	7.7	121	S	IS	SM	0.3	0.9	0.3	South Fork San Miguel
NGW837	3.2	312	MD	RM	UN	200.0	36.0	104.5	Hudson mine drainage
NGW839	3.0	204	S	RM	UN	95.0	17.6	49.7	Headwater Red Mtn Creek
NGW840	2.5	>2000	MD	RM	UN	1000.0	178.0	522.3	Guston mine drain
NGW841	2.9	284	S	RM	UN	180.0	34.0	94.3	Red mtn Creek middle

**Table 10.** Copper-zinc index for surface waters, CWS, Colorado—*Continued.*

Sample ID	pH	Conductivity ( $\mu\text{S/cm}$ )	Water type	District	Watershed	CuI*	ZnI*	CZI*	Site name
NGW842	2.9	762	MD	RM	UN	91.0	160.0	65.5	Joker tunnel drainage
NGW844	5.5	415	MD	IS	SM	3.4	7.2	2.6	Mine drain makes Fe floc
NGW845	3.7	325	S	IS	SM	9.2	5.6	5.3	Surface flow, much Fe floc
NGW850	7.3	51	S	TM	GU	0.4	2.8	0.6	Tomichi Creek near smelter
NGW852	7.3	131	MD	TM	GU	0.1	8.6	1.1	Akron tunnel drainage
NGW853	7.8	50	S	TM	GU	0.4	0.8	0.3	Tomichi Creek 200 m SW mill
NGW856	8.7	45	S	GB	GU	0.1	0.4	0.1	Gold Creek east of mines
NGW857	8.2	48	S	OU	UN	11.0	3.8	6.0	Dexter Creek below mines
NGW860	6.9	184	S	na	UN	6.0	5.2	3.7	Uncompahgre north side Ouray
NGW862	7.7	240	S	na	UN	3.8	1.0	2.0	Uncompahgre N. of Ridgeway
NGW864	6.4	42	S	SN	UN	1.0	1.7	0.7	Sneffels Creek above Atlas tails
NGW865	6.8	55	S	SN	UN	0.1	2.0	0.3	Sneffels Creek below Atlas tails
NGW866	7.0	64	S	SN	UN	2.0	5.6	1.7	Imogene Creek SW Camp Bird
NGW869	72.0	51	MD	SN	UN	1.3	5.4	1.3	Adit drain old CB#3
NGW870	5.8	55	MD	SN	UN	32.0	92.0	27.5	Adit drain Hidden Treasure, red
NGW871	6.6	124	S	TE	SM	3.6	164.0	22.3	Surface flow at Argentine dump
NGW873	7.1	190	S	SN	UN	1.1	4.0	1.1	Canyon Creek below tailings

## Geochemical Concepts for Water Quality

Some brief comments on geological-geochemical aspects of water compositions may provide helpful perspective on the huge array of analyses of water samples collected in this study. Discussion here will focus on 11 elements that are of most concern to human or aquatic-habitat health (table 11; figs. 8 and 9).

Some metals, such as Cu and Zn, have similar behavior in solids and in solutions and also tend to be enriched together, whereas some are essentially unique in their geochemical behavior (Smith and Huyck, 1999; Langmuir, 1997). Cadmium and zinc are very similar chemically and geochemically, yet they have different toxic properties. Iron and manganese are chemically similar and may or may not follow each other geochemically. Arsenic, molybdenum, and uranium are very different chemically and toxicologically, but respond similarly because they are mobile as oxy-anions under oxidizing, alkaline conditions. Copper, lead, and zinc generally are derived from sulfide minerals, are mobilized by acids, but respond differently as pH rises during mixing or buffering reactions.

*Aluminum* is abundant in most geologic settings but is mobile only under very acidic conditions (figs. 6, 9) or under very alkaline conditions that are not attained on the CWS. Significantly, there are many high to very high concentrations in sampled waters: 48 percent of samples exceed the ALWS of 100 ppb. The very highest concentrations of Al are in waters with pH values below 3, but some pH 6.6–7.1 waters contain high concentrations. High Al concentrations are most common in the Red Mountain, Ruby, and Iron Springs districts. High Al concentrations lead to the formation of colloids that, among

other things, can clog gills in fish.

*Iron and manganese* can come from silicate, sulfide, and oxide minerals, and both tend to be mobile in acidic waters (fig. 6). Iron can be enriched in waters from either mined or unmined sulfidic rock—the latter shown by the widespread deposits of ferricrete and bog iron, as in the Ophir, Silverton, and Ironton areas, which clearly are unrelated to historic mining (Purinton, 1898; Hanshaw, 1974). Manganese abundance is more variable than that of its geochemical cousin, Fe, and probably is related to Mn-enriched green-altered volcanic rocks and Mn-gangue minerals rhodonite and rhodochrosite (Burbank and Luedke, 1969). Concentrations of Fe exceed the ALWS (1,000 ppb) in 51 percent of samples, whereas this is less common for Mn (only 25 percent in excess of ALWS). The high to very high values of Fe and Mn are in acidic waters (pH < 3), but some pH 6.5 to 8.1 waters carry high concentrations of Fe and Mn. Iron and Mn are most commonly excessive in the Iron Springs, Red Mountain, and Ruby districts. Iron and Mn are involved in several kinds of reactions in surface waters as the pH rises, most commonly observed as red to black coatings on stream cobbles. Such coatings generally form from colloids in stream water (Church and others, 1997) and have a major impact on biological habitat.

*Copper, lead, zinc, and cadmium* tend to be associated with each other, but relative enrichments clearly differ from area to area according to deposit mineralogy. These metals are mostly derived from sulfide minerals, either in mine waste and tailings or unmined altered rocks. Concentrations of these metals are generally in proportion to acidity (fig. 6), and some waters with pH below 3 have concentrations that are 100× to 10,000× ALWS. Surface waters at many sites are in excess of ALWS: Cu, 57 percent of the samples have >10 ppb; Pb, 44

**Table 11.** Summary of ICP-MS analyses of surface waters, CWS, Colorado.

	pH	Conductivity ( $\mu$ S/cm)	Al (ppb)	As (ppb)	Cd (ppb)	Co (ppb)	Cu (ppb)	Fe (ppb)	Mn (ppb)	Mo (ppb)	Ni (ppb)	Pb (ppb)	Sb (ppb)	Se (ppb)	Te (ppb)	Tl (ppb)	U (ppb)	Zn (ppb)
<b>Streams and creeks (n=109)</b>																		
Minimum	2.5	12	0.0	0.1	0.0	0.0	0.3	52.0	0.3	0.0	0.2	0.1	0.0	0.2	0.4	0.0	0.0	0.7
Median	6.6	121	55.7	0.5	0.7	0.6	6.6	311	93.7	1.0	2.3	2.5	0.2	0.3	0.4	0.1	0.2	110
Maximum	8.7	1,035	33,000	100	138	79.6	5,862	63,499	4,328	180	53	1,229	2.7	344	0.4	3.1	887	15,629
<b>Springs and seeps (n=7)</b>																		
Minimum	2.9	90	1.5	0.2	0.1	0.1	2.7	76.4	1.0	0.1	1.7	0.6	0.0	0.5	0.0	0.1	0.0	4.8
Median	3.5	625	391	0.4	1.1	2.8	92.3	2,145	14.1	1.7	8.4	1.5	0.1	7.1	0.0	0.1	1.4	17.8
Maximum	7.9	>2,000	12,462	1,130	309	55.6	2,510	12,609	63,766	1,864	128	108	1	20,032	0.0	0.2	145	68,090
<b>Mine and dump drainage (n=56)</b>																		
Minimum	2.0	32	0.0	0.1	0.1	0.0	1.0	63.5	0.1	0.1	0.8	0.2	0.0	0.2	0.3	0.0	0.0	2.2
Median	3.6	375	1,474	5.0	7.0	14.9	297.8	5,576	1,207	1.7	11.4	28.9	0.4	1.2	0.5	0.5	1.1	772
Maximum	8.5	>2,000	469,703	1,000	165	1,000	28,000	1,500,000	57,836	174	780	1,374	13	194	12	6.4	555	27,648
<b>Tailings seep and pore water (n=14)</b>																		
Minimum	2.1	448	16.1	0.8	0.5	1.3	1.7	1,148.8	14.3	0.4	2.4	2.4	0.1	0.3	0.6	0.1	0.0	67.2
Median	3.0	980	6,546	7.0	39.9	33.6	515	7,552	2,863	1.3	19.3	370	1.4	3.0	0.6	0.2	1.1	8,233
Maximum	7.3	>2,000	190,941	5,910	567	2,674	31,734	4,731,542	105,535	1,908	405	2,775	38.8	244	24.5	5.7	51.5	58,050

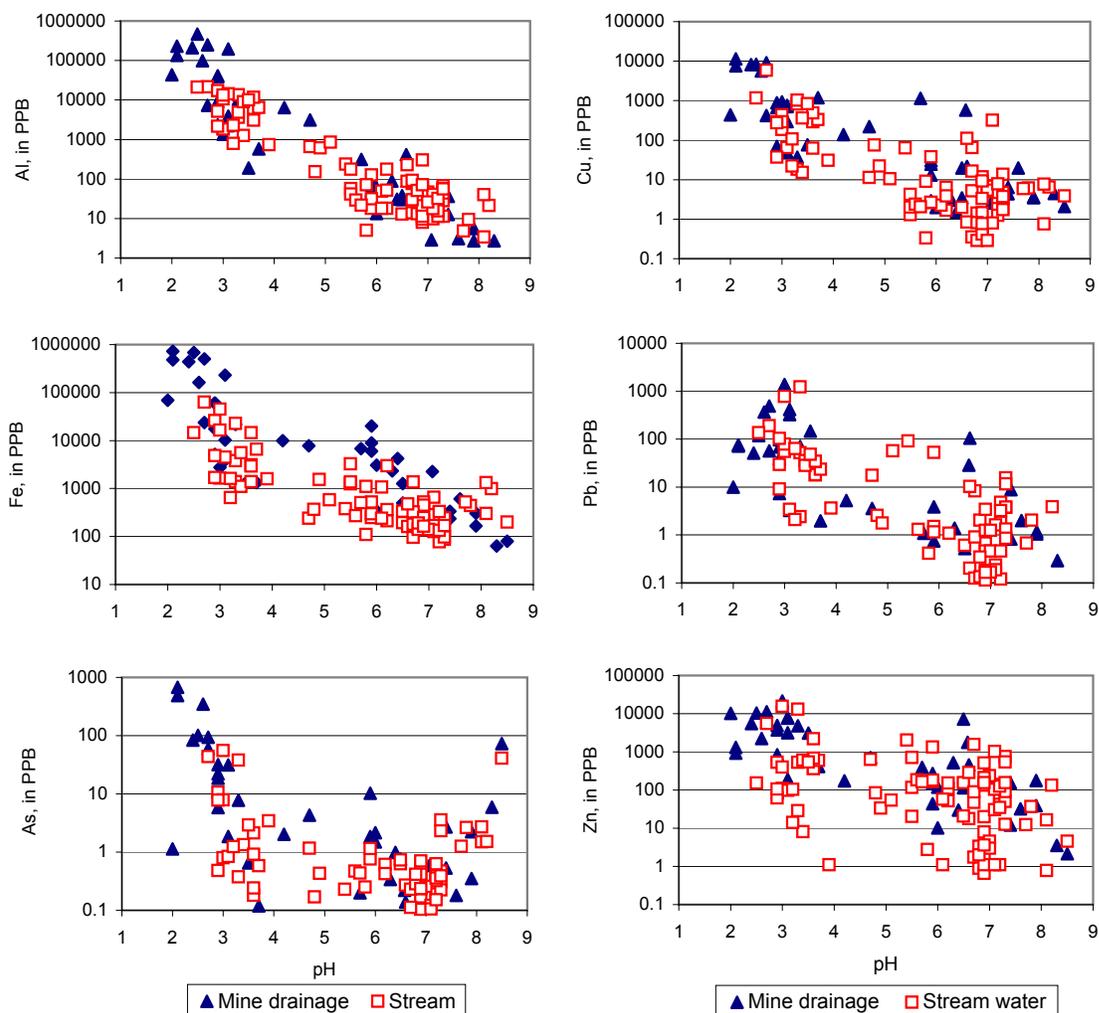


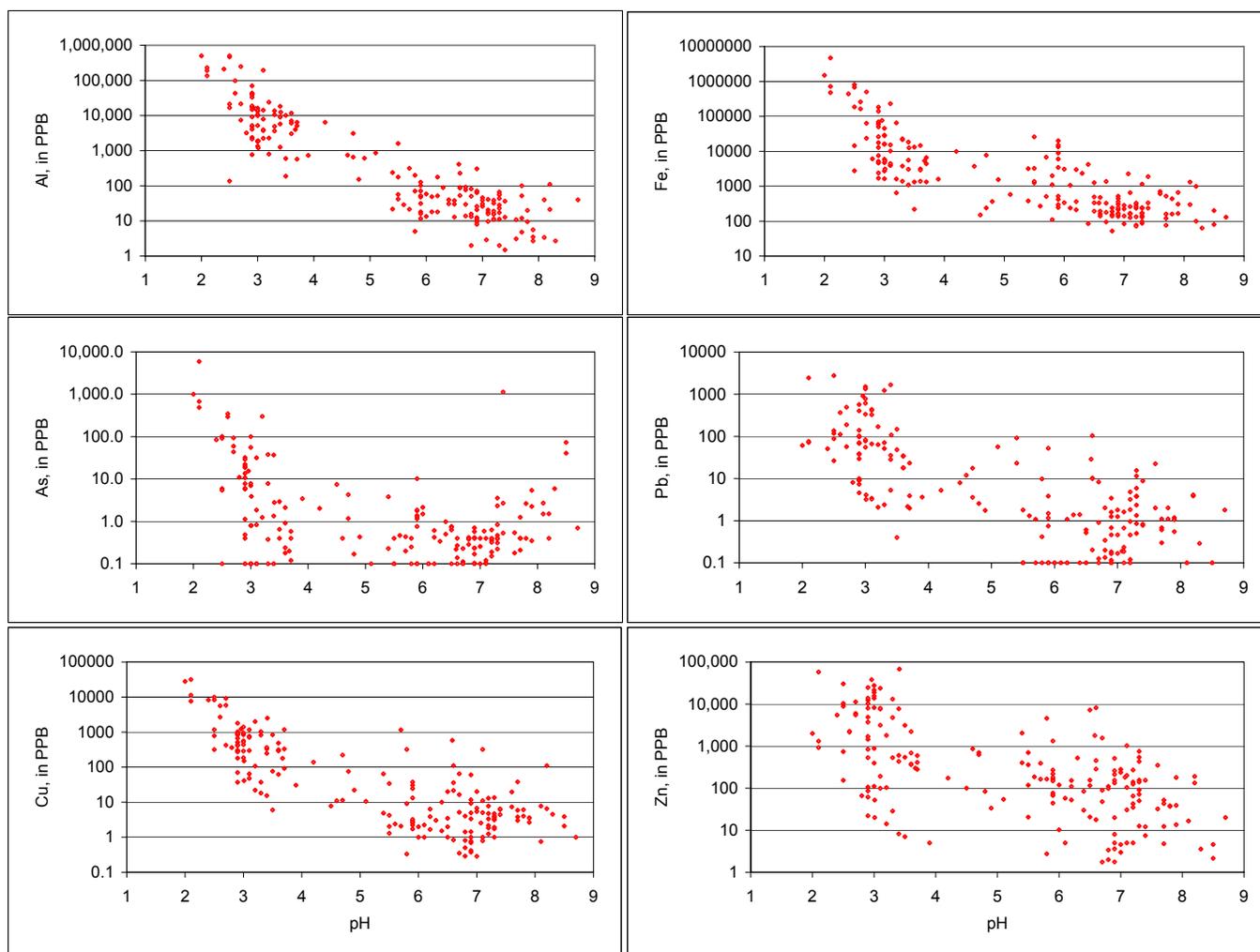
Figure 8. Comparison of mine drainage and stream-water compositions, central Western Slope, Colorado.

percent have >25 ppb; Zn, 74 percent have >50 ppb; Cd, 54 percent have >1 ppb. Lead concentrations in surface waters tend to be lower relative to the other base metals and less commonly are above the ALWS. The high base-metal concentrations typically are in acidic waters with pH < 3 to 4, but some waters with pH 6.6–7.1 can carry concentrations in excess of ALWS. Combinations of these base metals, such as Cu+Zn, may be especially problematic for fish (Nimmo and others, 1998; Besser, 2000).

*Cobalt and nickel* are commonly enriched in sulfide minerals of some ore deposits (Plumlee and others, 1999), but they are not abundant or significant on the CWS. Concentrations of these metals in this study area are generally low: only 8 percent of surface samples contain >100 ppb Ni (the ALWS for Ni is 100 ppb), and 9 percent contain >100 ppb Co. Cobalt and Ni generally behave like Cu and Zn in surface waters, thus geochemical studies or reclamation that focus on Cu and Zn will likely treat Co and Ni as well. These metals will not be discussed further.

*Arsenic* is a metal of concern chiefly for human health—thus, it is specified in drinking-water uses—and is less significant for aquatic biota (Smith and Huyck, 1999). It is rich in some ores on the CWS, such as the Red Mountain breccia pipes but generally is not elevated in waters. The median is 0.84 ppb, which is less than the average for world stream waters (Forstner and Wittman, 1979). Only 8 percent of samples exceed the drinking-water standard of 50 ppb. The high As concentrations are limited to acidic waters of the Red Mountain and Mineral Point districts and two sites in the Uravan district with alkaline pH. Highest As values are in a specific ore environment in the Red Mountain district (breccia pipe deposits) that are known to be rich in As (Burbank and others, 1972; Fisher and Leedy, 1973). Arsenic poses special problems because it tends to remain soluble as pH is neutralized by natural or engineered reactions, but that does not appear to apply to more than a few sites on the CWS.

*Selenium*, similar to As in some chemical and geochemical aspects, is not elevated in most surface waters on the CWS.



**Figure 9.** Compositions of surface waters, central Western Slope, Colorado, as a function of pH.

Only four sites have Se concentrations >50 ppb, a domestic drinking-water standard, and 13 samples contain more than the 5 ppb ALWS. The uranium deposits tend to be enriched in Se; a few water samples from those areas show elevated Se concentration. One characteristic of Se should be noted: it tends to accumulate by evaporative processes on the surface layers of mine dumps with the formation of efflorescent salts of sulfate and selenate. These salts are very soluble and the first rain after a dry period can flush the Se into puddles or creeks and thus become available to livestock or wildlife.

*Molybdenum* poses health threats such as molybdenosis in cattle, but it is not quantified on the Colorado list of toxic metals (CWQCC, 2000). Concentrations of Mo in water are generally low on the CWS, but 8 samples contain more than the aquatic life standard of 19 ppb. All but one of the Mo-rich surface waters are from the Uravan area. Acidic waters can transport Mo, but more commonly Mo is mobile as the oxyanion molybdate in oxidizing, alkaline waters. These conditions are found on the Colorado Plateau.

*Antimony (Sb), tellurium (Te), and thallium (Tl)* are enriched in some kinds of ore deposits, and minerals of Sb and Te are known in several of the districts studied. Concentrations of Sb, Te, and Tl in surface waters are generally low. More than 98 percent of samples had detectable amounts of Sb (>0.1 ppb), but the five highest values are only 5 to 13.1 ppb. Higher concentrations of Sb (14 to 39 ppb) were detected in tailings pore water samples. Only 8 percent of surface-water samples have detectable amounts of Te (>0.2 ppb); five samples have concentrations in the range of 3.1 to 11.9 ppb Te. Thallium is highly toxic at low concentration levels (the drinking-water standard is 13 ppb), but it is not significantly concentrated in these districts (only two samples are in the range of 5 to 12 ppb). The few high values of these metals are in strongly acidic waters that are even more problematic for several other metals; these trace metals probably decrease as other metals precipitate, such as by adsorption on iron oxyhydroxides.

*Uranium* (and associated radium) is listed as a hazardous material rather than toxic metal. The concern is for exposure

to radioactivity, especially by humans. The Colorado standard for U in surface and ground waters is 44 ppb (or 30 picocuries radioactivity) (CWQCC, 2000), and this is exceeded in only five surface waters. All five of the U-rich samples are from uranium mining areas and have pH values of 7.4 to 8.5. Four samples of mine drainage from the Uravan district exceed the uranium standard, but those flows are rare, of small volume, and are not typical of the notoriously dry Uravan district. Uranium is mobile in acidic waters and is noted in some highly acidic waters from deposits that contain low concentrations of uranium (Wanty and others, 1999), but U also is mobile in alkaline waters that are typical of sedimentary terranes, such as the Colorado Plateau. On the CWS, the alkaline scenario explains local U transport in the Uravan area. Thorium (Th), also radioactive, is less mobile than U; in this study, only one unusually acidic water contained more than 20 ppb Th (there are no toxicity standards for Th in water).

## Risks During Extreme Storms: Failure of Tailings Impoundments

Failure of a tailings impoundment or waste dump during an extreme storm event (flash flood) is a concept that is rarely mentioned in mine lands assessments, yet the dire consequences of such a failure are known in the past century and the probabilities are high enough to demand consideration in land planning. Major failures are known to have occurred, but little emphasized, in the Silverton area during historic flooding of October 4–6, 1911, and there was another major failure near Silverton in 1975 from which tailings were carried to New Mexico. There is evidence of significant erosion at many tailings impoundments in this study area, and recent studies of tailings in northern Nevada (Nash, 2001) revealed that nearly every substantial impoundment had experienced a major or partial failure in the past 100 years. Failure of mine-waste dumps, not mentioned in the literature, also is a possibility. The major concern is that tailings placed in lowlands, not far from streams, could be hit by the “wall of water” that forms during a cloudburst in a confined upland watershed. Geologic evidence shows that boulders the size of a car can be moved by rushing waters that rise 5 to 15 ft above normal levels and have flow volumes more than 100 times normal. Whether mined materials fail by erosion or sliding, mass movement in conjunction with unusual amounts of water can carry and spread both solids and dissolved metals.

Floods in mountain canyons of Colorado caused by cloudbursts or several days of summer rains are well known for their violence, property damage, and loss of life (Follinsbee and Sawyer, 1948). Witnesses and meteorologists describe intense rainfall of several inches an hour, or 5 to 10 inches in 48 hours; accumulations of water in fields sufficient to drown a horse; and “walls of water” several feet high in local creeks occurring within minutes during a downpour. There

are numerous historical reports and photographs of flooded homes, broken bridges, and eroded stream banks. Some of these reports are for flooding along creeks in the vicinity of Telluride, Ouray, Silverton, Rico, Creede, and Gunnison. Although generally not mentioned in newspapers at the time, damage to mining properties has happened on the CWS and has been interpreted years after the event (Vincent and others, 1999). Elsewhere in the Western United States—as at the Clark Fork River, Anaconda, Montana (Nimick and Moore, 1991; Moore and Luoma, 1990), at Eureka, Colorado, on the Animas River (Vincent and others, 1999), and at the Puerco River, New Mexico–Arizona (Wirt, 1994)—tailings failures impacted up to 100 mi of stream and flood plain. In Nevada (Nash, 2001), small- to medium-sized tailings impoundments hit by flash floods have lost 20 to 90 percent of their tailings, which were transported 2 to 10 mi, channeled down arroyos or spread across alluvial fans. Outside of the well-studied Animas watershed, I know of little information on flood-related tailings on the Western Slope. In this study, I observed that fluvial tailings along Imogene Creek and Canyon Creek in the Snefels district were deposited as overbank sediments where storm waters were about 3 ft higher (vertically) than normal flow level. I also observed that drainage-control devices used in mine reclamation, such as straw bales and plastic netting, were commonly ripped away within a few years by storm runoff. More systematic studies are needed to document the frequency and distribution of storm waters and flood deposits to help in the estimation of their behavior in storm events.

Because rainfall in upland watersheds generally is not measured by gages, and many floods are not witnessed, the database for thunderstorms and flash floods in these settings is not well developed. As an alternative to weather predictions and databases, the behavior of flood runoff can be modeled by evaluating physical characteristics of watersheds (Black, 1996). Factors that influence peak runoff in storms include (1) capacity for water storage; (2) amount of infiltration; (3) basin shape and size, and (4) elevation. A watershed is said to be “flashy” (prone to flash floods) if it produces high flows of runoff in a short time during a storm event or snowmelt. Flashy character is increased in watersheds that have geology that supports little water storage and low infiltration, and a shape that is elliptical. Smaller watersheds at high elevations tend to most conducive to “flashy” high flows; elevation influences temperature and the ways extreme storms develop. Many of the headwater basins on the CWS have these “flashy” characteristics.

The statistical framework for prediction and risk analysis of floods is complex but can be calculated (Cohon, 1988; Kite, 1988). Because processes in these modern events are similar to those in the geologic record, geomorphological and sedimentological information may be good guides for evaluating effects and risks of future storm damage. For instance, the 100-year flood level could be estimated from Quaternary alluvial terraces. Numerous risk analyses have been undertaken for failure of uranium mill tailings. Fluvial processes, geomorphic evidence, catastrophic floods, and extreme hydrologic events

were discussed by Schumm and others (1981) as a guide to planning uranium mill tailings sites. Another study (Shepherd and Nelson, 1978) concluded that engineering aspects of tailings impoundments are important for the short-term behavior, whereas geomorphological processes (especially natural events such as floods and earthquakes) are more significant for the long term. At present, it seems that scientists are monitoring the relatively predictable effluent from mining areas while giving little thought to the risks posed by infrequent natural events like flash floods that have the potential to cause great impact.

Canyons and arroyos in western Colorado are capable of focusing substantial amounts of storm water with incredible power that can transport large quantities of boulders and fine-grained materials, such as those commonly found in Quaternary alluvial deposits. Some of the deposits appear to be debris flows (Blatt and others, 1972). However, the debris flow model may not be appropriate for some transported tailings. The large volumes of transported tailings, where catastrophic dam failure seems to have occurred, are nearly pure tailings with weakly developed to massive bedding and minor sorting. Where seen, coarse clasts are rare and seemingly random rather than in distinct beds. These features are more akin to those of "hyperconcentrated flows," which Costa (1988) defined as having 40 to 70 percent sediment by weight. Technical properties are intermediate between normal fluvial sediments and debris flows (moderate viscosity and shear strength) controlled by a dominance of fine-grained materials (silt and clay). The sedimentary features in transported tailings might also be explained by one dominant source of materials (impounded tailings). The term "fluvial tailings," used by many to describe transported tailings, may be inappropriate for these massive deposits but will be used later in this report. Much more study is needed of these unusual deposits to properly evaluate the conditions of formation.

Physical characteristics of watersheds that create high runoff in floods, briefly outlined above, should help in the identification of stream segments that may pose special threats to tailings impoundments and predict where the normally stable fine sands in tailings impoundments may be ripped up and transported as a viscous mud or sand flow (depending upon particle size and flow velocity). There are several important consequences for the environment downstream: (1) tailings are spread as overbank and channel-filling sand deposits for as much as 2 to 10 mi down the canyon or arroyo. Redeposition of fluvial tailings can be expected at sites where channels widen or gradient decrease, either of which decreases flow velocities. (2) Water in the debris flow is likely to attain more extreme concentrations of metals than found in simple low-flow runoff. Depending upon the ratio of solids to water, the metal concentrations could be even higher than determined by passive leach tests because mechanical mixing and abrasion in a debris flow would stimulate reactions. For years after the flood, the tailings will continue to react either with stream water or with precipitation, slowly releasing metals to the flood plain. The tailings will be difficult to clean out of ripar-

ian zones and remove from sandbar or overbank deposits. For years after the flood, the tailings will continue to react either with stream water or with precipitation, slowly releasing metals to the flood plain.

A preliminary evaluation of tailings impoundments on the CWS suggests that as many as 19 sites may be at high risk to failure during an extreme storm event (table 7). These estimates are made from the considerations discussed above, chiefly the size and shape of the basin above the tailings site, the geometry of the stream or canyon at the site, and the position of the impoundment. These sites are discussed later by mining district. More work is needed on criteria for the flood model and collection of appropriate field data.

## Geochemical Investigations of Historic Mining Districts

The influences of historic mining on watersheds and biota are not easily quantified. Ideally, multidisciplinary studies are made by an integrated team with diverse skills in hydrology, biology, geology, and chemistry (e.g., Buxton and others, 1997; Nimick and von Guerard, 1998; Plumlee and Logsdon, 1999). However, to undertake studies in such breadth and depth would cost tens of millions of dollars for an area the size and complexity of the CWS. This reconnaissance study is an attempt to highlight the major problems so that FLMA's can focus limited funds and staff on specific scientific investigations and reclamation decisions on the most significant problems.

One of the more complex scientific issues is the objective and quantitative evaluation of contamination sources from unmined mineralized rocks, a topic that is mentioned here but inadequately treated in the time available. The perspective in this study is clearly that of an earth scientist whose observations and interpretations may be at odds with the work and interpretations of scientists in other fields. Much additional work may be needed to reach a balanced and reliable understanding of these undisturbed systems and to develop a clearer definition of how much reclamation must be done to enhance water quality and biologic health while not attempting to purify systems that were releasing contaminants before the arrival of men and machines.

The next section of this report describes mining areas in a format that provides a framework of history and geology for the mineral-environmental investigations and their inferences. The descriptions are organized by the major watersheds of the Gunnison, Uncompahgre, and San Miguel Rivers. For each mining district, information is provided on commodities (metals) mined, along with sketches of the mining history, which are not consistent because I was unable to locate the same amount of information for all districts. This history is needed to gain an understanding of how current conditions came to be and to recognize that the early mining was done

with few environmentally sensitive regulations in place. Production tonnages are reported to show that, in many districts, chemical contamination is roughly proportional to the amount of ore produced or to the amount of earth disturbed. Mills and tailings are described in what may seem to be disproportionate detail because this information is scattered through the literature, at best, and typically not reported at all. The theme of mill tailings in riparian zones is repeated throughout this report because this feature of historic mining is inadequately understood by scientists and FLMA's.

Geology is the essential framework for this investigation and is the major influence on processes that operate at sources, during transport, and in natural mitigation of contaminated waters. The description of geology here is unconventional in that it is nontechnical (and possibly appalling to geologists) in an effort to simplify and demystify this science for the majority of readers who are not trained in geology.

The composition of bedrock and alluvium plays an important role in the evolution of water compositions and thus has been appropriately highlighted by Geoff Plumlee in the phrase "geo-environmental models" (Plumlee and Nash, 1995; Plumlee, 1999). The mineralogy and chemistry of ores, waste rocks, and altered host rocks determine the "geoavailability" of metal contaminants in source areas (Smith and Huyck, 1999), and the composition of rocks and alluvium nearby in the watershed dictate the potential for natural mitigation of contamination by processes of acid neutralization and subsequent metal precipitation (Smith and others, 1994; Plumlee, 1999). The geologic framework of the deposits and watersheds is emphasized here, especially the influence of large areas of altered rocks that can either exacerbate problems from mined rocks or can provide significant natural mitigation of initially contaminated waters. Alteration of bedrock units is particularly important, in the author's experience, and will be emphasized as a major factor in watershed geochemistry.

## Mining Districts of the Gunnison River Watershed

Geology in this large watershed is a diverse mix of Precambrian metamorphic rocks, Paleozoic-Mesozoic sedimentary rocks, and Tertiary volcanic rocks and intrusions (Tweto and others, 1976). Quite logically, this diverse geology is host to a variety of ore-deposit types with substantial ranges in chemical composition. Mining districts are shown in figure 10. Some of the more important ore deposits and unmined resources include (1) gold-sulfide deposits in metavolcanic rocks of the Gunnison Gold Belt; (2) thorium and titanium resources associated with mafic alkalic intrusions at Powderhorn; (3) silver-rich base-metal deposits in carbonate rocks adjacent to intrusions as at Tincup and Whitepine; (4) large, unmined molybdenum deposits in a felsic intrusive complex

near Crested Butte; and (5) vein-like uranium deposits in sedimentary rocks at Marshall Pass and Los Ochos.

## Cochetopa District (Gunnison Gold Belt)

*Location.*—Gunnison Co., 10–15 mi southwest to southeast of Gunnison (figs. 10, 11). Climate is relatively dry for the study area, averaging about 12 to 16 inches of precipitation per year.

*Principal commodities.*—Silver, copper, zinc, gold.

*Mining history.*—Gold was discovered in Cochetopa Creek in 1880, but mining was not substantial until after the silver crash of 1893 brought miners looking for gold. Copper was the major target after 1900, but the mines closed after World War I and never reopened. Uranium was located and mined in the 1950's. In the 1970's there was substantial exploration for gold using a new volcanogenic model, but no mining resulted from that good work.

*Production and disturbance.*—Small, intermittent production came from adits and shafts. Largest production was from the Vulcan mine area, discussed in the next section. Production of native sulfur and selenium at Vulcan is notable. Recorded production during the Depression years was thousands of dollars per year (very small). The underground workings generally produced small amounts of disturbance with dumps measured in only thousands to tens of thousands of tons at individual sites. Production from these deposits is given as 55,650 short tons (Long and others, 1998).

*Geology.*—The Gunnison Gold Belt is coextensive with Precambrian metavolcanic rocks of the Dubois Greenstone, a sequence of metamorphosed submarine volcanic rocks and interbedded cherts. Compositions of the layered rocks vary greatly, from mafic to felsic, and some ore zones are associated with siliceous layers that were originally submarine chert. Although the rocks have undergone multiple stages of folding, shearing, and faulting, the ore zones follow specific rock units (this is termed "stratabound"). Some samples on mine dumps have discernable fine-grained layering that may be pre-metamorphic and characteristic of stratiform volcanogenic massive sulfide deposits that form on the sea floor, according to research around the world on analogous deposits.

Three varieties of ore are recognized: (1) stratiform massive sulfide lenses rich in pyrite and containing sphalerite and chalcopyrite; (2) siliceous zones and veins rich in fine-grained silica that contain Au-Ag-Cu, partly as telluride minerals; and (3) local zones of native selenium with native sulfur. Native selenium is very rare in world ore deposits. In the 1970's–80's, exploration focused on potential for economic amounts of gold or zinc.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Chemical analyses of eight dump samples confirms the high concentrations of base metals

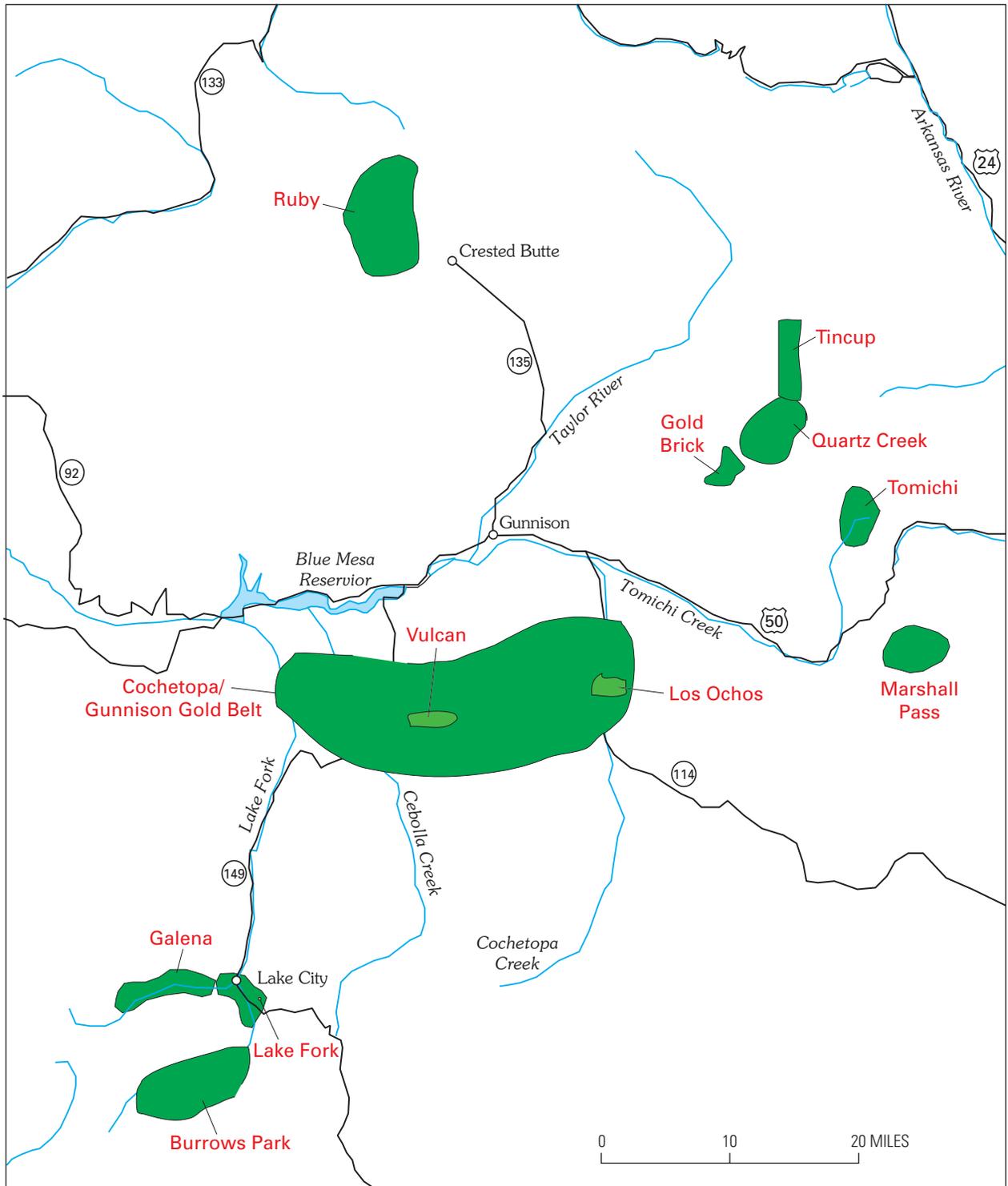


Figure 10. Location of studied mining districts in the Gunnison River watershed.

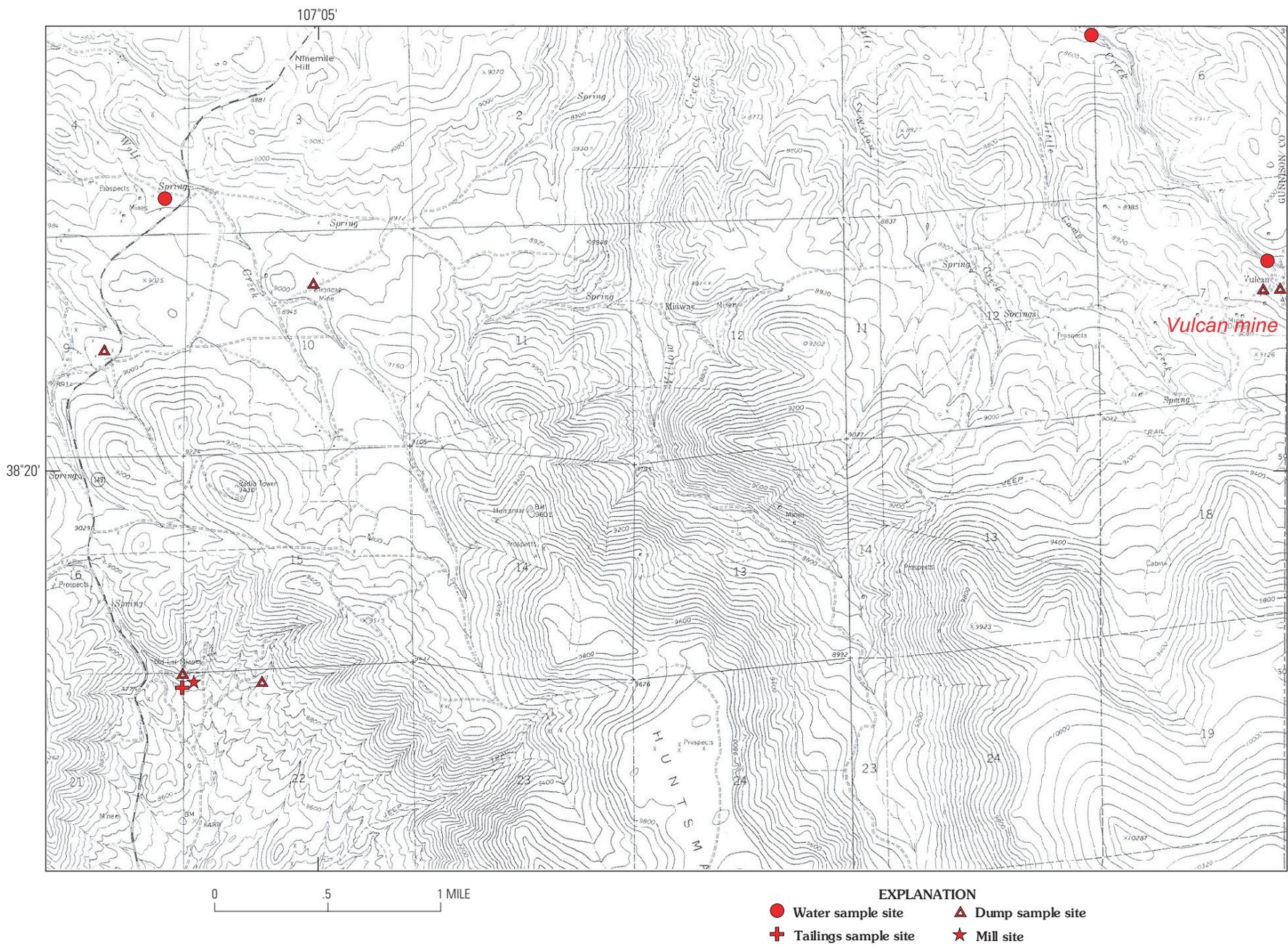


Figure 11. Features of part of the Cochetopa mining district, showing sample localities.



**Photograph 13.** Small mines and mills are typical of the Cochetopa district. The Old Lout mill has collapsed, and there is little evidence for production (small mine, small amount of tailings).

in these ores and waste rocks: very high concentrations were determined for Fe, Mg, As, Bi, Cd, Cu, and Zn (but Pb is not high). Leach tests on six samples produced some *extremely* high concentrations (>10,000 ppb Cu and Zn), chiefly from mineral desiccation crusts (photograph 18). Leach solutions had pH values in the range of 1.9 to 4.8, but most are < 3. Some dump samples from the Vulcan mine area, described next, yielded very high base-metal concentrations, such as Cu and Zn > 100,000 ppb. For leach solutions of pH < 3, concentrations of Al, Fe, Mn, Cd, Cu, and Zn are very high. Concentrations of Se are variable, and some are very high. Even the concentration of U is high, despite the fact that the ores contain only normal amounts of U. Under these extremely acidic conditions nearly all elements are mobile, and in this relatively arid climate many metals become enriched in desiccation crusts.

Metamorphosed mafic volcanic rocks in the Dubois Greenstone may have high acid-neutralizing capacity (ANC), as at other similar massive sulfide deposits (Nash and others, 1996). However, in this district many of the deposits are in siliceous layers, and these probably lack the ANC of the mafic layers. Mixed lithologies in alluvial units may provide good ANC to naturally attenuate acidic drainage from mines and dumps.

*Mills and tailings.*—Only one mill was located, and it served one mine with minor production. A smelter at the Vulcan mine treated a small amount of ore. Ore appears to have been chiefly the “direct shipping” type that was shipped elsewhere for milling and smelting.

*Surface waters.*—All of the mines examined were dry. The Gold Belt is in a semiarid climate zone typified by sage-

brush that appears to get much less snow and rain than the surrounding mountains. Surface waters do not appear to be influenced by these mines. The concern here is for possibly degraded waters in runoff from a brief summer thunderstorm that might carry a “flush” of acid and metals from soluble crusts on mine dumps.

*Summary.*—Studies of the Cochetopa district suggest that mine dumps from the massive sulfide deposits have the potential to generate acid and to release base metals, but, thanks to the arid conditions, that potential is rarely accomplished. There is little evidence for even local, episodic release of degraded waters sufficient to kill vegetation. The mined massive sulfide deposits and adjacent mineralized unmined rocks do not appear to be causing measurable contamination of surface water, except in the Vulcan mine area. If there is mine-related contamination, it probably would be for short periods during storms or snowmelt and is not expected to persist for more than about 1 mi.

*References.*—Sheridan and others (1981); Drobeck (1981); Hartley (1983).

### Vulcan Subdistrict (within Cochetopa District)

*Location.*—Gunnison Co., 15 mi SW. of Gunnison (fig. 12). Climate is relatively dry, about 12 to 16 inches of precipitation per year.

*Principal commodities.*—Sulfur, gold, silver, and selenium.

*Mining history.*—Gossans were identified in 1884, but most production of gold and silver was from 1889 and 1904,

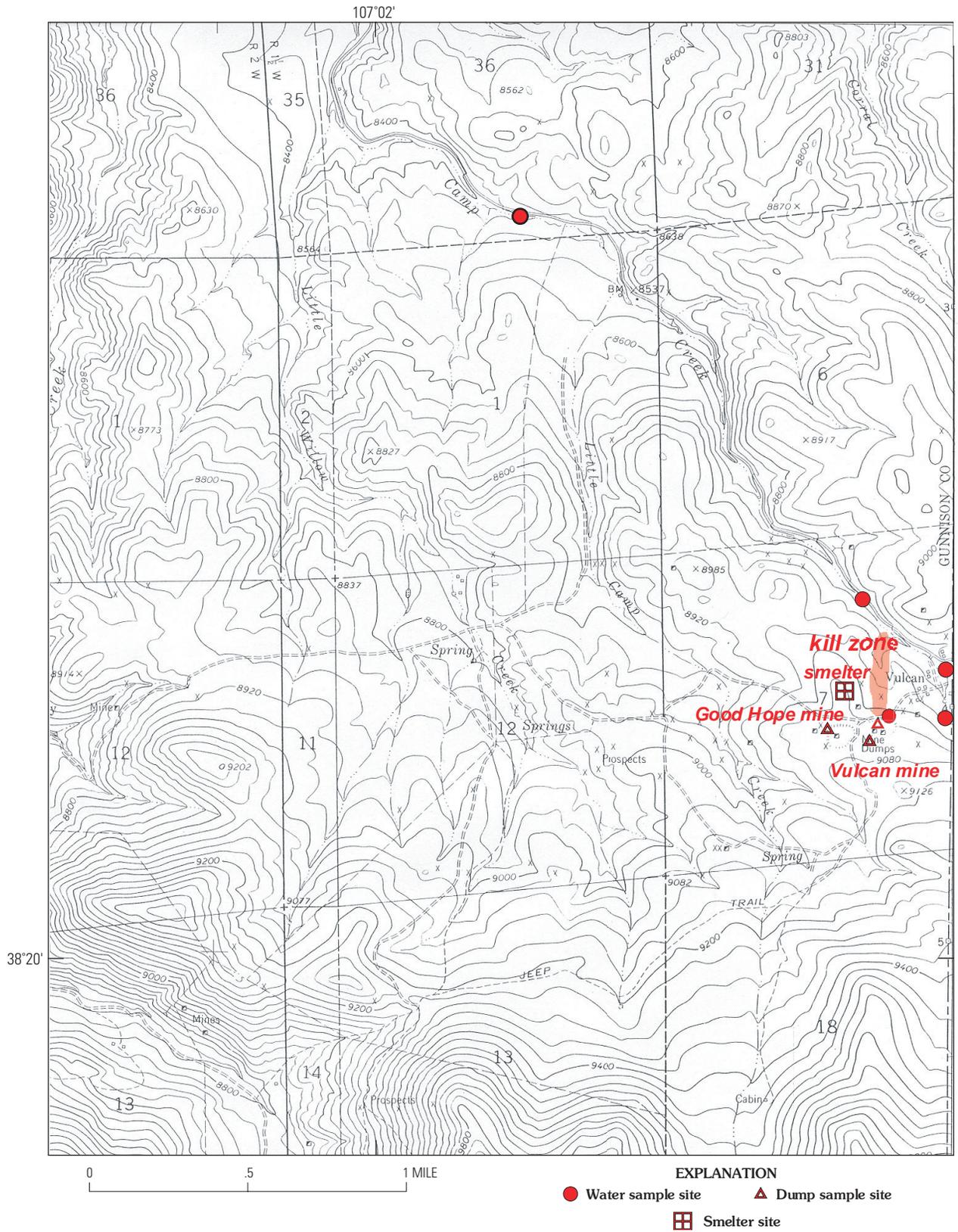


Figure 12. Features of the Vulcan mine area, showing sample localities. Some sites were sampled more than once.



**Photograph 14.** The Vulcan mine and nearby Good Hope mine created dumps of moderate size but very high sulfide content, in center of photograph. A small smelter is in the aspen grove (right side).



**Photograph 15.** North of the Vulcan mine and dump is this zone devoid of vegetation, termed a "kill zone." The zone reflects acidic waters related either to the mine dump or to outcropping mineralization. In most cases, kill zones are created after mining, but this one may be related to the gossan that helped prospectors discover the massive sulfide deposit.



**Photograph 16.** The remains of the smelter at the Vulcan mine are barely identifiable, but the black vitreous material to the left is unmistakably slag. This smelter probably operated before 1900.



**Photograph 17.** Vulcan mine dump, rich in sulfide minerals, develops this white crust of sulfate minerals that is very reactive in water. The crusts dissolve immediately to form an acidic solution with very high concentrations of zinc and copper. The gray material is pyrite that accumulates at the surface when clays wash away.



**Photograph 18.** This popcorn texture is created as sulfate crusts grow on pyritic waste rocks of the Vulcan mine dump. Scale is 6 inches long.

with some production of sulfur and selenium in 1917–18. Shafts developed ore to a depth of 600 ft and 1,800 ft on strike.

*Production.*—About \$500,000 worth of gold and silver (value not adjusted for inflation). Value of sulfur and selenium is not recorded.

*Geology.*—Lenses of pyrite-rich rock occur in Precambrian metavolcanic rocks of the Dubois Greenstone. The two major mines at Vulcan and Good Hope are part of the east-west alignment of deposits in the area that formed as part of the submarine volcanic stratigraphy. Unusual concentrations of selenium and tellurium occur in parts of the deposits (Hartley, 1983). Gold and silver have been the major commodities of interest, but copper and zinc have also been produced.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Rocks on the waste dumps are extremely rich in pyrite—more than 30 percent—and the trace metals Zn, Cu, Se, and Te are abundant enough to be of concern. Weathering of the pyritic rocks in the semiarid climate produces efflorescent sulfate crusts (photograph 17) that are soluble in water and create extremely acidic, metal-rich solutions in laboratory leach tests.

Outcrops north of the Vulcan mine are highly sericitized schist that are rusty colored and devoid of vegetation (photograph 15). Inspection of the gossan and soil profiles suggest that much of the rusty rock was produced by weathering prior to mining. Gossan zones such as these attracted the early prospectors in the 1880's (Hartley, 1983).

*Mills and tailings.*—The remains of a small smelter are located a short distance north of the Vulcan shaft (photograph 16). This old smelter (pre-1900?) probably was not effective because the amount of slag is very small (a few hundred tons). No mill is evident, so the pyritic ore probably was shipped directly to a distant smelter or mill. Hand-sorted ore rich in gold and silver is quoted by Hartley (1983) from unpublished company records.

*Surface waters.*—The mining areas were dry when visited in June of 1997 and 1998. The one stream that was flowing in June, and probably is perennial, is Camp Creek, which flows north from the Vulcan mine area to Blue Mesa Reservoir. Field measurements of pH in 1997 and 1998 were in the narrow range of 7.7 to 8.2, and conductivities ranged from 250 to 350  $\mu\text{S}/\text{cm}$ , which is fairly high. Chemical analyses of four stream-water samples showed low metal concentrations, with the exception of high zinc (2.6 times ALWS) found in the spring-fed creek south of the Vulcan mine and collected on-strike with the zone that contains the massive sulfide lenses (the water that was sampled may have been in contact with unmined mineralization related to Vulcan but probably not the Vulcan deposit itself or mine waste). Copper-zinc index (CZI) values for the four surface-water samples are below 0.5, which suggests relatively good aquatic health. Samples of Camp Creek north of the mining area, and the kill zone that goes downslope to the north of the Vulcan mine dumps, did not contain significant metal concentrations, although metal concentrations are elevated relative to surface waters elsewhere.

Areas with no vegetation (“kill zones”) adjacent to the Vulcan and Good Hope dumps resemble those produced by acidic drainage flowing overland from dumps. However, the

association of the large kill zone north of Vulcan with limonite-sericite-rich rocks suggests the possibility that some of the effect could be related to naturally outcropping mineralization (gossan) that existed prior to mining. Alternatively, episodic runoff during storm events or melting of snow would likely dissolve efflorescent salts on the dumps, as seen in laboratory leach tests, to create pulses of highly acidic, metal-rich waters. Judging from compositions of storm runoff collected in other mining areas, and the known high solubility of crusts on mine waste, there is high risk of a flush of initially acidic, metal-rich water during fairly rare runoff events that could be toxic to aquatic life and fish in Camp Creek. Additional sampling during storm events or snowmelt may be required to document mine-related contamination in Camp Creek.

*Summary.*—The highly pyritic waste dumps at Vulcan may pose a substantial threat to Camp Creek. However, the release of metals and acid probably is not as bad as imagined because the climate is semiarid and the coarse-grained pyrite is not highly reactive. Some of the pyrite does react to form crusts of efflorescent sulfate salts that are highly reactive, and these salts create very acidic and metal-rich waters in a “flush” during snowmelt or thunderstorms. Episodic releases of toxic waters is probable, but the frequency of these events and the spatial scale of the impact could not be measured. Laboratory tests demonstrate that sulfidic rocks on dumps from the Vulcan and Good Hope mines have high potential for acid generation and release of high concentrations of base metals. Under most conditions, there is insufficient water to initiate the reactions and transport the potentially toxic metals into nearby Camp Creek. Further studies may be required to sample runoff during storm events and to evaluate reclamation methods that would reduce the risk of runoff in storm events.

*References.*—Hartley (1983); Nelson and Riesmeyer (1983).

## Los Ochos Subdistrict (within Cochetopa District)

*Location.*—Saguache Co., 10 mi SE. of Gunnison (fig. 10). The climate is relatively dry, averaging about 12–16 inches of precipitation per year.

*Principal commodity.*—Uranium.

*Mining history.*—Uranium was discovered in 1954 and mined from 1955 to 1962. A series of shafts and adits were made along the east-west-trending zone for about 1.3 mi. Ore was trucked to a mill near Gunnison.

*Production and disturbance.*—Total production was about 1.2 million lb  $U_3O_8$  (McCulla, 1980). Moderate disturbance is present in the form of trenches and waste dumps. The dumps are substantial in size, on the order of a million tons.

*Geology.*—Veins rich in pyrite-marcasite-pitchblende occur along the high-angle Los Ochos fault zone. Sandstones in the Jurassic Morrison Formation are the favored wallrocks for the veins, and smaller amounts of ore occur in sandstones

in the Cretaceous Dakota Formation. Tertiary volcanic rocks formerly covered the area and are thought to have been important in ore formation.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Mined waste rocks contain high amounts of iron sulfide (pyrite and marcasite) in the range of 1 to 10 percent (photograph 20). Radioactivity is variable from about 3× to 20× background from place to place on the dumps, with some of the piles from test mining in the 1970’s being least weathered and most radioactive. Median concentrations (in parts per million) in six analyzed samples are: As, 290; Cu, 22; Mo, 37; Pb, 160; Se, 0.8; Th, 6; U, 58; V, 57; Zn, 66. Curiously, these dump samples contain some of the highest thallium concentrations in the study area: the median Tl concentration is 13 ppm, and the highest is 23 ppm. Chemically, The Los Ochos ore compositions share features of both the sandstone uranium ores of the Uravan district, described later, and polymetallic deposits of the study area.

Leach tests on three dump samples yielded solutions with pH’s in the range 3.5 to 5.4. Not surprisingly, the most acidic leachates carried the highest metal concentrations, including As-Cu-Zn values in the range of many polymetallic vein dumps. Uranium was very high in the pH 3.5 leachate (250 ppb) and much lower in the other two leachates. These limited results suggest that uranium is mobile in situations where sulfide minerals generate acid, and the acid solutions then carry uranium. Similar high uranium leachates were produced by sulfidic dump samples from vein-uranium deposits near Austin, Nevada (Nash, 2000b).

*Mills and tailings.*—Ore was trucked to the mill at Gunnison, thus there are no mill tailings at this site.

*Surface waters.*—The mines are dry at the surface and all mining probably was above the water table. No surface water was seen in June of 1997. A spring 1 mi north of the mine area, developed for livestock, has pH 7.9, elevated conductivity (680  $\mu S/cm$ ), and high concentrations of some metals that are mobile in alkaline waters: U and Se exceed ALWS and Se exceeds the standard for livestock. The CZI is low at 0.1. Concentrations of Zn and Mo are elevated, and the Mo may be a concern if livestock drank this water for long periods. These four elements in the spring water could be derived from Cretaceous shale or other sedimentary rocks, or from the fringes of mined or unmined uranium-bearing veins.

*Summary.*—The uranium mines are dry most of the year and do not appear to be creating surface drainage. Leach tests show that there is potential for uranium to be mobile in proportion to acidity: the more acid produced, the more uranium goes into solution. The status of the substantial tonnage of waste rock on mine dumps is not clear. There are no specific regulations for uranium mine dump reclamation, but radioactivity levels for much of the dump rock at Los Ochos are well above the standard for uranium mill tailings. Criteria need to be reviewed by State and Federal regulators. Further studies



**Photograph 19.** Uranium mines of the Los Ochos district are chiefly along an east-west fault zone in the dry hills of the Los Ochos district. Vegetation has not returned to the dumps in more than 30 years. This view is looking west from the east end of the mining area.



**Photograph 20.** Uranium of the Los Ochos district are rich in pyrite, causing them to turn red and yellow on the surface as they weather. Most dumps are considered waste, but these angular boulders are ore grade (mined in the 1970's but not taken to a mill) and the smoothed-out part is pre-1970 material that was low-grade ore (mineralized but not sufficiently rich in uranium to be worth milling). These rocks are highly radioactive.

should be undertaken, perhaps using specifications for uranium mill tailings as guidelines, to evaluate needs and methods for reclamation. The dumps are not an obvious threat to a few residents living about a mile distant. Materials mobilized in a storm event could spread some contaminants locally but are unlikely to contaminate Cochetopa Creek, a mile distant.

*References.*—Wright and Everhart (1960); McCulla (1980); Nash (unpub. field notes, 1978); Olson (1988).

## Marshall Pass District

*Location.*—Gunnison Co., 35 mi east of Gunnison (fig. 13). The climate is harsh with deep snowpack and about 25–40 inches of precipitation per year.

*Principal commodity.*—Uranium.

*Mining history.*—Radioactive zones were found in Harding Quartzite beds in 1955 and traced to deposits in fracture zones filled by yellow hexavalent uranium minerals. The same year, prospectors located several other kinds of uranium deposits in the area, including the larger ones along a major thrust fault in Leadville Limestone and Belden Formation shales at the site of the Pitch mine. Most of the prospects had little or no production in the 1950's, but substantial underground workings were made at the Pitch (Pinnacle) and Little Indian mines, with ore processing at the mill near Gunnison. The upsurge in uranium prices in the 1970's encouraged renewed exploration and subsequent mining of an open pit at the site of the former Pitch underground mine. The richest ore was trucked to a mill in New Mexico for about 4 years, through 1984. The Pitch mine is being reclaimed, and the owner has expressed interest in turning all claims (including several patents) back to the U.S. Forest Service.

*Production and disturbance.*—Production was about 1 million pounds  $U_3O_8$ . Early underground mines made small disturbances, but the 1980's open pit at Pitch mine created moderate disturbance and also is the site of a substantial landslide (slip) along the highly fractured Chester fault zone. Ongoing reclamation is addressing waste dumps and water discharge under a permit with the Colorado Division of Minerals and Geology (CDMG). The original Pitch mine adit has been sealed, with beneficial results.

*Geology.*—Rocks ranging in age from Precambrian to mid-Tertiary host ore deposits in this district. A thick section of Cambrian to Cretaceous sedimentary rocks is crunched against and under Precambrian schist and granite along the Chester fault. Tertiary volcanic rocks cover the sedimentary rocks and the fault zone, suggesting a Laramide age for the structure.

The uranium deposits of several types that are present are unconventional compared to major uranium deposits elsewhere in the United States. There are differences of opinion as to the source of the uranium and relations between deposit types, but the importance of the Chester fault in creating the fractures and structural permeability seems clear. Uranium is the only highly enriched element in these ores, and it occurs in both oxidized

and reduced states. The trace metals that are slightly enriched in the ores (Cu, Fe, Mo, Pb, Zn, V) are not economically significant but may provide clues to ore-forming processes. Fine-grained pyrite and marcasite are abundant in the Pitch mine ores.

## Mineral-Environmental Conditions

No new studies were made in this district because the mining company appeared to be making good progress on reclamation. Some general comments are made from my previous geochemical studies and from observations in a brief mine tour in 1998.

*Mineralized rocks and ores.*—Compositions are described in some detail by a previous study (Nash, 1988), which emphasized the fine-grained iron sulfide minerals in ore zones and generally low concentrations of base metals such as Cu, Pb, and Zn. The abundant limestone and dolomite in the mined and unmined rocks should be more than adequate to neutralize any local pockets of acidic waters if they are produced by iron sulfides. However, uranium is highly mobile in carbonate waters (Wanty and others, 1999), which is the chief concern here (along with decay products such as radium).

*Mills and tailings.*—Ores mined in the 1950's and 1980's were hauled to mills outside of the district. There are no tailings in the district.

*Surface waters.*—Unlike the Los Ochos district to the west, the Marshall Pass district receives large amounts of snow and summer rains. Water is a major factor in this district. The Pitch mine pit collects water, and the underground tunnel formerly had mine drainage. The mining company has undertaken millions of dollars worth of reclamation and is working with CDMG and other regulators to meet discharge standards. If unchecked, discharge of U and radionuclides would be a problem, but the reclamation activities, including a water-treatment plant, should bring the discharge into compliance.

No new samples were collected near the mines, and no water data for the Pitch mine are published. There is a small pond at the bottom of the pit, perhaps an acre in size. The mining company has a water-treatment plant onsite that treats water from mine workings and the pit pond before release into Indian Creek. The mining company monitors concentrations of U, and radionuclides such as radium, in the discharge.

A spring on the Chester fault zone south of the mine, developed for livestock, may be typical of waters in the area. The spring has a pH of 7.7 and conductivity of 90  $\mu S/cm$ , and base metals and U are low. The CZI for the spring water is only 0.11.

*Summary.*—High precipitation, combined with substantial disturbance created by the open-pit mine and the wide distribution of high- and low-grade uranium concentrations in mined and unmined rocks, create conditions that are unique within the eastern part of the study area. There is potential for water contamination caused by oxidation of iron sulfide minerals in the Pitch mine pit and in associated mine dumps

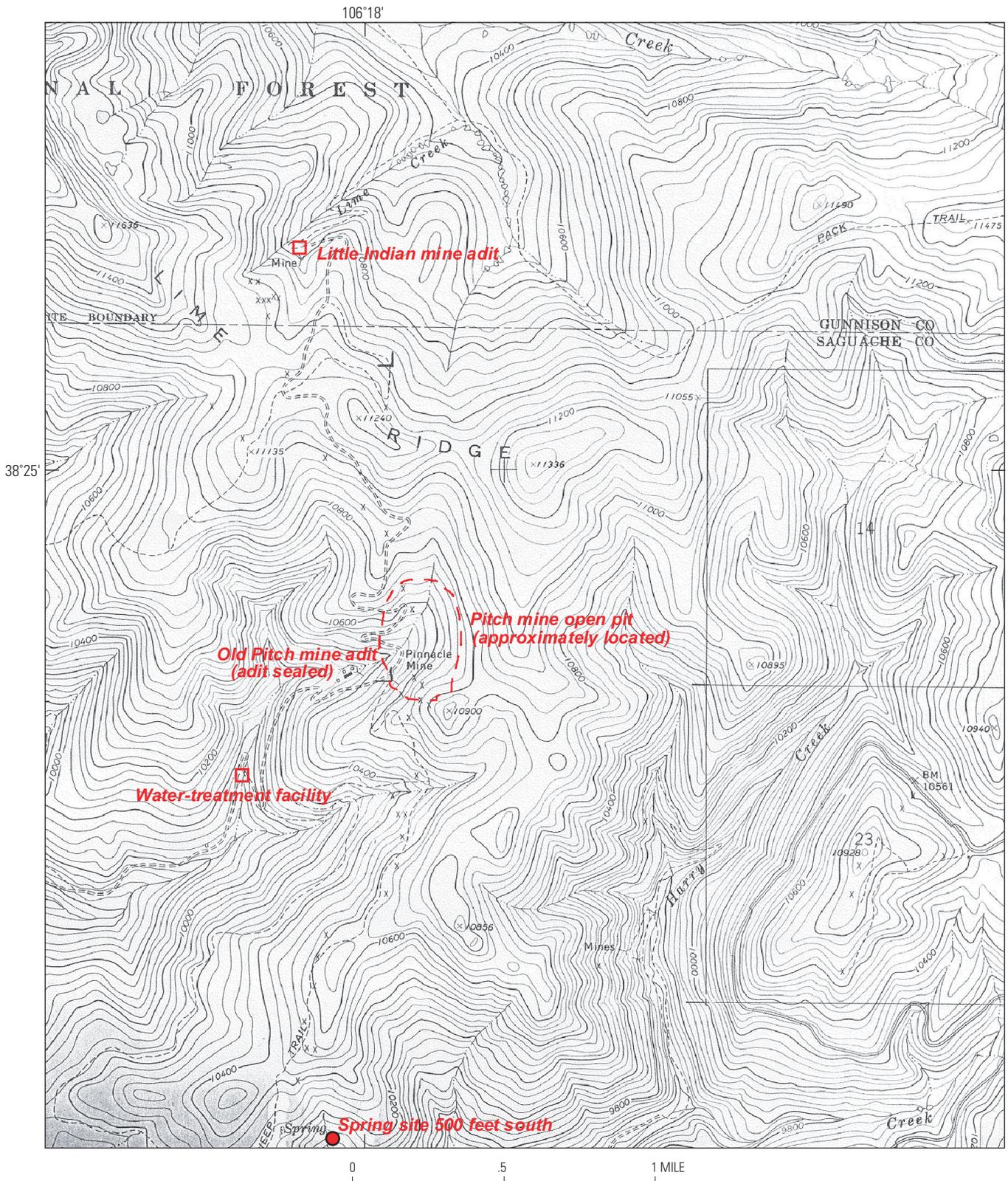


Figure 13. Features of part of the Marshall Pass mining district, showing major uranium mines.



**Photograph 21.** The Pitch uranium deposit formed along a major fault that placed gray shales and limestones (left side) against Precambrian granite (tan, right side), as seen in 1977 when excavation of the open pit was started. The heavy snowfall and abundant fracturing along the fault zone cause environmental problems that complicate reclamation.

and for mobility of uranium and associated radium. Acidic waters, if any, would be a local phenomenon in this area of carbonate rocks, but uranium is mobile in carbonate-rich water and could migrate for miles. A water-treatment plant and major reclamation work by the mine owner appear to be mitigating known and potential problems, but technical monitoring of water quality may be required to confirm tests by the company. The long-term future of the reclaimed lands and the stability of the slide on the Chester fault cannot be evaluated here.

*References.*—Wright and Everhart (1960); Nash (1981, 1988); Olson (1983, 1988).

## Gold Brick District

*Location.*—Gunnison Co., about 18 mi east of Gunnison (fig. 14). Climate is moderate for the study area, with about 20–30 inches of precipitation per year.

*Principal commodities.*—Silver, lead, gold.

*Mining history.*—Discovery was in 1879, and there was a burst of exploration activity thereafter, but discoveries were small in size. In the period 1905 to about 1925, three long tunnels were driven northward from the level of Gold Creek and resulted in more productive mines than the original shaft operations. Each mine was served by a mill with amalgamation plates and tables to concentrate gold and sulfide minerals (photograph 23). There was modest production in 1930's and little if any production after World War II despite exploration ventures in the 1950's.

*Production and disturbance.*—Total district production was less than \$500,000 from about 200,000 tons of ore (Long and others, 1998). At least three mines produced more than 10,000 oz Ag and 1,000 oz Au (which is not large for the study area or considering the size of the mine infrastructures). Mining disturbance was small, but three tailings impoundments are in the flood plain of Gold Creek (photograph 25).

*Geology.*—Precambrian metamorphic rocks dominate the geology of this district. Several varieties of schist containing muscovite, garnet, amphibole, and other minerals have been mapped. Beds of quartzite and impure quartzite also are common. Granitic and mafic igneous rocks, later metamorphosed to gneisses, intruded the bedded rocks. There were several periods of deformation and metamorphism in the Precambrian. Cambrian to Mississippian clastic rocks (chiefly quartzite and limestone) overlie the Precambrian rocks. Middle Tertiary dikes and plugs intrude the older rocks. The complex geology was unraveled with the aid of improved structural and geochronologic methods (DeWitt and others, 1985) that demonstrated the importance of mid-Tertiary plutonism for ore formation.

Quartz veins with sulfides and precious metals occur in Precambrian metamorphic rocks and as local replacement bodies in Paleozoic carbonate rocks. Galena was the most prominent ore mineral, but most ore values came from gold and from silver associated with galena. Most of the mined ore was oxidized “limonitic quartz” with residual galena; oxidation or partial oxidation extends to the deepest workings, about 1,300 ft below the surface. Some of the milled ore

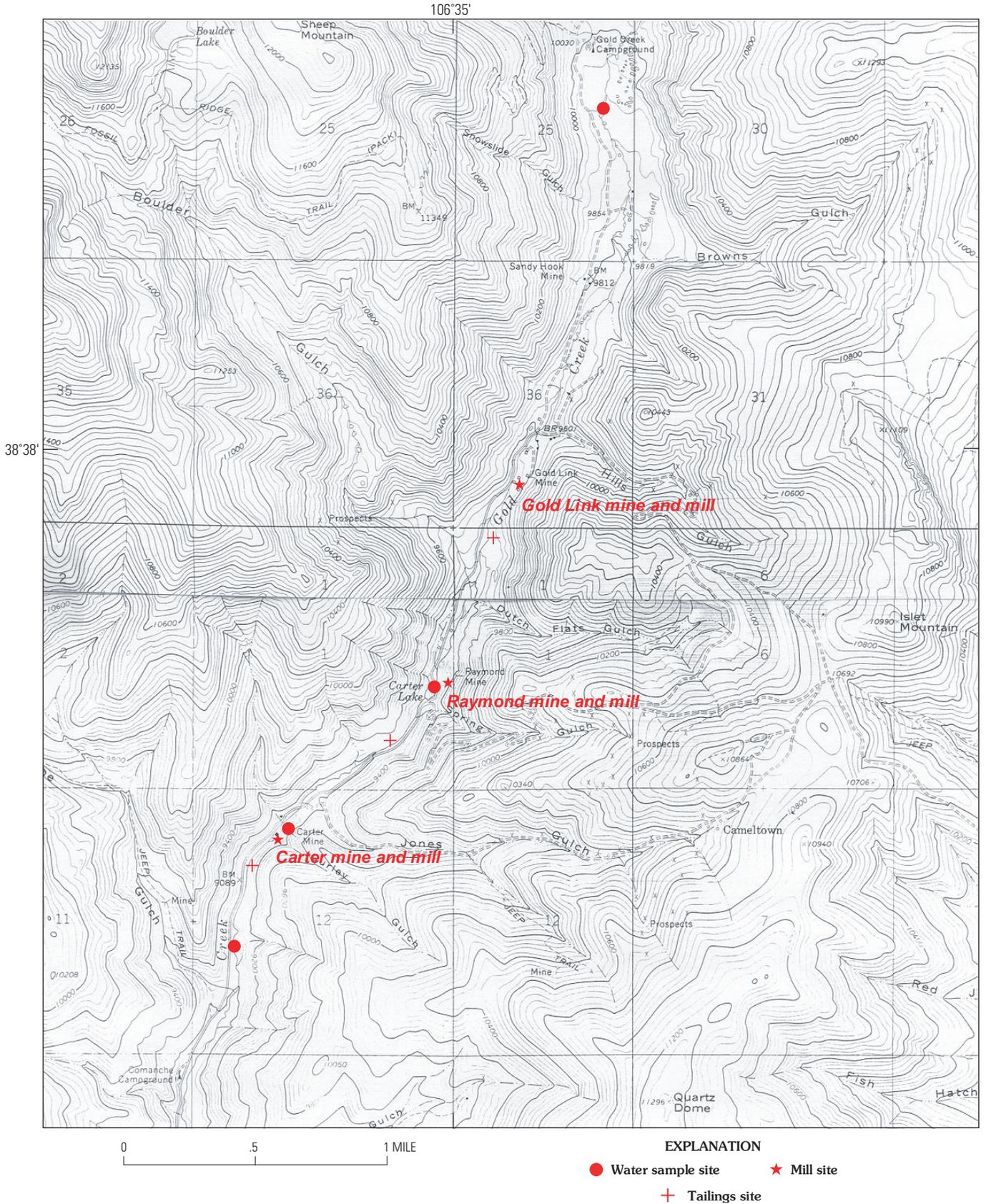


Figure 14. Features of the Gold Brick mining district, showing sample localities.



**Photograph 22.** The Raymond mill on Gold Creek, Gold Brick district, had 40 stamps and was active before 1915. Tailings were run through sluices and placed on a terrace close to the creek.



**Photograph 23.** The Gold Link mill was a large and impressive structure in 1915. Tailings were run eastward about 500 yards and placed on a terrace about 50 yards from the creek.



**Photograph 24.** Tailings from the Gold Link mill were placed in gullies east of the mill. The pine trees probably grew after the tailings were emplaced, possibly before 1920.



**Photograph 25.** Tailings from the Raymond mill (white in center of photograph) were sluiced to the east about ½ mi and placed on a bench above Gold Creek. Water infiltrating these tailings goes into alluvium and then into the creek.

contained sphalerite, bornite, and pyrite, and some pockets of molybdenite (Mo sulfide) were encountered. Oxidized Mo and Cu minerals on upper levels were not of economic importance. Ankerite (Ca-Mg-Fe carbonate) was locally abundant. These deposits are polymetallic veins, but in some details of mineralogy and texture they differ from other polymetallic veins in the study area (perhaps a reflection of greater depth of formation here relative to those in volcanic rocks).

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—The deposits in this district are polymetallic, but the quartz veins are not as rich in sulfide minerals as most polymetallic ores in the study area. Chemical analyses of dump and tailing samples show them to be rich in Ag, As, Cd, Pb, Sb, and Zn, and moderately rich in Cu relative to other metallic deposits in this study.

*Mills and tailings.*—At least three substantial mills are known, all along Gold Creek. The tailings from these mills were placed in an unusual way, as long and narrow “shoe-strings” hundreds of yards east of the mills. The light-colored tailings were carried in wooden sluices (still evident) and placed by design in lowlands within the valley, and in several places the tailings are less than 50 ft from Gold Creek and within the flood plain (photograph 25). Other mills at this time were dumping their tailings into streams, as allowed by law prior to 1935. Although production figures suggest small tonnages were produced by these mines, the tailings appear to be relatively large in volume, perhaps because they were placed in long, thin impoundments which actually have more surface area than the larger tonnages placed in thicker piles, as at Whitepine and Iron Springs.

Leach tests on four tailings samples show moderate to low acid-generating potential (pH 3.4–5.3), and generally low metal release. However, two samples (from Gold Link and Carter mills) released very high concentrations of Pb, Se, and Zn.

The tailings along Gold Creek merit additional work and evaluation, including surveys for land status. The tailings are in the flood plain and vulnerable to erosion in a major storm event. The reactivity of the tailings is not conclusively understood from the four leach tests here: some leach results suggest low potential for acid generation and metal release, but other results suggest high and potentially significant release of Pb, Zn, and Se. The land status also is complex: there are many patented mining claims (private land) in this district, but the tailings are sufficiently distant from the mines that all or parts could lie on the Gunnison National Forest.

*Surface waters.*—The surface waters in this district are clear and no iron stains are visible on cobbles. Measured pH values were in the range 6.9–7.9 in 1997 and 1998. Conductivities were very low (<100  $\mu\text{S}/\text{cm}$ ) in streams but reached 240  $\mu\text{S}/\text{cm}$  in one mine drainage. Chemical analyses of four water samples showed very low metal concentrations for a mining area; the highest Zn concentration was 0.9 $\times$  ALWS.

Other metals are elevated but well below ALWS. The CZI for samples of Gold Creek are <0.2 at and below the mines and tailings, suggesting low threat to aquatic life.

*Summary.*—Historic mining causes only minor degradation of water in this district. Contamination is limited to elevated concentrations of a few metals such as Zn but not in amounts that exceed ALWS. Mill tailings were placed in lowlands adjacent to Gold Creek, where they are potentially susceptible to erosion during a major storm event, such as a 100-year flood. The three tailings impoundments probably have a high risk of failure during an extreme storm. Metals such as Pb and Zn are possibly being leached from mill tailings. The effects are not apparent in the water quality of Gold Creek. Technical monitoring of contamination into shallow ground water may be called for, which would require that wells be drilled and carefully sampled.

*References.*—Crawford and Worcester (1916); DeWitt and others (1985).

## Quartz Creek District

*Location.*—Gunnison Co., about 20 mi east of Gunnison (fig. 15). Climate is moderate to severe, with about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver, lead, gold; molybdenum and tungsten; graphite.

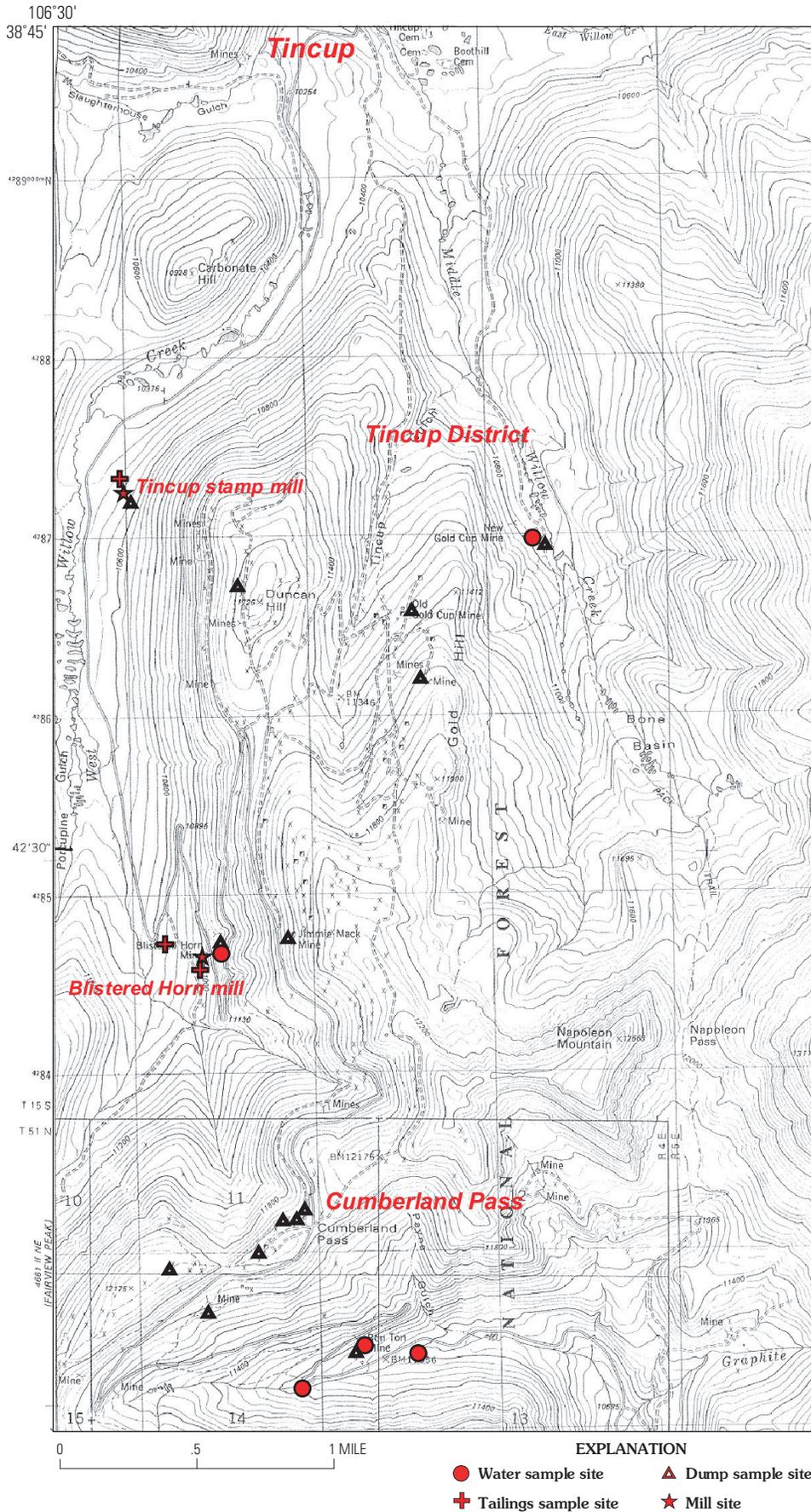
*Mining history.*—The gold-silver mines were first located in the 1890's. The Mo-W mines near Cumberland Pass were located somewhat later and mined chiefly during World War I. There was very minor production in the Depression years. None of the mines have been in operation since World War II, but a significant exploration program for molybdenum in the 1970's included the drilling of many core holes near Cumberland Pass. There has been no mining or exploration in recent years.

*Production and disturbance.*—Total less than \$500,000. Small disturbance from mining.

*Geology.*—The western part of the district above Pitkin is really a continuation of the adjacent Gold Brick district. In this area, quartz veins with base-metal sulfides and precious metals occur in Precambrian metamorphic rocks, and minor replacement bodies occur in Paleozoic carbonate rocks. The deposits in the northeastern part, near Cumberland Pass, are related to a Tertiary intrusive complex and include a type of vein deposit that is rich in Mo-W. A large pyritic alteration halo adjacent to the Tertiary diorite intrusion is visible from Cumberland Pass (photograph 28).

## Mineral-Environmental Conditions

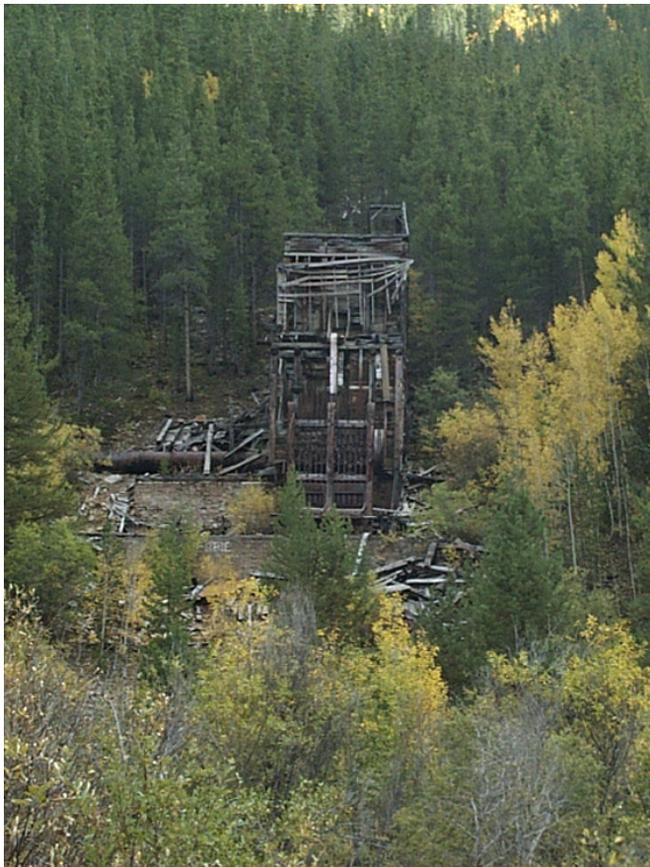
*Mineralized rocks and ores.*—The ores in this district differ from most others on the CWS in the major metals of economic interest. The ores near Cumberland Pass (Bon Ton mine) are rich in Mo and the associated trace metals Bi



**Figure 15.** Features of the Tincup and northern part of the Quartz Creek mining districts, showing sample localities.



**Photograph 26.** The Pitkin mill, in the village of Pitkin, was active before 1900. Stone foundations are suggestive of an early date as concrete foundations were used after about 1900. Tailings may be in the wetlands to the east of the mill, or washed down Quartz Creek.



**Photograph 27.** The Roosevelt mill, about 2 mi south of Pitkin, is remarkably intact; the vertical features are stamps that crushed and ground ore. No tailings are evident. The tunnel and mill venture reportedly failed in its attempt to reach ore to the east and mined through shafts at the Camp Bird and nearby mines.



**Photograph 28.** Altered rocks at Cumberland Pass formed in association with a diorite intrusion that also created several molybdenum-tungsten vein deposits. These bleached rocks with iron oxides on fractures formed from disseminated pyrite and are a guide to prospectors. The area in this photograph was explored by several diamond drill holes during a search for molybdenum in the late 1970's.



**Photograph 29.** The Bon Ton mine was a modest producer of molybdenum during World War I. Drainage from the tunnel (right center) flows over the waste rocks on the dump.



**Photograph 30.** Mine drainage from the Bon Ton mine is acidic and makes these tell-tale red colors. Reactions of mine water with sulfidic waste rocks on the dumps further degrades the water quality.

and W and have moderate concentrations of Cu, Pb, and Zn. Concentrations of Ag are relatively high, and As concentrations are variable (low to moderate).

*Mills and tailings.*—The status of mills and the condition of tailings in this district is unclear. A mill was reported by Hill (1909) to be 1 mi northeast of Pitkin, but was not located in this study. At least two large mills are visible today, one on the southwest edge of the town of Pitkin (photograph 26), and another 2 mi to the south on Quartz Creek at Roosevelt (photograph 27). The latter is described as part of a speculative venture to produce ore from a long tunnel to the east and appears to have produced very little ore; no tailings are evident in the willows along Quartz Creek. The mill at Pitkin was a significant producer, but no tailings could be identified in the adjoining area of private property. Tailings from the Pitkin mill are probably in the wetlands southeast of the town, less than a mile from the State fish hatchery.

*Surface waters.*—Water in this district is locally contaminated at small mines. A modest flow of water from the Bon Ton mine (photograph 29), just south of Cumberland Pass, may be typical: it flows at about 20 gpm, has a pH of 3.7 and conductivity of 210  $\mu\text{S}/\text{cm}$ , compared to pH 6.7 and conductivity of 40 in Quartz Creek below the mines. The acidic mine drainage contains high concentrations of many metals relative to ALWS (Al 6 $\times$ , Cd 8 $\times$ , Cu 230 $\times$ , Fe 6 $\times$ , and Zn 8 $\times$ ). The Bon Ton, which produced a small amount of molybdenum, is one of the few sources of Mo-rich waters in this study with 29 ppb Mo.

A sample was taken from Quartz Creek to determine the effects of the large alteration area near Cumberland Pass and

the several mines near Bon Ton: the pH of 6.7 and chemical analyses indicate that water has elevated metal concentrations and Cu is 6 $\times$  ALWS. The CZI values for the Bon Ton mine drainage waters are high (30.5 and 29.9). The CZI for Quartz Creek is elevated (1.7) and may represent a threat to aquatic life. Because the waters of Quartz Creek looked clear and made no iron deposits, no samples were collected farther south of the mining area at Cumberland Pass. Additional sampling of Quartz Creek would establish how far the metal contamination extends.

*Summary.*—The effects of historic mining are highly localized in this district and are not evident in the water quality of nearby streams. The large pyritic alteration zone near Cumberland Pass does not degrade stream-water chemistry.

*References.*—Crawford and Worcester (1916); Goddard (1936); Dings and Robinson (1957).

## Ruby District (Irwin)

*Location.*—Gunnison Co; 10 mi west of Crested Butte (fig. 16). The climate is relatively severe, with thick snowpack and about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver; lesser lead, zinc, copper, gold.

*Mining history.*—Silver was discovered in 1872, and Irwin was soon developed as a town. Railroad service was completed from Gunnison to Crested Butte in 1881. There was modest production in the 1880's and 1890's, but little other production until the 1940's–50's when the Keystone and

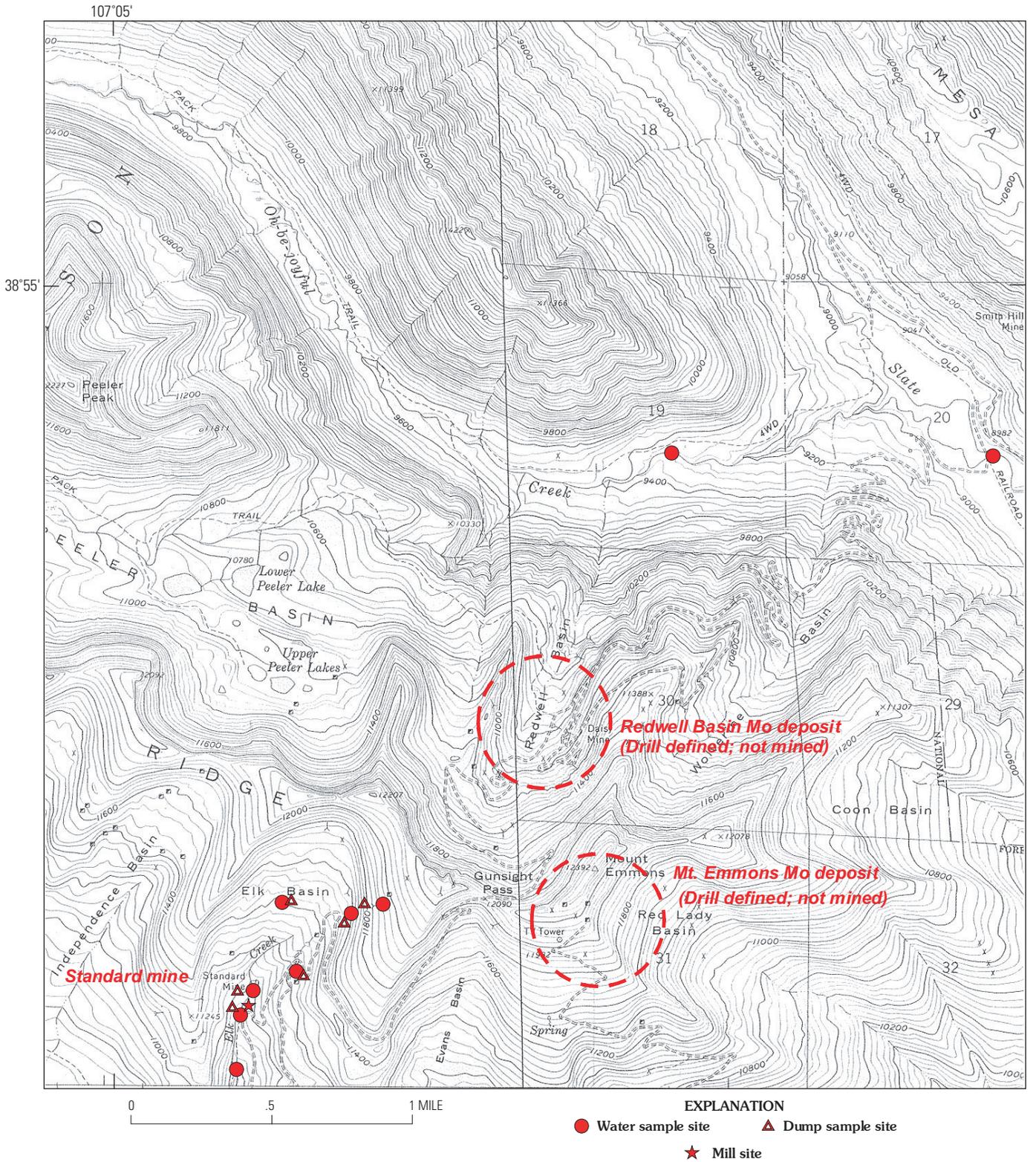


Figure 16. Features of the northeast part of the Ruby mining district, showing sample localities. Approximate locations of two known molybdenum deposits are shown.

Standard mines were developed. Exploration for molybdenum in the 1970's defined at least two substantial deposits beneath Mt. Emmons (Thomas and Galey, 1982), but there has been no mining of these deposits.

*Production and disturbance.*—Production, chiefly before 1900 and from 1955–69, was about 24,000 oz gold, 5.2 million oz silver, 6.6 million lb copper, 31 million lb lead, and 55 million lb zinc (Ellis, 1983). Large reserves and resources of molybdenum have been defined by drilling (more than 200 millions tons of ore). The underground mines created small to moderate disturbance. Dumps and tailings at the Keystone mine (polymetallic vein deposit) were effectively reclaimed in recent years.

*Geology.*—Sandstones and shales of Cretaceous and Tertiary age dominate the geology of this district, but intrusive sills, dikes, and stocks of Tertiary felsic igneous rock are keys to formation of the metal deposits. Moderate to intense pyritic alteration adjacent to the intrusions creates bright colors where weathered; some of the altered zones are given apt names such as Red Lady basin. Coal beds occur within the Tertiary sequence, and some have been mined north of the district.

## Mineral-Environmental Conditions

Part of this district in Elk basin (Standard mine and nearby mines) was investigated in moderate detail (2 days on site) because this is a priority concern of the USFS (D. Gusey, written commun., 1998). The rest of the district, including the east side (Slate Creek drainage) were investigated very briefly. Wanty and others of the USGS have more detailed hydrogeochemical studies underway in Redwell basin and other areas adjacent to Mount Emmons.

*Mineralized rocks and ores.*—The polymetallic ores of the Standard mine area were the only ones studied. Samples from mine dumps are very high in Pb, high in Ag, Cd, Cu, and Zn and moderately high in As and Bi relative to other polymetallic ores on the CWS. Concentrations of Se are notably high. Leach tests on five samples yield pH's in the range 2.9 to 3.3, which is more acidic than most in this study. The leachates contained very high concentrations of Pb, Zn, and Cd and high concentrations of Cu, Fe, Mn, and Se.

*Mills and tailings.*—There were three or more mills in this district, but only two are well known. The Keystone mill was a major operation through the 1950's and created large tailings piles that are shown on the USGS topographic map. A former source of contaminants, the tailings were reclaimed by AMAX (Cyprus AMAX) Mining Co. in the mid-1990's, including placement of a plastic liner on the downslope side to contain percolating waters. Any contaminated water flowing from the mine and tailings is now processed through the water-treatment plant.

The Standard mine and mill operated in the 1950's and appears to have had only small production. The tailings impoundment is intact and measures about 1 acre in size; the pond continues to receive runoff, and the waters seep

through the tailings and flow a few feet into the nearby creek (photograph 34).

The position of the tailings "pond" at the Standard mine, above and close to Elk Creek, makes it vulnerable to runoff and erosion during an extreme storm (photograph 35). My initial evaluation suggests a high risk of failure during an extreme storm. The tailings are saturated with water, the pond continues to collect water from overland flows, and the bank slopes steeply down to Elk Creek. These conditions may be conducive to failure by slippage (as in a landslide) or liquefaction during a major storm. Erosion and undercutting are also possible.

*Surface waters.*—Water quality in this district is variable and not reliably predicted from geology. Some basins of previously reported impacted water quality are, in my brief experience, quite good, whereas the area of small mines in Elk basin, expected to be mitigated by sedimentary rocks, has quality worse than expected. Waters in basins underlain by carbonate-bearing sedimentary rocks, and with few or no mines, has very good water quality at three sites on West Elk, Slate, and Oh-Be-Joyful Creeks. At these sites, pH values range from 5.8 to 7.3, conductivities are 20 to 40  $\mu\text{S}/\text{cm}$ , and zinc is 3 to 90 ppb. The highest zinc content, 90 ppb in Slate Creek, may reflect coal mining or associated Cretaceous black shales. The CZI values for these three sites range from 0.3 to 0.02, which suggests few problems for aquatic life.

Water in Elk basin is contaminated by mine waters and mine waste from several polymetallic mines. The Standard mine is the most obvious source, but there are others higher in the basin that also contribute acid and base metals (photograph 31). The distribution of samples and pH values is shown on figure 17. The sulfide-rich mine workings and waste dumps generate acidic waters, with pH values near 3, and high to very high concentrations of Al, Cd, Cu, Fe, Pb, and Zn. The Pb concentrations are among the highest in this study, and those of Cu and Zn are far in excess of ALWS. Mine water flowing from the partly collapsed Standard mine adit is similar to that from at least six other mines in the basin above. Significant, in my opinion, are the overland flows during runoff and storm events, which are nearly as acidic (pH 3.3), and that flow onto the Standard mine-waste dumps and tailings (photograph 33). There are many seeps and flows from the mine and waste that enter Elk Creek within 100 ft or less, before natural reactions can mitigate water quality. CZI values range from 104 to 27 for these waters, which indicate problems for aquatic life.

Water quality in Elk Creek is degraded by waters seeping from the Standard mill tailings pond. The pH of water in the pond is 3.1 and water seeping from the pond has pH values of 2.9–3.0; conductivity of pond water is high at 450  $\mu\text{S}/\text{cm}$  and is even higher (775  $\mu\text{S}/\text{cm}$ ) in the seeps. Chemical analyses show that the water seeping into Elk Creek has extremely high Zn (>200 $\times$  ALWS), Cu (>100 $\times$ ), and Pb (>15 $\times$ ). Concentrations of Fe, Mn, Se, and Cd also are very high. Pore water collected from the slushy tailings has a pH of 2.5, conductivity of 1,250  $\mu\text{S}/\text{cm}$ , and even higher metal concentrations than the seeps (Pb is 110 $\times$  ALWS).



**Photograph 31.** An old mine and waste dump in upper Elk basin creates acidic mine drainage. Drainage such as this flows down onto the Standard mine below.



**Photograph 32.** The Standard mine, from the 1950's, is relatively intact, and no reclamation work has been done. Overland flow of water adds to drainage from the tunnel to flood the waste dumps.



**Photograph 33.** Drainage from the Standard mine tunnel reacts with waste dump rocks, gaining more metals and acid, then runs into Elk Creek. Red coatings of iron minerals form at some times of the year if pH rises above 4, but when pH is below about 3.5 the water is clear and forms no precipitates.



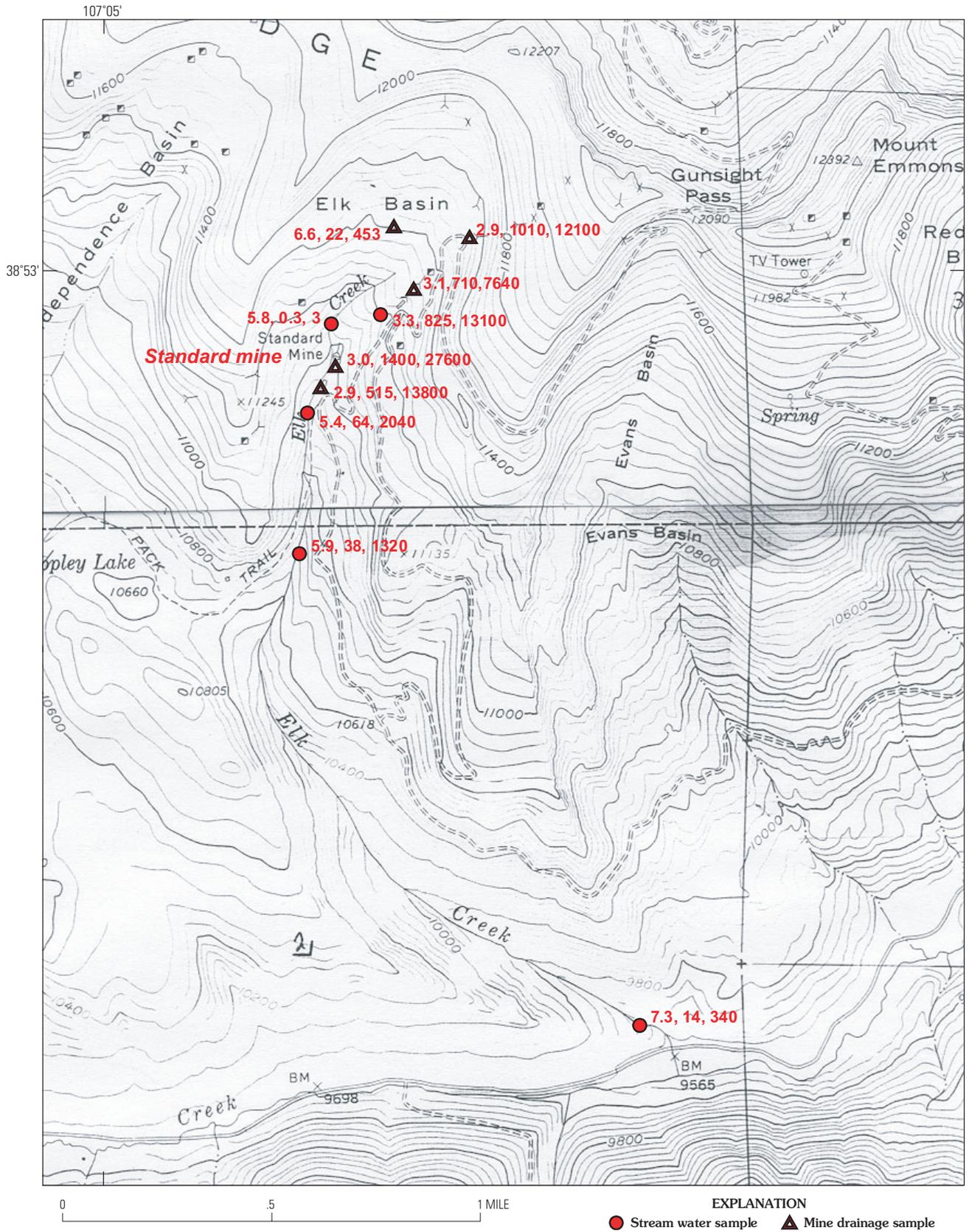
**Photograph 34.** Only the foundation remains from the Standard mine mill, and the tailings pond is intact to the south. Elk Creek is on the right side of this view.



**Photograph 35.** The Standard mine tailings pond collects runoff from waste dumps, making the water composition worse than it might be.



**Photograph 36.** The red material is a variety of ferricrete; the iron was deposited from acidic drainage from the Keystone vein. This is one of the largest volumes of ferricrete on the Western Slope. This material could have formed either prior to mining, from natural springs, or by flow along mine workings. The trees were killed by a forest fire.



**Figure 17.** Sample sites and water compositions in the Elk Creek watershed near the Standard mine. Explanation of composition (2.9, 1010, 12400): pH 2.9; 1010, Cu in ppb; 12400, Zn in ppb.

Water compositions were examined at several sites downstream (south) of the Standard mine. The pH of Elk Creek rises progressively (fig. 17) from 5.4 to 7.3, and zinc, lead, and copper concentrations decrease proportionately. However, zinc is still high (10× ALWS) 1.9 mi to the south, near the road along Coal Creek. The expectation was that the Zn, and also Pb and Cu, concentrations would decrease in a shorter distance by reactions with minerals such as calcite in sedimentary rocks and by dilution. The CZI values decrease from 6.7 to 1.7 below the mines, which shows improved water quality but persistent Zn and Cu that may be a threat to aquatic life. Natural mitigation is not as effective in these siliciclastic rocks as in carbonate and volcanic rocks elsewhere on the CWS. The mobility of base metals in Elk Creek basin is surprisingly high and underscores the need for reclamation that will minimize contact of surface waters with polymetallic-sulfide waste.

Twenty years ago, drainage from the Keystone mine tunnel and from the Keystone mill tailings were poorly controlled and contributed significant contamination to Coal Creek, the drinking-water supply for Crested Butte. Reclamation activities by Cyprus AMAX Mining Co. have made great improvements to the quality of Coal Creek. A treatment plant, the largest treating mine-related waters on the Western Slope, removes metals by a series of chemical reactions and processes, driven by the addition of lime to raise the pH to about 10. As of 1998, Coal Creek appears to meet water-quality standards because it is no longer on the 303(d) list of CWQCD.

*Summary.*—Recent reclamation at Keystone greatly improved drainage and reduced reactions in mine dumps and mill tailings. The \$40,000,000 water-treatment plant improves water quality to regulatory standards. Together, these actions have produced dramatic improvements in the quality of Coal Creek, the drinking-water supply for Crested Butte. Reclamation is needed at the Standard mine, chiefly to reduce surface-water interaction with sulfidic mine waste and mill tailings. In particular, moderately acidic waters from the East Fork of Elk Creek need to be redirected away from Standard mine-waste materials. Dilution and neutralization by sedimentary rocks mitigates acidic waters and lowers metal concentrations, but Zn-Cd remain higher than desired for aquatic life or drinking water. Reclamation is needed at the mine and mill site to reduce the concentrations of Zn-Cd-Cu leaving the mine area.

The location of the tailings impoundment at the Standard mine makes it vulnerable to failure by erosion, slippage, or liquefaction during an extreme storm. This tailings impoundment ranks among the top five sites for risk of failure during an extreme storm and should be investigated further for physical stability and possible methods of stabilization.

*References.*—Thomas and Galley (1982); Ellis (1983).

## Tincup District

*Location.*—Gunnison Co., 30 mi NE. of Gunnison (fig. 15). The climate is relatively severe with about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver; minor gold (placer), lead, zinc.

*Mining history.*—Gold was discovered in the Taylor River in 1861, but significant mining did not occur until 1880 when the lode deposits were developed. The most profitable mining in the district was from oxidized silver-lead deposits in replacement and vein deposits in limestone. The largest mine, the Gold Cup, is credited with more than \$1,000,000 worth of production. A small amount of iron was mined at the Cumberland mine near the pass, and relatively small amounts of placer gold have been mined from Willow Creek and Tincup Gulch. Production from 1901–35 was small, slightly more than a thousand tons of ore, and increased slightly during World War II. The mines have not been active for many years, and production has been minor since World War II.

*Production and disturbance.*—Total production was somewhat in excess of \$3 million, but the waste dumps and limited amount of surface disturbance caused by the underground mines do not suggest this magnitude. In the 1880's, there were two small smelters, and later there were at least two mills on the flank of Gold Mountain that utilized trams (photograph 38).

*Geology.*—The oldest rocks in this district are Precambrian gneiss, which are overlain unconformably by Paleozoic sedimentary rocks, chiefly limestone and dolomite. These rocks were intruded by Tertiary diorite porphyry (intermediate composition). Alteration produced by the intrusive body is most evident at Cumberland Pass, but effects in the carbonate rocks are subtle. Skarn-type deposits of magnetite and other minerals are clearly associated with the diorite intrusion, and the polymetallic vein and replacement deposits of silver and lead probably are also related to Tertiary intrusions. To the south, extending into the adjoining Quartz Creek district, prospectors found quartz veins containing molybdenite that are relatively small in size compared to the important Climax-type “moly” deposits near Leadville and Crested Butte that were discovered later.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—The polymetallic deposits of this district appear to be smaller than the value of production would imply, but the small volume of waste at the surface is characteristic of pre-1900 bonanza ores that were hand-sorted underground. The ores are unimpressive to the eye, probably because most of the ore minerals are oxidized varieties. Chemical analyses of nine samples from dumps of replacement deposits are very high in Ag, Pb, and Zn and high in As, Bi, Cu, Mo, and Sb (no results for Se and Te). Analyses of three samples from skarn prospects showed relatively low base-metal concentrations, but high Fe. Chemical analyses of two tailings samples are similar to those for the dumps, but show lower base-metal concentrations. The old stamp mills were not very efficient: two tailings samples contain 110 and 390 ppm Ag, and 1.2 and 0.2 ppm



**Photograph 37.** Tailings from the mill south of Tincup have been dug into, possibly to reprocess the tailings. The stamp mill created sand that is coarser than beach sand. The cable (center of photograph) is from the tram that ran up Gold Mountain to the southeast. This is one of the rare tailings piles that is not near a stream.



**Photograph 38.** The stamp mill south of Tincup is now a pile of beams. A tram cable runs up the hill to the right, and tailings remain nearby. This mill probably operated before 1910.



**Photograph 39.** The mill at the Blistered Horn silver mine is in relatively good condition. Several tons of tailings remain at the site, but more were probably generated and lost down the slope.

Au, respectively—levels that are considered high grade ore in modern mines elsewhere.

Leach tests on three dump and two tailings samples showed generally low reactivity, suggested by low conductivities, pH's in range 4.3 to 6.2, and low metal concentrations. One sample yielded a pH of 4.3 and very high Pb and high Fe and Zn concentrations. The high Pb may relate to the presence of anglesite (lead sulfate), which is more soluble than galena (lead sulfide). The leach results indicate low acid-generating potential for these mine wastes.

*Mills and tailings.*—The two mills on Gold Hill south of Tincup are small and produced only a few thousand tons of tailings. One small mill was below the Blistered Horn mine, and the structure is fairly well preserved (photograph 39). Tailings from the Blistered Horn mill are scant near the foundation, and some materials that appear to be tailings cover the hillside several hundred feet west of the mill (but the path of the washout could not be traced back to the mill). Another mill is located at the base of Gold Hill, but not near the creek (which is unusual), and was served by a tram. The collapsed wooden structure contains many signs that it was a mill, and coarse-grained tailings below it are typical of 1880's stamp mills. The tailings have been partially excavated, possibly for reprocessing (photograph 37).

Leach tests on two tailings samples show low to very low acid-generating potential, and one sample (Blistered Horn) released extremely high Pb, and high Cd, Cu, and Zn concentrations at pH 4.3. The Blistered Horn tailings do have the potential to release base metals (especially lead), but the

volume of these tailings is small and they lie hundreds of feet from flowing streams. Carbonate in the tailings and nearby rocks is abundant and should have high ANC to mitigate any local sources of acid. Although the Tincup tailings are rich in Pb and other base metals, they do not appear to be mobile; thus, they should not pose a threat to the aquatic environment.

*Surface waters.*—Most of the mines in the district release no water to the surface. Shafts do not create mine drainage, and the few adits that do drain water have near-neutral pH and carry low concentrations of base metals. No samples of stream waters were collected because their pH is neutral to weakly alkaline, conductivities are  $<100 \mu\text{S}/\text{cm}$ , and there are no visual signs of iron-oxide coatings on cobbles that typically warn of contaminated water. Two mine drainage samples have pH of 7.4 and 7.6, conductivity of 240 and 280  $\mu\text{S}/\text{cm}$ , and low metal concentrations (Zn is  $<0.7 \times \text{ALWS}$ ), but one contained elevated Cu ( $4 \times \text{ALWS}$ ). Aside from some locally poor water quality at mine portals, the surface waters in this district appear to be of good quality.

*Summary.*—The geologic setting and style of mining at Tincup do not generate significant amounts of acid waters or metals. The predominant shaft-style mining does not generate drainage to the surface, and the few tunnels that do drain to the surface have neutral-pH waters with low metal concentrations. The abundant carbonate rocks mitigate whatever small volumes of acidic drainage that is generated. Lead in the mine waste is mobile for only short distances in this environment.

*References.*—Goddard (1936); Dings and Robinson (1957).

## Tomichi District (Whitepine)

*Location.*—Gunnison Co, about 27 mi east of Gunnison (fig. 18). The climate is moderate to severe with about 16–30 inches of precipitation per year.

*Principal commodities.*—Silver, lead, zinc, copper, gold.

*Mining history.*—The first deposit was discovered in 1879, and several others were located shortly thereafter. As in many parts of Gunnison County, there was some mining in the 1880's and 1890's when most production was from oxidized portions of deposits. Mining was active and prosperous from 1885 to 1893. At first, rich ore was sorted and carried by wagon to the railroad at Sargents, then by rail to a smelter at Pueblo. Two smelters in the district (photograph 40) had mixed success. Mining of deeper, sulfidic ores was made possible by property consolidation in 1901 and construction of the Akron tunnel (photograph 42) at the elevation of Tomichi Creek. A floatation mill was built at the portal, on the southeast edge of the town of Whitepine. Mining expanded in the early 1940's and continued through 1956. Despite favorable similarities to the Leadville district, production never came close to that giant district.

*Production and disturbance.*—Total value was about \$6.6 million. Disturbance is moderate, with chief concerns being sulfidic waste dumps and mill tailings at the Akron portal (mill site).

*Geology.*—A large body of Precambrian granite is the basement rock in this area, overlain unconformably by Paleozoic quartzite and limestone beds that are faulted and folded. A body of Tertiary granite and associated dikes intrude the folded sequence and was the presumed source of heat and metals that formed the ores.

Base-metal sulfide replacement deposits in Paleozoic carbonate rocks have been the major producers. Several different stratigraphic horizons are favored for replacement by sphalerite-galena-pyrite and lesser chalcopryrite. The galena contains silver, and, in some areas, pyrite contains gold. Veins in Tertiary intrusive rocks have had modest production in the northern part of the district; the veins contain chalcopryrite, sphalerite, galena, silver-sulfosalts, and some local streaks of gold.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Chemical analyses confirm the polymetallic character of these ores. Four samples from dumps contain very high Pb, Zn, and Cd and high Mn, Ag, As, Bi, and Cu concentrations. Analyses of four tailings samples were similar, with metal concentrations about one-third to one-half that of dump samples. A fluvial tailing sample from Tomichi Creek (photograph 46) contained very high metal concentrations, suggesting it is from an old stamp mill (pre-Akron mill).

Chemical analyses by two methods of a sample of slag, yielded improbably low results for base metals, suggesting that

the analytical methods did not properly dissolve the vitreous material. Most slags are very rich in Pb, Zn, Cu, and associated ore metals.

Leach tests were made on seven samples. Six dump samples yielded pH values of 5.3 to 7.9, variable high to low concentrations of Mn-Fe-Zn, and some notable amounts of Se; Cu and Pb concentrations were low. A sample of slag was more reactive than most slags that I have tested, creating a pH of 3.1 and releasing moderate amounts of Fe, Cu, Mo, Pb, and Zn to the leachate.

*Mills and tailings.*—Two smelters operated in the 1880's, one of which I located near the road southwest of Whitepine. Crawford (1913) reported "A large mill south of Whitepine seems not to have treated much ore." A small tailings impoundment on Tomichi Creek near a stone foundation seems to be that old mill site. Mill tailings in Tomichi Creek, about 2 mi to the southwest of Whitepine village, appear to come from the breached tailings pond. The fluvial tailings have a distinctive composition that is much richer in metals than tailings from efficient mid-1900's mills. The Akron mine and mill complex, with its large dumps and two mill tailings piles, is conspicuous on the south edge of the village. The Akron mill tailings (photograph 43, photograph 44) are stacked high above the creek with no retaining structures, but appear to have been relatively stable for nearly 50 years. The steep slope on the two tailings piles at the Akron mill, and their position high above Tomichi Creek, suggest that, under unusually wet conditions of a major storm, the materials might become unstable and fail by slippage or liquefaction or be undercut and eroded by floodwaters. The Akron mill tailings merit further technical study. Among the many tailings impoundments in the study area, these two appear to be among the highest risks of failure in an extreme storm event.

Leach tests on two samples of tailings produced leachate solutions of pH 5.4 and 5.9, suggesting low acid-generation potential, but concentrations of Zn and Cd were high and Fe, Mn, and Pb were moderate. A white efflorescent crust on tailings yielded a leach pH of 4.5 and substantially higher concentrations of the metals just mentioned, as well as a notably high Se concentration. Despite the near-neutral pH of the tailings leachate solutions, there are suggestions that the tailings have the potential to release significant amounts of Zn and Cd, and lesser amounts of Pb, Mn, and Se, to Tomichi Creek.

*Surface waters.*—The carbonate rocks in this district are evident in the water quality, which is much better than expected for the substantial amount of mining, sulfidic mine waste, and tailings that are present. The pH values measured in the field were consistently near neutral (pH 7.0 to 7.9), and conductivity measurements were very low (<100  $\mu\text{S}/\text{cm}$ ), except for mine drainage of 410 and 180  $\mu\text{S}/\text{cm}$  at two sites. Chemical analyses of five samples (three stream, one mine, one dump drainage) showed generally low metal concentrations, but the mine and dump drainage carried high zinc concentrations (2 $\times$  to 7 $\times$  ALWS). Water from Tomichi Creek north of the town of Whitepine has very low metal content, but two samples of the creek downstream (south) of the Akron mine

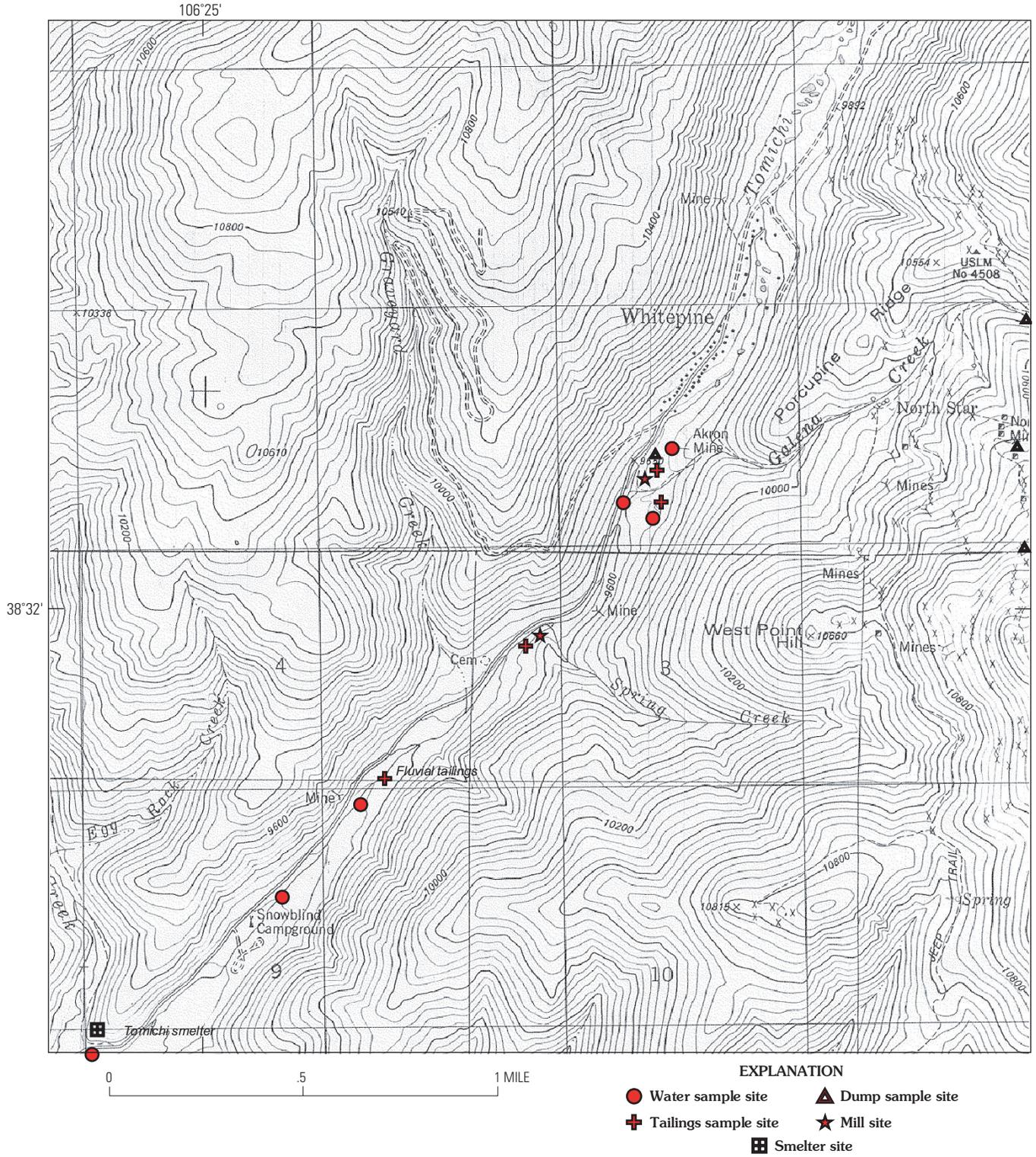


Figure 18. Features of the Tomichi mining district, showing sample localities.



**Photograph 40.** The Tomichi smelter, southwest of Whitepine, was part way up the hillside and slag was poured out onto the flat area below. The slag is not very reactive; thus, it probably is not a threat to the nearby creek, but the high concentration of base metals could be toxic if ingested.



**Photograph 41.** The Akron mill on the south edge of Whitepine was productive through 1956. Sulfidic waste and mill tailings are piled above the mill.



**Photograph 42.** The Akron mine tunnel served both for rail haulage and mine dewatering. Drainage from the tunnel today has near-neutral pH because of the carbonate rocks underground. However, the drainage carries some zinc and also reacts with sulfidic mine waste before entering Tomichi Creek.



**Photograph 43.** This tailings impoundment is 200 yards south of the Akron mill. These tailings are above Tomichi Creek and erosion or failure in an extreme storm event would carry them into the stream.



**Photograph 44.** This view is from the south tailings impoundment to the Akron millsite and reddish tailings piles above it. Numerous tracks from vehicles show the dumps and tailings are popular with visitors.



**Photograph 45.** This small tailings pond was built across Tomichi Creek before 1900. The stone foundation for the mill is a short distance to the north and about ½ mi south of Whitepine. The pond was breached and tailings were eroded years ago.



**Photograph 46.** These gray sands are mill tailings that were eroded and carried south along Tomichi Creek and came to rest among willows. The likely source is a pre-1900 stamp mill about 1 mi to the north.

contain zinc that is above ALWS ( $2\times$  and  $2.5\times$ ), although a Zn content of 125 ppb is not very high relative to some other contaminated streams on the Western Slope. Samples collected in 1999 south of the Akron mill showed low metal concentrations (although the Zn value of 140 ppb at site NG850, 2.3 mi to the southwest, is higher than at sites closer to the mines and mill and might reflect an error of some kind). The CZI values of 0.6 to 0.2 for Tomichi Creek samples below the mines and mill suggest low threat to aquatic life. The carbonate rocks appear to be effectively neutralizing pH, which limits the solubility of most base metals. However, Zn is still slightly soluble and enriched downstream from mining, as is well known from other studies (Smith and others, 1994).

*Summary.*—The zinc enrichment in Tomichi Creek is clearly related to the Akron mine drainage and waste piles. Further study of base metals in Tomichi Creek by more detailed methods may be warranted. No reclamation or water management has been undertaken, and even a partial reclamation would likely improve the somewhat contaminated waters of Tomichi Creek. The two tailings impoundments at the Akron mine are located in situations that appear to make them vulnerable to failure by erosion, slippage, or liquefaction during an extreme storm event. These tailings rank among the top five sites for risk of failure during an extreme storm and should be investigated further for physical stability and possible methods of stabilization.

*References.*—Crawford (1913); Dings and Robinson (1957).

## Mining Districts of the Lake Fork Watershed

Mining in the area of Lake City (figs. 19, 20) started in the 1870's and soon became famous when very rich silver-gold zones were encountered at the Ute-Ulay and Golden Fleece mines. However, the boom was short-lived and did not attain the magnitude of adjacent San Juan mining districts (production was only \$550,000 after 1930). The veins in volcanic rocks associated with the Lake City caldera are relatively short and narrow, and the mining complexes are smaller and produced less disturbance than the later and larger mines elsewhere because the miners used small equipment to focus on the rich ore zones. This simple mining technology, combined with favorable volcanic rock characteristics, leaves this watershed with healthier habitat than some adjoining districts.

Geology of this mining area is dominated by the nested Lake City and San Juan caldera complexes that have been thoroughly studied (Lipman, 1976; Steven and Lipman, 1976; Slack, 1980; Hon, 1987).

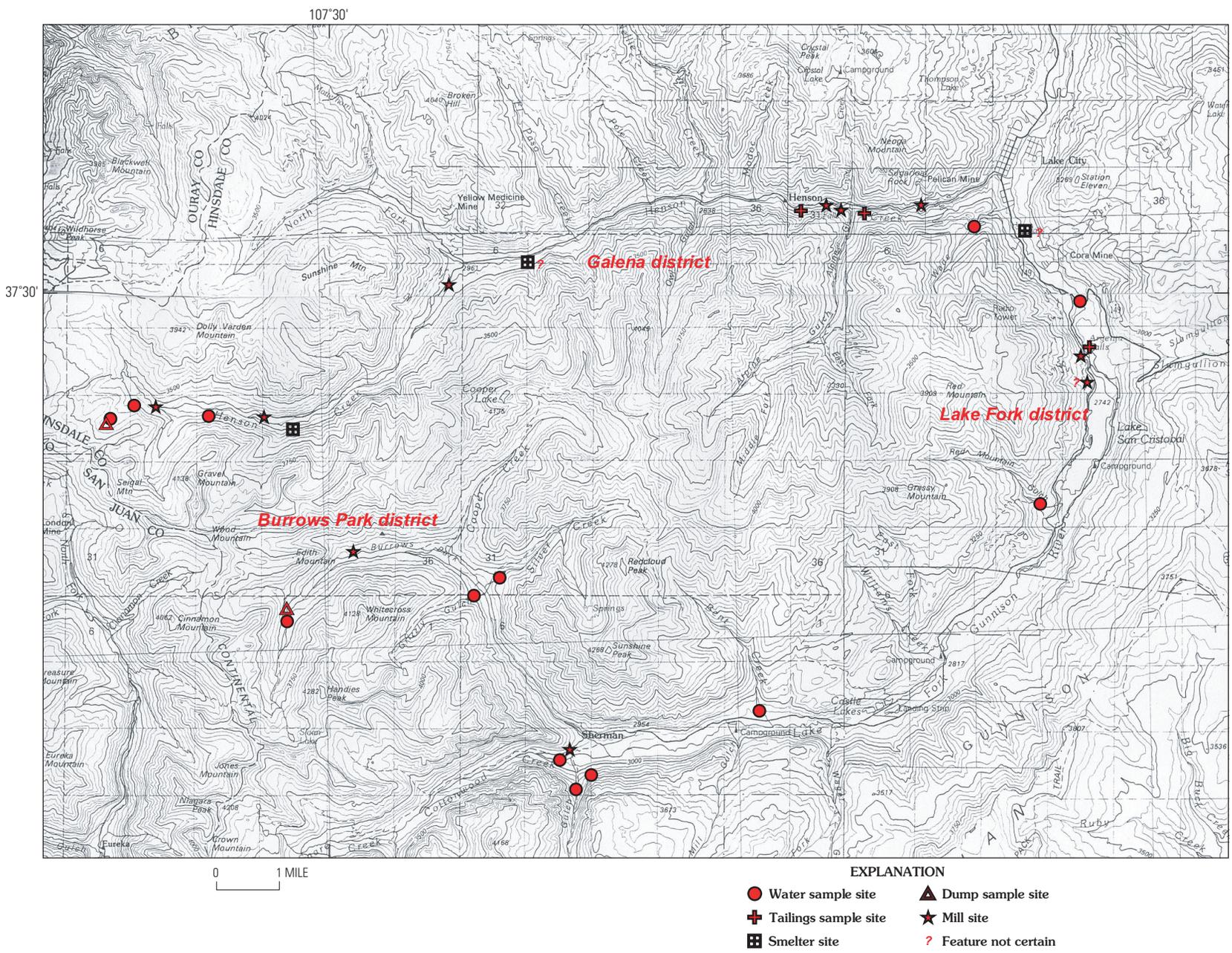


Figure 19. Features of the Galena, Lake Fork, and Burrows Park mining districts, Lake City area, showing sample localities.

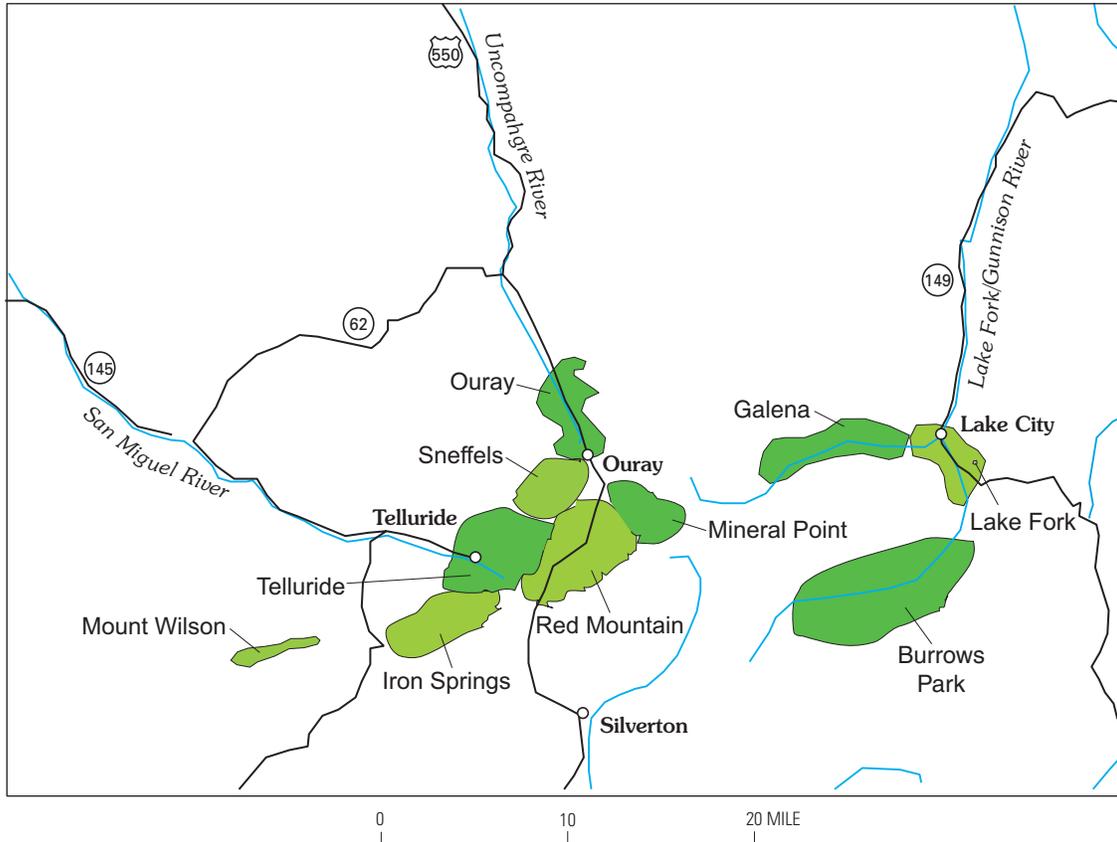


Figure 20. Location of mining districts in the San Juan Mountains region.

### Burrows Park District (Whitecross, including Sherman)

*Location.*—Hinsdale Co., 20 mi south of Lake City (fig. 19). The climate is relatively severe with about 25–40 inches of precipitation per year.

*Principal commodities.*—Ag, Au, minor base metals.

*Mining history.*—Little has been written about this district. There was not sufficient activity in the early years to warrant mention in the review by Henderson (1926). Presumably, earliest discoveries were made in the 1870's, during the rushes in adjoining districts. A study of the mines in 1906 suggested that most mines had closed by that time and that production was limited by the distance to railroads (Woolsey, 1907). A small mill was erected on the Lake Fork at the townsite of Sherman, probably in the 1880's (judging from the style of stone construction); this mill was named the Black Wonder according to Southworth (1997), who provides an historical photograph. None of the mines in the district attained significant production, and most have been inactive for the past 50 years. There have been a few attempts to reopen older mines on patents.

*Production and disturbance.*—Small production, a minor contribution to the \$11.2 million for Hinsdale County. Small disturbance.

*Geology.*—The ore geology of this area can be considered to be an extension of the much larger and better understood Eureka district in the Animas watershed to the southwest. In the Burrows Park district, the veins in Tertiary volcanic rocks have tended to be smaller than to the south but are mineralogically similar in their content of galena, sphalerite, with variable amounts of chalcopyrite, gold, and silver. Some veins occur in Precambrian granite-gneiss but are of the same age and character as those in Tertiary volcanic rocks.

### Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Polymetallic ores from this district are mineralogically and chemically similar to the Eureka district, for which there is abundant new information on rock alteration and dump compositions (Nash, 1999b; and others). One sample was analyzed for this study. Dump compositions may have large local variability in composition, depending upon the style of mining. Some dump piles are likely to have very high Pb, Zn, and Cd concentrations; high Ag, Cu, and Mn; and locally high Bi, Mo, or Se concentrations.

*Mills and tailings.*—The small stamp mill at Sherman townsite is presumed to have had small production; only the stone foundation remains visible today. Tailings from this mill

probably were placed on the flood plain and subsequently carried away by floodwaters. No sample of tailings could be collected. The Sherman mill does not appear to pose any problems today, and there is no material remaining to be reclaimed. The Tobasco mill is located about 1.7 mi east of Cinnamon Pass, at the west edge of Burrows Park; it is on private property and was not investigated.

*Surface waters.*—During the brief reconnaissance study, three samples were collected for analysis. Field tests of pH and conductivity indicated good quality with pH near neutral (6.9–7.4) and conductivity very low (<50 in streams, but 240  $\mu\text{S}/\text{cm}$  for one mine drainage). The surface waters appear to be of good quality, and two analyses show very low metal concentrations, well below ALWS. One mine drainage sample from a small adit in American basin had a pH of 7.4 and high zinc concentration (3 $\times$ ALWS). The CZI values for two stream samples were very low (0.1 and 0.01), suggesting low threat to aquatic life. The predominant green alteration in volcanic rocks is expected to provide ANC adequate to naturally attenuate the locally contaminated acidic waters at small mines such as site 287.

Water composition baselines and geologic influences on them were studied in fair detail by McHugh and others (1995) and Miller (1998), who found anomalous amounts of metals and sulfate in several tributaries to the Lake Fork, such as Silver and Copper Creeks. Miller found similar results to those reported here for the main stem of the Lake Fork and described how neutralization reactions improved the quality of water downstream from the mineralized areas.

*Summary.*—The water quality in the main stem of the Lake Fork is good, judging from the standard indicators of pH, conductivity, and trace metals. Some small mine workings may release degraded water, but this evidently is mitigated within about a mile by dilution and the neutralizing effects associated with the predominant green-altered volcanic rocks. Some mining properties may merit reclamation work to minimize release of metals, but the ability of natural processes to mitigate mine-related contamination suggests that these reclamation efforts have lower priority than in several other mining areas on the CWS.

*References.*—Lipman (1976); Woolsey (1907).

## Galena District (Henson Creek)

*Location.*—Hinsdale Co, 2–10 mi west of Lake City (fig. 19). The climate is moderate to severe, with about 20–40 inches of precipitation per year.

*Principal commodities.*—Silver, lead, zinc, copper, gold.

*Mining history.*—Gold was discovered in 1842 by a member of the Fremont Expedition, but the locality was lost. The Ute and Ulay veins were discovered in 1871, when the land still belonged to the Indians. Development started after the treaty of 1874, and by 1876 there was a short-lived boom. The Palmetto and Frank Hough mines near Engineer Pass were located in 1881 and were productive in the 1880's. The collapse of silver prices in 1893 depressed mining in the area, but it

regained activity in the late 1890's, sooner than most districts on the Western Slope. However, mining declined in the early 1900's and never attained the stature of districts in the western San Juan Mountains, which became highly productive in a second stage of mining so-called "low-grade ore" with the use of modern flotation and cyanide mills in the period 1915–80.

*Production and disturbance.*—Production from this district was the largest in Hinsdale County, for which the total recorded is about \$11.2 million. However, records differ because production from the Ute and Ulay mine alone through 1906 is stated to be \$12 million (Irving and Bancroft, 1911). Production from the district through 1959 was about 126,800 tons and has been small to nil since then. Disturbance is small from the underground mines that were highly selective in their mining, but several substantial impoundments of mill tailings are close to Henson Creek.

*Geology.*—Polymetallic vein deposits containing base-metal sulfides with variable amounts of precious metals occur in Tertiary volcanic rocks on the northern margin of the Lake City caldera, similar to the larger and more famous deposits of the western San Juan Mountains near Silverton and Telluride (Burbank and Luedke, 1968). Most veins contain chiefly galena and sphalerite, with lesser amounts of chalcopyrite, silver minerals, and gold. Some of the veins are famous for locally rich pockets of gold and silver, some of which is in the form of telluride minerals. Alteration halos of pyritic rock are relatively narrow. The southwestern part of the district, near Engineer Pass, contains veins that are structurally continuous over the divide into the Eureka and Mineral Point districts. A few of these vein deposits, such as at the Hough mine, were most productive in the 1880's. In the eastern part of the district, a cluster of veins was worked by the Ute-Ulay mines and were the first to be discovered in the Lake City area. A large area of acid-sulfate alteration occurring at Red Mountain contains a subeconomic deposit of alunite (an ore mineral of aluminum) but no important associated metal deposits.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Polymetallic ores from this district are mineralogically and chemically similar to the Eureka district, for which there is abundant new information on rock alteration and dump compositions (Nash, 1999b, Fey and others, 2000b). Three samples were analyzed for this study. Dump compositions may have large local variability in composition, depending upon the style and history of mining. Dumps from small mines tend to have piles of ore-grade material that is separated from waste but not of sufficient tonnage to ship to a mill. Dumps at large mines, generally on patented land, tend to have chiefly low-grade piles, including waste from crosscuts that is low in metals. Some dump piles are likely to have very high Pb, Zn, and Cd concentrations; high Ag, Cu, and Mn; and locally high Bi, Mo, or Se concentrations. Pyrite is abundant on most dumps, as can be seen at the Hough mine (photograph 50).



**Photograph 47.** Mill tailings from the Hidden Treasure mill are on the south bank of Henson Creek about 3 mi west of Lake City. These tailings would be eroded in a major storm.



**Photograph 48.** The Ute-Ulay mine was the largest in the lower Galena district. The tailings from an early stage of milling, lower left, were later covered with mine waste. All of this waste is within tens of feet of the Henson Creek. Tailings from later periods of milling were placed in impoundments west of the mill, also close to the creek.



**Photograph 49.** Tailings and mine waste at the Ute-Ulay mine are stacked on a narrow bench very close to Henson Creek, which is in the gorge on the far side. These materials appear to be unstable and at risk in a major storm event.



**Photograph 50.** The large dumps from the Hough mine are rich in sulfide minerals and pose problems for the alpine ecology. Sulfidic dumps and draining mine adits in Palmetto Gulch pollute the headwaters of Henson Creek.



**Photograph 51.** The Palmetto mill at treeline in the headwaters of Henson Creek processed ore from several mines in the Engineer Pass area. It probably used water power in the 1880's. No tailings could be identified.



**Photograph 52.** This brick smokestack is the remains of the Lee smelter, located southwest of Capitol City. It was an early (1880's?) operation that attempted to refine silver and gold from sulfidic ore.



**Photograph 53.** The Empire Chief mill is a popular tourist stop on the Engineer Pass road, southwest of Capitol City. No tailings could be found nearby.

*Mills and tailings.*—There were six or more mills on Henson Creek, some of which used water power. Tailings are preserved at the Ute-Ulay mine-mill complex (photograph 48), but most other tailings appear to have been lost to the stream. The Hidden Treasure mill foundation is visible near the foot-bridge over Henson Creek to Alpine Gulch. A small pocket of tailings that is visible on the south side of the creek a short distance downstream may be related to this mill, but other tailings are presumed to have been washed downstream (photograph 47). No structures or waste materials could be found at the sites reported (Irving and Bancroft, 1911) for the Ocean Wave smelter and Red River mill about 2 mi east of Capitol City. The structure at Capitol City has the appearance of a mill, but residents say that it was a water-powered electricity generator. A concrete foundation with the form of a mill is evident in Henson Canyon slightly less than 1 mi west of Lake City, below the Pelican mine. The mill is not reported by Irving and Bancroft (1911); thus, it would seem to be younger than that date. No tailings could be located at this probable mill site.

Four mill sites are evident west of Capitol city, one being a popular tourist site (Empire Chief mill, photograph 53). The Palmetto mill utilized water power on Henson Creek just below treeline (photograph 51). A brick chimney from the Lee smelter is visible on the south side of the creek, 2 mi southwest of Capitol City (photograph 52). Another mill is reported to

have operated on Henson Creek about a mile west of Capitol City, and that location was confirmed by the BLM (B. Hite, BLM, written commun., 1999). No tailings or slag are evident at these sites.

Leach tests on two samples of tailings show that they produce relatively low amounts of acid and metals, much lower than dump samples from the district.

The concern in the vicinity of the Ute-Ulay mine is the stability of several tailings impoundments that are within 100 yards of Henson Creek. Some are supported by gravel berms but others appear from a distance to be supported only by wooden structures. The mixed private and public land status prevented thorough onsite inspection. Release of these tailings during a major storm event or by failure of an impoundment structure would certainly put these materials into Henson Creek, an important fishery.

*Surface waters.*—Waters in this drainage generally are clear and produce no iron coatings on stream cobbles, except in the westernmost headwaters near Palmetto basin. Six water samples were collected for analysis. Three stream waters in the lower part of the basin had neutral pH (6.9–7.4) with low conductivity values. Analyses show low metal concentrations. A sample from Henson Creek at the east edge of Palmetto basin, near several small mines, had a pH of 7.1, notably

higher conductivity (240  $\mu\text{S}/\text{cm}$ ) and enrichments in some metals (Zn 20 $\times$ , Cu 64 $\times$ , and Cd 9 $\times$ ) relative to ALWS. These metals most likely are related to the drainage from small adits that produce modest flows of about 10–20 gpm with pH values of 2.9 to 3.3 and very high metal concentrations. Samples from Henson Creek east of Capitol City have good quality and do not have significant concentrations of metals (Zn less than 0.7 $\times$  ALWS). CZI values are high (10 to 14) in headwaters near Palmetto basin, but are low (0.3 to 0.03) in the lower reaches of Henson Creek east of Capitol City. This reconnaissance sampling suggests that there are local sources of acidic, metal-rich waters, but these waters are naturally mitigated by dilution and reactions with green-altered or unaltered volcanic rocks.

Miller (1998) studied the surface waters of the area and the influences of geology on water compositions. His results are similar to those described here, and he reached a similar conclusion that water quality in the main stem of Hensen Creek is good. Miller sampled side streams with sources in altered rocks and found what he termed “high” or “anomalous” concentrations of Zn, Mo, U, F, and rare earth elements—but the concentrations found by Miller are defined relative to other samples in his study and are much lower than from most of the mining districts described here for the CWS.

*Summary.*—Water quality is better than most would predict from the district’s mining history. There are some local point sources of acidic waters in the headwater region near Engineer Pass that deserve further study, but these waters are mitigated by the prevalent green-altered volcanic rocks within a few miles of the mine sources. The uppermost part of Henson Creek, the alpine zone above treeline, has several mine dumps that seem to be more problematic than similar mine dumps at lower elevations. From observations in this brief reconnaissance, the largest risk to the watershed might be physical failure of the mill tailings at the Ute-Ulay mine during an extreme storm.

Further studies should be made of the mill tailings at the Ute-Ulay mine and mill to determine their land status and physical stability. Based on limited information and my preliminary model of tailings failure, these tailings appear to have high risk of failure during an extreme storm flood. Studies are needed to improve models for (1) extreme flood-stage flow in the main stem of Henson Creek (how high would water rise in the narrow gorge at the mine?) and (2) estimates of possible overland flow in gullies on the north side of the canyon that might direct floodwaters to the tailings.

*References.*—Irving and Bancroft (1911); Slack (1980).

## Lake Fork District (Lake San Cristobal)

*Location.*—Hinsdale Co., about 5 mi southwest of Lake City (fig. 19). The climate is more moderate than adjacent districts, but has about 16–30 inches of precipitation per year.

*Principal commodities.*—Silver, lead, zinc, copper, gold.

*Mining history.*—The first discovery, in 1874, turned out to be the largest producer—the Golden Fleece mine overlooking Lake San Cristobal. The Golden Fleece is famous for very

high grade pockets of gold telluride (petzite) in the veins. The Golden Fleece and adjoining Black Crook mine were well developed on many levels by 1906 (Irving and Bancroft, 1911) but appear to have had little production in later years.

*Production and disturbance.*—This district had the second largest production in Hinsdale County, which had a total of \$11.2 million. Disturbance is small. One pile of tailings is close to the Lake Fork.

*Geology.*—Similar to that of the adjoining Galena district but lying on the southern structural margin of the Lake City caldera. Veins in Tertiary volcanic rocks contain base-metal sulfide minerals, with locally rich streaks of silver and gold. Wallrock alteration is intense locally along some veins and is widespread, but some is not associated with mined ore to the east, notably at the Slumgullion landslide.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Mine dumps were not sampled because most are on private property (mine patents are now summer home sites). The ores are known to be rich in base-metal sulfides such as galena and sphalerite; thus, they have high concentrations of Pb, Zn, and Cd. Dumps may be locally rich in Se and Te because there were Ag-Au ore minerals of these elements, but the concentrations of these potentially toxic elements are highly diluted by others kinds of ore and waste rock. Judging from similar ores in adjoining districts (Nash, 1999b), the mine waste may have high acid-generating potential in small local areas.

*Mills and tailings.*—There were at least two mills in the district, both near the lake. The Golden Fleece mill may have disposed of tailings into the lake (the mill is so close to the lake that there was little other space). The Contention mill placed its tailings a short distance north of the outlet from the lake, near Argenta Falls. No samples could be obtained from these private sites. The impounded tailings may merit study because they are within the flood plain of the Lake Fork of Gunnison River and are vulnerable to erosion in an extreme flood.

The Black Crook smelter, frequently mentioned in the literature, could not be identified. A likely location would be at Crook Falls, 1 mi southeast of the town of Lake City, an area of private property that could not be investigated. Slag from this smelter probably does not release metals to the Lake Fork.

*Surface waters.*—Streams in this district appear to be of good quality, judging from their clarity, absence of iron coatings on stream cobbles, and field measurements. Indicators of good quality are the consistently very low conductivities, most less than 50  $\mu\text{S}/\text{cm}$ , and pH near neutral (pH 6.8 to 7.0). Chemical analyses show the waters to be very low in metals. Zinc, which generally is the most mobile metal released from polymetallic waste, is far below ALWS. Zinc concentrations of 18 ppb or less found in these samples are well below the median concentration for the CWS. CZI values for seven

stream samples are very low (<0.1), which suggests low threat to aquatic life. There may be some local contamination in the district, such as on private sites that were not accessible, but these do not have significant influence on the larger streams or the main stem of the Lake Fork.

*Summary.*—Water quality appears to be good in this district. Although there was substantial mining at some sites, disturbance was not extensive, and highly degraded mine drainage is not evident. There are substantial natural sources of metals in unmined, highly altered rocks, including those in the Slumgullion Slide, thus the geochemical background in this area was not pristine. The good water quality, shown by geochemical indicators, appears to reflect the ability of weakly altered volcanic rocks to mitigate local sources of acid and metals. Construction of homes on mine dumps may aggravate conditions at some localities. One tailings pile near the Lake Fork may need to be stabilized; further study of this private site is recommended.

*References.*—Irving and Bancroft (1911); Slack (1980).

## Mining Districts of the Uncompahgre River Watershed

The headwater region of the Uncompahgre River drains four mining districts with major production that spanned slightly more than 100 years (fig. 20). This area on the northwest flank of the Silverton Caldera is noted for numerous large veins that are radial to the caldera and formed some 10 million years after the volcanism (Burbank and Luedke, 1968; Lipman and others, 1976). A different style of small, but very rich orebodies formed in breccia pipes associated with post-caldera intrusions in the ring fracture zone, most notably near Red Mountain Pass. An older stage of ores formed near Ouray in sedimentary rocks adjacent to mid-Tertiary intrusions. Production from these three kinds of ore in Ouray County had a gross value of about \$111,000,000, chiefly from gold and silver, but the values of copper, lead, and zinc were much higher than in most of the districts on the CWS. The high relief and deep canyons of the region were favorable for the construction of haulage tunnels several miles deep into the mountains that allowed efficient haulage of ore, much of which actually came from San Miguel County (Telluride district). This style of mining had the consequence of creating major complexes for milling of ore, dumping of waste, and draining of mine water on the Ouray County side of the mountain divides. These major mine-mill complexes, including the Idarado mine (Treasury Tunnel), the Revenue Tunnel, and the Camp Bird mine, are mostly on private (patented) lands but pose questions for managers of adjacent public lands administered by the USFS. Reclamation in parts of this watershed will have to be considered in the context of the Permit for the Idarado Mine Site Remedial Action Plan (CDH, 1992), which addresses issues far beyond the geochemical scope of this study.

## Ouray District (Uncompahgre)

*Location.*—Ouray County, adjacent to Ouray (fig. 21). The climate is relatively moderate for the CWS, with about 20–30 inches of precipitation per year.

*Principal commodities.*—Silver, gold, lead, copper, zinc.

*Mining history.*—Ore was discovered in 1875 and mining commenced shortly thereafter. The first lodes to be discovered were rich in gold, but subsequent discoveries were rich in silver and lead. Some of the most productive years were in the 1890's. Mining has not been productive since the 1950's, but the major mill in the district treated ore from other districts through 1952.

*Production and disturbance.*—The total value was about \$12.7 million from about 392,500 tons of ore (Long and others, 1998). Generally small disturbance from underground mines. Two mill tailings impoundments near the Uncompahgre River are potential sources of contaminants.

*Geology.*—Ores in this district are unusual for the study area for their Laramide age (Late Cretaceous to early Tertiary). This is clearly shown by the erosional surface above altered rocks and contrasting overlying unaltered mid-Tertiary rocks (this is visible from downtown Ouray if one looks to the northeast). The geologic section, from Precambrian through Tertiary, is generally similar to nearby districts. More than 4,000 ft of Devonian through Cretaceous sedimentary rocks are well exposed in the canyons, and this is the type locality for the red sandstones of the Cutler Formation (Cutler Creek). Dikes and sills of early Tertiary porphyries intrude and alter the sedimentary rocks, most notably at the “Blowout,” well exposed in the canyon wall east of Ouray (Luedke and Burbank, 1962). Ore deposits are associated with the porphyry intrusions.

Ore deposits are polymetallic veins and replacements that in detail vary in mineralogy. The most important deposits contain base-metal sulfides with silver tellurides and gold. Sulfide veins tended to expand outward into replacement zones when they encountered reactive carbonate rocks. The replacement or “manto” deposits thus resemble ore deposits near the Continental Divide (e.g., Leadville and Tomichi districts) rather than the nearby Sneffels and Red Mountain districts, which are chiefly in volcanic rocks.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Analyses of three samples of dump rocks and three samples of tailings confirm the polymetallic character of this district. The mine waste tends to have high concentrations of Mn and Pb relative to other districts. Some dump compositions are high in Bi and Mo, and moderate (near the study median) in Ag, As, Cu, and Zn. Laboratory leach tests on four samples (one dump, three tailings) yielded solutions with pH values of 3.2 to 4.1, but concentrations of base metals in the leachates were lower than from most polymetallic deposits on the CWS. One sample

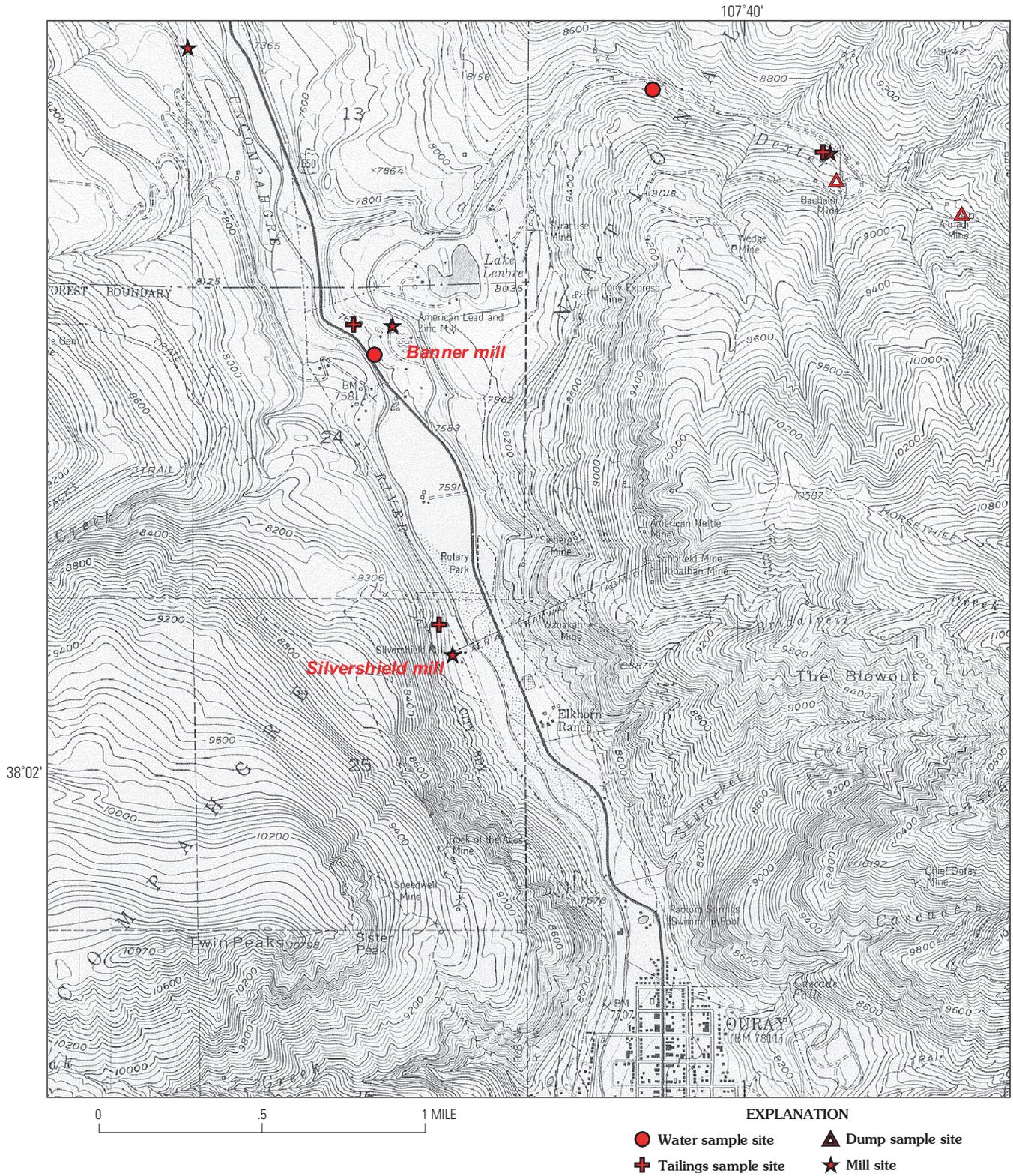


Figure 21. Features of the Ouray mining district, showing sample localities.



**Photograph 54.** This small stamp mill on Dexter Creek probably operated before 1900. Traces of tailings are in the foundation, but others were carried down the creek.



**Photograph 55.** This stone and concrete foundation probably was for a mill located just below the old railroad near Corbett Creek about 4 mi north of Ouray.



**Photograph 56.** The Silvershield mill processed ore carried in by a tram to mines above the red rock cliffs north of Ouray.



**Photograph 57.** Tailings from the Silvershield mill are about 100 yards west of the Uncompahgre River. They are relatively stable here but are accessed by visitors.



**Photograph 58.** The Banner mill site, 2 mi north of Ouray, was developed for housing in the 1970's, but the tailings were left in place. Homes are built on the tailings. Storms wash tailings down the gulch toward the Uncompahgre River.

of tailings from the foundation of the Bachelor mill yielded extremely high concentrations of Pb, Zn, Cd, Cu, and other metals, suggesting either that this old mill was very ineffective at removing metals, or that my sample was spilled concentrate, not tailings. These tests suggest that the mine waste generates more acid and releases more metals than expected for typical carbonate-hosted ores.

*Mills and tailings.*—I have located five mills in this district, but only two are described in the literature. One small mill is on Dexter Creek (photograph 54), a short distance northwest of the Bachelor mine; no tailings remain. A concrete foundation remains adjacent to the old railroad grade north of Ouray near Corbett Creek (photograph 55); no tailings are evident. A “modern” custom mill stands next to U.S. Highway 550, 3 mi north of Ouray; the tailings piles are small if any at this private site. The Silvershield and the Banner (American Lead and Zinc) mills are shown on the USGS Ouray topographic map. The Silvershield mill (photograph 56) received ore by tram from mines to the east; tailings cover about an acre on the west side of the Uncompahgre River (photograph 57). This site is being developed for homes. The Banner mill (photograph 58) started operations in the 1930's, was expanded to 350 tons/day in the 1940's and was closed in 1952. The mill was dismantled and the site has been developed for homes; about a dozen homes have been built on the tailings, placed on a bench of red rocks east of the river. Storm events carry some tailings and also acidic drainage over the edge of the bench toward U.S. Highway 550 and the river. Unreclaimed mill tailings at the Banner and Silvershield mill-

sites may be vulnerable to erosion in extreme flood conditions.

*Surface waters.*—Water in Dexter Creek below some of the historic mines is clear and produces no red iron stains or coatings. Field measurements showed neutral pH values (6.9 to 8.2) and low to moderate conductivity (120 to 230  $\mu\text{S}/\text{cm}$ ). A water sample collected below the mill and mines had surprisingly high concentrations of Cd, Cu, and Zn (1 $\times$ , 11 $\times$ , and 4 $\times$  ALWS, respectively) for a pH value of 8.2. The CZI value of 6 suggests this water could be a threat to aquatic life.

The Uncompahgre River in the vicinity of Ouray tends to be turbid and leaves deposits of tan to ochre materials on stream cobbles. My tests of water chemistry are not sufficiently specific to distinguish contamination from local sources from known sources upstream. The pH of 6.9 and conductivity of 210  $\mu\text{S}/\text{cm}$  in the Uncompahgre River at Ouray suggests that quality is improved from that in the Red Mountain Creek source area to the south (described later), but chemical analyses show that Fe and Zn concentrations remain high (1 $\times$  and 3.3 $\times$  ALWS). Two samples from the Uncompahgre River collected in September of 1999 on the north side of Ouray, and north of Ridgeway (near inlet of Ridgeway Reservoir) contained concentrations of Cd, Cu, and Zn greater than ALWS. The CZI values for these river samples are 3.7 and 2.0, suggesting a possible threat to aquatic life.

Water seeping over the sandstone bench at the former Banner mill site leaves red coatings, and sample 309 taken during a rainy week in June had a pH of 3.4, conductivity > 2,000  $\mu\text{S}/\text{cm}$ , and very high concentrations of Al, Cd, Cu, Fe, and Zn. This particular seep flowed at only a few gallons/

minute but with very high concentrations ( $> 100\times$  ALWS) that could pose problems for wildlife.

My reconnaissance water samples in the Ouray area suggest that more detailed work on water quality and biota on the middle Uncompahgre River and its tributaries are needed to evaluate the aquatic habitat.

*Summary.*—Mines in this district appear to have low impact on surface-water quality, but more base metals may be carried in surface waters that one would expect for pH 7 to 8 waters. Mill tailings in this district may be significant sources of contamination. The turbidity and elevated metal concentrations in the Uncompahgre River in the vicinity of Ouray is chiefly from the Red Mountain district, but there also is some input from the Ouray district. Also, mill tailings at two sites are reactive and were not effectively reclaimed as of 1999. Construction of homes on mill tailings may be adding domestic water that compounds the input from tailings to the Uncompahgre River.

*References.*—Burbank (1940, 1947a); Luedke and Burbank (1962).

## Sneffels District

*Location.*—Ouray Co., 10 mi southwest of Ouray (fig. 22). The climate is severe, with about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver, gold, lead, copper, zinc

*Mining history.*—Several deposits in the district were located in 1875 and substantial mines were operating in the 1880's. The Revenue Tunnel was one of the first in the region to be driven deep into a mountain (7,600 ft) in 1896 to facilitate mining of the Virginus and other veins high on the ridge and beyond to the Telluride side. Chronic problems with avalanches and water pumping from shafts were alleviated by the tunnel. The mill at the Revenue Tunnel has huge for the time with 350 tons/day capacity. It began production during the era of depressed silver prices, but the discovery of the gold-rich Montana vein in 1909 boosted its revenues. However, in 1911 the Ophir tunnel was developed to carry the ore west to the Tomboy mill on the Telluride side. The Revenue struggled a few more years, and the mill burned in the 1920's, marking the end of its activity.

The most famous mine in the district, the Camp Bird, was not located until 1896 but soon developed into a large operation. In 1898, a 9,000-ft-long tram was built to carry ore down to a stamp mill at Canyon Creek to avoid avalanche problems above treeline in Imogene basin. The mill burned in 1906 but was replaced by a bigger one. In 1916, the 14th-level tunnel, more than 10,000 ft long, was developed to carry ore to the mill and drain the mine workings. Mining continued intermittently for the next 60 years, with expansion of the mill on Canyon Creek to 500 tons/day in 1956 and development of deep Cu-rich replacement ores in the Telluride Conglomerate in the 1970's. The mine ceased operations in 1977, but the mill processed a limited amount of custom ore in the 1980's. A

gold-exploration venture reopened upper mine workings in the late 1980's. The mine was inactive in the 1990's and reclamation work was underway in 1999.

*Production and disturbance.*—Because figures tend to be aggregated by county or by place of milling rather than mining, the totals for production are unclear. Total value of ore produced was very high, about \$50 million, much of which came from the Camp Bird mine. Total production from the district is estimated to have been 3,108,000 tons through 1978 (Long and others, 1998). Disturbance is moderate.

*Geology.*—This district lies just northwest of the margin of the mid-Tertiary Silverton caldera and adjacent to the Stony Mountain intrusive complex (under Mount Sneffels). Most exposed rocks are volcanic flows and tuffs, but important intrusions are present singly and in clusters or dike swarms. A thick section of Devonian to Jurassic sedimentary rocks lies between the Tertiary rocks and Precambrian basement (Burbank and Luedke, 1964).

Most ore came from vein deposits in the volcanic rocks, but important replacement ores rich in base metals were found in the Telluride Conglomerate adjacent to the veins. The radial aspect of the veins relative to the San Juan caldera has received careful attention and mapping (Burbank and Luedke, 1968). The early mining, through about 1900, focused on the upper portions of veins that were oxidized and rich in gold (Fisher, 1990). Later mining at depths of 1,000–2,000 ft worked polymetallic sulfide veins in volcanic rocks and polymetallic replacements in calcareous parts of the Telluride conglomerate.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Ores from this district are known to be rich in base and precious metals; values are highly variable in detail along a vein, and gold in particular is erratic. Only one dump sample was collected, from the Eclipse mine, and it is probably not representative of the district in general. Concentrations of base metals are moderate in the one sample. The polymetallic ores probably are similar to those from the Silverton area (Nash, 1999a, 1999b). Another approach to estimating composition and input to the aquatic environment is information from a stream sediment collected from Canyon Creek below the Camp Bird mine. The metal concentrations are high in this stream-sediment sample. In particular, concentrations of As, Ag, Bi, Pb, and Zn are higher than in the nearby Camp Bird tailings, suggesting the likelihood of multiple sources of mineral contaminants to the streambed. Stream-sediment compositions were systematically studied as part of the GMUG project (Smith, in press).

*Mills and tailings.*—This very productive district has been supported by many mills that have created tailings piles that range in size from small at the Eclipse mine (tens of thousands of tons?) to very large at Camp Bird (several millions of tons) (photograph 59). As in other districts, the mills and tailings share the common feature of being located close

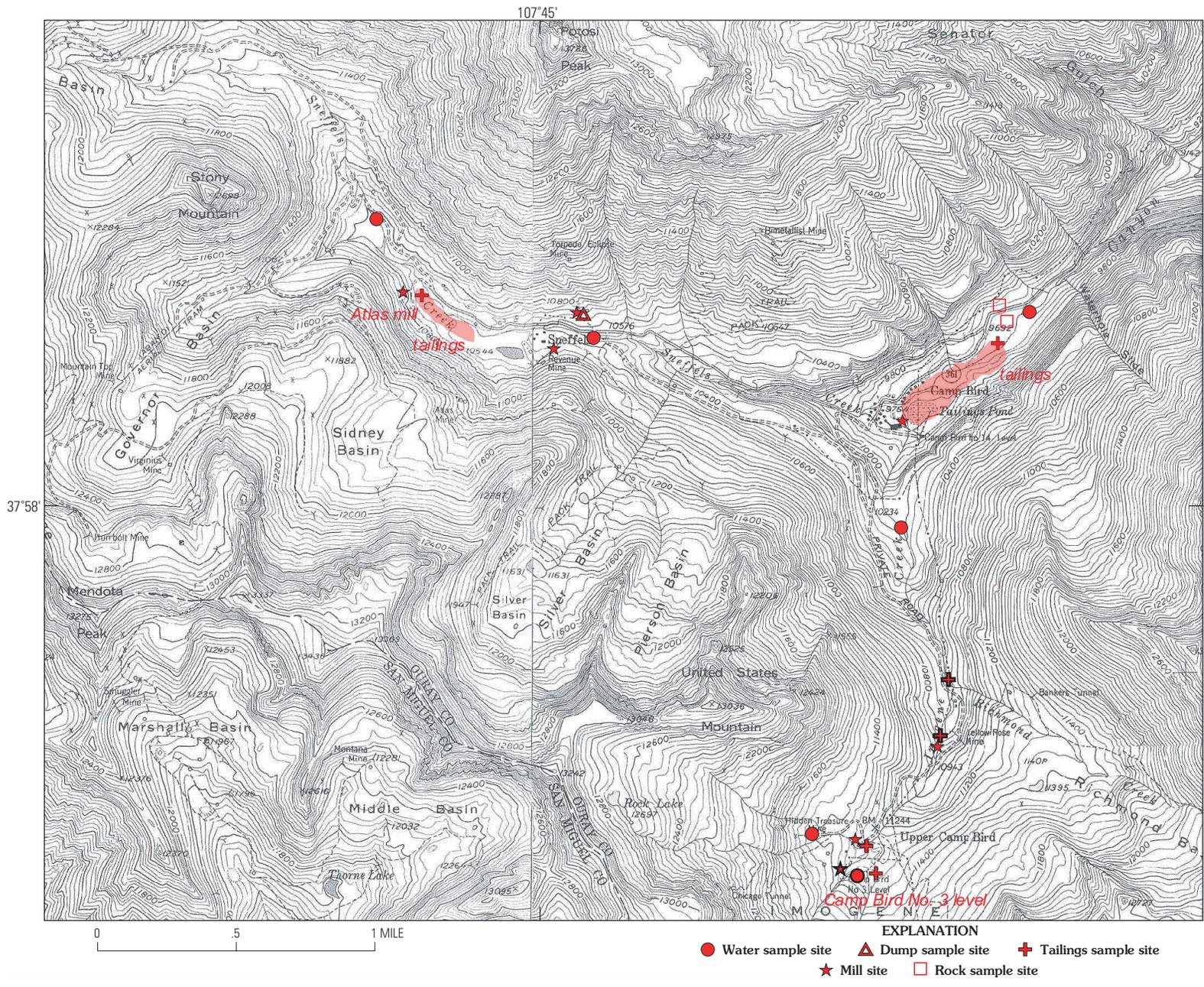
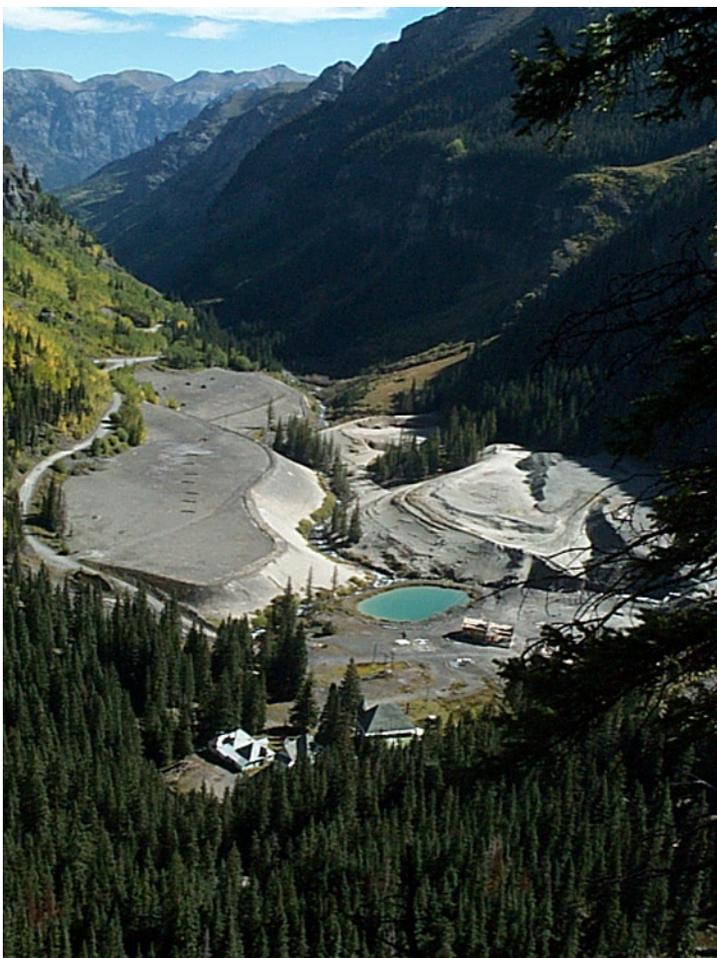
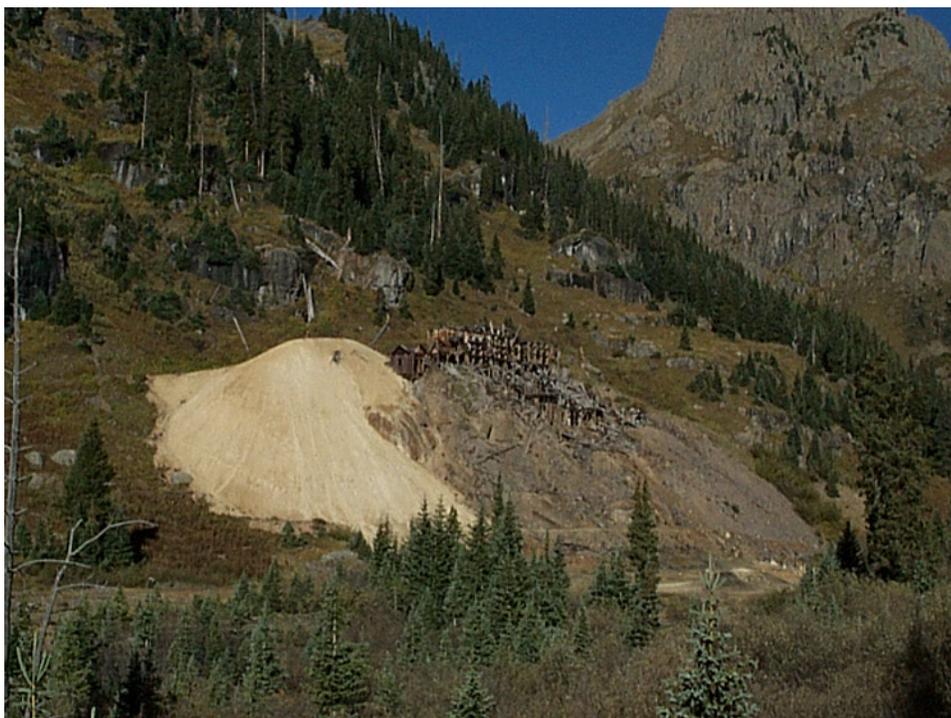


Figure 22. Features of the Sneffels mining district, showing sample localities.



**Photograph 59.** The Camp Bird mine was the major producer in the Sneffels district and created large waste and tailings piles along Canyon Creek. The high acid-neutralizing capacity of the green-altered volcanic rocks is important in the natural mitigation of effects from these materials.



**Photograph 60.** The Atlas stamp mill was a significant producer in the early years of mining in the Sneffels district. Tailings are on the left side of the mill, and also on the flood plain below.



**Photograph 61.** Mill tailings from the Atlas mill were placed along and in Sneffels Creek with no concern for their impact on water quality prior to regulations against this practice, which were enacted in 1935.



**Photograph 62.** These cement pieces of a foundation are all that remains of the Camp Bird 3-level mill. The green rocks were dumped on top of the mill site in recent years (1980's?). Other concrete foundations are above the dump, near the wooden building.



**Photograph 63.** The Camp Bird 3-level mill produced this small volume of tailings before problems with avalanches forced the construction of a new mill on the 14th level near Canyon Creek.



**Photograph 64.** These eroded gray tailings are from a small mill, ca. 1900, on Imogene Creek between the Camp Bird 3- and 14-level mills. The mill, at the base of the cliff and waterfalls, probably used water power.



**Photograph 65.** Overbank deposits of tailings can be seen at several places along Imogene and Canyon Creeks. These are in the middle reach of Imogene Creek, probably from the Richmond mill 1 mi to the southwest.



**Photograph 66.** The Hidden Treasure mine is one of the few in the Sneffels district that produces acidic mine drainage. The light-colored rocks on the dump with pyrite-sericite alteration may be the reason for the acid generation.



**Photograph 67.** The Hidden Treasure mill was only 100 yards north of the Camp Bird 3-level mill. The mill was located on the cut (center of photograph) and placed the tailings on a small flat area and presumably in Imogene Creek. These ocher tailings release more acid and metals, according to laboratory tests, than the Camp Bird tailings and are a bigger threat to Imogene Creek.

to streams. (photograph 60). One of the earlier mills (below Richmond basin) probably utilized water power, and to do so the mill was located just below a water fall. The tailings from several sites are generally rich in base metals, but there are some notable differences in composition between the tailings. A sample from the Hidden Treasure mill tailings (next to the Camp Bird No. 3 mill in the upper Imogene Creek watershed) has unusually high concentrations of Pb, Ag, Cd, Mo, and Zn relative to the district and to other tailings analyzed in this study. At the Atlas mill tailings impoundment (photograph 60), there are two distinct varieties of tailings, ocher and gray. Those of ocher color have notably high As but relatively low base-metal concentrations relative to the gray tailings. The colors probably reflect ore mineralogy rather than weathering: ocher from ore that was oxidized prior to mining and gray from unoxidized, sulfidic ore. Two samples from the very large Camp Bird tailings impoundment, which are nearly as pale as ocean beach sand, have low base-metal concentrations relative to other tailings. However, Pb and Zn contents are substantial and higher than in unmined mineralized rocks.

Leach tests on the six tailings samples from the Sneffels district generated solutions of significantly different composition. Leachate pH values ranged from 2.8 to 5.6. The most acidic leachate, from the Hidden Treasure tailings, carried the highest metal concentrations, including notably high concentrations of Pb, Cu, and Zn. The Atlas tailings generated pH

values of 4.3 and 5.2; the gray sulfidic tailings yielded less acid but higher metal concentrations than the ocher tailings. The Camp Bird tailings yielded pH values of 5.5 and 5.6 and low metal concentrations, including low values for Cu and Zn, metals of prime concern for fisheries. The tailings with the highest potential to generate acid and release metals are from the old mills: these have small volumes and probably have a proportionately low input to the nearby streams. The metal input from fluvial tailings washed away in streams could not be evaluated.

The physical settings of tailings in this district suggest that all but those at the Eclipse mill have high risk of erosion in an extreme storm event. Storm events have eroded tailings in this watershed in the past century, shown by overbank deposits of tailings along both Imogene Creek and Canyon Creek. The fluvial sediments are about 3 ft vertically above normal flow levels in these streams (photograph 65), suggesting that the flow during the storm events must have been quite spectacular. The geometry of tailings and stream channels vary significantly among this group of four sites, and each should be evaluated more thoroughly to obtain a more reliable risk assessment. The very large volume of tailings at the main Camp Bird site, and their proximity to Canyon Creek, suggest there could be major problems if this impoundment was hit by an extreme storm event. One could argue that the Camp Bird tailings are relatively small sources of acidity and metals,

but others can argue that the aquatic health of Canyon Creek requires minimum contact of water with these materials.

*Surface waters.*—Canyon Creek and its headwater tributaries (Sneffels, Yankee Boy, and Imogene Creeks) cut through this district and are popular for fishing and hiking. The water is clear and leaves no red deposits on stream cobbles, quite unlike the Uncompahgre River headwaters just to the east in the Red Mountain district. The streams were checked several times in an effort to obtain reliable information. Measurements of pH in the field ranged from 5.7 to 7.2, and conductivities ranged from 44 to 190  $\mu\text{S}/\text{cm}$ ; I consider these to be indicative of fairly good water quality, although some would say that the pH of 5.7 and conductivity of 190  $\mu\text{S}/\text{cm}$  are indications of contamination.

Chemical analyses of eight surface-water samples, some of which are replicates, show low metal concentrations. Iron concentration is elevated, but the maximum value at 0.5 $\times$  ALWS is not bad. Zinc is the element of chief concern here, with concentrations in the range 110 to 185 ppb (relative to ALWS at 50 ppb). Some Cu concentrations also exceed ALWS. Zinc concentrations are at highest in Sneffels Creek (sampled east of the Revenue mine and Atlas mill tailings). Zinc concentration in Canyon Creek is 3 $\times$  ALWS below the Camp Bird tailings, but that improves to 2 $\times$  ALWS over the next 2.5 mi. The CZI values for the streams range from 1.7 to 0.02. Most of the stream reaches have CZI below 1.0, but a more detailed study is required to evaluate the threat to aquatic life. Geochemically, it appears that the green alteration prevalent in the basin minimizes acid production at mine sources and mitigates pH and Zn concentrations by in-stream reactions below mine sources. Water compositions in Canyon Creek are not pristine, but they would be far worse without the attenuation provided by geologic processes.

Only three mine drainage samples were collected in this basin where my access was limited. The small Eclipse mine produces a small flow of water with pH 6 and only 10 ppb Zn. The upper Camp Bird portal (number 3 level) yields a small flow at pH 7.2 with somewhat elevated metal concentrations, but only the Zn concentration exceeds ALWS. Only at one adit, the Hidden Treasure mine, is surface water visibly contaminated (it creates a red precipitate) (photograph 66) and the chemical analysis shows high concentrations of Al, Cd, Cu, Fe, Mn, and Zn. The most problematic concentrations are those of Cd, Cu, and Zn (22 $\times$ , 32 $\times$ , and 90 $\times$  ALWS, respectively). The CZI for the Hidden Treasure mine drainage is 28, indicating a potential threat to aquatic life.

*Summary.*—Water quality and physical habitat in this district are surprisingly good considering the large amount of mining and disturbance in the basin. Mining methods and waste disposal techniques are basically the same as in the adjoining Red Mountain district, so the difference in conditions appears to reflect the ANC provided by green-altered volcanic rocks. Only at the Hidden Treasure mine did I see evidence for acid mine drainage. The geology of this mine may be unusual for this district as the partially oxidized vein appears to be in a felsic dike rather than in green volcanic rock.

Large volumes of tailings and waste rocks were placed near creeks, yet the downstream water quality is quite good. The concern for these near-neutral waters is the mobility of Zn, Cd, and Cu and their concentrations near or somewhat above the standards for aquatic life.

Mine-waste rock piles and tailings in riparian zones should be evaluated for physical stability and for release of metals in more detail than was possible here. Acid generation does not appear to be a problem, but these polymetallic ores could release enough Zn, Cd, and Cu to be detrimental to aquatic life or human health.

There may be high risk of failure during an extreme flood at four tailings sites in this district. Each of these sites has different setting relative to headwater basins and stream channels, and each deserves independent review. Based on the preliminary model, the very large tailings impoundment at the Camp Bird mine ranks among the top five sites on the CWS in need of additional technical information and possible reclamation to minimize exposure to floodwaters. Reclamation activities at the Camp Bird mine, observed at a distance in 1999, included capping by unmineralized rocks, which should reduce infiltration and runoff. It is not clear how much this work will reduce the risk of erosion in an extreme storm.

*References.*—Burbank (1947b); Fischer and others (1968); Fisher (1990); Benham (1980).

## Mineral Point District (including Upper Uncompahgre)

*Location.*—Ouray County, 10–15 mi south of Ouray, at the headwaters of the Uncompahgre River (a small part is in San Juan County). (fig. 23) The climate is severe, with about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver, gold, lead, copper, zinc.

*Mining history.*—Ore was discovered in 1874 and the most valuable production was prior to 1900 from about a dozen mines, none of which produced more than \$500,000 worth of ore. There was intermittent activity over the years. Some mines such as the Mickey Breen and Mountain Monarch mines shipped ore to the Banner mill near Ouray from the late 1930's to early 1950's.

*Production and disturbance.*—Total value of ore produced was about \$2 million. The individual mines were small to moderate in size and created relatively small waste dumps. The mills in the district impounded very small amounts of tailings—most ore was hauled to mills outside the district.

*Geology.*—This district on the flank of the Silverton caldera worked large, through-going polymetallic veins in volcanic rocks. Some of the major veins, as near the San Juan Chief mine, can be traced for several miles and extend into the Eureka district to the south and the Galena district to the northeast. Volcanic rocks in the area are much less altered than the adjacent Red Mountain district and are mostly the green, propylitic variety. The deep canyon of the Uncompahgre River

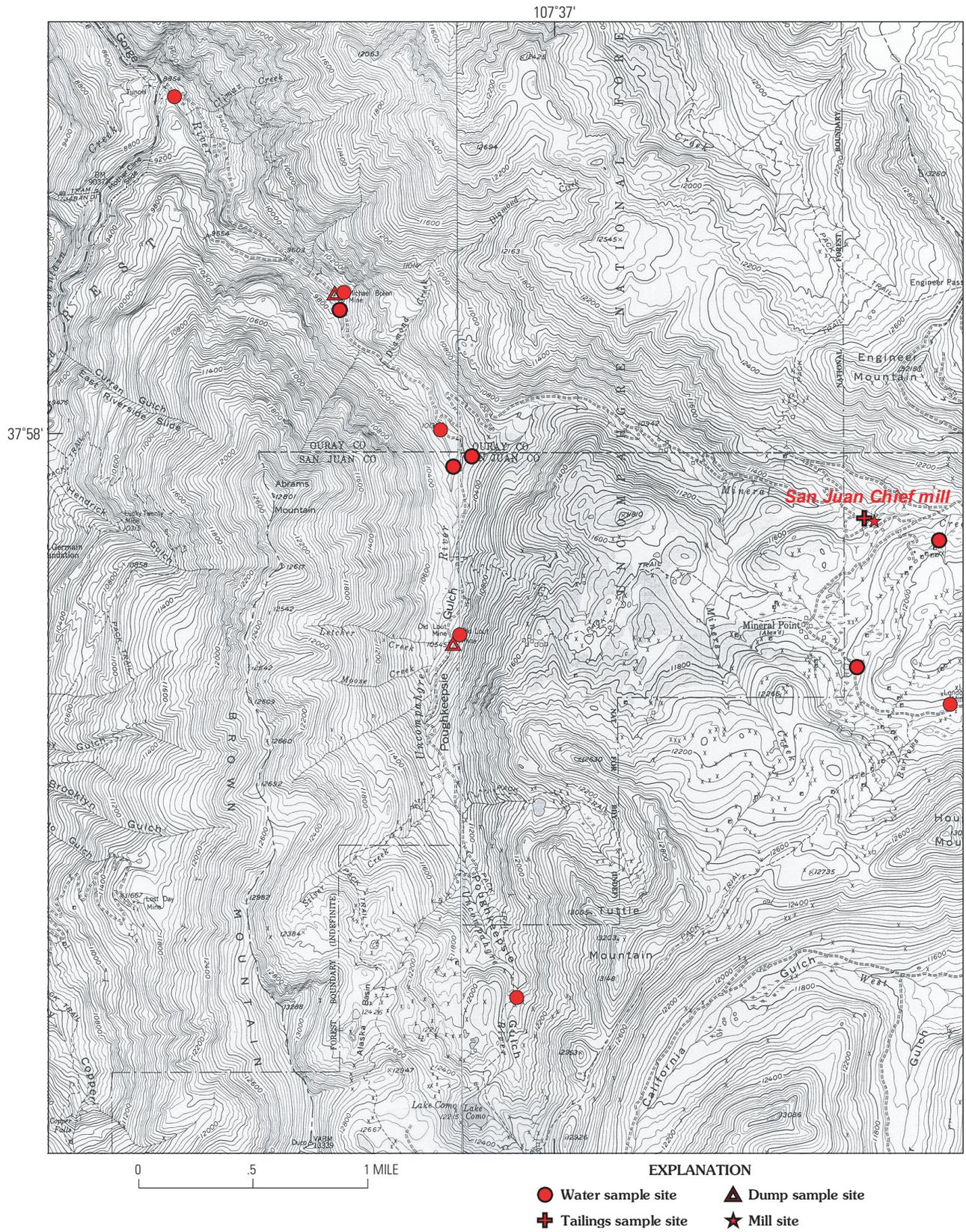


Figure 23. Features of part of the Mineral Point mining district, showing sample localities.



**Photograph 68.** The San Juan Chief mine created a series of workings up this hill, following a major vein. Drainage from the lowest level has a pH of 2.7. The acid rock drainage kills alpine flora for tens of feet beyond the waste dumps.

at the northwest end of the district cuts into Precambrian rocks and a thin sequence of Paleozoic sedimentary rocks, which are north of the major mines. The pre-Tertiary rocks in this area do not contain significant ore.

### Mineral-Environmental Conditions

Numerous chemical analyses of similar ores in the Animas watershed to the south (Nash, 1999a, 1999b) show that those ores are rich in base metals: Ag, Cd, Cu, Pb, and Zn tend to be moderate to high in concentration. Three new analyses of samples from this district confirm this trend and suggest that Bi and Mo concentrations are high locally and that As is moderate to below the median in this study. The

abundant sulfide minerals in the mined materials create a high potential for acid generation, according to many laboratory leach tests (Nash, 1999a, 1999b).

*Mineralized rocks and ores.*—Mineralization in this district is very similar to that of the Eureka district to the southeast, which has been studied in detail (Nash, 1999b). The vein deposits contain abundant coarse sulfide minerals (pyrite, galena, chalcopyrite, sphalerite) as well as complex silver-bearing minerals and pink manganese minerals (rhodnite,  $\text{MnSiO}_3$ ; and rhodochrosite,  $\text{MnCO}_3$ ). According to leach tests, these ores have the potential to generate acid and release metals. These test results are consistent with acid mine waters in some (but not all) adits in the district.

*Mills and tailings.*—Most of the ore from this district appears to have been milled at sites outside the district. A mill in the canyon near Mickey Breen mine is not recognizable today, and the San Juan Chief mill is photogenic but appears to have produced very little. As elsewhere, tailings from pre-1900 mills may have been lost to the nearby streams. Tailings do not appear to affect water quality.

*Surface waters.*—The headwaters of the Uncompahgre River are just to the northwest of the sources of the Animas River and Henson Creek, and the geology and ore deposits are continuous across the drainage divide (in fact, some veins were mined on different sides of the drainage divide). Abundant precipitation and steep topography create rapidly flowing streams with numerous cascades. The water is clear and, in most places, creates no red iron stains or coatings. Mineral Creek is a short tributary that is sourced east of the Uncompahgre River. Field determinations of pH and conductivity showed most sites have near-neutral, low-conductivity ( $<150 \mu\text{S}/\text{cm}$ ) compositions. Chemical analyses of the near-neutral (pH 6.2 to 7.0) waters show mostly low metal concentrations, but Zn concentrations are elevated ( $1\times$  to  $6\times$  ALWS). Acidic waters (pH 2.7 to 4.7) at and near some mine adits carry high metal concentrations, including very high concentrations of Cd, Cu, Pb, and Zn (which locally reaches 15,000 ppb, or  $300\times$  ALWS) at the San Juan Chief mine adit. For the first few miles of headwater streams, near mines, quality is degraded (CZI values are 0.73 to 7.1), but downstream, near the confluence with Red Mountain Creek at U.S. Highway 550, the quality improves to a CZI value of 0.5, although Zn still is high ( $3\times$  ALWS).

*Summary.*—In this headwater basin underlain by green propylitic-altered volcanic rocks, acidic conditions do not persist for more than about a mile. In both Mineral Creek and the uppermost Uncompahgre River, local segments with pH near 3 improve to pH 7, and with that loss of acidity comes much improved water quality. For example, zinc decreases by orders of magnitude, but still is elevated ( $3\times$  ALWS) at pH 6.9. This is an excellent example of natural attenuation, but also a warning that high concentrations of Zn can persist after neutralization. Further study is needed for more precise details on Zn mobility and its potential effects on aquatic life.

*References.*—Burbank (1947c); Fischer and others (1968).

## Red Mountain District

*Location.*—Ouray County, 20 mi south of Ouray (fig. 24). The climate is probably the most severe in the study area, with 30–50 inches (or more) of precipitation per year.

*Principal commodities.*—Silver, lead, zinc, copper, gold.

*Mining history.*—A boulder of galena was discovered in 1881, and soon there was a rush and discovery of significant mines including the Robinson, Yankee Girl, and National Belle. The most productive years for these silver mines in chimney or breccia pipe deposits were prior to 1890. These mines struggled through years of depressed silver prices in the mid-1890's, and all closed in 1896 and 1897. The Barstow mine on the west side of the district was an exception in that it started in 1899 and continued operating until about 1917; the Barstow vein was like those on the Telluride side of the mountain and was later mined from the Treasury Tunnel. Starting in 1907, the Joker Tunnel was driven nearly 5,000 ft into Red Mountain to access lower levels and dewater the Genessee, Guston, and other mines; this ambitious project was intermittently active, eventually reached some targets, but produced relatively little ore through the 1940's.

With success at the Barstow mine, attention shifted to veins on the western side of the district. The Treasury tunnel was driven to the west under the Barstow and other veins; started in 1896, it advanced in stages and by the 1930's allowed significant production from polymetallic veins on the west side of the district. Consolidation of diverse properties in 1939 and the mid-1940's, forming the Idarado Mining Co., allowed efficient mining on veins that had not been worked since the 1920's. Development of improved milling techniques allowed recovery of major amounts of copper, lead, and zinc and byproduct bismuth and other metals from ores that earlier were considered too low in grade or too complex metallurgically to be mined profitably. The Treasury Tunnel of the Idarado mine, whose portal was near Red Mountain Pass, extended west of the divide into San Miguel County. It provided access to major veins such as the Black Bear and Tomboy in the Telluride district, allowing higher production from the mid-1940's through 1956. The Idarado mill at Treasury Tunnel portal, with a capacity of 800 tons/day, closed in 1956 when all ore was diverted to the Pandora Mill east of Telluride. Production after 1956 used the Mill Tunnel, which is 1,600 ft lower in elevation than the Treasury Tunnel. Mining at the Idarado mine ceased in 1978. There appears to have been very little production from the Red Mountain district in the past 40 years, but there have been underground and drilling exploration programs, as on Red Mountain, in the 1970's and 1980's.

*Production and disturbance.*—Production data tend to be mixed with adjacent districts and aggregated by county and tends to be based upon where ore was processed rather than where ore was mined. Dollar value is very high, more than \$50 million. One estimate of total production for the Red Mountain district, 1,685,000 tons (Long and others, 1998), appears to include only the eastern pipe deposits, not production from the

Treasury Tunnel (Idarado mine). There is a large amount of disturbance in this district, including hundreds of substantial mine dumps (photograph 69). There are five mill tailings impoundments plus substantial areas of unconfined tailings along Red Mountain Creek. Reclamation in the mid-1990's concentrated on mine drainage and the impounded tailings.

*Geology.*—This district lies astride the margin of the mid-Tertiary Silverton caldera. Most exposed rocks are volcanic flows and tuffs. Small but important intrusions occur along the caldera margin. A thick section of Devonian to Jurassic sedimentary rocks lies between the Tertiary rocks and Precambrian basement (Burbank and Luedke, 1964); although mineralization is known in these sedimentary rocks (Mayor and Fisher, 1993), very little has been mined because of the great depth.

Most ore came from vein deposits in the volcanic rocks, but the famous high-grade pipe ores are in and near small intrusive plugs. The Telluride Conglomerate, at the base of the Tertiary section, hosted important replacement ores rich in base metals. Polymetallic replacement deposits and skarn in underlying limestones were drilled but are too deep to be mined (Mayor and Fisher, 1993). A few small veins mined from Precambrian host rocks are the roots of mid-Tertiary veins, and there has been some production from veins in pre-Tertiary sedimentary rocks.

Economic geologists distinguish many different kinds of ore in this district, but for this study we can lump the ores into three classes, some of which are related genetically: (1) polymetallic veins and replacements rich in Pb-Zn-Cu-Ag; (2) gold-quartz veins that generally formed along veins of the polymetallic type (the two vein types are mined together); and (3) acid-sulfate type disseminated deposits under Red Mountain, containing gold and pervasive alteration that added much pyrite and alunite while destroying silicate minerals; these deposits are similar to those mined at Summitville and have a tendency to produce highly acidic drainage because the buffering capacity of the host rocks has been destroyed (King, 1995; Plumlee and others, 1999).

## Mineral-Environmental Conditions

*Preface.*—The framework of geology-based evaluation and ranking of districts and problem sites used in this report may not be relevant to this district. Standards for the stream of concern, Red Mountain Creek, appear to be set by a court settlement with a performance objective of 1.25 ppm dissolved zinc (CDH, 1992). This standard, 1,250 ppb (the scale used in this report) is far higher than aquatic life standards of about 50 or 100 ppb Zn applied to many streams in Colorado. Also, most of the mining properties in this basin are on private property and reclamation of those lands is not done by FLMA's because of complex liability, legal, and political issues. The following discussions are in the same style as for other districts and do not address past or ongoing legal issues.

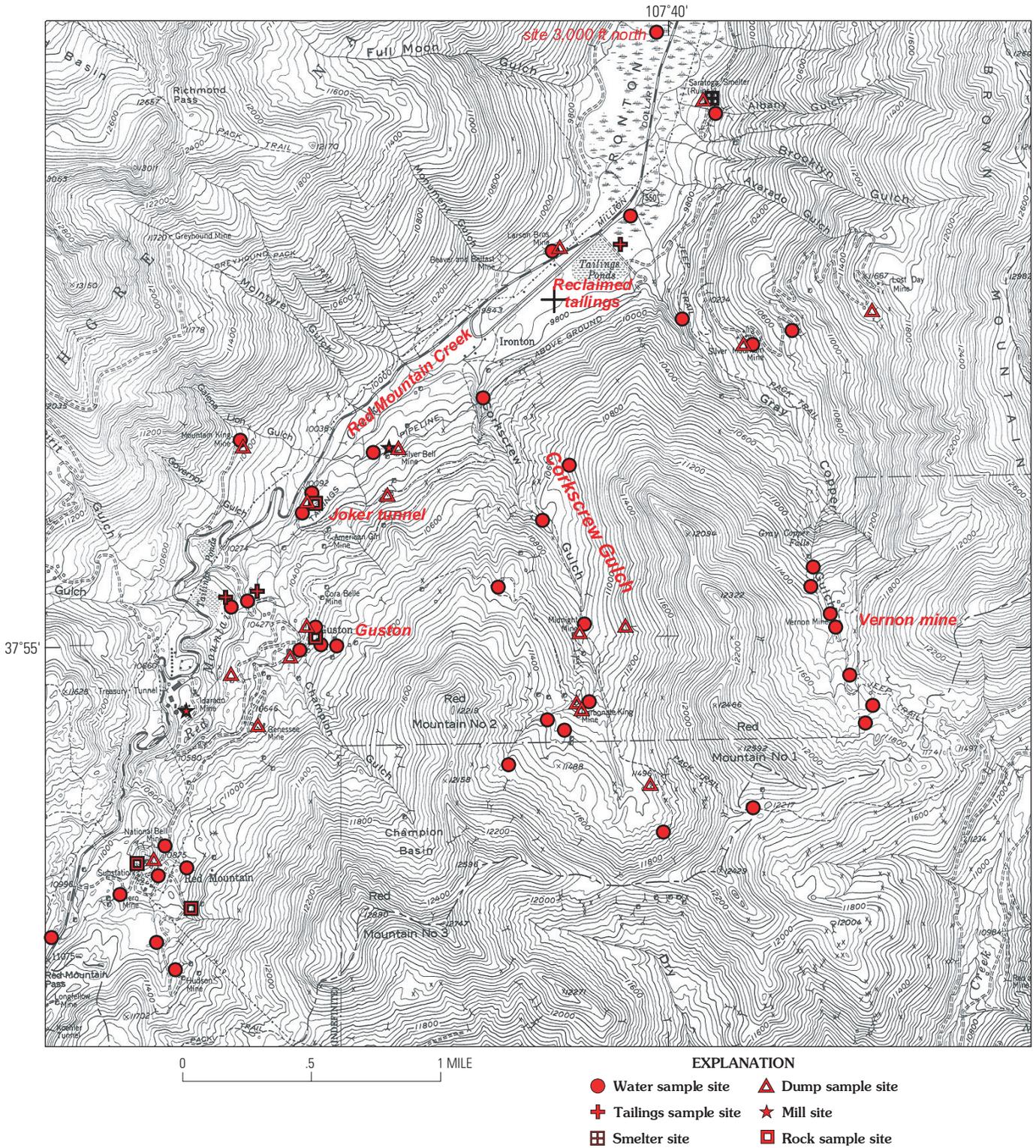


Figure 24. Features of the Red Mountain mining district, showing sample localities.



**Photograph 69.** The Red Mountain complex of highly altered rocks and associated mineral deposits is at the geographic center of three watersheds that are among the most contaminated by acid and metals in all of Colorado. Natural weathering of disseminated pyrite creates the red hues and releases acid and metals.



**Photograph 70.** This colorful, silicified knob encloses the National Belle pipe deposit. This was one of the richest silver mines in the district. These altered, unmined rocks are probably releasing acid and metals to ground water, but this is difficult to quantify.



**Photograph 71.** Brown iron deposits (ferricrete) guided prospectors to the Guston and other deposits in the Red Mountain district. The mine tunnel was excavated into the ferricrete, demonstrating that the iron is pre-mining. The pipe carries mine drainage that is extremely acidic (pH 2) and rich in metals into a concrete gutter that was created recently as part of reclamation work.



**Photograph 72.** This ferricrete deposit formed at the site of the Hudson deposit, Red Mountain district, just like the larger one at Guston.



**Photograph 73.** The Genessee mine, at the base of Red Mountain Number 2 and 3, produced a large dump of pyritic waste rock. The swath of dead trees below the Genessee dump is a reflection of acid rock drainage from the sulfidic waste.



**Photograph 74.** The highly sulfidic dump of the Genessee mine is slowly releasing acid and metals. Laboratory tests show waste rocks like these generate acidic and metal-rich leach solutions. The concrete gutter on the left was installed to carry mine drainage and runoff beyond the waste rocks.



**Photograph 75.** Acidic waters from the Guston mine and dump have killed vegetation below it. Some of this damage may be related to natural conditions that created the ferricrete prior to mining, but mining and mine waters reacting with waste rocks on the dump probably made the condition worse.



**Photograph 76.** Red Mountain Creek is highly degraded by many sources. This view shows the prominent red deposits of iron minerals that still form as very acidic waters mix with other waters and pH rises to about 3.5 to 4. Overbank deposits of tailings were deposited during a flood event or a tailings spill.



**Photograph 77.** Tailings contribute to problems along Red Mountain Creek. These tailings may have been released during early milling (before regulations) or during a flood event.



**Photograph 78.** Seepage from mill tailings is acidic and as high in metals as mine drainage. This seepage flows into Red Mountain Creek.



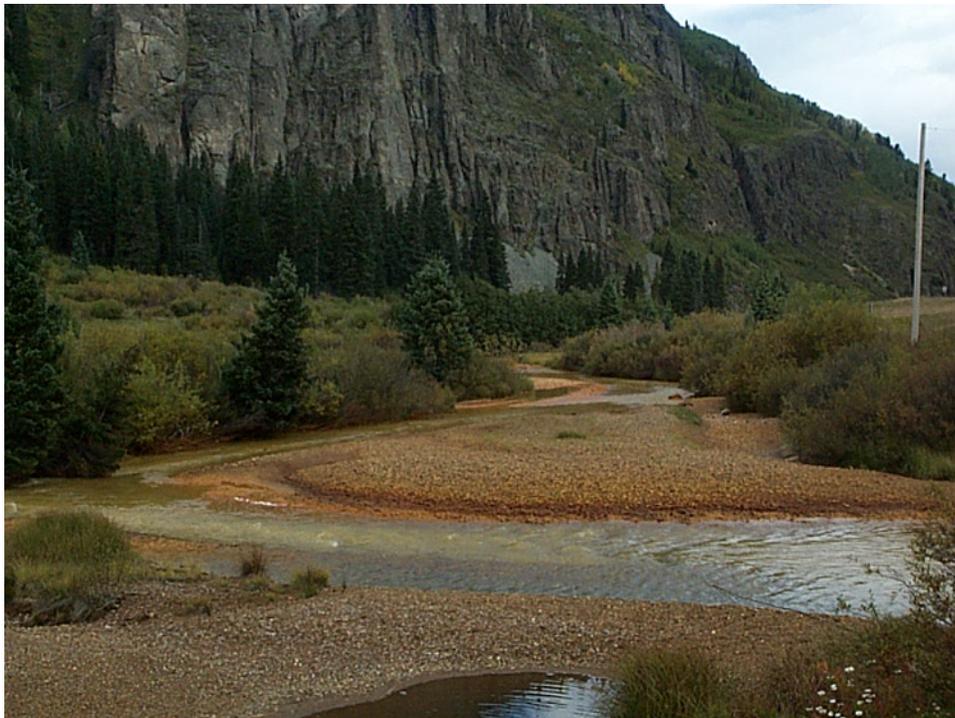
**Photograph 79.** The Larson Brothers mine tunnel is one of the few on the west side of the Red Mountain district that creates acidic drainage. Although the working was mostly in green-altered volcanic rocks, the drainage is moderately acidic and carries substantial iron and zinc. This water precipitated red iron coatings as the mine waters pick up oxygen outside the mine.



**Photograph 80.** Tailings impoundments of the Idarado mill have been reclaimed and concrete gutters put in place to control drainage.



**Photograph 81.** The Idarado mine complex at the portal to the Treasury tunnel is near the summit of Red Mountain Pass; U.S. Highway 550 is in middle of photograph. The mill was situated near the lower left corner of this photograph. The ridge on the skyline shows the green and purple-gray colors of propylitic-altered volcanic rocks.



**Photograph 82.** The quality of Red Mountain Creek improves as it mixes with other streams, especially those coming in from the west. But the concentrations of iron and other metals remain high, as shown by the red iron coatings on the stream bed at the north end of Ironton Park.



**Photograph 83.** The Vernon mine in the headwaters of Gray Copper Gulch Creek is a concern because the highly sulfidic waste and mine drainage are degrading the creek. Sulfidic waste slumps into the creek, and drainage from the dump flows directly into the creek.



**Photograph 84.** The Vernon mine on Gray Copper Gulch Creek creates several problems as shown here. Sulfidic waste is very close to the creek, and mine drainage (the flow on right) reacts with the sulfidic waste to create even more acidic metal-rich water, which flows into the creek.



**Photograph 85.** The headwaters of Gray Copper Gulch creek received contaminated waters from several sources, including small mine dumps like this. Springs on fractures in altered rocks also are sources of acidic waters. The small spring on the left side of the dump has a pH of 3.



**Photograph 86.** The highly altered rocks on Red Mountain No. 3 on the west side of Gray Copper Gulch are a likely source of much of the acid and metals in the creek. The alteration is the acid-sulfate type (pyrite-alunite) that is well known as a source of acid and metals.



**Photograph 87.** Corkscrew Gulch Creek on the flank of Red Mountain No. 1 is highly degraded for a watershed containing only a few small mines. One adit at the head of Corkscrew Gulch (small dump, center of photograph) releases pH 2 water with extremely high metal concentrations. This extremely acidic water may reflect both mine workings and unmined altered rocks.

*Mineralized rocks and ores.*—Several kinds of ore were mined in this district, but chemical analyses show them to be generally similar in their polymetallic character, being rich in Ag, As, Bi, Cd, Cu, Mo, Pb, Sb, and Zn. Comparing the results for 12 dump samples from breccia pipes with results for 14 samples from polymetallic veins, the pipe ores are richer in Ag, As, and possibly Au, whereas the vein ores and waste materials have substantially higher concentrations of Mn, Mo, Pb, and Zn. Analyses of three tailings samples from the Idarado mill (which processed vein ore) show their base-metal concentrations are relatively low, roughly one-fourth to one-half that of dump samples, but substantially higher than unmined rocks. Altered red rocks on Red Mountain (photograph 86) are well known for their unusual geochemical signature, including high arsenic concentrations (Fisher and Leedy, 1973).

Leach tests made on 20 samples (10 pipe deposits, 10 vein deposits) show a strong tendency for mine waste to generate acid: leach solutions averaged pH 3.0 for pipe samples and 3.1 for vein samples, and pH values ranged from 2.5 to 3.7. The acidic leachate solutions carried high amounts of Fe, Cu, Zn, and Pb, but other metals were moderate to low. Although most analytical results for Ag in leachate solutions were not valid because of analytical difficulties, some apparently reliable results for Ag in these 20 leachate solutions are very high. This suggests that there is potential for Ag toxicity, which is

rarely an issue in waters on the CWS. Comparing the two groups, Fe and Pb tended to be higher from the pipe-deposit leachates, whereas Mn and Zn tended to be higher in the vein-sample leachates. The high Pb in pipe-sample leachates is notable and is not predicted by the bulk composition of the samples (Pb tends to be higher in the vein samples). The high amount of Pb in leach solutions may reflect the presence of anglesite in pipe deposit samples, whereas galena in the vein samples would not yield much Pb to those leach solutions (Desborough, *in* Nash and others, 1998). The differences in detail between the pipe- and vein-deposit sample test results do not appear to be significant for this reconnaissance evaluation: waste from both deposit types has the potential to generate acid and release many base metals.

*Mills and tailings.*—The early production from this district was direct-shipping ore that was sent to a smelter. A small amount of the pre-1900 production went to the Saratoga smelter in this district, but most was smelted outside the district. A small amount of ore from the district was milled at Chattanooga (south of Red Mountain Pass) prior to about 1920. The mill at the Barstow mine is described in the literature, but was not accessible. A small volume of tailings at the Silver Bell mine suggest there was a mill at that site, and it probably would have been active prior to 1900. The Idarado mill east of the Treasury Tunnel seems to have been the only major mill in the Red Mountain district, and it was expanded

several times to about 500 tons/day prior to closure in 1956. Tailings were placed in at least four impoundments (CDPH, 1992). Also tailings from some stage of activity, or perhaps some major spills, are in the middle reach of Red Mountain Creek above the Joker Tunnel. The large Idarado tailings impoundments were reclaimed in the mid-1990's by recontouring and planting of grasses.

Two estimates of what metals may be mobilized from the tailings can be made from a laboratory leach test and from the composition of waters seeping from the tailings. The laboratory leach test yielded a solution with a pH of 3.5 and elevated concentrations of Cu, Fe, and Zn. The metal concentrations were higher than from normal rocks, but lower than from most dump samples. Seeps from the tailings piles, sampled at two locations in 1997 and 1998, had pH values in the range 2.5 to 2.9, very high conductivities (>2,000  $\mu\text{S}/\text{cm}$ ), and analyses showed extremely high concentrations of Al, Cd, Cu, Fe, Mn, and Zn. The individual seeps flow only a few gallons per minute, but the total infiltration into alluvium underlying the tailings could be substantial. The total amount of flow, and the metal loadings, requires information from wells. Drilling of wells should be considered for the monitoring phase of the Red Mountain (Idarado) reclamation.

*Surface waters.*—Water from heavy snow and summer storms is evident in this district as flows from many mine adits, countless springs, and numerous streams. The main stream that cuts through the center of the district is Red Mountain Creek, and there are substantial streams in side canyons such as Corkscrew and Gray Copper (fig. 24). Essentially all of the streams in the eastern half of the district, east of U.S. Highway 550, are visibly contaminated with suspended tan or ocher material and ubiquitous red to ocher coatings on cobbles, streambed, and refuse such as logs. Because these coatings form on recent trash such as logs that were not present in 1998, I believe that the coatings continued to form in 1999. However, I do not have observations that can define the time of formation relative to high and low flow regimes. Streams on the west side tend to be clear, but a few do deposit tan and ocher coatings on streambeds. The Red Mountain Creek watershed in the headwaters of Uncompahgre River is the most complex in the study area and is of high concern for contamination (CWQCD, 1998, 303(b) list). My geochemical studies show that it contains many mine-related sources of extremely contaminated water, and natural attenuation of pH and metal concentrations in the streams is not as effective as in other districts. Quite simply, the streams on the east side of this district are overwhelmed by the input from unmined and mined sources.

The remarkably high metal concentrations of waters in this district are summarized in table 12. The general trends in this district are possibly more significant than the details of contamination at specific mine sites.

1. Mine tunnels driven into Red Mountain, as at Joker and Guston (photograph 71), individually produce hundreds of gallons per minute of extremely degraded, acidic waters that are 10× to 100× or more in excess

**Table 12.** Summary of surface-water compositions, Red Mountain district.

[Metal compositions in parts per billion; analyses by ICP-MS. Cond., conductivity ( $\mu\text{S}/\text{cm}$ )]

	Mine and dump drainage (n=20)		Streams (n=24)	
	Median	Maximum	Median	Maximum
pH	2.9	2.0	3.4	2.5
Cond.	1,375	>2,000	366	1,035
Al	40,200	470,000	5,600	21,600
As	20	675	0.7	44
Cd	9.4	152	0.8	36
Cu	775	11,400	229	5,860
Fe	116,600	721,000	3,000	63,500
Mn	3,800	106,000	291	3,460
Pb	38	367	26	189
Zn	1,800	30,400	132	5,500

of ALWS. The Joker tunnel has a large flow of pH 2.9 water with very high metal concentrations (e.g., Fe 60× and Zn 75× ALWS), and metal concentrations in some other mine drainages are even higher.

2. Mine tunnels on the western side, driven to access veins in green-altered volcanic rocks, release water that is variable in pH and metal content. At some mines the water quality is relatively good, and at others the high Fe concentrations produce conspicuous amounts of ocher to orange precipitate (photograph 76). However, none of the water samples from the western side is degraded to the extent of the acidic waters on the Red Mountain side. A rare example of high metal content in a west-side water is the drainage from the Mountain King mine; this high volume of pH 6.3 water forms orange iron floc outside the portal. Metal concentrations are high but not extreme (Fe 2.3×, Zn 10× ALWS).
3. Unmined sources create highly acidic waters in some places, and some of these waters can be very rich in metals. One example is the creek in Champion Gulch, an area on the west flank of Red Mountain where there are few mines. The creek has a pH of 2.9 and very high metal concentrations (e.g., Fe 29×, Zn 11× ALWS). Not all streams and springs on the flank of Red Mountain are this acidic—some have pH values of 5 to 6, and at those pH's metal concentrations are below ALWS.
4. Many of the mine drainages are arguably of mixed sources, reflecting the influence of both mining and natural processes. A good example is the unnamed adit at the head of Corkscrew Gulch, which possibly extends 300–500 ft into red pyritic rocks: it yields a flow of pH 2.1 water with extreme metal concentrations (Fe 720×, Zn 26×, ALWS, and the highest As concentration in the study at 675 ppb). Based on my

experience with hundreds of other mine workings, my estimate is that the composition of this mine drainage is too extreme to be the result of reactions in the small mine tunnel. Reactions in unmined altered rocks also appear to contribute acid and metals to this water. Quantitative evaluation of anthropogenic and natural sources of acid and metals in water such as this is extremely complicated, and even the best scientific methods may produce ambiguous results.

5. Mill tailings in Red Mountain Creek (photograph 77) and in four large impoundments contribute acid and metals to the creek. For example, water seeping from a tailings impoundment along the middle part of the creek (photograph 77) has a pH of 2.5 and elevated base-metal concentrations: the concentration of Fe in this water is moderate at 2.5 $\times$ , but other metal concentrations are much higher (Zn is 26 $\times$ , Cu is 63 $\times$ , and Pb is 110 $\times$  ALWS). The contributions from the tailings are impossible to determine without wells to sample shallow ground waters that eventually flow into the creek.
6. The degree of recovery after reclamation cannot be assessed here. My observations and samples were from 1997 and 1998—after the reclamation activities in the Red Mountain Creek basin that included work on the large tailings impoundments and construction of concrete gutters to carry mine drainage over and around mine waste (photograph 74). Surface-water quality was not good in 1997 but may have been worse previously. A logging operation on hundreds of acres of private property in 1997 and 1998 probably also contributed to suspended sediment and metals to surface waters; this logging operation disturbed large areas, and few measures were used to contain or control surface runoff.

The CZI values for stream waters in this district range from 0.02 to 160 and only six of 28 stream-water samples are below the “healthy” value of 1. Many mine-adit drainage waters and seeps from dumps or tailings have high to extremely high CZI values (14 to 1,405). These values, which indicate high to severe threat to aquatic life, come from both copper and zinc. In detail, some sources are higher in Zn concentrations than Cu, but the problem at many of these sources is the high Cu concentrations (and the higher toxicity of Cu to aquatic life such as fish and invertebrates described earlier). These water compositions may be only part of the bad news for these streams. A possibly even greater problem may be the widespread deposits of ocher schwertmanite (iron oxyhydroxide-sulfate) and other complex coatings on streambeds. The geochemistry of these materials and their implications for biota should be studied in detail as they are in the Cement Creek basin, on the east side of Red Mountain (USGS, 2000). One of the dire aspects of these complex iron deposits is their high reactivity: they precipitate and adsorb toxic metals such as Cu and Zn during low flow regimes and dissolve them during high flow stages, releasing acid and the adsorbed base metals (Desborough and others, 2000).

Some features indicate that geologic processes are working to naturally attenuate some of the acidic, metal-rich waters in this district. For example, mixing of stream waters has beneficial effects. Tributaries entering Red Mountain Creek from the west mix with the acidic surface waters from the east to produce water with pH of about 3.6, conductivity of about 400  $\mu$ S/cm, and much lower metal concentrations (Zn content drops to about 600 ppb). The quality of Red Mountain Creek leaving Ironton Park (photograph 82) is further improved (Zn is about 300 ppb) but pH remains near 3.6. The mixing reactions cause iron oxyhydroxide materials to precipitate, and trace metals coprecipitate with the iron (Smith and others, 1994; Plumlee and others, 1999). Similar mixing occurs in some of the side tributaries, such as Corkscrew Creek, but in Corkscrew the pH and water quality do not improve much, which suggests there is low buffering capacity in this basin. It appears that natural processes in Red Mountain Creek basin are inadequate to improve quality to levels required for aquatic life. Possibly the chief reason for the limited recovery of Red Mountain Creek is the large amount of material that precipitates within the streambed (photograph 76). Some of this material probably is schwertmanite, as in Cement Creek near Silverton (Desborough and others, 2000; Nash, 1999b). Other kinds of coatings, such as clay, generally coats rocks, cobbles, and streambeds, preventing water-rock reactions that otherwise might be beneficial (Plumlee, 1999).

More specific description is needed for several mines or mining areas that are sources of degraded water that are significant, yet seemingly small sources when compared to those near Red Mountain Creek. The mining areas that will be discussed next are in two tributaries, Gray Copper Gulch and Corkscrew Gulch, that are east of Red Mountain Creek. The headwaters of Gray Copper Gulch are on the flank of a large ridge of red, pyritic alteration (photograph 86). Small mines and prospects expose highly pyritic rocks and create moderately acidic drainage (pH values are 3 to 4) (photograph 85). The largest mine in the upper basin, the Vernon mine, has sulfidic dumps of moderate size and an adit that releases acidic water (pH 2.9) (photographs 83, 84) containing very high metal concentrations: Cu, Fe, Zn, and are 60 $\times$ , 5 $\times$ , and 1.7 $\times$  ALWS, respectively. The creek is acidic (pH 3.2) above the Vernon mine, presumably from both small mines and from unmined red-altered rocks. Water quality is much worse below the Vernon dumps and mine drainage: the pH of the stream is 2.9, and Cu, Fe, and Zn concentrations increase eight- to twelve-fold to levels well above ALWS. A stream sample taken 1.5 mi west of the Vernon mine shows that the composition has improved by natural attenuation, but the pH is still acidic (4.9) and several metals remain highly concentrated (Al 6 $\times$ , Cu 2 $\times$ , Fe 1.5 $\times$ , and Zn 0.6 $\times$  ALWS). The ZCI values for Gray Copper Gulch creek drop from 7 below the mine to 0.6 near Red Mountain Creek. The major source, the Vernon mine, lies on several patents, but much of the waste and water release appear to be on Uncompahgre National Forest lands. By tonnage, this site is a fraction of the size of Idarado mine complex, and by metal loading it is small compared to other

sources, but the Vernon property clearly contributes to water-quality problems in the upper Uncompahgre River watershed and merits further study for reclamation options.

Corkscrew Gulch creek contributes acidic, metal-rich waters to Red Mountain Creek. This small watershed is mostly on public lands, with scattered mine patents such as at the Midnight mine, the largest of the small mines in the basin. Sources of acid and metal are not easily quantified in this basin. The productive mines had shafts that produce no surface drainage, and their dumps do not appear to contribute much to nearby small tributaries. One mine tunnel at the head of Corkscrew Gulch (site N587, mentioned earlier) produces 20–40 gpm of pH 2.1 water that is extremely rich in metals. This mine is a major contributor to Corkscrew Gulch creek, but there must be numerous other ground water sources that contribute to, or sustain, the low pH and high metal concentrations. The CZI values for samples of the creek decrease from 30 at the eastern headwaters to 5 near the Midnight mine then rise to 12 at the gorge. The increase in Cu and Zn in the western part of Corkscrew Gulch is not associated with mines, but underlying volcanic rocks are highly altered. Water quality does not improve downstream in this basin, probably because underlying volcanic rocks are highly altered and lack minerals that can neutralize the acidity.

The Silver Mountain mine has workings of moderate size that produce ugly orange water of pH 6 and rich in several metals (Fe 3×, Mn 1.2×, Zn 2.3×). These waters are seemingly a small contribution to the upper Uncompahgre River, but if this mine were in another district or another State, it probably would be highlighted as a significant problem.

*Summary.*—Field observations, measurements of pH and conductivity, and chemical analyses of surface waters in the Red Mountain Creek basin suggest that this basin of about 10 mi<sup>2</sup> is the most chemically and physically contaminated on the CWS. It is clear to the author that substantial amounts of this contamination is derived from unmined (natural) sources on Red Mountain. Detailed studies such as are underway in the upper Animas River watershed above Silverton (USGS, 2000) would be required to quantify the numerous mined and unmined contributions to the Red Mountain Creek watershed. Geochemical background values in this area clearly were very high prior to mining. One indicator of that high pre-mining background is the abundant ferricrete in the basin and the fact that many early ore discoveries were made on ferricrete outcrops. The ferricrete at the Guston and Hudson mines is spectacular.

Reclamation in compliance with a State Court consent decree has been undertaken on many waste dumps and on the four Idarado mill tailings piles. My reconnaissance observations and sampling in 1997 and 1998, after the reclamation, indicate that waters draining from the reclaimed areas were among the most acidic and metal rich of any found in this study. Although not a biologist, even I can conclude with fair certainty that many of these waters are toxic to aquatic life. Others will have to evaluate the amount of contamination attributable to unmined sources and whether the discharges

meet the water-quality objectives of the Remedial Action Plan (CDH, 1992).

The tailings impoundments in this district were established in stable areas and at a fair distance from the major streams; thus, they seem to be relatively low risks in an extreme storm event. The channel for much of Red Mountain Creek is not natural but appears to be of good design. It might be evaluated for capacity to handle a 100-year flood. If the creek rose out of this constructed channel, the overflow would have access to the lower (largest) tailings impoundment. Fluvial tailings in and along Red Mountain Creek are highly vulnerable to erosion in an extreme storm, but the volume of these tailings is relatively small in comparison with the impounded tailings.

*References.*—Ransome (1901); Fischer and others (1968); Burbank and others (1972); Mayor (1978); Fisher and Leedy (1973); Plumlee and others (1999).

## Mining Districts of the Upper San Miguel River Watershed

The headwaters of the San Miguel River on the west flank of the San Juan mountains are the sites of some of the larger historic mines in Colorado as well as countless smaller mines that were developed in slightly more than 100 years. The major mines closed by 1980, and there was no significant mining here in the late 1990's. The legacy of mining carries both positive and negative values to the area around Telluride. This relatively small mining area was the most productive of the metal mining areas on the CWS, with a gross value of more than \$250 million coming from the hard-rock mines of San Miguel County. There also was production of gold from placer operations, which will not be considered here.

As we evaluate the impact of mining on this watershed, it is useful to consider three stages in the mining history. First, there was relatively small scale mining of locally rich veins near the surface. Production during this stage, from 1875 to about 1895, was approximately \$20,000,000 for the watershed, chiefly from silver. The second stage was one of increased usage of electricity, long haulage tunnels, and long wire-cable tramways, to increasingly large, consolidated mining complexes with milling centers at Tomboy, Ophir, and Pandora. This second stage, from about 1895 to 1945, yielded about \$110,000,000, with silver values about 10 times those of gold. The final stage was the further consolidation of the mines, with the Idarado Mining Co. emerging as the giant operator, and the construction of the large Mill Tunnel and a new mill at Pandora that eventually was expanded to treat more than 3,000 tons per day. The value of ore produced during the final 35 years equaled that of the prior 70 years, and the tonnage of rock broken and ore milled certainly exceeded that of the previous years. Many of the early-mined veins were deemed too small to mine by the new methods and remained idle after World War II.

The ore deposits within the upper San Miguel River basin are basically similar in their vein character, sulfide mineralogy, and polymetallic composition. There are technical differences that were important for exploration and mining, such as (1) the presence of distinct quartz-adularia-gold veins high in many of the mines, (2) tabular zones of copper-rich ore replacing calcareous parts of the Telluride Conglomerate deep in the Idarado mine, and (3) veins rich in molybdenum and tungsten in granitic intrusions in the Ophir district. If each of these were mined and processed separately, the ore distinctions might have significance to this environmental analysis, but in practice the mine wastes tended to become mixed and rarely are distinct today on the surface.

## Telluride District

*Location.*—San Miguel County, east of Telluride (fig. 25). The climate is severe, with about 25–40 inches of precipitation per year.

*Principal commodities.*—Gold, silver, lead, copper, zinc.

*Mining history.*—Gold and silver were discovered in 1875 and by 1880 several substantial mines were operating on veins, such as the Smuggler and Argentine, that would be mined by incrementally larger and deeper operations. The first mills were small and simple (photograph 88). As mining progressed, larger mills were erected in the basins above treeline, such as at Tomboy (photographs 89, 90). Then trams several thousand feet long were used to haul ore down to the valley near Pandora to minimize problems with avalanches. Eventually, an elaborate network of long tunnels was created to haul ore more efficiently than by lifting in shafts, and also to drain water from workings 2,000 or more feet deep. The complex history of this major district is well summarized by Henderson (1926), who provides information on the many mills in the district. The larger mining companies, such as Tomboy Gold Mines Co. and Smuggler Union Mining Co., produced about \$600,000 to \$1,100,00 gross value per year from the 1890's to 1920's. Production decreased in the late 1920's and 1930's but increased in the 1940's when major haulage tunnels were developed. The Mill Level Tunnel to the Pandora mill was completed in 1948, allowing internal, gravity-assisted handling of ore in the Idarado mine. The Pandora mill was expanded several times to more than 3,000 tons/day capacity. In the 1960's and 1970's base-metal replacement ores were mined from the Telluride Conglomerate at the base of the Tertiary section. Deep drilling in the 1970's indicated the presence of geologically significant skarn mineralization in Paleozoic sedimentary rocks about 1,000 ft below the Mill Tunnel level adjacent to the Black Bear vein system (Mayor and Fisher, 1993), too deep to pursue. Mining ceased in 1978.

*Production and disturbance.*—Production figures may include ores located east of the drainage divide but carried west through long haulage tunnels. Total value through 1964 was about \$229 million. The value through 1980 has not been published. Total production from the district is estimated at

23,076,000 tons (Long and others, 1998), but this probably includes some Idarado Mining Co. ore processed in the Red Mountain district. The early production from veins in the district was about 13,364,000 tons (Mayor, 1978; this may include some adjoining districts). The production by Idarado Mining Co. from 1946 to 1976 was 10,508,000 tons (Mayor, 1978). The total output to closure was about 14,000,000 tons. Thus, this district had the largest production from metal mines in the study area (excluding the Uravan district). The major producer was the Idarado mine complex. Disturbance is the largest in study area (excluding Uravan). Numerous old mill sites had unconstrained tailings, which allowed the tailings to move downstream. The very large tailings impoundments at Pandora have been reclaimed.

*Geology.*—This district is on the western flank of the Silverton caldera and adjacent to the Tertiary Stony Mountain intrusive complex. Deeply incised canyons expose a thick section of Tertiary volcanic flows and tuffs above a thick section of older sedimentary rocks that include prominent Permian red-bed sandstones. Most of the worked veins are within Tertiary volcanic rocks, but some of the production in later years came from replacement zones in the Tertiary Telluride Conglomerate, and similar replacement ores were drilled in deeper, underlying sedimentary units. The well-mapped geology of veins (Burbank and Luedke, 1966) looks like fractures in a broken pane of glass, but there is a fundamental radial pattern related to the caldera and to the intrusion under Mt. Sneffels (Burbank and Luedke, 1968). Hydrothermal alteration is locally intense sericite-pyrite within a few yards of the veins, but, more generally, it is of the low-intensity propylitic type that is green to purple-gray from introduced Fe-Mg minerals such as chlorite and, in many places, calcite.

## Mineral-Environmental Conditions

Access to mining sites in many parts of this district is restricted, notably the upland basins in the headwaters of Marshall, Bridal Veil, and Ingram Creeks. Mines and mills in these areas are described briefly by Henderson (1926) and in other reports but could not be visited or sampled in this study.

*Mineralized rocks and ores.*—The mines in this district worked ores that were rich in Ag, Pb, Zn, and Cu, and local vein portions (often at higher elevations) that were rich in Au (Fisher, 1990). Consistent with the well-described ore minerals and production, samples of mine dumps tend to be rich in Cd, Cu, Pb, Zn, with variable high to moderate amounts of Mn, Ag, and As. Leach tests were made on two samples from Tomboy basin, yielding similar pH values (pH 3.6 and 4.0) but considerably different compositions; the pH 4 leachate carried very high concentrations of Cd, Cu, Mn, Pb, and Zn, whereas the pH 3.6 solution contained moderate concentrations of these metals. The fact that surface waters of this district have near-neutral pH, as described later, indicates that

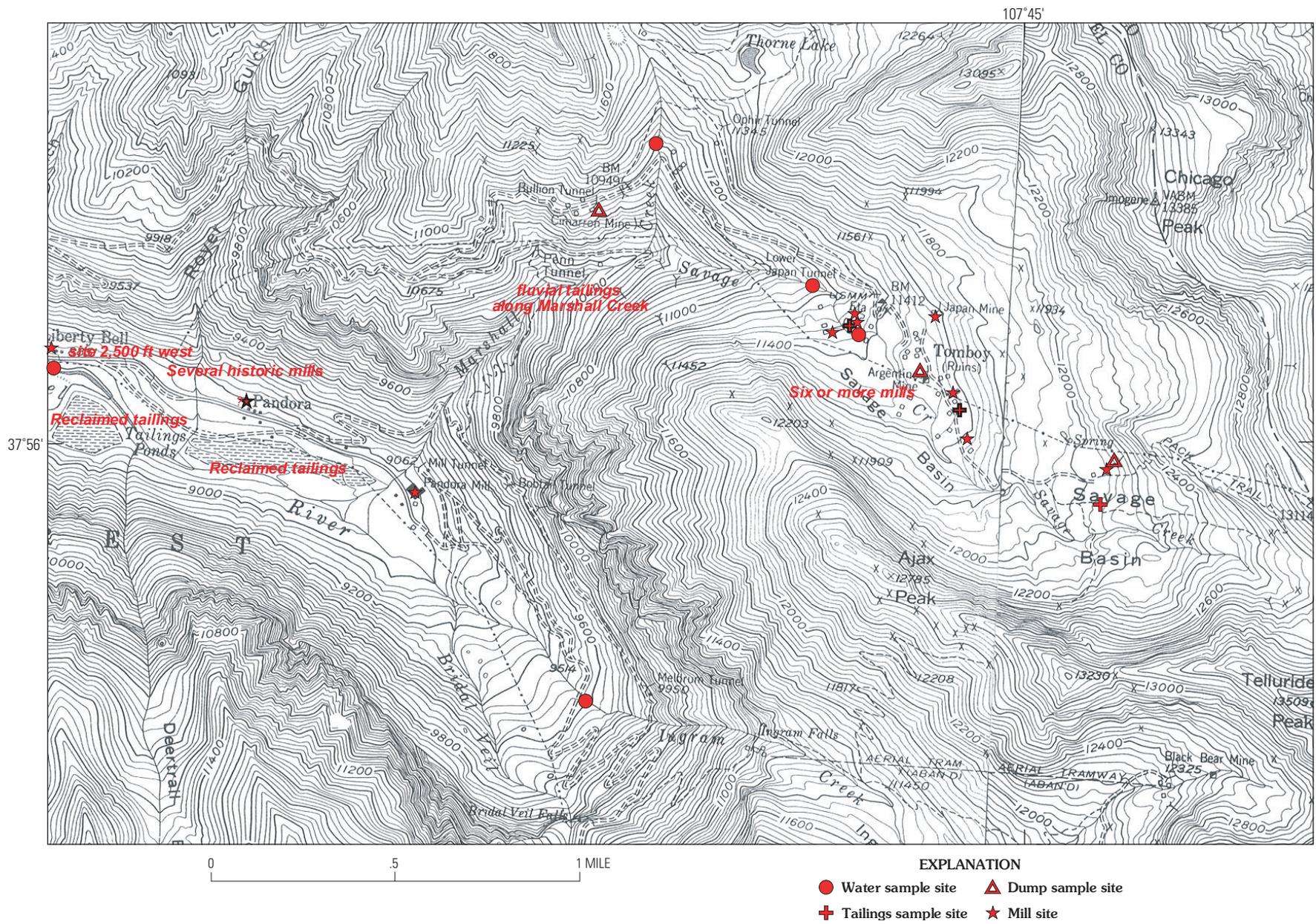


Figure 25. Features of part of the Telluride mining district, east of Telluride, showing sample localities.



**Photograph 88.** These are the sparse remains of a pre-1900 stamp mill in Savage basin, east of Telluride. Some of the metal parts are from jig tables used to concentrate gold.



**Photograph 89.** Tomboy basin above Telluride was the site of several large mills between 1890 and 1910, before ore was carried down the mountain to Pandora in tunnels or trams. There was very little space for mining and milling facilities, and a small amount of tailings can be found near these mills, a small percentage of their production.



**Photograph 90.** Four or more mills at Tomboy were located at the west edge of the upland basin at the edge of the cliffs. The burned remains appear to be from at least two mills, built very close together. There was no space for waste rock or tailings.



**Photograph 91.** Mills require strong foundations for the heavy equipment, and often that is all that remains after a fire. This is a closer view of part of photograph 90.



**Photograph 92.** These ruins are another mill southwest of those shown in photograph 90. No tailings could be retained at the edge of the cliff.

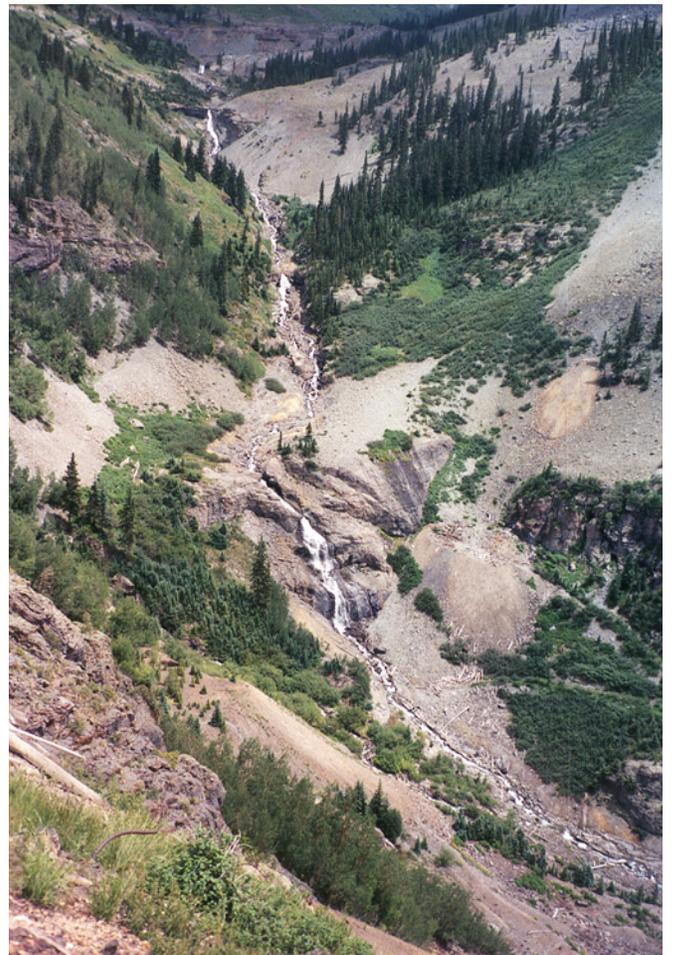


**Photograph 93.** The Japan mill was a medium-sized producer in Tomboy basin through about 1910. Jig tables can be identified in the collapsed ruins.



**Photograph 94.** The Argentine mine and mill complex in Tomboy basin was one of the large producers in that area from about 1890 to 1920. The mill probably was in the area supported by beams and logs (center), and tailings were poured out on the surface (foreground). A few hundred tons of tailings remain, but much more must have been lost from this large producer.

**Photograph 95.** This view of Marshall Creek, west of the junction with Savage Creek, shows the rugged character of the canyon below Tomboy. Patches of rusty colored mill tailings can be seen along the banks of the creek, remnants of the much larger volume of tailings dumped into Savage creek prior to 1920.





**Photograph 96.** Mill tailings are in and along the San Miguel River at Society Turn, about 4 mi west of Telluride. The mill may have been located on the old railroad, near here, but the exact site could not be located with certainty. There is a plan to reclaim this area.

the acid-generating potential of mine waste does not actually result in acidic surface waters; this may be a consequence of buffering by the widespread green-altered volcanic rocks.

*Mills and tailings.*—Mills were numerous, and tailings were produced in abundance, to support the large production from this district. I was able to identify the remains of 12 mills, and this does not include many other mills that were burned beyond recognition, replaced by a newer one at the same site, or located beyond public access. There may have been about 20 mills in the Tomboy and Pandora areas alone. The record of tailings is considerably less complete: partial remains can be found for some of the pre-1920's mill tailings, but one is left to surmise that most of the tailings were never impounded (they were poured into running streams) or the impoundments failed during storms. Looking at the six or more mills at Tomboy, it seems clear that there was very little room to stack the tailings because the mills were within a few hundred feet of a steep precipice. It was most convenient to pour the tailings over the edge and let them be carried away by Marshall Creek, and prior to 1935 there were no regulations against this practice. The four or more mills of several ages near Pandora appear to have consolidated their tailings in six impoundments. In the mid-1990's these were recontoured and vegetated and now are two large, flat-topped mounds, as specified by the Remedial Action Plan (CDH, 1992). West of Telluride, the mill at Society Turn created three tailings impoundments along the San Miguel River; these were not reclaimed as of 1998. Because milling after 1935 processed more tons of ore than prior to

that date, much of the total tailings production probably is retained in the Pandora impoundments. However, a very large amount of tailings cannot be identified, and presumably have been washed down streams of the San Miguel watershed. The amount of tailings lost may be on the order of 2 to 5 million tons (judging from the production figures given by Henderson (1926) and an estimate of the portion impounded in the Pandora area. A estimate of similar magnitude was made for "lost" tailings from the more than 50 mills in the Silverton area (Nash, 2000a).

Total chemical analyses of five tailings samples from the district suggest a wide range in compositions, with very high metal concentrations in some tailings from inefficient stamp mills. The average of the five samples suggests generally high metal concentrations relative to altered or unaltered unmined rocks (results in ppm): Ag, 73; As, 134; Cd, 97; Cu, 2,400; Mo, 125; Pb, 4,000; Se, 2.6, and Zn, 22,000. Leach tests on a sample from Tomboy yielded a pH of 4.2, very high Zn, and high Cu and Pb concentrations. A leach test on tailings from Society Turn yielded a solution with pH 7.7 and relatively high concentrations of Cd, Fe, and Zn, and moderately high Cu.

No samples could be collected from the Pandora tailings impoundments. Chemical information from the Red Mountain Idarado mill tailings may be relevant. A leach test yielded a pH of 3.5 and elevated concentrations of Cu, Fe, and Zn—higher than normal rocks, but lower than most dump samples, as previously discussed. Seeps from the tailings impoundments at Red Mountain have pH 2.5 and 2.9, very high conductivities

(>2,000  $\mu\text{S}/\text{cm}$ ), and extremely high concentrations of Al, Cd, Cu, Fe, Mn, and Zn. Presumably, monitoring by others of water quality in the San Miguel River near the Pandora tailings would determine if metal-rich waters are seeping from the tailings there.

Additional work may be appropriate to evaluate the stability of tailings in an extreme flood. Only a small volume of tailings remain in Tomboy basin (photograph 94); many of these appear to be susceptible to erosion by higher water flow in a major storm. Tailings that were put into Marshall Creek at Tomboy years ago may have been redeposited on sandbar and slackwater sites, and these might be vulnerable to erosion under the high-flow, high-energy conditions of an extreme flood. The large impoundments at Pandora appear to be stable but may deserve closer inspection. The Society Turn tailings are near the San Miguel River (photograph 96) and probably would be eroded under flood conditions.

*Surface waters.*—Water is abundant in this district, especially in the upper basins that receive large amounts of snow. The extensive network of tunnels and shafts in the Idarado mine has redirected water into the Mill Tunnel, and that water is treated prior to release into the San Miguel River. No mine drainage could be sampled. Surface waters in this district are clear and generally produce no iron stains or coatings on stream cobbles, similar to the Sneffels district. Field measurements of pH and conductivity show the streams to consistently have near-neutral pH (pH values are 6.5 to 7.3) and low conductivity (<150  $\mu\text{S}/\text{cm}$ ), both suggesting good quality. The stream waters appear to be of good quality, but chemical analyses are required to measure concentrations of metals of concern. Chemical analyses of eight stream samples show generally low metal concentrations, but several metals are elevated or exceed ALWS. Four samples contain high Cd (1 $\times$  to 7 $\times$  ALWS), three samples have high Cu (2 $\times$  to 4 $\times$  ALWS), and seven samples contain high Zn (3 $\times$  to 160 $\times$  ALWS). As in other mining areas, zinc is the metal most likely to be mobilized from polymetallic mines. The CZI values for streams are 4.3 to 1.5 near mines and mills in the vicinity of Tomboy, but the CZI values are 0.5 to 0.04 in lower stream segments more distant from mine sources. High Zn values, not Cu, create the high CZI. Zinc remains elevated in the upper San Miguel at the junction with the South Fork, 8 mi west of Telluride (140 ppb or 2.8 $\times$  ALWS), but this is apparently in compliance with the performance objective of 0.23 ppm (230 ppb) Zn in the Remedial Action Plan (CDH, 1992).

Reclamation activities over the past 10 years appear to have improved water quality, but I do not have data to monitor the changes with time.

*Summary.*—Considering the large amount of mining and milling in this district, the headwaters of the San Miguel River are in remarkably good condition. Water quality is surprisingly good for an area with more than 100 years of major mining activity. Mine development in the early years was no different than in other districts on the CWS, but here the consequences of unregulated practices are much less evident. The explanation appears to be geologic mitigation: good ANC is provided

by green propylitic-altered volcanic rocks and calcareous sedimentary rocks. Reclamation activities on the Idarado mine-mill complex seem to be helping. The near-neutral pH waters have a tendency to carry low but possibly significant concentrations of Cd-Cu-Fe-Zn, and sources of these metals may need to be minimized if healthy fisheries are to be maintained.

Substantial amounts of ore was milled at many upland sites, as in Tomboy basin, in the early years of mining and prior to laws regulating disposal of tailings. Reconnaissance observations of about 10 early mills suggests that only a small percentage of the tailings are retained near those mill sites. One must conclude that millions of tons of tailings were lost to streams such as Marshall Creek through a combination of unconfined tailings disposal and storm events. These tailings presumably now lie in streambeds of the San Miguel watershed. These fluvial tailings should be considered part of the geochemical baseline of the area. Presumably these fluvial tailings slowly release metals to stream waters, in a manner similar to the weathering of unmined sulfidic rocks. Reclamation of these fluvial tailings would be costly, would destroy physical habitat, and probably would produce only slight improvements in water quality.

Tailings in this district seem to be at lower risk of failure in an extreme storm than many districts on the CWS. An extreme flood in Tomboy basin or lower in the San Miguel River headwaters would likely erode enough tailings and fluvial tailings (previously eroded) to cause chemical degradation of the San Miguel River. Tailings in fluvial deposits (sandbars, slackwater deposits, overbank deposits) cannot be protected from a flood, but tailings in the small impoundments at Society Turn and Tomboy possibly could be reclaimed in ways that would minimize the risk of erosion in an extreme storm.

*References.*—Purinton (1898); Henderson (1926); Burbank (1947b); Mayor (1978); Fischer and others (1968); Mayor and Fisher (1993).

## Iron Springs District (Ophir)

*Location.*—San Miguel County, 6–10 mi south of Telluride. Included here are the Alta, San Bernardo, and Trout Lake mining areas (fig. 26). The climate is severe, with about 25–40 inches of precipitation per year.

*Principal commodities.*—Silver, gold, and lead; minor iron (for pigment) and tungsten.

*Mining history.*—The first vein in this area was discovered in 1877, and soon thereafter many mines were operating. In the early years of mining the ore had to be carried by pack animals to Silverton for smelting. The first mill was built in 1881, but major production did not occur until after completion of the railroad to Ophir Loop in 1890. Some of the most productive years were in the 1890's, chiefly from the Carbonero and Carribeau mines. Mining was intermittent for the next 60 years. The Alta mine was productive in the 1930's and forced to close

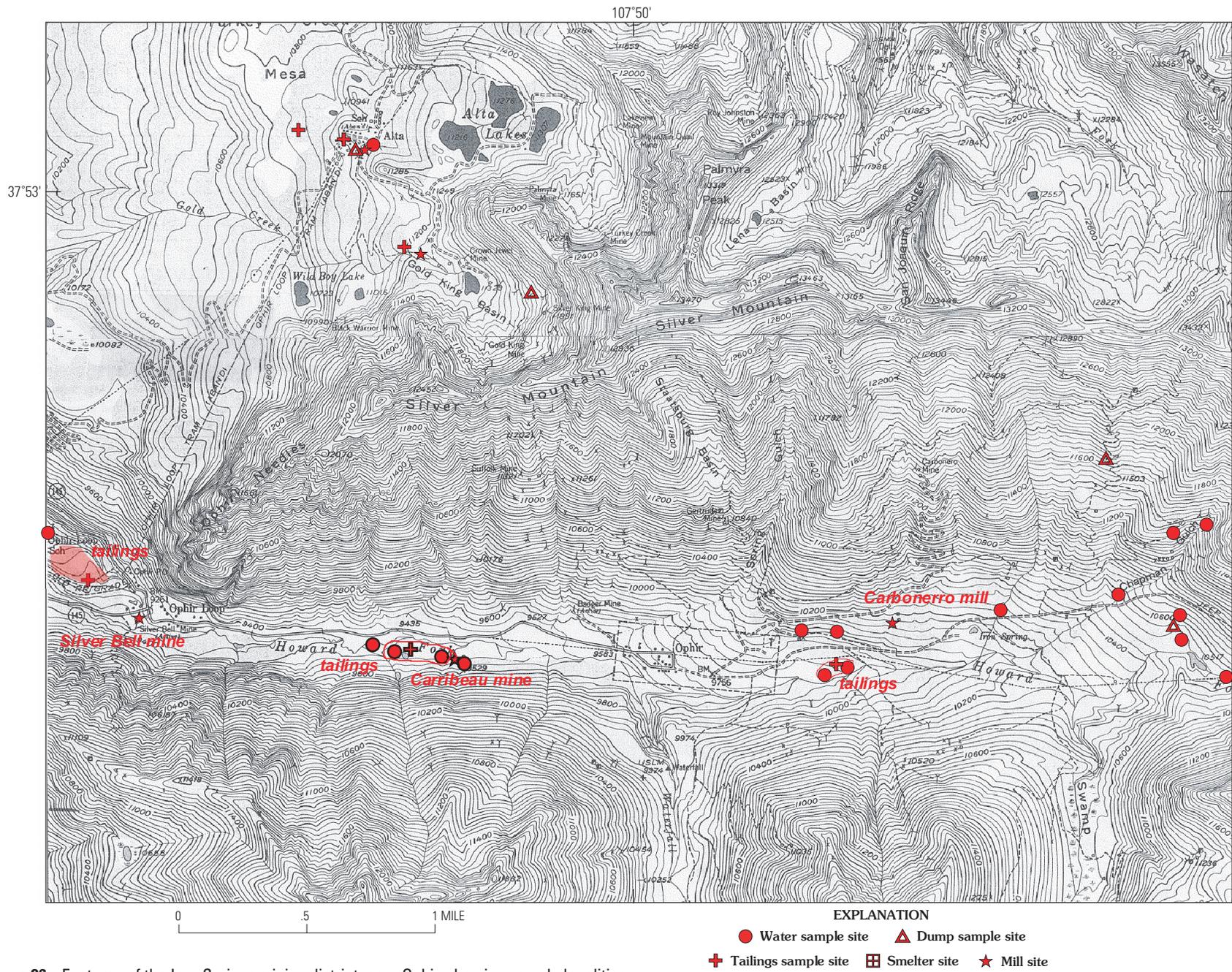


Figure 26. Features of the Iron Springs mining district, near Ophir, showing sample localities.



**Photograph 97.** Tailings from the Carbonero mill were placed close to the Howard Fork and are eroded by smaller side streams.



**Photograph 98.** View of the Carbonero tailings from the east edge of Ophir. This broad area of tailings is in wetlands of the Howard Fork and may be at risk of severe erosion in an extreme storm (flash flood).



**Photograph 99.** Tailings from the Carribeau mill were placed on a flat area on the south side of the Howard Fork. These tailings cover several acres.



**Photograph 100.** Drainage from the Carribeau mine has high flow and carries fairly high concentrations of iron and other metals; the composition and pH of 6 suggests that it may be more acidic deep in the mine. Remediation of such a high flow of metal-rich water is complex.



**Photograph 101.** Tailings from the Carribeau mill are flooded by mine drainage from the adit. The view is from the mill site, which is about 200 ft south (behind) this location. Several acres of tailings (rust color) are flooded by the mine drainage; the Howard Fork is about 10 ft beyond the upper right corner of the photograph.



**Photograph 102.** This closer view of flooded Carribeau mill tailings shows the influence of active beaver dams. Mixed ownership of the mine and tailings impoundment complicates remediation.



**Photograph 103.** Tailings from the Silver Bell mill were stacked high above the Howard Fork. Reclamation work was underway in September of 1999. Part of the tailings impoundment is on USFS lands.



**Photograph 104.** Large iron deposits (ferricrete) formed from iron-rich springs after glaciers carved the Ophir valley. Waters that formed these deposits are thought to be geochemically similar to acid mine drainage but clearly are thousands of years older than any mining in the area.



**Photograph 105.** Iron was mined at Iron Springs east of Ophir from this cut that shows the layering in the iron deposit. Springs are active today and continue to carry high iron concentrations at about pH 3.



**Photograph 106.** Artesian flow of water carries high concentrations of iron and other metals from a drill hole made for metal exploration in the 1960's(?). The drill hole is on the hillside north of the Iron Springs iron mine.

when the mill burned in 1948. Mines in the Ophir Valley had later periods of production in the 1950's and 1960's, with ores processed through the Silver Bell mill. There has been little activity for the past 20 years. Mining in this area was greatly aided by hydroelectric power generated at Ames.

*Production and disturbance.*—Total value was about \$11 to \$15 million. The Silver Bell mine is credited with about \$1.6 million production through 1970, and the San Bernardo mine produced about \$700,000 worth of ore (chiefly Ag, less value in Au, Cu, and Pb) through 1928. Total production from the district is estimated at 873,000 tons (Long and others, 1998; this figure may not include the Alta and San Bernardo mining areas). The underground mines (chiefly adits) produced relatively small dumps and workings, but several substantial impoundments of tailings were made and are located close to the Howard Fork (photographs 97, 99).

*Geology.*—Veins containing base-metal sulfides and precious metals occur in Tertiary volcanic rocks on the western flank of the Silverton caldera and adjacent to a Tertiary intrusive complex. Pre-Tertiary sedimentary rocks underlie the volcanic section and should be good ore hosts, but ore has not been described in these rocks. The intrusions are larger in size and coarser in grain size than at most districts in the San Juan Mountains, and the alteration halos around these intrusions are large (as at Silver Mountain and Yellow Mountain). Some of the alteration is pyritic and appear to be sources of acidic, iron-rich waters such as at the artesian springs that gave the district its name (Iron Springs). The springs that issue near the valley floor have deposited iron as thinly laminated bog-type deposits, now hard and crystallized as goethite, which have been mined for use as a pigment (photograph 104).

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Chemical analyses of three dump, nine tailings, and two ferricrete samples supports the classification of the ores in this district as polymetallic. The dump samples have high Ag, Cu, Mn, Mo, and Pb concentrations relative to other dump samples in the study, and As and Zn values are near the median concentration for all dumps. Nine samples of tailings show a range in compositions, but the tendency is to be high in Ag, Mn, and Pb relative to other mineralized rocks, moderately high in As, Cu, and Mo, and low in Zn (but the Zn concentration is higher than in unmined altered rocks). In leach tests, two dump samples both yielded solutions of pH 3.1, carrying high concentrations of Fe, Mn, Cu, and Zn; one sample released very high concentrations of Mo. Two samples of ferricrete were not very reactive in leach tests, as shown by low conductivity values and low metal concentrations in the solutions, which had pH values of 3.5 and 3.8; moderate amounts of Cu and Zn were released. Tailings samples had differing reactions, with pH values of leach solutions in the range of 2.6 to 4.1. The acidic solutions carried high concentrations of Al, Fe, Mn, Cu, and Zn. A sample of tailings from the Silver Bell impoundment released very high

concentrations of Mo, Pb, and Se.

*Mills and tailings.*—The long history of production in this district is reflected in the six or more mills and associated tailings. The first mill may have been the Gold King in the Alta area that operated in the 1880's. The preserved tailings from this mill are tens of tons on a flat area that appears to be quite stable; I could not determine if tailings were sluiced into the creek and washed away. The Gold King mill probably was replaced by a mill at Ophir Loop, connected by the tram shown on the Ophir topographic map. Other early mills were erected to service the Carbonero and Carribeau mines, a mile east and west of Ophir. The Carbonero mill probably was served by a tram, but no physical remains, other than the tailings impound and tailings spilled from the sluice, could be found. The Carribeau ore was handled inside the mountain and brought out through a low-level tunnel. Tailings from the Carbonero mill are spread wide and thin in wetlands north of the Howard Fork (photograph 97), and those from the Carribeau are on the bank south of the river (photograph 99). The volume of tailings in these two sites is moderate, and an undetermined amount was lost by erosion or sluiced directly into the streams. These mills probably ceased operating prior to the 1935 regulation for impounding tailings. The Alta mill operated chiefly in the 1930's–40's, until it burned in 1948. Tailings were piped to a small impoundment near the mill and to a larger area to the west on an upland bench; the western Alta tailings were recontoured and revegetated in 1997. Ore from the San Bernardo mine was trammed across the Lake Fork to the mill near the railroad, and tailings placed nearby, on the east side of the stream. The moderate volume of tailings had not been reclaimed as of 1998. The largest volume of tailings is from the Silver Bell mill, which may have been placed on top of tailings from the 1910-era mill at Ophir Loop because only one tailings pile is evident (photograph 103). The Silver Bell mill probably did not operate after about 1970. The moderate to large volume of tailings were stacked like a hay pile, above and to the south of the Howard Fork. The San Bernardo and Silver Bell tailings are not wetted by the adjacent streams, but they do collect precipitation and must react with that water. No diversions or liners are evident that could minimize runoff, erosion, or infiltration.

Leach tests on six samples of tailings from this district show a wide range in potential for acid and metal release. Tailings from the Silver Bell tailings pile contain layers that differ in color, suggesting ores of differing mineralogy were processed at various times. The majority are ocher (from oxidized ore?), and this material releases high concentrations of Al, Fe, Mn, and Zn, whereas gray layers (sulfidic ore?) release very high concentrations of Cu, Fe, Mo, and Pb, and more As than the ocher material. Both tailings types generate acid, yielding solutions of pH 2.8 from the ocher tailings and 2.6 from the gray. Leach tests on the Carbonero tailings yielded pH 2.7 and high metal concentrations (especially Cd, Fe, Pb, and Zn). Two samples of the Carribeau tailings yielded solutions of pH 3.3 and 4.1, much lower conductivity than others in the district, and moderate concentrations of Cu and Zn. A sample of tailings from the Alta mill yielded a solution with

pH 3.5 and relatively low metal concentrations, but a very high Pb concentration.

Tailings at most, if not all, of the sites in this district are involved with, and react with, surface waters. These waters were sampled at eight sites to determine the reactivity of the tailings and metals released. One sample type is unconventional: pore waters. I collected these waters where possible by making a hole in the tailings with an auger and collecting the water that seeped into the hole in about 10 minutes. Most tailings are damp but do not produce water for this test. Compositions of waters in contact with these tailings are highly variable: pH value ranged from 2.1 to 7.3, and conductivity values ranged from 660 to >2,000  $\mu\text{S}/\text{cm}$ . Pore fluids in the saturated tailings tend to be most acidic and metal rich. Concentrations of Al, Cd, Cu, Fe, Mn, Pb, and Zn range to high and extremely high values (10 $\times$  to 100 $\times$  ALWS). Tailings in the Iron Springs district clearly lie in the flow of surface waters, are reacting, and are releasing significant concentrations of base metals. I cannot determine the volume of flow involved, thus cannot compute metal loadings. Also, wells are required to determine infiltration into shallow ground water in alluvium beneath the tailings. A puddle on the Silver Bell tailings pile had a pH of 2.1 and extremely high metal concentrations, possibly concentrated by cycles of evaporation. Puddle waters are not typical of surface-water compositions, but I believe that in general they provide an estimate of the first flush of water in storm runoff.

At least three of the tailings impoundments in this district rank among the sites on the CWS having highest risk to failure in an extreme storm. The tailings from the Carbonero, Carribeau, and Silver Bell mills are poorly constrained and are in close proximity to the Howard Fork. The Carbonero and Carribeau tailings commonly are wet and eroded by side streams, and the Howard Fork would not have to rise much to cause major erosion. The Silver Bell tailings are high enough to be above likely flood levels on the Howard Fork, but their shape and position suggest that they may be vulnerable to slumping or liquefaction in a protracted storm with exceptional amounts of precipitation. The engineering properties of the Silver Bell tailings, such as shear strength, could be determined by geotechnical consultants to better evaluate risk of slumping into the Howard Fork. Reclamation work underway in the summer of 1999 should reduce runoff from the Silver Bell tailings but may not change the physical condition of the pile or resistance to slumping. I could not inspect the San Bernardo tailings impoundment to evaluate its setting, and this should be done by others. A worst-case scenario for these tailings would be mass failure during a storm event, creating a slide or slurry of tailings into the stream. This seems a low probability risk, but the consequences to the Howard Fork would be devastating to aquatic life and impossible to clean up.

*Surface waters.*—Water quality in this district has tendencies that are not so clearly expressed elsewhere: waters from unmined altered rocks in the eastern part are highly mineralized and give the name to the district, but weakly altered rocks in the western part provide natural attenuation. Mines and mill tailings are sources of metals, but the waters are not as

acidic or metal rich as in some districts. Thirty-three samples of surface waters and tailings pore waters were collected for chemical analyses, and many additional sites were checked for pH and conductivity.

Waters in the eastern part of the district continue to deposit iron in the vicinity of historic iron springs and substantial mined deposits of iron (goethite) (photograph 105) (Hanshaw, 1974). The spring waters have locally variable pH and conductivity, in the general range of 3–5 and 300–1,000  $\mu\text{S}/\text{cm}$ , respectively, which I interpret as indicative of mixing of ground waters of different compositions. A nearby drill hole has artesian flow (photograph 106); it has pH 5.9 and conductivity of 1,540  $\mu\text{S}/\text{cm}$ . These spring and drill-hole waters are very rich in Fe (3 $\times$  to 15 $\times$  ALWS), Cu (0.3 $\times$  to 50 $\times$ ), and Zn (1.4 $\times$  to 9 $\times$ ). There are indications that an acidic, metal-rich water flows from the mountain of pyritic-altered volcanic rocks and mixes with unmineralized surface and ground water. These reactions at the source and at sites of iron deposition have been operating for at least 8,000 years, according to radiocarbon dating (Hanshaw, 1974) and clearly are not related to man and mining. There also is evidence that these metal-rich waters may have been thermal springs at one time (W. Miller, written commun., 1999). The iron-rich deposits are chemically related to iron-cemented rock (“ferricrete”) that is common in the San Juan Mountains. Numerous ferricrete deposits are known in the Red Mountain and Eureka districts (Nash, 1999b), and an especially large one formed at the site of the Keystone mine, Ruby district. Ferricrete was used by prospectors as a guide to ore, and tunnels were often made through the ferricrete to find veins, as described previously for the Red Mountain district. Waters in some San Juan ferricretes have high concentrations of Cu-Zn-Cd, but many are rich in only Fe or Fe and Al (Nash, unpub. data, 1999).

Mine drainage, investigated at seven sites, is highly variable with pH values in the range 3.7 to 7.1. The worst water quality was found at a small mine across the valley west of the USFS Matterhorn campground. Water draining from the Alta mine tunnel has a pH of 6.6, which normally means low metals, but this site is exceptional for its very high base metals (Cd 18 $\times$ , Cu 58 $\times$ , and Zn 36 $\times$  ALWS). The Carribeau tunnel issues a huge flow of water that deposits iron and reacts with tailings (photograph 100); the pH is 5.6 to 6 but metal concentrations are quite high (Cu 5 $\times$ , Fe 6 $\times$ , Mn 1.7 $\times$ , Zn 4 $\times$  ALWS). Some mine adits on the north side of the canyon, not far from the county road, issue similar high flows of pH 3.7–5.5 water that create large amounts of red iron precipitate. The Al, Cu, Fe, and Zn concentrations in these waters are about 2 $\times$  to 9 $\times$  ALWS.

Despite these inputs of acid and metals from natural and mine sources, the quality of water in the Howard Fork is quite good. Small streams and the eastern headwaters of the Howard Fork have pH 5.1 to 6.2, and moderate conductivity (300 to 600  $\mu\text{S}/\text{cm}$ ). The altered rocks and iron springs in the area would logically explain the elevated conductivity, as well as elevated Fe and Zn values (each near the ALWS). The composition of the Howard Fork does not change much downstream (west) of the

Carbonero and Carribeau tailings: pH is 5.5, conductivity 350  $\mu\text{S}/\text{cm}$ , and the Zn concentration rises to  $2.2\times\text{ALWS}$  at a site 1.5 mi west of Ophir. The Howard Fork at Ames, west of the Silver Bell mine and mill complex, has pH of 6.7 to 7.2 (measured in different years), conductivity of about 350  $\mu\text{S}/\text{cm}$ , and Zn concentrations of  $1.3\times$  to  $1.8\times\text{ALWS}$  in two analyzed samples. Mixing with the Lake Fork and other streams improves the quality of the resulting stream (the South Fork of San Miguel River) to pH 7.1 and conductivity of 270  $\mu\text{S}/\text{cm}$ , and Cu and Zn concentrations are below ALWS. The CZI value for the Howard Fork east of the mines and iron springs is 0.16, and four samples downstream from the tailings are in the range 0.22 to 0.48, showing the input from mining and mill tailings. The increase in pH probably reflects the input from tributaries that flow through green-altered rocks. The rise in pH for the Howard Fork at Ames might reflect alkalinity from the Silver Bell tailings (lime is typically added in mills). Water quality in the Lake Fork is very good (CZI is 0.02), and mixing with the Howard Fork creates good water in the South Fork (Zn is 43 ppb and CZI is 0.26, both suggesting low threat to aquatic life).

*Summary.*—Water quality in the Howard Fork within this district is somewhat degraded by inputs from mines, several mill tailings impoundments, and from unmined sources, including iron-rich springs. The good acid-neutralizing qualities of green-altered volcanic rocks appear to minimize the effects of acidic inputs. However, the neutralized waters that emerge within the district tend to have elevated Zn, Cd, Fe, and Cu concentrations. There are numerous discharges from collapsed mine tunnels that have pH values of 4 to 7 and deposit large amounts of red iron floc (which probably looks worse than it is chemically), but also carry elevated concentrations of Cd-Cu-Zn. Reclamation of some of the mine and tailing sites mentioned here would reduce the base-metal concentrations in the Howard Fork, but the significant input of metals and acid from unmined altered rocks cannot be mitigated. Further work to evaluate mill tailings in riparian zones along the Howard Fork and reclamation options for the iron-rich mine waters are needed to maintain or improve water quality in the Howard Fork.

Possibly the highest environmental risk in this district is that a rare, extreme storm and flood that could cause significant erosion of the Carribeau and Carbonero mill tailings. The physical properties of the Silver Bell tailings cannot be determined by reconnaissance inspection, but if they have low shear strength when saturated with water, they would slump into the Howard Fork and create chemical conditions that would be catastrophic to aquatic life. Geotechnical studies of these tailings, and also the San Bernardo site, are suggested as a step toward preventative reclamation.

*References.*—Varnes (1947a); Vhay (1962); Bromfield and others (1972).

## Mount Wilson District

*Location.*—San Miguel Co, about 10 mi west of Telluride (fig. 20). The climate is severe, with more than 30 inches of

precipitation per year.

*Principal commodities.*—Chiefly Au and Ag; minor values of Cu and Pb.

*Mining history.*—The Silver Pick deposit was located in 1882 and mined for a short period, 1883–98, with most of the production coming from an extremely rich bonanza vein in parts of just 3 years. The vein was mined in very rugged terrain at 12,100 ft to 13,200 ft elevation. The ore was trammed several miles to the north to a mill at treeline. The main mine, the Silver Pick, produced only small amounts of ore from 1930 to 1960. A few small mines nearby were productive before 1900.

*Production and disturbance.*—Production was valued at between \$600,000 and \$1 million, mostly prior to 1900. Chief values were in Au and Ag, with a little from Cu and Pb. Disturbance was small in the vicinity of the Silver Pick mine. The small mill has very little waste today; tailings may have been reprocessed in the 1930's or dumped in the nearby stream.

*Geology.*—This mining area is in rugged mountains that are west of the main San Juan Mountains, and adjacent to the Colorado Plateau Province. The geology is dominated by several intrusions, including the Wilson Peak stock. A sequence of more than 5,000 ft of Triassic to Tertiary sedimentary rocks is intruded and altered by middle- to late-Tertiary stocks and laccoliths. The flat-dipping laccolithic bodies resemble the famous ones of eastern Utah. The intrusive bodies are altered and cut by veins. The ore deposits are thought to be related to later stages of the intrusive complex.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—The ores were mined chiefly for gold, but the veins probably carried base-metal oxide and sulfide minerals as well. The Silver Pick mine was not visited in this study, but earlier I had visited mines on the southern side of the divide (San Juan Forest) and noted the oxidized character and narrow width of those veins. As described in reports, the Silver Pick mine operated under arduous conditions, for a few months at a time, and probably was even more selective than other mines of the time. The dumps probably are small and oxidized, and, if these attributes are correct, there would be low potential for acidic drainage and low mobility of base metals.

*Mills and tailings.*—The small stamp mill for the Silver Pick mine has no visible tailings. Presumably, the tailings produced prior to 1900 were carried away in the stream.

*Surface waters.*—No samples were collected. On brief hikes through the area, north and south of the Silver Pick mine, there were no visible suggestions of significant transport of contaminants.

*Summary.*—No problems are evident or expected. Water quality does not appear to be influenced by the small mines.

*References.*—Varnes (1947b); Bromfield (1967); Bromfield and others (1972).

## Placerville District (Lower San Miguel)

*Location.*—San Miguel County, about 10 mi west of Telluride. (fig. 1) The climate is moderate, with about 16–20 inches of precipitation per year.

*Principal commodities.*—Vanadium, lesser uranium; some placer gold.

*Mining history.*—The vanadium deposits were discovered long before the uranium boom on the Colorado Plateau. These deposits were mined from 1909 to 1920, and again in the 1940's. Mills near Vanadium and Sawpit, north of the San Miguel River, processed ores that were carried by trams from mines in the canyon walls. The mill, which ceased operations in 1920, was reclaimed some years ago. Several small placer gold operations were run during the Depression years.

*Production and disturbance.*—Production was modest, with about 195,000 tons of ore milled to recover 3,700,000 lb of vanadium. The mines created small disturbance and are difficult to locate today.

*Geology.*—Stratabound deposits rich in vanadium occur in continental sandstones of the Jurassic Entrada Formation near Placerville. Copper minerals are present adjacent to the vanadium-rich lenses and suggest a similarity to the sandstone-hosted copper deposits farther west at the Cashin mine.

### Mineral-Environmental Conditions

*Mineralized rocks and ores.*—No samples could be collected for this study. The oxidized nature of the ore minerals suggests that there should be low potential for acid generation, and most metals are probably stable. Radioactivity could be a concern in the vicinity of mine workings and ore stockpiles, although most seem to have been reclaimed.

*Mills and tailings.*—The mill sites and tailings have been reclaimed.

*Surface waters.*—I collected no new data. These mines are not expected to pose problems, but radon could be a problem locally in ground water (if used for domestic wells).

*Summary.*—The major mill sites have been reclaimed. The BLM supervised reclamation of a mining area in 1997. No significant problems are likely, but there could be local accumulations of uranium, radium, or radon in ground water from either unmined mineralization or from mined rocks.

*References.*—Fischer and others (1947).

## Mining Districts of the Lower San Miguel River Watershed

### Uravan District

The Uravan district<sup>5</sup> is also called Uravan mineral belt

and includes many subdistricts, such as Slick Rock, Bull Canyon, and Uravan (fig. 1).

*Location.*—Western San Miguel County, extending into Montrose and Dolores Counties, and adjacent Utah. The climate is semiarid, with about 10–16 inches of precipitation per year.

*Principal commodities.*—Uranium and vanadium, radium.

*Mining history.*—This area has a long history as a leading producer in the uranium and vanadium industries. Uranium was recognized and mined in a small way in 1898, but the yellow carnotite ores first became important as a source of radium in 1910. In the 1930's and in World War II the ores were valued for vanadium. During World War II the Manhattan Project recovered uranium from the U-V ores. After World War II the Atomic Energy Commission (AEC) encouraged production of uranium, and vanadium was a byproduct. Mining declined in the 1960's during the era of limited quotas, but returned as prices rose in the 1970's. All mines were closed in the 1980's. The mills at Uravan and Vancorum (Naturita) have been reclaimed as part of Federal efforts to reclaim uranium mills and tailings through the Uranium Mill Tailings Reclamation Act (UMTRA) and Superfund processes. As of 1998, some mines and dumps had been reclaimed and several of the mines were “mothballed” and could be reopened on short notice.

The AEC was granted mineral rights to large tracts of land on the Colorado Plateau in the 1940's (Chenoweth, 1981), and they explored for uranium with technical assistance from the USGS. If uranium in significant amounts was located, it was leased for mining. Leasing continued in the 1970's. The successor to the AEC, the Department of Energy (DOE), administers those lands. In recent years, the DOE has been negotiating the transfer of the Uranium Reserves to the BLM. The uncertain status of these Federal lands, unique in the Western United States, appears to complicate reclamation of mining sites on those lands.

*Production and disturbance.*—Production was very large, representing one of the giant uranium mining areas of the United States and the world. Production figures were obscured to protect confidentiality. Through 1964, the majority of the Colorado production of 8 million tons of uranium ore, valued at about \$176 million, came from this mining area. The total production through 1978 (close to the end of mining) was about 60,000 tons U<sub>3</sub>O<sub>8</sub> (not tons of ore) (Shawe and others, 1991); this comprises nearly 13 percent of the total U.S. uranium production in the 20th century. Disturbance in this area really cannot be compared directly with the hardrock mining to the east because the district is actually a region, the style of mining is different, the climate and topography differ from the mountain districts, and so forth. The uranium mines ranged

<sup>5</sup> Only brief investigations were made in this district and much more work is needed. The information that follows is highly generalized and is included here to permit some comparisons with other mined lands of the CWS.



**Photograph 107.** Uranium mines of the Uravan district commonly tunneled into rim sandstones on steep canyon walls, as here on the edge of Atkinson Mesa. Mine waste that was dumped down the steep escarpment is difficult to reclaim. The San Miguel River is at the base of the slope; ponds to the south are near Uravan.



**Photograph 108.** Mine waste in mesa terrain is subject to erosion in storm events, as shown here. Stormwaters have carried uranium mine waste across Highway 141 to the Dolores River.



**Photograph 109.** Widespread mining across the Uruan district created thousands of small waste dumps like this one at the Cougar mine near Slickrock. Thunderstorms have severely eroded the mine waste at this site. Reclamation standards are vague for sites like this.



**Photograph 110.** The ore-buying station at Poverty Flat, near Slickrock, was thoroughly reclaimed in the 1990's, possibly using the standards for mill tailings. The Dolores River is in the foreground.

from small to large in size, utilizing mainly adits and shafts, and a few were open pits. The total amount of rock mined and placed on the surface was very large (many millions of tons). There was one very large mill at Uravan, a buying station at Slickrock, and smaller vanadium mills at Naturita (Vanorum) and Gateway. The uranium mills and tailings have been carefully reclaimed under stringent UMTRA and Superfund standards. There are no Federal standards for uranium mine and dump reclamation, so the work on mines and dumps is inconsistent. Some reclamation allows highly radioactive rock to remain exposed at the neatly groomed surface.

*Geology.*—Uranium and associated vanadium occur in tabular lenses in sandstones of the Salt Wash Sandstone Member of the Jurassic Morrison Formation across a broad arc in western Colorado and adjacent Utah, hence the term Uravan mineral belt. The sandstones are not continuously mineralized, but the zones of U-V became predictable after an unprecedented amount of geologic study for the Atomic Energy Commission and countless exploration drill holes. The ores were originally dark colored (reduced) U-V minerals with pyrite and other sulfides, but some zones were oxidized by near-surface weathering processes to form yellow carnotite. Subtle bedding-plane features, warps, and faults influence the distribution of ore. Some of the controlling structural features are related to the salt anticlines of the Paradox Basin.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Numerous analyses are available for uranium-vanadium ores of this area. For this investigation, 13 samples were collected for analysis, sufficient to confirm the generalization of previous research that these ores are enriched in a distinctive suite of elements that includes V, Mo, and Se, but very low concentrations of Th and base metals. Median concentrations for this set of samples are (in parts per million): As, 20; Cu, 21; Mo, 23; Pb, 53; Se, 49; Th 3; U, 490; V, 2,700; and Zn, 49. The ore compositions thus are very distinct from the metal deposits to the east.

Leach tests on six samples of sandstone-hosted ores from waste dumps produced leachates with pH in the range 6.0 to 7.6 and generally very low metal concentrations. Some samples released elevated amounts of As, Mo, and Se, but concentrations of U were low. Concentrations of As, Mo, and Se are potentially of concern because they are relatively high and probably would be transported for long distances in these near-neutral waters (in contrast to the decreases in metal concentrations caused by neutralization of acidic waters).

Uranium mines and waste are generally measured for radioactivity because exposure to radiation is one of the health concerns. Not surprisingly, the radioactivity of mine dumps can be high to very high. Reclamation practice as of 1998 was to create a flat surface on waste dumps, if accessible, but there was no requirement for cover by clay or barren rocks to minimize radiation or water infiltration. Radiation standards for uranium mill tailings are very specific. Recon-

naissance measurements of some dumps (reclaimed and unreclaimed) with a gamma ray spectrometer, calibrated on “pads” at USDOE in Grand Junction, show radiation levels are commonly 2× to 10× those of the mill-tailings standards. Much more work is needed on this aspect of uranium mine dumps and their reclamation standards. Guidance from EPA or other Federal experts is needed to provide consistent standards for reclamation of uranium mines and their waste dumps.

*Mills and tailings.*—Unlike practices in the metal mining industries, uranium ores throughout the Western United States have been trucked to centrally located mills to recover U and V. This has the important consequence of creating a large central tailings impoundment, rather than numerous smaller ones close to the mines. In most of the Western United States, uranium mill tailings have been reclaimed, often by trucking to new repositories, under provisions of UMTRA. The mill and tailings site at Uravan was declared a Superfund site for reclamation of all associated liquid and solid wastes. Ore-processing facilities at Slick Rock (called a buying station) and at the older vanadium mill near Naturita have been reclaimed. Tailings from the vanadium mill near Gateway have not been reclaimed but are much smaller in volume and are being monitored by the BLM Grand Junction office.

*Surface waters.*—There is very little surface water in this area, but the Dolores and San Miguel Rivers traverse the mining area. No mines in the study area discharge water to the surface, but some adits just to the west in Utah do drain water that is enriched in U, Mo, As, and Se. These mine waters have some of the highest concentrations seen for these elements in this study: U (2× and 20× relative to the ALWS of 44 ppb U), As (0.1× to >20× relative to drinking-water standard of 50 ppb), Se (4× and 69× relative ALWS at 5 ppb), and Mo (2× and 9× relative to ALWS at 19 ppb). The stream below these mines is also highly enriched in As, Mo, Se, and U (1.5×, 9×, 39×, and 13× ALWS, respectively). Vanadium is enriched in these mine waters, but vanadium is generally not listed as a toxic element outside of some industrial plants. One seep near a small mine has some extremely high As, Mo, Se, and U concentrations (As >20×, Mo 100×, Se >400×, and U 3.3× ALWS); cattle drink this water, but probably should not. Atkinson Creek, below Atkinson Mesa and numerous substantial uranium mines, has elevated but less dire concentrations of these elements (As 0.1×, Mo 3.6×, Se 0.8×, and U 0.4× ALWS). No chemical determinations were made of radioelements related to U (radon, radium, etc.). These few samples confirm the known solubilities of the elements in neutral to alkaline waters that carry As, Mo, and Se as oxyanions. No studies were made of the San Miguel and Dolores Rivers, but previous investigations had shown them to have compositions that exceed standards in the vicinity of the Uravan mill (CWQCD, 1989).

*Summary.*—Conditions and concerns for these uranium-vanadium mine lands are very different from the metal-mining areas. The lack of a national standard for uranium mine waste complicates evaluation and recommendations for reclamation. Currently (2001) it is permissible to reclaim mine waste and

leave uranium-rich rocks at the surface, even though the radioactivity is 2 to 10 times the regulatory limit for mill tailings. Mine drainage probably is not a significant problem. Drainage from a few uranium mines may be enriched in As, Mo, Se, and U, but the amount of water discharged is so small as to pose no threats to the main streams, such as the Dolores and San Miguel Rivers. There could be local contamination of water to the point that it should not be used for stock, irrigation, or domestic drinking water.

Further study is needed of the risk of erosion of mine dumps during flash floods, especially along the canyon walls near the Dolores and San Miguel Rivers. There are indications of recent storm-induced erosion and transport of mine waste to within a hundred yards of these important rivers. In an extreme storm event, the aqueous composition of the debris flow would certainly exceed standards for turbidity, uranium, and possibly a few other elements (Ra, Rn, As, Mo, Se).

*References.*—Shawe and others (1991); Fischer (1942); Fisher and Hilpert (1952); Chenoweth (1981); Shawe and others (1959).

### Cashin District (La Sal Creek)

*Location.*—Westernmost Montrose County, 20 mi west of Naturita (fig. 1). The climate is semiarid, with about 12–20 inches of precipitation per year.

*Principal commodities.*—Copper, silver.

*Mining history.*—Veins of copper were located in 1880's,

and for the next 20 years several small mines produced very rich copper-silver ore. In later years, production was only a few thousand dollars per year (gross) but increased to more than \$70,000 in the mid 1940's. Silver was an important coproduct. There has been no mining since 1945, but a thorough exploration program based on a new geologic concept defined substantial copper resources in the 1990's (more than 11 million tons of ore with an average grade of 0.55 percent Cu) (Hahn and Thorson, 1995).

*Production and disturbance.*—Production was about 1,700,000 pounds copper and 350,000 oz silver. The Cashin mine is the only substantial mine in a cluster of small mines and prospects. Disturbance is small and produced just a few small mine dumps. Only a small volume of mill tailings could be identified at the Cashin mill site.

*Geology.*—Red sandstones of the Triassic Wingate Formation host copper-silver mineralization that has mixed oxide and sulfide mineralogy. The early workers focused on the high-grade ore in faults and called it “vein-type” and hydrothermal, but new concepts, based in part on knowledge gained in the past decade on other sediment-hosted copper deposits of the world, guided drilling that shows much wider and thicker Cu-rich zones than the vein concept. The ores are now believed to be related to normal faults along salt anticlines: reducing gases and metal-rich brines moved up the fault structures and moved outward into permeable sandstone beds. Pyrite produced by the sulfidation of iron in sandstones is present in a large halo around the faults, but recent oxidation has destroyed the pyrite and makes recognition difficult by nonspecialists. Copper is



**Photograph 111.** The Cashin mine, located in the meandering canyon of La Sal Creek, had underground workings to the right (north) of the buildings. The largest building (green roof) is the mill. The creek is in lower center of photograph; mill tailings probably were placed in the creek east of the mill but are difficult to identify today.

present in both sulfide and oxide minerals.

Studies by a Summo USA Corporation are underway to determine the economics of bulk mining the copper zone, possibly in an open-cut mine. This company completed engineering studies of an identical deposit in Lisbon Valley, Utah, received a permit to mine it, but the permit process was appealed in court. The mine plan calls for recovering the copper by sulfuric acid leaching on sealed pads and electro-winning the dissolved copper.

## Mineral-Environmental Conditions

*Mineralized rocks and ores.*—Mineralized samples from the Cashin mine area are enriched chiefly in copper, with lower but important amounts of silver. Arsenic is slightly enriched, but uranium is low. No samples were tested by the leach method, but by analogy, I expect that there would be little acid generation and modest copper solubility. Existing mined rocks do not appear to pose chemical problems for Cashin Creek.

*Mills and tailings.*—There is a small mill at the Cashin mine, but tailings from the mill are difficult to distinguish from fine sand derived from nearby sedimentary rocks. The amount of mill tailings is uncertain; some or most may have been washed down La Sal Creek.

*Surface waters.*—No samples were collected. The copper and associated metals are not expected to be highly mobile under the likely neutral pH conditions. Calcite in the sedimentary rocks provides high ANC. New studies should examine the mobility of copper in surface waters because even low concentrations of copper are toxic to fish and associated biota.

*Summary.*—This variety of copper deposit appears to be relatively stable and benign in the semiarid environment. The relatively small amount of disturbance by historic mining does not appear to contaminate surface waters. However, copper has modest solubility in near-neutral pH waters and could pose a threat to aquatic life.

*References.*—Fischer (1936); Hahn and Thorson (1995).

## Discussion and Summary

Observations and interpretations have been summarized for the individual districts on the central Western Slope (CWS) and will not be repeated here. These should be considered as interim interpretations. This reconnaissance study demonstrates the need for further work, some of which will be done in ongoing efforts that use GIS (geographic information system) techniques to combine the results presented here with digital geology, regional geochemistry, and mineral-deposit information. This spatial analysis will allow comparison with the geo-environmental map of Colorado developed by Plumlee and others (1995d).

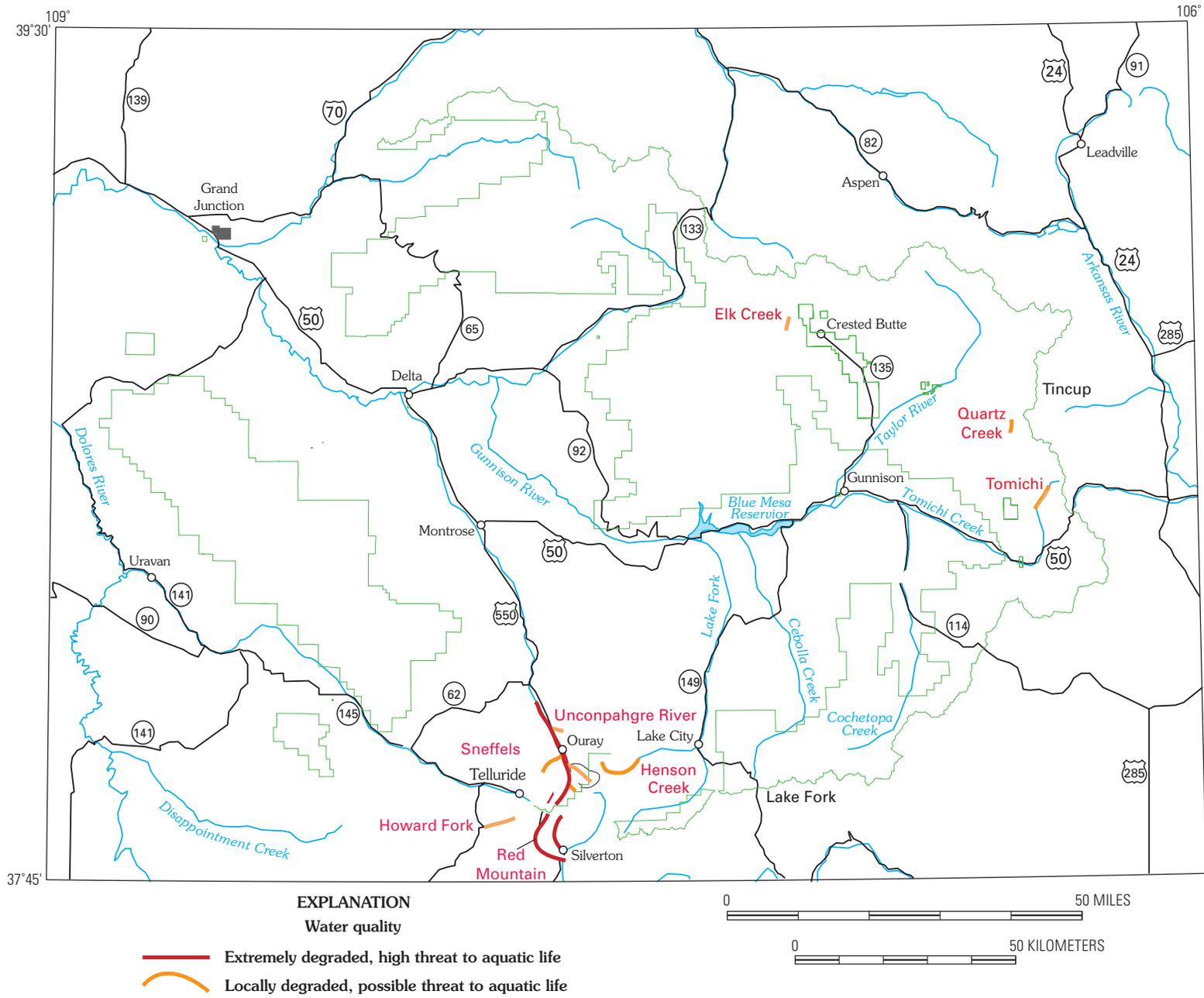
*Water quality in relation to mining.*—These geochemical investigations show that contamination from historic mines is

generally less than asserted by some (for example, CWQCD, 1989) or suggested by some geochemical models. Most of the studied mining districts have at least a few sources of contaminants, but, in the majority of cases, the metals and acidity of concern are naturally mitigated to trace levels within about 1 to 3 mi of the sources. For these self-mitigating areas, the major concern is zinc and cadmium concentrations that are at or somewhat higher ( $1\times$  to  $2\times$ ) than the chronic exposure standard for aquatic life. A few districts having a combination of stronger sources and limited buffering capacity have impacts on water quality, and presumably on aquatic health, beyond the area of mining and downstream for distances of about 3 to 10 mi. In these headwater streams, metals (such as Al, Cu, Fe, Mn, and Pb in addition to Zn and Cd) also tend to be above standards for aquatic life. In such situations, reclamation activities are more complicated—engineers may be asked to do what nature cannot do—reduce the discharge from mining sources or design treatment systems that remove the toxic metals. Several areas on the CWS have unmined sources of acid and metals that complicate mitigation planning. Additional specific studies will be needed to define the baselines produced by the unmined mineralized rocks and to determine if water-quality standards for aquatic life (fisheries) can be met without treatment systems for contaminants from unmined sources.

In this reconnaissance study 22 mining districts were examined and sampled briefly. Of these districts, one stands out as a significant threat to watershed health: the Red Mountain district in the headwaters of the Uncompahgre River watershed (fig. 27). Red Mountain Creek in the Red Mountain district appears to be the most contaminated of those studied here, and there are indications of multiple sources, including unmined rocks of Red Mountain. The problems here are comparable in magnitude, spatial extent, and geochemistry to those in the adjoining Cement Creek of the upper Animas watershed. The upper Animas watershed is being studied by several agencies, with close attention by the EPA, and water-quality standards have not yet been assigned to those headwater streams.

Several other watersheds have mining-impacted waters, with smaller magnitude than at Red Mountain Creek and smaller than implied by the 1989 point-source assessment (CWQCD, 1989). My findings are consistent with the revised point-source assessment or “303(d)” list of 1998 (CWQCD, 1998). Several streams have water quality that may be harmful to fish and their supporting food web. Elevated zinc concentrations persist 1 to 5 mi downstream from mine sources in Elk Creek (Ruby district), the uppermost Uncompahgre River (Mineral Point district), Dexter Creek (Ouray district), Tomichi Creek (Tomichi district), Gray Copper Creek (Red Mountain district), Howard Fork of the San Miguel River (Iron Springs district), and the upper part of Henson Creek (Galena district). More studies are needed of these headwater basins, including the mobility of metals (especially Zn and Cu) and the effect that these metals may have on aquatic life.

*Copper-zinc index of threats to aquatic life.*—Copper and zinc concentrations of surface waters vary greatly across the



**Figure 27.** Location of streams, central Western Slope, Colorado, that have degraded water quality. Location of degraded streams approximate; line width exaggerated. Streams near Silverton from USGS (2000).

CWS and are very high for many local sources such as adit drainages, tailings-pond seeps, and puddles on mine-waste dumps. The CZI summarizes in a simple number the water-quality concerns in relation to aquatic life, such as fish or their supporting food chain. Several streams have CZI values of 1 to 10 that suggest degraded quality and health, and some streams have values in the range 10 to >100 that suggest very bad quality and an especially high threat to aquatic life. The streams with very high CZI values include Red Mountain Creek and two side creeks in the Red Mountain district, Elk Creek (Ruby district), and the headwaters of Henson Creek (Galena district). Marginal CZI values, 1 to 0.3, for five other creeks suggest possible threats to aquatic life. Copper appears to be the chief element of concern in several of these districts, but zinc is the concern in a few, such as Elk Creek (fig. 7).

*Geologic attenuation of mine contamination.*—In most districts there is geochemical evidence for attenuation of acidity and high metal concentrations by processes controlled by lithologies having high acid-neutralizing capacity (ANC). The most important and widespread lithologies on the CWS are propylitic-altered (green to purple) volcanic rocks, limestone, and sandstones containing calcite or other minerals that consume acid. Although these lithologies are the major factors, the beneficial reactions actually occur when surface or ground water flowing from those lithologies mix with acidic water. A simple field test of these mixing processes is conductivity measurements that show high variability over tens of feet in a stream. Reactions of acidic waters with rocks having high ANC also occur, but in many of the most degraded streams or mine discharges, the channel is coated with clay and other precipitates that minimize reactions with the rocks. In the district discussions, it was repeatedly pointed out that waters leaving the district probably would have much lower quality if there were no attenuating reactions. The one district that does not benefit much from these processes is Red Mountain, where the many mined and unmined sources of acid and metals

exceed the neutralizing capacity of the highly altered rocks. In the Red Mountain district the sources from the eastern side (red rocks) are partially mitigated by waters from the western side (green rocks).

The natural attenuation reactions are helpful but incomplete. In many basins, the waters leaving the mining area carry concentrations of zinc, and in places cadmium or copper, that are high enough to be threats to fish and the food chain that supports them. Concentrations of Zn in the range of 50 to 200 ppb are found in many streams 1 to 5 mi below mining areas (table 13). The good news is that these Zn concentrations are quite low relative to the upstream sources; the bad news is that this much Zn can stay in the streams for many miles. After the waters have been through the neutralizing reactions, causing various Fe and Al phases to precipitate and adsorb trace metals, not much else happens to decrease metal concentrations. Water-treatment facilities raise the pH to more than 9 to remove these persistent metals, but this does not happen in natural systems. More research is needed on these metal-rich circum-neutral waters, the stability of toxic metals in them, and the geologic processes that could remove those metals.

*Threats from mill tailings.*—Tailings are identified here as important contributors of contamination in many mining areas on the CWS. Also, many tailings impoundments appear to be at risk of catastrophic failure in an extreme storm. Most of the damage from a tailings failure cannot be reversed once the tailings are in a stream because removing them is costly and probably would further damage riparian habitat. Preventative maintenance may be appropriate in many situations. Many tailings impoundments are collapsing after 50 or more years with no maintenance, and those located in flood plains are at risk of collapse in a major storm. Locally focused flash floods are fairly common in Colorado, especially in the summer, and they are capable of moving millions of tons of earth materials—sufficient to close Interstate highways several times in 1999. A case can be made that failure of a tailings impoundment in a

**Table 13.** Summary of persistent zinc concentrations in near-neutral surface waters, central Western Slope, Colorado.

[District abbreviations as in table 1; distance to one or more likely sources; metal analyses by ICP-MS; CZI, copper-zinc index]

Stream	District	Site	Distance (mi)	pH	Zn (ppb)	CZI (ppb)
Elk Creek	RB	558	1.8	7.3	540	6.0
Red Mountain Creek	RM	347	1 to 4?	3.6	360	8.2
Dexter Creek	OU	857	0.6	8.2	360	6.0
Upper Uncompahgre	MP	641	2.5?	6.2	152	0.48
Tomichi Creek	TM	565	1.7	7.2	125	0.43
Canyon Creek	SN	371	2.7	6.2	115	0.32
Howard Fork	IS	637	1-3	6.7	89	0.35
Gray Copper Creek	RM	605	1.5	4.9	33	0.64
Henson Creek	GN	552	2?	7.3	13	0.07

storm event can cause more intensive and extensive damage to a watershed than the more predictable effects of steady acid mine drainage or release of metals from mine waste. Further study of tailings stability in extreme storms is needed to make an appropriate risk analysis.

*Site problems for land management.*—This study has identified some sites that appear to be significant sources of contamination on or very close to public lands. Many of these are tens of acres in size (a mining claim is 10.3 acres) and do not show clearly on watershed-scale maps. At watershed or 1:24,000 scales, property lines of patented claims (private property) relative to public lands administered by USFS and USBLM are important but difficult to identify without a cadastral survey. Many of the problem sites, such as the Standard mine, the Vernon mine, and the Akron mill site, appear to involve mixed ownership: the mine workings tend to be on patented lands, whereas the mine dumps or mill tailings may lie on public lands. Also, there are questions of impacts of contaminated waters that flow from patented claims onto public lands. A prime example is the Red Mountain district, which has about a thousand mine patents. The relations of those private lands to slivers of public lands between patents or relative to lands of the Uncompahgre Forest to the north is complex.

Most of my observations and new geochemical data on the CWS suggest that the land-management problems involve water and the need to control the contact of surface waters with mine waste (dump or tailings) to minimize reactions at these sources. The general concepts are fairly simple, but the technical details and legal issues can be complex and require skillful coordination in land reclamation. Few, if any, of the waste materials are so terrible in composition or in location that they require excavation and placement in more safely engineered repositories. However, many reclamation projects could be done more effectively, with greatly reduced future risk, by moving mine waste or mill tailings to safer sites. The major obstacles to this reclamation activity are not geology or engineering but complex legal and liability issues that should be streamlined to encourage cooperation among private owners, concerned citizens, FLMA's, and regulatory agencies.

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- Manuscript approved for publication January 14, 2002  
Published in the Central Region, Denver, Colorado  
Editing, layout, photocomposition—Richard W. Scott, Jr.  
Graphics by the author and Gayle A. Dumonceaux  
Cover and jewel-case design, graphics consultation—  
Carol A. Quesenberry

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## **Appendix I: Water-Sampling and Analysis Methods**

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Water samples were collected from mines and streams where the water was deemed to be representative of a geologic or mine setting and would yield information on the mobility of metals in that environment. The sampling methods have been used by the author since 1994 for reconnaissance characterization of geologic units, rock alteration, mine drainage, and reactions with tailings or dump materials (Nash and others, 1996); these methods are similar to but simplified from those described by Ficklin and Mosier (1999). First, flow rate and water characteristics (color, suspended material, bed colors, or mineralogy) are recorded, and pH and conductivity are measured using portable instruments. The pocket-sized conductivity meter (Corning CD-55), with an upper limit of 2,000  $\mu\text{S}/\text{cm}$ , responded consistently and showed no drift after calibration. The pH meter (Orion 250), with built-in temperature electrode, required frequent calibration during the day, and at most sites the calibration was checked on a standard solution after the field measurement. The field standards were buffered solutions of pH 4.0, 7.0, and 10.0. Based on experience with portable pH meters applied to diverse water compositions, I consider the measurements of pH to carry an uncertainty of about  $\pm 0.05$  standard unit, even though the meter reports to 0.01 units. The water sample for analysis was collected with a disposable 60-mL syringe, then pushed through a disposable 0.45- $\mu\text{m}$  cellulose filter. The syringe and the 60- or 120-mL polyethylene bottle were rinsed twice in the sampled water prior to collection. The filtered sample was acidified to a pH of about 2 at the site with 5 drops of ultrapure 1:1  $\text{HNO}_3$  per 60 mL. The acid stabilizes metals for periods of more than 2 years without the use of refrigeration, based on replicate analyses over a 2-year period. At appropriate localities (with pH  $>4$ ), an unfiltered sample was collected for determination of alkalinity. Laboratory and field blank tests using deionized water, indicated that contamination introduced by the sampling procedure and equipment is in the low parts per billion level (1–10 ppb), which is considered adequate in the search for metal concentrations orders of magnitude greater than the sampling error.

The filtered and acidified water samples collected in 1997 and 1998 were analyzed by a commercial laboratory in Denver, Colo., within 20 days of collection, using ICP-MS (inductively coupled plasma–mass spectrometry); the general method is described by Lamothe and others (1999). Data for more than 60 elements are reported, generally to levels below 1 ppb (table 14), but in detail the limit of determination varies with sample composition and element mass. For many samples, the actual limit of determination is lower than that shown in table 14. In this report, only 10 to 15 elements of prime environmental concern are tabulated and discussed. This analytical method is appropriate for determination of many metals of potential concern that are expected to have a wide range in concentrations; the method is especially good for trace metals in water such as As, Cu, Pb, and Zn (Crock and others, 1999). If water compositions reported here are a concern or are in question, follow-up work should be done using trace-metal sampling protocols and more precise analytical methods.

Leachate solutions from laboratory tests were analyzed

by the commercial laboratory using the same ICP-MS methods just described. The analytical errors are judged to be similar to those for surface-water samples.

A small number of samples collected in 1999 were analyzed by A.L. Meier of the USGS in the Lakewood, Colo., ICP-MS laboratory, using methods generally similar to those described above. The 1999 ICP-MS results should be distinguished from the 1997–98 results because there are differences in detail; the 1999 analyses are done by research protocol and have better precision and accuracy, and Ag values are reliable.

Experience with the laboratory from 1996 to 1999, analysis of replicate samples (repeat analyses of water from the same bottle or duplicate bottles collected at the same time), and of reference standards allows the following generalized comments on precision (reproducibility) and accuracy (approximation of the correct value).

1. Results are reported for five or six orders of magnitude for most elements of interest in these natural waters and leachates, from less than 0.1 ppb (part per billion

**Table 14.** Elements and lower limits of determination for analyses of water and leachate samples by ICP-MS.

[Elements are listed by atomic weight, and all determinations are in parts per billion; the lower limit of determination shown here is rounded by Nash and is somewhat higher than reported by the commercial laboratory for the 1997 and 1998 analyses]

Element	Lower limit of determination	Element	Lower limit of determination
Li	0.1	Nb	0.01
Be	0.5	Mo	0.04
Na	0.8	Ag	0.01*
Mg	0.5	Cd	0.01
Al	0.4	Sn	0.01
Si	1	Sb	0.04
K	1	Te	0.2
Ca	1	Cs	0.01
Sc	0.1	Ba	0.01
Ti	0.04	La	0.01
V	0.04	Ce	0.01
Cr	0.04	Eu	0.01
Mn	0.04	Yb	0.01
Fe	0.3	Ta	0.01
Co	0.01	W	0.01
Ni	0.01	Re	0.01
Cu	0.01	Os	0.01
Zn	0.01	Pt	0.01
Ga	0.01	Au	0.01
As	0.04	Hg	**
Se	0.04	Tl	0.01
Br	1	Pb	0.02
Rb	0.01	Bi	0.01
Sr	0.01	Th	0.01
Y	0.01	U	0.01

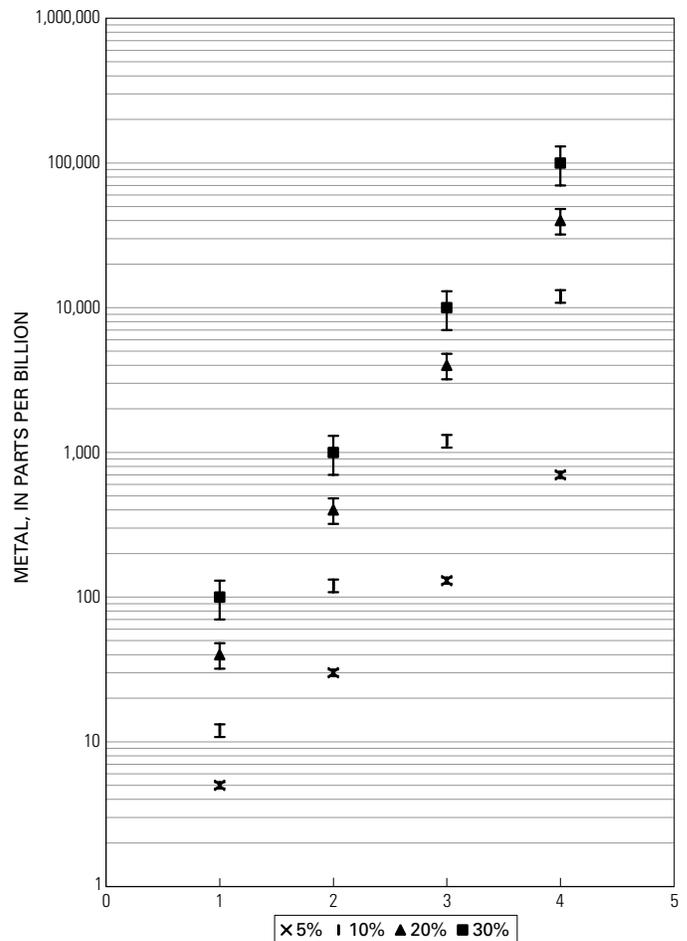
\* unreliable.

\*\* not appropriate.

or microgram per liter) to 100,000 to more than 1,000,000 ppb in some unusual waters. This is a difficult challenge for any method or chemist. These samples often required dilution by amounts of 1:10 to 1:100 to keep the analyte within the calibration range.

2. Precision and accuracy are complex issues and are not the same for all elements and all concentration levels. I assume that analytical error is greater at the very high concentration levels of some of my unusual mine or experimental waters because there are no standards at this level, but documentation of that error is difficult and is incomplete at this date. Replicate samples suggest that precision at very high concentration levels is not much worse than at normal concentrations. Because reliable standards do not exist for waters with very high “trace” metal concentrations (>10,000 ppb), it is not possible to evaluate accuracy at high levels the way it is done at more normal levels. For reasons that I cannot explain, the precision and accuracy for “major” elements (Al, Ca, Mg, etc.) by this method are much lower than for trace metals. The user of these results should consider them semiquantitative with errors of about 50 percent in some cases (but less in others). Another method, ICP-AES, provides more reliable results for major elements (Crock and others, 1999).
3. Based on standards and replicate samples, the precision for trace metals (such as As, Cd, Co, Cr, Cu, Mn, Mo, Ni, Pb, Sb, Se, V, and Zn) appears to be about 10 percent (1 standard deviation) for normal water concentrations, and possibly 20 percent at very high concentrations. The accuracy of determinations on these trace metals seems to be about 10 percent for normal concentrations. The precision and accuracy for Fe, which generally is not a trace metal in these waters, is not as good as most others, and possibly is about 20 percent.
4. Some trace metals pose special problems. Mercury is determined by ICP-MS, but it is well known that Hg is not stable unless preserved by special methods; thus, for my samples the Hg originally present was not in the analyte (the analysis is valid, but the sample is not). Silver should be high in many of my samples, but analytical results are erratic and nearly always very low for unknown reasons (despite good results for neighboring masses); therefore, results for Ag are not considered reliable. Results for Se, Te, and Tl appear to be reasonable, but the standards used did not contain these elements; precision is about the same as for other trace elements.
5. These ICP-MS results are deemed satisfactory for the objectives of this study: classification or ranking of water compositions. The protocol used for 4 years in Arizona, Nevada, and Colorado AML studies (sampling, storage, analysis, and quality control) is reproducible from day to day, year to year, at a total error of about 20 percent for most trace elements. Replicate analyses show that the composition of these acidified water samples change

by less than a few percent after a year of storage at room temperature (D. Fey, oral commun., 1999). Many replicate samples have an error of about 10 percent, and a few have a higher error. These error numbers are best understood by examination of figure 28. The error bars show that an analysis for a metal with a concentration of about 10,000 ppb is reliably different from one having 1,000 ppb or 100 ppb. This is the real goal of these studies, characterization of sites with high to very high metal concentrations, not testing for Zn at 220 vs. 240 ppb to determine compliance with regulatory standards. For sample compositions within the error bar of 20 percent, relative to a standard of concern, further sampling and analysis is recommended.



**Figure 28.** Diagram of analytical error for various metal concentrations. This diagram shows graphically the magnitude of 5, 10, 20, and 30 percent error bars over an wide range of hypothetical metal compositions. Many of the trace metal determinations (for example, Cu, Pb, Zn) have error bars like those shown for 10 percent example, a few (Cd) may be closer to the 5 percent example, and some (e.g., Fe) are like the 20 percent example. This plot shows that, even with high analytical error (20 or 30 percent), the reported values for about 10,000 ppb do not overlap those of a sample with 1,000 or 100,000. Thus the ICP-MS results are appropriate for ranking of water compositions spanning 4 to 6 orders of magnitude (1 to 1,000,000 ppb).

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## **Appendix 2: Tables 1–3**

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**Table 1.** Description and locations of sample sites and media, CWS, Colorado.

[Sample ID: the third character specifies the medium at the site: C, crust; D, dump; G, slag; R, rock; S, soil; T, tailings; W, water; X, other. Analysis: LC, leachate chemistry by ICP-MS; TC, total chemistry (solids) by ICP-AES; WC, water chemistry by ICP-MS. District: CA, Cashin; CO, Cochetopa; CO-V, Vulcan sub-district; GB, Gold Brick; GN, Galena; IS, Iron Spring; LO, Los Ochos; MP, Mineral Point; OU, Ouray; QC, Quartz, Creek; RB, Ruby; RM, Red Mountain; TE, Telluride; TM, Tomichi; TC, Tincup; UV, Uravan. Deposit type: BP, Breccia pipe; FE, iron deposit (ferricrete or magnetite); PMV, polymetallic vein; PMVR, polymetallic vein and replacement; SU, sandstone-type uranium; TV, Tertiary volcanic-hosted vein; UV, uranium vein; VMS, volcanogenic massive sulfide. Description abbreviations: cps, counts per second radioactivity; na, not appropriate; SS, sandstone, Ls, limestone; FeOx, iron oxide minerals;ppt, precipitate]

Site no.	Sample ID	pH	Conductivity ( $\mu$ S/cm)	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG101	NGD101	na	na	38.2211	108.7161	TC, LC	UV	SU	Uranium mine, stockpile?	Sandstone, py, black U, 2500 cps
NG101B	NGD101B	na	na	38.2211	108.7161	TC	UV	SU	Uranium mine dump	Sandstone, typical dump, 1800 cps
NG102	NGD102	na	na	38.2194	108.7186	TC, LC	UV	SU	Thunderbolt mine dump	Sandstone, typical dump, 1500 cps
NG103	NGD103	na	na	38.3117	108.9489	LC	CA	SC	Cashion mine main dump	Sandstone, Cu and Fe oxides
NG104	NGD104	na	na	38.3108	108.9494	TC, LC	CA	SC	Cashion, south small adit	Sandstone, Cu and Fe oxides
NG105	NGW105	8.5	440	38.3361	109.0246	WC	UV	SU	Lion Creek below U mines	Stream, clear, no ppt
NG106	NGW106	8.3	310	38.3364	109.0151	WC	UV	SU	Uranium mine adit water	Adit drainage, no ppt
NG107	NGW107	8.5	410	38.3369	107.0252	WC	UV	SU	Uranium mine adit water	Adit drainage, no ppt
NG108	NGD108	na	na	38.0592	108.9256	TC, LC	UV	SU	Cougar mine dump	Tan SS, black and yellow U
NG109	NGD109	na	na	38.0503	108.9033	TC, LC	UV	SU	Uranium mine dump	Shale and SS, 1500 cps
NG110	NGD110	na	na	38.0444	108.8906	TC, LC	UV	SU	Uranium mine dump	Tan sandstone, 3500 cps
NG111	NGD111	na	na	38.0797	108.8222	TC, LC	UV	SU	West Sunday mine dump	Tan sandstone, 4000 cps
NG112	NGD112	na	na	38.2119	108.7669	TC, LC	UV	SU	Uranium mine dump, reclaimed?	Tan sandstone, 1800 cps
NG113	NGD113	na	na	38.2411	108.7878	TC, LC	UV	SU	Uranium mine dump, reclaimed?	Tan sandstone, 2000 cps
NG114	NGD114	na	na	38.2294	108.7469	TC, LC	UV	SU	Uranium mine dump, reclaimed?	Tan sandstone, 2000 cps
NG115	NGW115	7.4	1,740	38.3089	108.1263	WC	UV	SU	Seep into mine dump	Spring, low flow
NG116	NGD116	na	na	38.3083	108.7383	TC, LC	UV	SU	Uranium mine dump	Tan sandstone, 1200 cps
NG117	NGW117	8.1	840	38.3908	108.0112	WC	UV	SU	Atkinson Creek	Stream, clear, no ppt
NG118	NGD118	na	na	38.4058	108.8003	TC, LC	UV	SU	Golden Cycle mine dump, reclaimed	Tan sandstone, black U, 1500 cps
NG119	NGD119	na	na	38.4058	108.8003	TC	UV	SU	Golden Cycle, reclaimed	Picks with visible black U
NG200	NGG200	na	na	38.5169	106.4233	TC, LC	TM	PMR	Tomichi smelter	Black glassy slag
NG201	NGT201	na	na	38.5378	106.3956	TC	TM	PMV	Akron mill tailings, upper	Reddish fine sand
NG203	NGT203	na	na	38.5378	106.3956	TC, LC	TM	PMV	Akron mill tailings, south	Ocher fine sand
NG204	NGT204	na	na	38.5375	106.3964	LC	TM	PMR	Akron tailings, south	White crust on ocher tails
NG205	NGW205	7.6	410	38.5372	106.1348	WC	TM	PMVR	Akron tailings pond	Small pond on south impound
NG206	NGD206	na	na	38.5428	106.3814	TC, LC	TM	PMVR	Erie tunnel dump	Mixed Qtzite-LS, oxide-sulfide
NG207	NGD207	na	na	38.5319	106.3842	TC, LC	TM	PMVR	Victor mine dump	Mostly LS, low sulfide-oxide
NG209	NGD209	na	na	38.5378	106.3831	TC, LC	TM	PMVR	Northstar(?) mine dump	Mixed Qtzite-LS, moderate FeOx
NG212	NGT212	na	na	38.5378	106.3958	TC, LC	TM	PMV	Akron mill north tailing mound	Reddish fine sand
NG213	NGD213	na	na	38.5394	106.3964	TC, LC	TM	PMVR	Akron mine dump at mill	Mixed rocks, mixed oxide-sulfide
NG214	NGW214	7.9	180	38.5392	106.0156	WC	TM	PMVR	Akron adit at portal	Adit drainage, clear, no ppt.
NG216	NGT216	na	na	38.5247	106.4122	TC	TM	PMV	Tailings in Tomichi Creek	Tan fine sands in willows
NG217	NGD217	na	na	38.3414	107.1019	TC, LC	CO	VMS	Anaconda mine dump	Ocher oxides in greenstone
NG219	NGT219	na	na	38.3194	107.0942	TC	CO	VMS	Lot mill tailings	Gray to reddish fine sand
NG220	NGD220	na	na	38.3203	107.0947	TC	CO	VMS	Lot mine dump	Greenstone with low sulfides
NG221	NGD221	na	na	38.3217	107.0872	TC	CO	VMS	Little Johnny mine dump	Thorium vein in greenstone, 3000 cps

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG222	NGR222	na	na	38.3211	107.0881	TC	CO	VMS	Alt. Rock along Th vein	Red alteration along vein, 4000 cps
NG223	NGC223	na	na	38.3461	107.0839	LC	CO	VMS	Copper King dump, crust	White crust on oxide dump
NG223	NGD223	na	na	38.3461	107.0839	TC, LC	CO	VMS	Copper King dump	Schist, mixed sulfides-oxides
NG224	NGD224	na	na	38.3433	107.0072	TC, LC	CO-V	VMS	Good Hope mine dump, dozed	Hi pyrite in rusty schist
NG225	NGC225	na	na	38.3433	107.0072	LC	CO-V	VMS	Good Hope, crust on dump	White crust on sulfidic dump
NG226	NGD226	na	na	38.3428	107.0050	TC, LC	CO-V	VMS	Vulcan mine dump	Very high pyrite, high clay
NG227	NGC227	na	na	38.3400	107.0047	LC	CO-V	VMS	Vulcan dump crust	White crust on sulfidic dump
NG228C	NGC228	na	na	38.3442	107.0042	LC	CO-V	VMS	Vulcan dump crust	White crust in dump seep
NG229	NGD229	na	na	38.3444	107.0036	LC	CO-V	VMS	Ditch E of Vulcan mine	Pyrite sand in drain ditch
NG230	NGW230	7.8	280	38.3469	107.0051	WC	CO-V	VMS	Camp Creek	Stream, clear, no mine input
NG231	NGD231	na	na	38.3481	0.0053	LC	CO-V	VMS	Transported waste, crust	White crust in kill zone
NG231	NGW231	7.7	310	38.3481	107.0067	WC	CO-V	VMS	Camp Creek	Stream, clear, below mine
NG232	NGW232	7.9	240	38.6128	106.0188	WC	GB	PMV	Carter mine	Adit drainage, clear, no ppt.
NG235	NGD235	na	na	38.6308	106.5672	TC, LC	GB	PMV	Sacramento mine dump	Gneiss with some FeOx
NG236	NGD236	na	na	38.6317	106.5703	TC	GB	PMV	Dump from adit, no name	Gneiss with low % FeOx
NG237	NGT237	na	na	38.6297	106.5811	TC	GB	PMV	Gold Link mill tailings	Ocher to tan fine sand
NG238	NGT238	na	na	38.6164	106.5892	TC, LC	GB	PMV	Raymond mill tailings	Light colored fine sand
NG239	NGT239	na	na	38.6111	106.5972	TC, LC	GB	PMV	Carter mill tailings	Light colored, some erosion
NG240	NGD240	na	na	38.7197	106.4889	TC, LC	TC	PMR	Duncan hill, small adit/dump	Carbonate rocks, no sulfide
NG241	NGD241	na	na	38.7186	106.4789	TC, LC	TC	PMR	Gold Cup mine shafts	Carbonate rocks, vuggy FeOx
NG242	NGD242	na	na	38.7186	106.4789	TC, LC	TC	PMR	Gold Cup mine shafts	Picks rich in FeOx
NG243	NGD243	na	na	38.7125	106.4794	TC	TC	PMR	Tincup mine dump	Sulfidic, galena, low oxides
NG244	NGD244	na	na	38.7125	106.4794	TC	TC	PMR	Tincup mine dump	Picks FeOx-jarosite
NG245	NGD245	na	na	38.7125	106.4794	TC	TC	PMR	Jimmy Mack shaft	Carbonate rx, high FeOx
NG246	NGD246	na	na	38.7242	106.4703	TC	TC	PMR	New Gold Cup mine adit	Picks of sparse FeOx
NG247	NGW247	7.4	240	38.7242	106.0061	WC	TC	PMV	New Gold Cup mine	Adit drainage, clear, no ppt.
NG249	NGW249	7.7	90	38.3781	106.0111	WC	MP	VU	Spring for stock	Spring, clear
NG250	NGD250	na	na	38.3700	106.7467	TC, LC	LO	VU	T1 dump, large, lowest adit	Pyritic sandstone, 2500 cps
NG251	NGD251	na	na	38.3694	106.7467	TC, LC	LO	VU	T1 dump, fresh piles	Pyritic sandstone, 3500 cps
NG252	NGD252	na	na	38.3692	106.7456	TC, LC	LO	VU	T1 dump, east, trestle	Pyritic sandstone, 3000 cps
NG253	NGD253	na	na	38.3689	106.7492	TC, LC	LO	VU	Upper T1 portal	Limonitic sandstone, 2500 cps
NG255	NGD255	na	na	38.3708	106.7336	TC, LC	LO	VU	T2 dump, eastern-most, adit	Pyritic sandstone, 2700 cps
NG256	NGD256	na	na	38.3697	106.7397	TC, LC	LO	VU	T2 dump, adit bears east	Pyritic sandstone, 2000 cps
NG259	NGW259	7.9	680	38.3764	106.0074	WC	LO	VU	Spring for stock	Spring, clear
NG261	NGD261	na	na	38.3978	106.7489	TC, LC	CO	VMS	Alaska shaft dump	High Feox, low py, white crust
NG262	NGD262	na	na	38.3961	106.7511	TC, LC	CO	VMS	Yukon mine dump	Soft, high FeOx, white crust
NG263	NGW263	3.7	210	38.6831	106.1422	WC	QC	PMV	Bon Ton mine, below dump	Adit drainage, makes red ppt
NG264	NGW264	5.7	120	38.6831	106.1377	WC	QC	PMV	Bon Ton mine, at portal	Adit drainage, makes red ppt
NG265	NGD265	na	na	38.6831	106.4811	TC, LC	QC	PMV	Bon Ton mine dump	Gneiss, ocher, sulfidic
NG266	NGS266	na	na	38.6833	106.4828	SLC	QC	PMV	Soil SW of Bon Ton adit	Red soil is pre-mining, under dump
NG269	NGR269	na	na	38.6883	106.4869	TC	QC	PMV	Cumberland pass prospects	Gneiss, veined, FeOx joints
NG270	NGD270	na	na	38.6881	106.4869	TC, LC	QC	PMV	Ida May dump	Qtz-py-ser alt gneiss, Mo?
NG271	NGR271	na	na	38.6872	106.4917	TC	QC	PMV	West of Cumberland Pass	Qtz-Mo veinlets?, ser alt gneiss

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG272	NGT272	na	na	37.9750	107.7222	TC, LC	TC	PMV	Blistered Horn mill	Dark gray to reddish sand
NG274	NGD274	na	na	38.7044	106.4908	TC, LC	TC	PMR	Blistered Horn upper adit dump	Mixed sulfide and oxide, LS
NG275	NGW275	7.6	280	38.7044	106.0099	WC	TC	PMV	Blistered Horn mine	Adit drainage, clear, no ppt.
NG277	NGD277	na	na	38.6958	106.4817	TC	TC	SK	Cumberland mine	Dark dump, biotite-rich, pyritic
NG278	NGD278	na	na	38.6969	106.4819	TC	TC	SK	Cumberland mine	Skarn? Marble with magnetite
NG280	NGD280	na	na	38.6808	106.4983	TC, LC	TC	SK	Mammoth mine, upper level	Mixed sulfide and oxide, gneiss
NG283	NGT283	na	na	38.0172	107.3819	TC, LC	GN	TV	Tailings impound W of Ute mine	Gray fine sand, large area
NG284	NGT284	na	na	38.0217	107.3836	TC, LC	GN	TV	Tailings impound W of Ute mine	Gray and ocher tails
NG287	NGW287	7.4	260	37.9206	107.0209	WC	BP	TV	American Basin	Adit drainage, clear, low FeOx
NG288	NGD288	na	na	37.9206	107.5178	TC, LC	BP	TV	American Basin	Py-rich in gray volcanics
NG296	NGW296	4.7	80	37.9522	107.0849	WC	MP	TV	Headwaters Uncompahgre River	Stream, ponded at mine dump
NG298	NGW298	3.5	200	37.9492	107.3879	WC	MP	TV	London mine adit	Adit drainage, makes FeOx
NG299	NGW299	3.6	410	37.9492	107.0395	WC	RM	TV	Red Mtn Creek	Stream, turbid, red ppt
NG302	NGT302	na	na	38.0622	107.6622	TC, LC	OU	PMR	Bachelor mill tailings	Gray and ocher course sand
NG303	NGD303	na	na	38.0622	107.6617	TC, LC	OU	PMVR	Bachelor(?) mine dump	LS-Shale rx, oxide-sulfide
NG303S	NGD303S	na	na	38.0622	107.6617	TC	OU	PMVR	Bachelor(?) mine dump	Picks of sulfidic pieces
NG305	NGD305	na	na	37.0600	107.6575	TC	OU	PMVR	Almadi mine dump	Dark Mn-rich, oxide-rich
NG307	NGT307	na	na	38.0547	107.6886	TC, LC	OU	PMR	Banner mill tailings	Ocher tails in outwash near 550
NG308	NGT308	na	na	38.0408	107.6825	TC, LC	OU	PMR	Silvershield mill tailings	Yellow-ocher sandy tails
NG309	NGW309	3.4	>2000	38.0539	112.1589	WC	OU	PMV	Seeps from tailings at US550	Low flow, makes red ppt
NG310	NGW310	3.7	570	38.0542	107.0432	WC	IS	FE	Iron Spring mine	Spring in ferricrete mine cut
NG311	NGR311	na	na	37.8583	107.8106	TC, LC	IS	FE	Iron Spring mine	Layered ferricrete in open cut
NG313	NGT313	na	na	37.8561	107.8214	TC, LC	IS	TV	Carbonero mill tailings	Yellow-ocher sandy tails
NG315	NGW315	3.0	1,660	37.8564	110.4192	WC	IS	TV	Carbonero mill tailings area	Water flowing on tailings
NG316	NGT316	na	na	37.8575	107.8492	TC, LC	IS	TV	Caribou mill tailings	Ocher to reddish fine sand
NG317	NGW317	7.3	980	37.8575	107.0256	WC	IS	TV	Caribou mill tailings	Water on tailings
NG318	NGW318	7.1	790	37.8569	107.0335	WC	IS	TV	Caribou mine	Adit drainage, makes red ppt
NG320	NGW320	6.6	290	37.8844	107.3080	WC	IS	TV	Alta mine adit	Adit drainage, makes white ppt
NG321	NGD321	na	na	37.8850	107.8542	TC, LC	IS	TV	Alta mine large upper dump	Mixed ocher oxides, green volc
NG322	NGD322	na	na	37.8778	107.8411	TC, LC	IS	TV	Silver King mine dump	Ocher oxides, some ser-py
NG323	NGT323	na	na	37.8850	107.8556	TC	IS	TV	Alta mill tailings	Tan to med. Brown fine sand
NG325	NGT325	na	na	37.8856	107.8572	TC, LC	IS	TV	Alta mill tails, main west area	Ocher to tan fine sand
NG326	NGR326	na	na	37.9050	107.7056	TC, LC	RM	BP	Alteration at National Belle	Silica alteration, FeOx
NG327	NGD327	na	na	37.9042	107.7039	TC, LC	RM	BP	National Belle mine dump	Silica-sulfide breccia
NG328	NGW328	4.8	250	37.9033	107.0124	WC	RM	TV	National Belle mine area	Stream, has red ppt on bed
NG329	NGR329	na	na	37.9019	107.7011	TC, LC	RM	BP	Talus of red alt. Volcanics	Silica-py-ser alt volcanic rx
NG329S	NGR329S	na	na	37.9019	107.7011	TC	RM	BP	Talus of red alt. Volcanics	Picks of sulfidic pieces
NG330	NGD330	na	na	37.9119	107.7128	TC, LC	RM	BP	Genessee mine dump	Gray and yellow, high py-clay
NG331	NGD331	na	na	37.9131	107.6958	TC, LC	RM	BP	Genessee mine dump	Average, high pyrite rock
NG332	NGD332	na	na	37.9147	107.6967	TC, LC	RM	BP	Shaft mine dump, old	Weathered yellow, hi clay
NG333	NGW333	2.9	1,340	37.9139	107.1132	WC	RM	TV	Champion Gulch creek	Stream near mines, no ppt
NG334	NGR334	na	na	37.9167	107.6939	TC	RM	FE	Ferricrete near Guston mine	Brown goethite cemented talus
NG335	NGW335	2.4	>2000	37.9175	107.7535	WC	RM	TV	Small flow below Guston	Flows on ferricrete, red ppt

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued*.

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG336	NGW336	3.1	>2000	37.9172	107.3791	WC	RM	TV	New adit below Guston	Adit drainage from pipe
NG338	NGW338	2.6	>2000	37.9211	107.2467	WC	RM	TV	Newer mine dump, hi clay	Puddle on dump
NG339	NGD339	na	na	37.9164	107.6917	TC, LC	RM	BP	New mine dump (adit)	Hi py-clay and FeOx
NG340	NGW340	2.7	>2000	37.9164	107.8033	WC	RM	TV	Guston mine adit	Adit drainage, red ppt below
NG341	NGD341	na	na	37.9167	107.6942	TC, LC	RM	BP	Breccia pipe dump	Silica-py-clay alt breccia
NG342	NGW342	3.5	640	37.9181	107.0707	WC	RM	TV	Red Mtn Creek	Stream, much ocher on bed
NG343	NGT343	na	na	37.9183	107.8306	TC	RM	TV	Tailings east of R M Creek	Ocher sands, older tails?
NG344	NGW344	2.9	>2000	37.9189	108.3199	WC	RM	TV	Tailings seep, large impound	Tailings seep, makes red ppt
NG345	NGT345	na	na	37.9192	107.8317	TC, LC	RM	TV	Tailings spill at creek	Pale fine sands
NG347	NGW347	3.6	410	37.9600	107.0508	WC	RM	TV	Red Mtn Creek, north end	Stream, turbid; orange bed
NG348	NGW348	2.9	1,650	37.9372	109.5354	WC	RM	TV	Larson mine at portal	Adit drainage, red ppt
NG350	NGD350	na	na	37.9386	107.6758	TC, LC	RM	TV	Larson mine dump	Yel-ocher, hi py-clay
NG351	NGD351	na	na	37.9278	107.6969	TC, LC	RM	TV	Mountain King mine dump	Mixed ocher and green volcanics
NG352	NGD352	na	na	37.9275	107.6967	TC, LC	RM	TV	Mountain King mine dump	Pick area of green alt volcanics
NG353	NGW353	6.3	740	37.9278	107.0720	WC	RM	TV	Mountain King mine adit	Hi flow, makes ocher ppt
NG355	NGD355	na	na	37.9242	107.6931	TC, LC	RM	BP?	Joker Tunnel dump	Mixed green and red alt volcanics
NG356	NGR356	na	na	37.9242	107.6925	TC, LC	RM	FE	Joker Tunnel dump	Bog iron deposit, mine drainage?
NG357	NGW357	2.9	1,410	37.9242	107.2946	WC	RM	TV	Joker Tunnel drainage	Hi flow, makes red ppt
NG358	NGD358	na	na	37.9244	107.6853	TC, LC	RM	BP	Silver Belle mine dump	Sulfidic, clay altered volcanics
NG361	NGD361	na	na	37.9264	107.6858	TC, LC	RM	BP	Silver Belle shaft dump, big	Mixed alt rocks and hi py-clay
NG363	NGW363	5.9	1,260	37.7442	107.0100	WC	RM	TV	Silver Belle(?) lower adit	Adit drainage, makes red ppt
NG363	NGX363	na	na	37.9275	107.6881	TC	RM	TV	Small adit drainage	FeOx sludge from drainage
NG365	NGW365	3.7	390	37.9250	107.0680	WC	RM	TV	Red Mtn Creek	Stream, makes ocher ppt
NG366	NGD366	na	na	37.9169	107.8069	TC, LC	RM	TV	Midnite mine dump	Variable clay and sulfide content
NG367	NGW367	3.0	290	37.9175	107.0184	WC	RM	TV	Corkscrew Gulch creek	Stream, clear, no ppt
NG368S	NGD368S	na	na	37.9122	107.6728	TC	RM	TV	Carbonate King mine dump	Picks of sulfidic pieces
NG369S	NGD369S	na	na	37.9103	107.6717	TC	RM	TV	W. Carbonate King shaft	Picks sulfides, galena, sphalerite
NG370	NGT370	na	na	37.9394	107.6708	TC, LC	RM	TV	Idarado large tailings impound	Spilled sands, light colored
NG371	NGW371	6.2	190	38.0028	107.0210	WC	SN	TV	Canyon Creek	Stream, clear, no ppt
NG372	NGT372	na	na	37.9750	107.7222	TC, LC	SN	TV	Camp Bird tailings impound	Fine sand like beach sand
NG373	NGW373	6.5	190	37.9747	107.0198	WC	SN	TV	Canyon Creek	Stream, clear, no ppt
NG374	NGS374	na	na	37.9747	107.7214	TC	SN	TV	Canyon Creek alluvium	Stream sediment, below tails
NG375	NGT375	na	na	37.9761	107.7569	TC, LC	SN	TV	Atlas mill tailings	Ocher layer of tailings
NG376	NGT376	na	na	37.9758	107.7572	TC, LC	SN	TV	Atlas mill tailings	Gray layer of tailings
NG377	NGT377	na	na	37.9736	107.7242	TC, LC	SN	TV	Camp Bird tailings impound	Pale tan fine sand
NG379	NGD379	na	na	37.9758	107.6347	TC, LC	MP	TV	Micky Breen mine dump	Mixed oxides and sulfidic rocks
NG380	NGW380	6.9	120	37.9747	107.0207	WC	MP	TV	Upr. Uncompahgre River	Stream
NG381	NGD381	na	na	37.9528	107.6250	TC, LC	MP	TV	Old Lout mine dump	Mixed oxides and sulfidic rocks
NG382	NGW382	6.5	480	37.9531	107.0175	WC	MP	TV	Old Lout mine	Adit drainage, makes tan ppt
NG385	NGW385	7.2	360	37.8661	107.0224	WC	IS	TV	Howard Fork	Stream, turbid, no ppt
NG386	NGT386	na	na	37.9478	107.8717	TC, LC	TE	TV	Society Turn mill tailings	Tan fine sands
NG387	NGT387	na	na	37.9386	107.7556	TC, LC	TE	TV	Tomboy mill tailings in mill	Ocher fine sand, hi clay?
NG388	NGW388	7.3	150	37.9383	107.0915	WC	TE	TV	Savage Creek	Stream, turbid, no ppt

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG389	NGD389	na	na	37.9356	107.7539	TC, LC	TE	TV	Argentine shaft(?) dump	Mostly green alt volcanics
NG390	NGD390	na	na	37.9417	107.7703	TC, LC	TE	TV	Cimaron mine dump	Ocher oxide part of dump
NG391	NGW391	7.2	110	37.9356	107.0226	WC	TE	TV	San Miguel River	Stream, turbid, no ppt
NG393	NGW393	3.3	1,510	37.9736	107.5084	WC	GN	TV	Palmetto basin, small adit	Adit drainage, makes ocher ppt
NG394	NGW394	2.9	1,500	37.9736	107.5237	WC	GN	TV	Palmetto basin, small adit	Adit drainage after dump
NG395	NGD395	na	na	37.9733	107.5758	LC	GN	TV	Palmetto basin, small adit	Yellow py-ser alt volcanics
NG395S	NGD395S	na	na	37.9733	107.5758	TC	GN	TV	Palmetto basin, small adit	Picks of sulfidic pieces
NG450	NGW450	7.1	240	37.9772	107.1677	WC	GN	TV	Head Henson Creek	Stream, clear, no ppt
NG455	NGW455	2.7	nd	37.9600	109.7537	WC	MP	TV	San Juan Chief	Adit drainage, red ppt below
NG492	NGW492	5.9	910	37.9675	107.7211	WC	SN	TV	Canyon Creek	Stream; replicate NGW681
NG493	NGW493	5.9	1,540	37.8508	107.8494	WC	IS	TV	Caribou mine adit	Adit drainage; replicate NGW632
NG494	NGW494	5.9	1,540	37.9006	107.6675	WC	IS	na	Drill hole flow	Replicate of NGW614
NG495	NGW495	2.1	>2000	37.8508	107.8494	WC	RM	TV	Headwall mine adit	Adit drainage, replicate NGW587
NG499	NGW499	3.0	285	38.8680	107.0739	WC	RB	TV	Standard mine adit	Adit drainage, replicate NGW512
NG501	NGG501	na	na	38.3453	107.0058	TC	CO-V	VMS	Vulcan smelter	Black glassy slag
NG504	NGC504	na	na	38.3442	107.0047	LC	CO-V	VMS	Vulcan mine dump	Popcorn-like crust on sulfides
NG505	NGW505	8.2	251	38.3346	107.0014	WC	CO-V	VMS	Cabin Creek, above Vulcan	Stream, spring source?
NG506	NGW506	8.1	354	38.3349	107.0053	WC	CO-V	VMS	Cabin Creek, below Vulcan	Stream, clear, no ppt
NG507	NGD507	na	na	38.8819	107.0706	TC, LC	RB	PMV	Small adit, Elk Basin	Gray SS, FeOx, and sulfides
NG507	NGD507	na	na	38.8811	107.0728	LC	RB	TV	Small adit, Elk Basin	Gray SS, FeOx, and sulfides
NG507	NGW507	3.1	200	38.8681	107.0728	WC	RB	PMV	Mid-Elk Basin, adit	Adit flow, makes ocher ppt
NG509	NGD509	na	na	38.8839	107.0678	TC	RB	PMV	Upper Elk Basin adit	SS with mixed oxides, sulfides
NG510	NGW510	3.1	320	38.8834	107.0683	WC	RB	PMV	Upper Elk Basin adit	Adit drainage, clear
NG511	NGW511	2.9	354	38.8682	107.0714	WC	RB	PMV	Upper Elk Basin	Drainage over dump
NG512	NGW512	3.0	285	38.8680	107.0739	WC	RB	PMV	Standard mine	Adit drainage, makes ocher ppt
NG514	NGW514	3.3	301	38.8681	107.0739	WC	RB	PMV	E Elk Creek, N of mine	Stream, clear
NG515	NGW515	5.8	20	38.8680	107.0742	WC	RB	PMV	W Elk Creek, NW of mine	Stream, clear
NG518	NGW518	3.0	169	38.8679	107.0744	WC	RB	PMV	Standard mine, surface	Drainage over dump
NG519	NGW519	3.1	448	38.8678	107.0744	WC	RB	PMV	Standard mine, tailings pond	Tailings pond water
NG520	NGW520	2.9	765	38.8678	107.0744	WC	RB	PMV	Standard mine, pond seep	Tailings seep, makes red ppt
NG521	NGW521	5.4	62	38.8678	107.0753	WC	RB	PMV	Elk Creek, below mine	Stream, red ppt on bed
NG523	NGW523	5.9	60	38.8672	107.0753	WC	RB	PMV	Elk Creek, below mine	Stream, some red on bed
NG524	NGD524	na	na	38.8792	107.0742	TC, LC	RB	PMV	Standard mine dump	Mixed oxide, high sulfide
NG525	NGD525	na	na	38.8789	107.0744	TC, LC	RB	PMV	Standard mine dump	Mixed oxide, high sulfide
NG526	NGT526	na	na	38.8781	107.0750	TC	RB	PMV	Standard mill tailings	Red-brown sand
NG527	NGT527	na	na	38.8781	107.0750	TC	RB	PMV	Standard mill tailings	Deeper part, red-brown sand
NG527	NGTW527	2.5	1,295	38.8678	107.0750	WC	RB	TV	Standard mill tailings	Tailings pore water
NG528	NGD528	na	na	38.8794	107.0756	TC, LC	RB	PMV	Standard mine dump west	Mixed SS and ocher waste
NG529	NGW529	7.2	97	38.6009	106.5989	WC	GB	PMV	Gold Creek below mines	Stream, clear, no ppt
NG530	NGT530	na	na	38.6083	106.5986	TC, LC	GB	PMV	Carter mill tailings	Pale tan fine sand
NG531	NGW531	6.9	90	38.6170	106.5853	WC	GB	PMV	Gold Creek at Carter Lake	Stream, clear
NG532	NGW532	6.9	56	38.6501	106.5742	WC	GB	PMV	Gold Creek, west of mines	Stream, clear
NG533	NGW533	6.7	40	38.6682	106.4792	WC	QC	PMV	Tomichi Creek, E of Bon Ton	Stream, clear, no ppt

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG534	NGD534	na	na	38.7278	106.4972	TC	TC	PMR	Ore spilled from tram	Brown, gossanous Fe-Mn-rich?
NG534	NGT534	na	na	38.7197	106.4972	TC, LC	TC	PMR	Tincup mill tailings	Brown, coarse sand
NG536	NGT536	na	na	38.7028	106.4917	TC, LC	TC	PMR	Blistered Horn mill tailings	Brown-red sand
NG537	NGW537	6.8	66	38.0167	107.3239	WC	LF	TV	Wade Gulch creek	Stream, clear, no ppt
NG540	NGW540	6.9	96	37.9505	107.3064	WC	LF	TV	Red Mtn Gulch creek	Stream, clear, no ppt
NG541	NGW541	6.6	43	37.9003	107.4264	WC	LF	TV	Lake Fork Gunnison	Stream, clear, no ppt
NG543	NGW543	6.7	46	37.9000	107.4331	WC	LF	TV	Cottonwood Creek	Stream, clear, no ppt
NG544	NGW544	6.8	34	37.9001	107.4342	WC	LF	TV	Cattaract Creek	Stream, clear, no ppt
NG545	NGW545	6.9	44	37.9336	107.4631	WC	BP	TV	Grizzly Creek	Stream, clear, no ppt
NG546	NGW546	7.1	47	37.9340	107.4581	WC	BP	TV	Silver Creek	Stream, clear, no ppt
NG547	NGW547	7.0	46	37.9007	107.3806	WC	LF	TV	Bent Creek	Stream, clear, no ppt
NG548	NGW548	6.8	76	37.9847	107.2958	WC	LF	TV	Lake Fork Gunnison.	Stream, clear, no ppt
NG549	NGW549	6.9	46	38.0007	107.4675	WC	GN	TV	W Fork Henson Creek	Stream, clear, no ppt
NG550	NGW550	7.2	42	37.9676	107.5350	WC	GN	TV	Upper Henson Creek	Stream, clear, no ppt
NG552	NGW552	7.3	46	38.0171	107.3319	WC	GN	TV	Lower Henson Creek	Stream, clear, no ppt
NG555	NGW555	3.0	779	38.8678	107.0753	WC	RB	PMV	Standard tailings pond	Tailings seep; repeat NGW520
NG556	NGD556	na	na	38.8844	107.0708	TC, LC	RB	PMV	Upper Elk Basin trench	Hi py in sandstone
NG556	NGW556	6.6	32	38.8834	107.0708	WC	RB	PMV	Upper Elk Basin trench	Drainage over dump
NG557	NGR557	na	na	38.8689	107.0414	TC	RB	FE	Keystone mine area	Layered ferricrete
NG558	NGW558	7.3	59	38.8507	107.0600	WC	RB	PMV	Lower Elk Creek	Stream, clear, no ppt
NG559	NGW559	7.3	37	38.9008	107.0253	WC	RB	PMV	Slate River	Stream, clear, no ppt
NG560	NGW560	7.3	24	38.9008	107.0467	WC	RB	PMV	Oh-Be-Joyful Creek	Stream, clear, no ppt
NG561	NGW561	7.0	62	38.5349	106.3892	WC	TM	PMV	Tomichi Creek, above Akron	Stream, clear, no ppt
NG562	NGT562	na	na	38.5389	106.3961	TC, LC	TM	PMR	Akron mill tailings north	Red-brown fine sand
NG563	NGW563	7.1	83	38.5178	106.4061	WC	TM	PMV	Tomichi Creek, S of Akron	Stream, clear, no ppt
NG565	NGW565	7.2	81	38.5172	106.4147	WC	TM	PMV	Tomichi Creek	Stream, clear, no ppt
NG570	NGW570	2.7	1,025	38.8678	107.0750	WC	RM	TV	Red Mtn Creek headwaters	Stream, makes red ppt
NG571	NGW571	2.6	1,140	37.9004	107.7031	WC	RM	TV	Small adit into ferricrete	Adit drainage
NG574	NGW574	3.3	436	37.9167	107.6931	WC	RM	TV	Red Mtn Creek	Stream, red ppt on bed
NG575	NGW575	2.9	385	37.9167	107.6919	WC	RM	TV	Champion Gulch	Stream, looks ok but is acid
NG576	NGW576	2.5	>2000	37.9167	107.6919	WC	RM	TV	Guston mine adit	Adit drainage, from pipe
NG577	NGW577	4.2	122	37.9169	107.6981	WC	RM	TV	Prospect trench	Water in trench
NG578	NGW578	2.5	>2000	37.9175	107.6928	WC	RM	TV	Seep from large impound	Tailings seep, makes red ppt
NG580	NGW580	2.9	885	37.9174	107.6928	WC	RM	TV	Joker Tunnel dump	Dump seep relates to adit
NG581	NGW581	3.6	319	37.9172	107.6758	WC	RM	TV	Red Mtn Creek at Joker	Stream, abundant ocher ppt
NG582	NGW582	3.0	565	37.9172	107.6758	WC	RM	TV	Corkscrew Gulch, middle	Stream, looks ok but is acid
NG583	NGW583	3.3	248	37.9008	107.6694	WC	RM	TV	Small crk drains Red Mtn	Stream, looks ok but is acid
NG584	NGW584	2.5	1,035	37.9006	107.6675	WC	RM	TV	Small crk drains red alt rx	Stream, looks ok but is acid
NG585	NGD585	na	na	37.9083	107.6689	TC, LC	RM	TV	Small adit and dump	Pyritic altered, red weathering
NG587	NGW587	2.1	>2000	37.9006	107.6622	WC	RM	TV	Headwall adit, Corkscrew	Adit drainage, very acidic
NG589	NGW589	3.9	39	37.9013	107.6742	WC	RM	TV	Lake at divide, red alteration	Lake, no mines, acid runoff?
NG590	NGD590	na	na	37.9158	107.6700	TC	RM	TV	Dump from adit	Hi sulfide, clay, galena, sphalerite
NG591	NGD591	na	na	37.9128	107.6739	TC, LC	RM	TV	Carbonate King shaft dump	Moderate sulfide, galena, sphal

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG592	NGW592	3.4	154	37.9180	107.6794	WC	RM	TV	Small creek, south fork	Stream, clear, no ppt
NG593	NGW593	3.1	563	37.9008	107.6744	WC	RM	TV	Small creek, few mines	Stream, looks ok but is acid
NG595	NGW595	2.9	492	37.9014	107.6736	WC	RM	TV	Small creek, small mines	Stream, looks ok but is acid
NG598	NGD598			37.9122	107.6739	LC	RM	TV	Carbonate King adit dump	Low sulfide and oxide content
NG599	NGW599	3.2	215	37.9337	107.6675	WC	RM	TV	Small creek, west fork	Stream, looks ok but is acid
NG605	NGW605	4.9	278	37.9183	107.6611	WC	RM	TV	Gray Copper Creek	Stream, abundant FeOx on bed
NG606	NGD606	na	na	37.9328	107.6611	TC, LC	RM	TV	Mountain Silver mine dump	High pyrite, high clay
NG606	NGW606	6.0	375	38.0834	107.6575	WC	RM	TV	Mountain Silver mine adit	Adit drainage, abundant ocher ppt
NG608	NGD608	na	na	37.9358	107.6519	TC	RM	TV	Lost Day mine dump	Abundant ocher and clays
NG609	NGW609	5.9	74	37.9509	107.6617	WC	RM	TV	Upper Mt Silver adit	Adit drainage, ocher ppt
NG610	NGW610	3.4	390	37.8511	107.7981	WC	RM	TV	Red Mtn Creek at US550	Stream, turbid, red ppt
NG612	NGW612	6.2	680	37.8511	107.7986	WC	IS	TV	Chapman Gulch creek	Stream, turbid, tan clay ppt
NG613	NGW613	6.4	620	37.8512	107.8019	WC	IS	TV	Small adit, small dump	Adit drainage, ocher ppt
NG614	NGW614	5.9	1,540	37.8508	107.8494	WC	IS	TV	Flow from drill hole	Spring makes red floc
NG615	NGD615	na	na	37.8697	107.8008	TC	IS	TV	Dump from trench and adit	Clay alt volcanics, low py
NG616	NGW616	5.1	95	37.8508	107.8097	WC	IS	TV	Small creek, drains red rx	Stream, clear, no ppt
NG617	NGR617	na	na	37.8575	107.8097	TC, LC	IS	FE	Trench in iron deposit	Layered ferricrete, hard
NG617	NGW617	3.4	955	37.8507	107.8089	WC	IS	FE	Iron Spring mine	Spring in ferricrete
NG618	NGW618	5.5	365	37.8508	107.8144	WC	IS	TV	E Howard Fork	Stream, looks good
NG619	NGT619	na	na	37.8561	107.8206	TC	IS	TV	Carbinero mill tailings	Yellow to ocher sand
NG619	NGTW619	3.4	1,830	37.8506	107.8206	WC	IS	TV	Carbonero mill tailings	Tailings pore water
NG620	NGTW620	3.0	1,710	37.8506	107.8203	WC	IS	TV	Carbonero mill tailings	Tailings pore water
NG621	NGW621	5.5	890	37.8839	107.9072	WC	IS	TV	Side creek, drains alt rx	Stream, turbid, clay on bed
NG625	NGT625	na	na	37.8617	107.8733	TC, LC	IS	TV	Silver Bell mill tailings	Ocher fine sand
NG625	NGTW625	2.1	>2000	37.8512	107.8733	WC	IS	TV	Caribou mill tailings	Puddle on tailings
NG626	NGT626	na	na	37.8614	107.8731	TC, LC	IS	TV	Silver Bell mill tailings	Gray and yellow fine sand
NG627	NGW627	6.1	330	37.8506	107.8467	WC	IS	TV	S Fork San Miguel	Stream, looks good
NG629	NGTW629	4.5	660	37.8507	107.8492	WC	IS	TV	Caribou mill tailings	Tailings pore water
NG631	NGT631	na	na	37.8567	107.8483	TC, LC	IS	TV	Caribou mill tailings	Rusty to yellow sand
NG631	NGTW631	5.4	nd	37.8507	107.8483	WC	IS	TV	Caribou mill tailings	Tailings pore water
NG632	NGW632	5.9	910	37.8508	107.8494	WC	IS	TV	Caribou mine adit	Adit drainage, tan ppt
NG633	NGW633	5.9	820	37.8508	107.8497	WC	IS	TV	Flow on Caribou tailings	Water on tailings
NG634	NGW634	5.5	348	37.8510	107.8561	WC	IS	TV	Howard Fork	Stream, looks good
NG635	NGW635	6.9	135	37.8343	107.8858	WC	IS	TV	Lake Fork San Miguel	Stream, looks good
NG636	NGW636	4.7	450	37.8514	107.8811	WC	IS	TV	Adit W of Campground	Adit drainage, ugly ocher ppt
NG637	NGW637	6.7	348	37.8514	107.8811	WC	IS	TV	Howard Fork	Stream, looks good
NG639	NGT639	na	na	37.8803	107.8497	TC	IS	TV	Gold King stamp mill	Tan coarse sandy tails
NG641	NGW641	6.2	118	37.9677	107.6375	WC	MP	TV	Upper Uncompahgre River	Stream, looks good
NG642	NGW642	6.5	472	37.9515	107.6250	WC	MP	TV	Micky Breen adit	Adit drainage, some clay ppt
NG644	NGW644	7.0	83	37.9514	107.6269	WC	MP	TV	Mineral Creek	Stream, looks good
NG645	NGW645	6.9	128	37.9510	107.5864	WC	MP	TV	Upper Uncompahgre River	Stream, looks good
NG649	NGT649	na	na	37.9606	107.5925	TC	MP	TV	San Juan Chief mill	Red tailings at mill
NG651	NGW651	3.0	707	37.9176	107.6214	WC	MP	TV	San Juan Chief adit	Adit flow, makes red ppt

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu\text{S}/\text{cm}$ )	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG652	NGW652	3.6	170	37.9014	107.6539	WC	MP	TV	Upper Uncompahgre River	Stream, llooks good
NG654	NGW654	3.1	443	37.9144	107.6806	WC	RM	TV	Small mine adit	Adit drainage, low flow
NG655	NGD655	na	na	37.9122	107.6536	TC	MP	TV	Dump from adit, 1970's?	Very high pyrite, some galena
NG656	NGW656	5.8	202	37.9672	107.7464	WC	na	TV	Uncompahgre River	Stream, leaves tan ppt
NG657	NGW657	5.9	69	37.9516	107.7278	WC	SN	TV	Sneffels Creek	Stream, looks ok
NG658	NGW658	6.9	48	37.9183	107.7486	WC	SN	TV	Imogene Creek	Stream, looks ok
NG659	NGT659	na	na	37.9525	107.7264	TC, LC	SN	TV	Mill tailings at Richmond Jct	Gray sand tailings
NG660	NGT660	na	na	37.9494	107.7297	TC, LC	SN	TV	Hidden Treasure mill	Ocher tailings
NG661	NGD661	na	na	37.9331	107.7456	TC, LC	TE	TV	Small dump E. Savage basin	Rusty to ocher, alt volcanics
NG661	NGT661	na	na	37.9331	107.7456	TC, LC	TE	TV	Stamp mill E Savage basin	Pyritic and rusty coarse sand
NG662	NGW662	6.5	48	37.9339	107.7586	WC	TE	TV	Savage Creek, E of Tomboy	Stream, looks ok
NG663	NGT663	na	na	37.9339	107.7525	TC	TE	TV	Mill tailings SE Tomboy	Gray sand tailings
NG670	NGW670	6.7	90	37.9344	107.7672	WC	TE	TV	Savage Creek below Tomboy	Stream, no ppt
NG671	NGW671	7.3	48	37.9173	107.7706	WC	TE	TV	Marshall Creek	Stream, some tan clay on bed
NG673	NGW673	6.9	53	37.9337	107.8036	WC	TE	TV	Ingrahm Creek below Meldrum	Stream, looks good
NG677	NGW677	7.2	73	37.9675	107.7467	WC	TE	TV	Bear Creek, at San Miguel R	Stream, looks ok
NG678	NGW678	5.6	66	37.9676	107.7472	WC	SN	TV	Sneffels Creek	Stream, repeat NGW657
NG679	NGD679	na	na	37.9764	107.7472	TC, LC	SN	TV	Torpedo Eclipse dump	Picks of rusty, CuOx in bin
NG679	NGW679	6.0	94	37.9681	107.7611	WC	SN	TV	Eclipse mine	Adit drainage, no ppt
NG680	NGW680	6.1	44	37.9675	107.7211	WC	SN	TV	Yankee Boy Creek	Stream, looks good
NG681	NGW681	5.7	121	37.8846	107.7161	WC	SN	TV	Canyon Creek	Stream, looks good
NG682	NGR682	na	na	37.9747	107.7225	LC	SN	TV	Alt. Volcanic rock in roadcut	Green and purple volcanics
NG683	NGW683	7.1	12	37.9346	107.6644	WC	RM	TV	Mineral Creek ditch	Stream, looks good
NG684	NGG684	na	na	37.9461	107.6636	TC	RM	BP	Saratoga smelter	Black glassy slag
NG685	NGW685	6.6	150	38.8846	107.7161	WC	RM	TV	Albany Creek	Stream, no ppt
NG699	NGT699	na	na	37.9339	107.7528	LC	SN	TV	Mill tailings at Tomboy	Gray fine sand
NG801	NGW801	2.9	382	37.9206	107.6567	WC	RM	TV	Vernon mine adit	Adit drainage, clear
NG802	NGW802	2.9	346	37.9192	107.6561	WC	RM	TV	Gray Copper Creek below Vernon	Stream, red ppt on bed
NG804	NGW804	3.2	136	37.9169	107.6550	WC	RM	TV	Gray Copper Creek above Vernon	Stream, some red ppt
NG820	NGW820	2.9	130	37.8956	107.6533	WC	MP	TV	Acidic spring on strike with vein	Killzone, make red ppt
NG823	NGW823	4.6	15	37.9128	107.6539	WC	MP	TV	Small creek drains green alt volcanics	Clear, looks ok
NG827	NGW827	2.8	189	37.9178	107.6558	WC	MP	TV	Vernon mine adit, moderate flow	Starts clear, turns red
NG828	NGW828	3.5	33	37.9206	107.6556	WC	RM	TV	Side creek, drains alt rx	Clear, trace red on bed
NG829	NGW829	3.0	121	37.9194	107.6561	WC	RM	TV	Gray Copper Creek below Vernon mine	Clear, some yellow on bed
NG830	NGW830	3.0	92	37.9172	107.6550	WC	RM	TV	Gray Copper Creek above Vernon mine	Clear, trace red on bed
NG831	NGW831	2.0	>2000	37.9053	107.6675	WC	RM	TV	Headwall prospect E end Corkscrew Gulch	Clear, no ppt from flow
NG834	NGW834	6.8	172	37.8652	107.8670	WC	IS	TV	Howard Fork San Miguel River	Clear, no ppt
NG835	NGW835	7.7	121	37.8722	107.8933	WC	IS	TV	South Fork San Miguel	Clear, looks ok
NG837	NGW837	3.2	312	37.8997	107.7019	WC	RM	TV	Hudson mine drainage, reclaimed	Makes red ppt

**Table 1.** Description and locations of sample sites and media, CWS, Colorado—*Continued.*

Site no.	Sample ID	pH	Conductivity ( $\mu$ S/cm)	Lat (N.) (degrees)	Long (W.) (degrees)	Analysis	District	Deposit type	Feature	Description
NG839	NGW839	3.0	204	37.9000	107.7061	WC	RM	TV	Headwaters Red Mtn Creek, no mines	Clear, some red on bed
NG840	NGW840	2.5	>2000	37.9164	107.6917	WC	RM	TV	Guston mine adit drainage	Clear, no ppt evident
NG841	NGW841	2.9	284	37.9186	107.6969	WC	RM	TV	Middle reach Red Mtn Creek	Turbid; makes orange ppt
NG842	NGW842	2.9	762	37.9233	107.6928	WC	RM	TV	Joker Tunnel drainage	Clear, no ppt evident
NG844	NGW844	5.5	415	37.8608	107.8299	WC	IS	TV	Fe-rich water from adit	Makes thick red floc
NG845	NGW845	3.7	325	37.8600	107.8299	WC	IS	TV	Fe-rich surface water, no mining	Makes thick red floc
NG850	NGW850	7.3	51	38.5169	106.4232	WC	TM	PMVR	Tomichi Creek at smelter	Clear, looks good
NG852	NGW852	7.3	131	38.5394	106.3953	WC	TM	PMVR	Akron mine tunnel drain	Clear, looks good
NG853	NGW853	7.8	50	38.5378	106.3972	WC	GB	PMVR	Tomichi Creek below Akron mill	Clear, looks good
NG856	NGW856	8.7	45	38.6064	106.5994	WC	GB	PMV	Gold Creek east of mills	Clear, looks good
NG857	NGW857	8.2	48	38.0655	107.6718	WC	OU	PMVR	Dexter Creek west of mines	Clear, looks good
NG860	NGW860	6.9	184	38.0483	107.6862	WC	OU	na	Uncompahgre River N of Ouray	Turbid, makes tan ppt
NG862	NGW862	7.7	240	38.1889	107.7472	WC	OU	na	Uncompahgre River N of Ridgeway	Clear, no ppt evident
NG864	NGW864	6.4	42	37.9769	107.7589	WC	SN	TV	Sneffels Creeks above Atlas mill	Clear, looks good
NG865	NGW865	6.8	55	37.9753	107.7461	WC	SN	TV	Sneffels Crk east of Revenue mine	Clear, looks good
NG866	NGW866	7.0	64	37.9658	107.7275	WC	SN	TV	Imogene Creek SW of Camp Bird	Clear, looks good
NG869	NGW869	72.0	51	37.9483	107.7311	WC	SN	TV	Mine drainage Camp Bird No 3 adit	Clear, no ppt evident
NG870	NGW870	5.8	55	37.9500	107.7339	WC	SN	TV	Hidden Treasure adit drainage, red	Turbid, makes red floc
NG871	NGW871	6.6	124	37.9347	107.7528	WC	TE	TV	Surface flow from Argentine dump	Clear, looks good
NG873	NGW873	7.1	190	37.9758	107.7195	WC	SN	TV	Canyon Creek east of Camp Bird	Clear, looks good

**Table 2.** Analytical results for samples of mineralized materials from mines, mills, and outcrops, central Western Slope, Colorado.

[Analyses: Headings with % and ppm, by ICP with "total digestion;" headings with /P (e.g., As/P) are results from partial digestion and organic extraction, with analysis by ICP. (nd, not determined). Districts: CA, Cashin; CO, Cochetopa; GB, Gold Brick; GN, Galena; IS, Iron Spring; LO, Los Ochos; MP, Mineral Point; OU, Ouray; QC, Quartz, Creek; RB, Ruby; RM, Red Mountain; TE, Telluride; TM, Tomichi; TN, Tincup; UV, Uravan. Deposit type: BP, breccia pipe; FE, ferricrete; PMR, polymetallic replacement; PMV, polymetallic vein; SC, sediment-hosted copper; SK, skarn; SU, sandstone uranium; TV, volcanic-hosted vein; VMS, volcanogenic massive sulfide. Sample type: the third letter in sample ID indicates sample type: D, dump; T, tailing; R, rock (unmined); X, other or unusual rock type]

Sample ID	District	Deposit type	Lat (N.) (degrees)	Long (W.) (degrees)	Al (%)	Ca (%)	Fe (%)	K (%)	Mg (%)	Mn (ppm)	Ag (ppm)	As/P (ppm)	Au/P (ppm)	Bi/P (ppm)	Cd/P (ppm)	Cu (ppm)	Hg (ppm)	Mo (ppm)	Pb (ppm)	Sb/P (ppm)	Se (ppm)	U (ppm)	Zn (ppm)
NGD101	UV	SU	38.2211	108.7161	2.40	2.19	0.78	1.12	0.84	137	<.5	17	0.00	0.4	0.8	11	0.146	7	30	1.29	49.9	975	49
NGD101B	UV	SU	38.2211	108.7161	2.30	2.16	0.70	1.10	0.83	130	<.5	18	0.00	0.4	0.8	12	0.192	8	31	1.2	49.1	1005	49
NGD102	UV	SU	38.2194	108.7186	3.00	2.49	0.90	1.54	1.18	220	<.5	14	0.00	0.4	0.3	13	0.124	6	29	1.12	40.6	365	28
NGD103	CA	SC	38.3117	108.9489	2.62	3.07	3.51	2.00	1.78	156	49	246	0.00	0.5	18.1	33326	3.67	5	145	8.6	0	< 10	788
NGD104	CA	SC	38.3108	108.9494	2.62	0.07	0.27	2.07	0.13	9	1.6	160	0.00	0.4	1.3	4351	0.218	<2	18	1.52	0.182	< 10	340
NGD108	UV	SU	38.0592	108.9256	4.75	3.86	1.27	1.96	1.19	373	5.6	20	0.00	0.6	10.2	1403	0.48	23	326	0.915	12.3	495	232
NGD109	UV	SU	38.0503	108.9033	3.47	2.75	1.00	1.33	1.00	237	2	10	0.00	0.6	3.3	409	0.313	24	207	0.672	22.4	370	64
NGD110	UV	SU	38.0444	108.8906	3.14	2.17	0.96	0.95	1.04	214	3.4	27	0.00	3.4	9.2	273	0.326	285	323	1.57	59.2	980	114
NGD111	UV	SU	38.0797	108.8222	3.39	2.16	1.32	1.20	0.94	199	0.8	18	0.00	1.1	2.9	28	0.358	58	53	1.7	98.6	1885	87
NGD112	UV	SU	38.2119	108.7669	2.19	1.70	0.85	1.18	0.82	129	0.8	24	0.00	0.8	0.7	22	0.11	38	57	1.34	15.1	445	48
NGD113	UV	SU	38.2411	108.7878	2.89	1.56	0.75	1.46	0.98	122	0.9	18	0.00	1.2	0.3	16	0.148	60	37	1.29	11.4	490	27
NGD114	UV	SU	38.2294	108.7469	2.00	2.63	0.60	1.18	0.79	142	<.5	6	0.00	0.7	0.3	8	0.098	31	12	0.824	31.4	230	33
NGD116	UV	SU	38.3083	108.7383	2.31	1.14	0.67	1.20	0.65	113	<.5	37	0.00	0.4	0.3	8	0.18	5	50	0.906	67	340	28
NGD118	UV	SU	38.4058	108.8003	3.86	2.52	1.34	1.92	1.19	235	0.5	42	0.00	0.5	1.4	20	0.113	6	238	1.22	73.4	455	177
NGD119	UV	SU	38.4058	108.8003	3.21	0.32	1.09	0.70	1.48	54	1.4	97	0.00	0.2	2.6	69	0.129	7	409	1.39	99	2680	391
NGD206	TM	PMVR	38.5428	106.3814	3.10	6.30	10.00	0.97	1.20	2190	46	20	0.30	31.0	24.3	750	nd	5	11200	5.0	nd	nd	4780
NGD207	TM	PMVR	38.5319	106.3842	5.90	3.10	1.90	2.90	1.90	4940	112	113	0.10	15.0	144.0	770	nd	12	25800	89.0	nd	nd	19800
NGD209	TM	PMVR	38.5378	106.3831	3.30	3.40	11.00	1.00	3.10	4280	91	219	0.60	60.0	135.0	1400	nd	52	43000	130	nd	nd	23500
NGD213	TM	PMVR	38.5394	106.3964	4.50	5.00	7.80	1.40	1.70	2490	54	7	0.20	51.0	76.7	253	nd	17	18500	26.0	nd	nd	12200
NGD217	CO	VMS	38.3414	107.1019	5.20	1.80	10.00	0.65	1.50	513	11	69	0.70	10.0	14.7	3940	nd	5	791	< 1	nd	nd	7040
NGD220	CO	VMS	38.3203	107.0947	5.60	6.00	6.10	3.00	3.60	1530	3	10	0.90	1.0	4.9	276	nd	5	577	3.0	nd	nd	365
NGD221	CO	VMS	38.3217	107.0872	6.80	6.90	6.60	1.70	3.80	1390	<2	< 1	<0.1	< 1	0.2	58	nd	<2	41	< 1	nd	nd	132
NGD223	CO	VMS	38.3461	107.0839	6.40	1.10	12.00	0.36	6.00	1960	2	< 1	0.30	1.0	1.1	9700	nd	<2	22	< 1	nd	nd	394
NGD224	CO	VMS	38.3433	107.0072	5.10	0.54	17.00	1.10	1.50	136	13	54	1.90	4.0	11.0	747	nd	29	113	< 1	nd	nd	2810
NGD226	CO	VMS	38.3428	107.0050	1.70	0.05	21.00	0.58	0.15	14	15	58	1.10	3.0	2.3	208	nd	46	135	1.0	nd	nd	578
NGD235	GB	PMV	38.6308	106.5672	6.00	0.43	5.80	2.60	0.88	3480	76	1000	2.00	1.0	44.6	345	nd	4	8350	80.0	nd	nd	3880
NGD236	GB	PMV	38.6317	106.5703	2.20	0.04	8.80	1.10	0.20	319	58	1410	8.20	1.0	16.9	952	nd	6	16500	66.0	nd	nd	1430
NGD240	TN	PMR	38.7197	106.4889	2.50	3.40	5.90	0.75	1.90	4800	300	55	6.50	< 1	5.2	709	nd	4	4210	286	nd	nd	2150
NGD241	TN	PMR	38.7186	106.4789	1.00	1.10	16.00	0.14	0.45	5860	69	60	0.50	1.0	35.4	3090	nd	24	35000	1360	nd	nd	22100
NGD242	TN	PMR	38.7186	106.4789	0.17	1.60	11.00	<0.005	0.40	2280	93	437	0.30	2.0	20.5	1560	nd	50	26100	916	nd	nd	10200
NGD243	TN	PMR	38.7125	106.4794	0.03	15.00	14.00	<0.005	6.50	5950	23	677	0.60	58.0	9.0	261	nd	21	9990	440	nd	nd	1960
NGD244	TN	PMR	38.7125	106.4794	0.23	0.29	14.00	<0.005	0.21	250	172	1540	1.40	380.0	40.5	1390	nd	256	26700	1550	nd	nd	5010
NGD245	TN	PMR	38.7125	106.4794	0.43	1.70	10.00	0.13	0.27	2480	124	579	1.90	9.0	52.1	5110	nd	17	57000	696	nd	nd	15500
NGD246	TN	PMR	38.7242	106.4703	4.50	0.44	3.60	1.60	0.14	5180	37	130	<0.1	< 1	61.8	364	nd	6	1090	21.0	nd	nd	31300
NGD250	LO	VU	38.3700	106.7467	5.85	0.09	2.15	1.72	0.20	41	<.5	171	0.00	1.1	0.2	19	0.162	41	157	9.97	0	27	50
NGD251	LO	VU	38.3694	106.7467	4.08	0.15	2.04	1.14	0.21	39	<.5	162	0.00	2.1	1.9	29	0.144	107	305	19.9	0.185	140	91
NGD252	LO	VU	38.3692	106.7456	7.29	0.03	2.54	0.89	0.18	17	<.5	319	0.00	0.9	0.2	11	0.227	29	91	4.34	1.44	55	27
NGD253	LO	VU	38.3689	106.7492	6.24	0.15	2.60	0.72	0.20	43	1.3	610	0.00	4.1	6.6	24	0.403	251	1080	83.8	1.15	355	112
NGD255	LO	VU	38.3708	106.7336	1.04	0.02	1.51	0.27	0.04	14	1.3	428	0.00	1.2	0.3	100	0.316	34	161	22.9	4.82	60	81
NGD256	LO	VU	38.3697	106.7397	5.87	0.06	2.16	2.38	0.15	27	<.5	44	0.00	0.9	0.1	19	0.174	14	56	3	0.356	< 10	29
NGD261	CO	VMS	38.3978	106.7489	6.70	0.86	5.80	3.30	0.75	864	11	1580	0.30	8.0	121.0	3390	nd	23	1570	20.0	nd	nd	19000
NGD262	CO	VMS	38.3961	106.7511	5.20	0.52	10.00	2.50	2.10	1010	35	958	2.00	65.0	60.0	10600	nd	32	7570	28.0	nd	nd	18600
NGD265	QC	PMV	38.6831	106.4811	6.10	0.46	3.10	3.30	0.45	96	31	56	0.90	26.0	0.9	159	nd	1190	321	7.0	nd	nd	155
NGD270	QC	PMV	38.6881	106.4869	4.90	0.05	0.92	3.20	0.26	250	73	194	0.50	11.0	43.0	388	nd	312	348	262	nd	nd	4110
NGD274	TN	PMR	38.7044	106.4908	4.80	1.60	3.40	2.80	1.30	2150	86	140	0.40	78.0	20.5	1310	nd	57	3020	58.0	nd	nd	3540
NGD277	TN	SK	38.6958	106.4817	3.40	0.24	4.50	2.70	1.20	394	16	205	<0.1	6.0	3.4	202	nd	12	185	190	nd	nd	321
NGD278	TN	SK	38.6969	106.4819	0.61	0.67	25.00	<0.005	7.40	2870	<2	1	0.80	< 1	1.1	<2	nd	2	56	1.0	nd	nd	228

**Table 2.** Analytical results for samples of mineralized materials from mines, mills, and outcrops, central Western Slope, Colorado—*Continued.*

Sample ID	District	Deposit type	Lat (N.) (degrees)	Long (W.) (degrees)	Al (%)	Ca (%)	Fe (%)	K (%)	Mg (%)	Mn (ppm)	Ag (ppm)	As/P (ppm)	Au/P (ppm)	Bi/P (ppm)	Cd/P (ppm)	Cu (ppm)	Hg (ppm)	Mo (ppm)	Pb (ppm)	Sb/P (ppm)	Se (ppm)	U (ppm)	Zn (ppm)
NGD280	TN	SK	38.6808	106.4983	6.50	0.12	1.10	4.20	0.35	41	17	12	0.20	11.0	3.4	243	nd	273	859	20.0	nd	nd	394
NGD288	BP	TV	37.9206	107.5178	5.40	0.29	4.60	2.80	0.68	554	14	91	0.20	14.0	11.7	494	nd	259	6050	14.0	nd	nd	1920
NGD303	OU	PMVR	38.0622	107.6617	7.70	0.13	3.60	3.20	0.35	3650	170	205	<0.1	5.0	3.3	394	nd	10	2810	196	nd	nd	652
NGD303S	OU	PMVR	38.0622	107.6617	2.00	0.07	1.80	0.80	0.18	874	<2	43	<0.1	<1	0.3	11	nd	4	209	3.0	nd	nd	85
NGD305	OU	PMVR	37.0600	107.6575	2.40	0.14	1.70	0.98	0.06	1360	34	53	<0.1	<1	3.1	100	nd	10	899	11.0	nd	nd	605
NGD321	IS	TV	37.8850	107.8542	4.90	0.72	9.40	2.20	0.41	2650	151	1280	2.70	251.0	4.4	1180	nd	89	4840	147	nd	nd	1070
NGD322	IS	TV	37.8778	107.8411	7.10	0.31	7.20	3.20	0.40	501	150	406	1.20	128.0	11.4	1270	nd	127	20600	486	nd	nd	2020
NGD327	RM	BP	37.9042	107.7039	7.20	0.04	5.40	2.10	0.20	62	103	4760	1.30	72.0	72.4	15200	nd	9	19400	1020	nd	nd	8030
NGD330	RM	BP	37.9119	107.7128	2.80	0.03	16.00	0.22	0.04	14	102	386	0.40	140.0	35.8	5670	nd	3	9770	2620	nd	nd	13900
NGD331	RM	BP	37.9131	107.6958	8.30	0.15	2.20	2.20	0.30	62	51	465	0.50	52.0	9.0	830	nd	5	5190	125	nd	nd	1770
NGD332	RM	BP	37.9147	107.6967	8.70	0.11	3.90	2.60	0.46	148	63	1210	1.90	66.0	9.2	1870	nd	3	13500	97.0	nd	nd	2820
NGD339	RM	BP	37.9164	107.6917	7.70	0.04	8.20	1.90	0.21	40	4	170	0.30	14.0	1.3	183	nd	3	1950	100	nd	nd	353
NGD341	RM	BP	37.9167	107.6942	5.70	0.01	12.00	0.97	0.12	24	422	13	1.40	111.0	16.6	893	nd	10	38600	168	nd	nd	4380
NGD350	RM	TV	37.9386	107.6758	7.80	0.19	3.60	3.00	0.85	603	9	4	<0.1	12.0	25.7	367	nd	32	8100	3.0	nd	nd	6380
NGD351	RM	TV	37.9278	107.6969	11.00	2.60	3.40	4.60	0.31	286	29	7	0.20	50.0	76.5	362	nd	135	8300	<1	nd	nd	14500
NGD352	RM	TV	37.9275	107.6967	9.10	0.88	3.70	3.30	0.88	845	5	<1	<0.1	15.0	7.8	51	nd	18	1520	2.0	nd	nd	1610
NGD355	RM	BP	37.9242	107.6931	12.00	0.07	2.60	2.10	0.39	46	356	43	1.70	100.0	3.6	126	nd	6	11100	143	nd	nd	1100
NGD358	RM	BP	37.9244	107.6853	9.10	0.06	1.50	2.20	0.22	51	19	137	0.10	11.0	2.5	38	nd	<2	1650	26.0	nd	nd	722
NGD361	RM	BP	37.9264	107.6858	6.30	0.03	3.80	1.50	0.15	53	163	1160	2.70	1820.0	3.0	2510	nd	5	1910	957	nd	nd	1190
NGD366	RM	TV	37.9169	107.8069	6.90	0.07	1.20	1.70	0.16	43	59	496	0.10	91.0	6.5	878	nd	5	5190	169	nd	nd	1240
NGD368S	RM	TV	37.9122	107.6728	1.10	<0.005	9.60	0.27	0.02	64	336	8	0.10	138.0	51.5	7970	nd	20	49000	1790	nd	nd	30700
NGD369S	RM	TV	37.9103	107.6717	0.37	<0.005	11.00	0.02	0.01	55	135	44	0.20	881.0	49.2	16700	nd	7	24500	1930	nd	nd	14000
NGD379	RM?	TV	37.9758	107.6347	6.80	0.17	7.20	3.10	0.48	1900	41	35	0.30	15.0	20.1	939	nd	35	13500	29.0	nd	nd	3190
NGD381	MP	TV	37.9528	107.6250	6.90	1.30	5.20	3.40	0.31	530	107	49	0.20	325.0	18.3	1290	nd	45	9160	37.0	nd	nd	2630
NGD389	TE	TV	37.9356	107.7539	6.50	0.30	6.20	2.80	0.50	11700	125	6	1.00	7.0	129.0	1480	nd	75	48600	48.0	nd	nd	33800
NGD390	TE	TV	37.9417	107.7703	7.40	0.39	4.80	3.40	0.57	2990	44	371	0.70	3.0	6.3	103	nd	67	1620	15.0	nd	nd	1110
NGD395S	GN-P	TV	37.9733	107.5758	1.30	0.05	4.70	0.50	0.05	242	231	6	0.90	<1	305.0	1600	nd	38	44900	51.0	nd	nd	66000
NGD507	RB	PMV	38.8819	107.0706	3.86	0.07	10.39	1.60	0.15	1685	98.7	505	0.01	65.9	141.0	1519	0	3	40351	32.8	55.3	nd	23413
NGD509	RB	PMV	38.8839	107.0678	6.50	0.04	11.48	2.48	0.25	1841	70.7	214	0.01	60.7	4.5	716	0.575	3	10384	15.9	76	nd	1648
NGD524	RB	PMV	38.8792	107.0742	4.28	0.15	11.50	1.84	0.22	2199	86.4	243	0.02	62.1	61.9	1337	0.194	3	44065	32.1	55.1	nd	9478
NGD525	RB	PMV	38.8789	107.0744	3.00	0.02	18.06	1.34	0.12	503	136.7	691	0.03	120.0	36.3	1502	0.65	5	44019	52.8	99.2	nd	6622
NGD528	RB	PMV	38.8794	107.0756	5.01	0.04	5.94	2.31	0.22	807	25.9	309	0.02	9.7	3.5	415	0.014	4	8756	14	21.4	nd	979
NGD534	TN	PMR	38.7278	106.4972	0.80	1.24	38.46	0.20	0.36	1204	167.1	5039	5.61	207.0	6.5	2664	0.112	14	2298	147	4.97	nd	2071
NGD556	RB	PMV	38.8844	107.0708	4.44	0.04	7.97	2.02	0.19	707	60.4	198	0.05	0.0	8.6	936	0.92	7	45177	53.5	21.7	nd	2479
NGD585	RM	TV	37.9083	107.6689	8.70	0.09	6.37	1.70	0.13	41	28.8	1584	0.04	26.5	2.2	4611	2.35	25	2954	253	5.12	nd	533
NGD590	RM	TV	37.9158	107.6700	8.41	0.06	3.82	3.48	0.70	711	59.9	272	0.09	33.9	40.1	449	1.33	53	11415	202	4.22	nd	8969
NGD591	RM	TV	37.9128	107.6739	7.44	0.05	6.25	2.13	0.27	109	247.3	1204	0.32	630.0	11.7	1737	6.54	9	6046	512	11.8	nd	2482
NGD606	RM	TV	37.9328	107.6611	6.86	0.20	6.01	3.34	0.27	462	19.9	42	0.04	19.5	3.8	244	0.476	283	932	46.9	4.64	nd	916
NGD608	RM	TV	37.9358	107.6519	7.22	0.02	2.71	3.13	0.30	382	24.7	81	0.05	0.4	1.1	43	0.166	10	26195	27.1	4.24	nd	194
NGD615	IS	TV	37.8697	107.8008	8.26	0.16	4.98	3.18	0.71	193	7.7	23	0.07	18.2	0.2	53	0.269	10	2605	12.1	11.9	nd	55
NGD655	MP	TV	37.9122	107.6536	1.66	0.01	37.92	0.04	<.01	8	8.1	318	0.02	2.5	0.4	9002	0.163	7	112	29.7	13.8	nd	53
NGD661	TE	TV	37.9331	107.7456	2.35	4.32	1.58	1.16	0.13	5989	17.3	43	15.90	1.0	72.7	1128	6.5	33	5505	4.88	0.717	nd	12744
NGD679	SN	TV	37.9764	107.7472	6.82	1.65	4.53	2.61	0.53	1125	7.8	276	0.65	1.9	1.8	28	0.75	75	482	9.8	1.32	nd	422
NGG200	TM	PMR	38.5169	106.4233	1.60	13.00	12.00	0.36	1.90	2200	31	5	0.30	<1	1.1	358	nd	13	7270	16.0	nd	nd	11400
NGG501	CO	VMS	38.3453	107.0058	3.95	9.94	20.02	0.82	0.86	105	1.8	4	0.21	0.0	0.2	1805	0	35	<5	0.222	3.36	nd	10372
NGG684	RM	BP	37.9461	107.6636	2.25	13.61	22.83	0.46	0.64	4483	24	12	0.11	3.2	0.5	910	0.016	63	350	11.7	1.04	nd	4522
NGL266	QC	PMV	38.6833	106.4828	6.10	0.36	6.80	2.50	0.50	404	6	2	0.30	21.0	0.5	931	nd	499	238	7.0	nd	nd	140
NGR222	CO	VMS	38.3211	107.0881	5.40	1.20	6.00	3.00	0.30	160	<2	292	0.60	2.0	0.2	24	nd	44	419	4.0	nd	nd	242
NGR269	QC	PMV	38.6883	106.4869	3.70	0.01	2.90	2.20	0.19	50	30	243	0.20	80.0	0.3	438	nd	2190	679	175	nd	nd	32
NGR271	QC	PMV	38.6872	106.4917	6.00	0.05	2.30	4.30	0.38	295	16	37	<0.1	24.0	0.7	581	nd	870	2060	45.0	nd	nd	182
NGR311	IS	FE	37.8583	107.8106	1.50	0.01	24.00	0.11	0.04	23	<2	5	1.60	2.0	1.1	1160	nd	4	99	2.0	nd	nd	200

**Table 2.** Analytical results for samples of mineralized materials from mines, mills, and outcrops, central Western Slope, Colorado—*Continued.*

Sample ID	District	Deposit type	Lat (N.) (degrees)	Long (W.) (degrees)	Al (%)	Ca (%)	Fe (%)	K (%)	Mg (%)	Mn (ppm)	Ag (ppm)	As/P (ppm)	Au/P (ppm)	Bi/P (ppm)	Cd/P (ppm)	Cu (ppm)	Hg (ppm)	Mo (ppm)	Pb (ppm)	Sb/P (ppm)	Se (ppm)	U (ppm)	Zn (ppm)
NGR326	RM	BP	37.9050	107.7056	4.50	0.08	5.80	0.45	0.13	83	52	2880	4.00	118.0	1.8	152	nd	8	2330	482	nd	nd	602
NGR329	RM	BP	37.9019	107.7011	6.50	0.11	4.70	0.82	0.23	80	24	12	1.00	34.0	0.6	512	nd	6	1190	85.0	nd	nd	237
NGR329S	RM	BP	37.9019	107.7011	3.20	0.04	4.00	0.04	0.01	13	5	59	0.60	7.0	0.2	320	nd	< 2	630	12.0	nd	nd	46
NGR334	RM	FE	37.9167	107.6939	0.98	0.03	24.00	0.26	0.05	18	< 2	96	1.80	4.0	0.5	370	nd	< 2	313	5.0	nd	nd	56
NGR356	RM	FE	37.9242	107.6925	0.16	0.01	24.00	0.04	0.02	13	< 2	192	1.00	2.0	0.3	21	nd	< 2	135	8.0	nd	nd	31
NGR557	RB	FE	38.8689	107.0414	0.73	0.01	56.01	0.18	0.04	141	< .5	9	0.01	0.4	0.8	24	0.23	< 2	56	2.92	1	nd	587
NGR617	IS	FE	37.8575	107.8097	1.39	0.01	55.23	0.11	0.03	17	< .5	4	0.00	0.3	0.5	1145	0.104	10	74	2.33	0.644	nd	180
NGS374	SN	TV	37.9747	107.7214	6.40	1.80	5.00	2.20	0.91	4370	19	47	0.60	13.0	13.1	464	nd	20	1370	11.0	nd	nd	2590
NGT201	TM	PMV	38.5378	106.3956	2.40	9.70	7.30	0.81	4.20	15800	24	13	0.20	20.0	85.0	324	nd	4	6790	18.0	nd	nd	11800
NGT203	TM	PMV	38.5378	106.3956	1.90	7.70	13.00	0.65	1.00	3130	27	79	0.50	30.0	25.6	221	nd	< 2	4970	20.0	nd	nd	4730
NGT212	TM	PMV	38.5378	106.3958	2.20	6.90	10.00	0.69	2.40	8410	25	38	0.10	25.0	49.4	303	nd	5	7460	13.0	nd	nd	7410
NGT216	TM	PMV	38.5247	106.4122	5.40	2.40	4.60	1.70	2.10	4130	175	147	0.30	88.0	36.9	3210	nd	42	28800	233	nd	nd	8790
NGT219	CO	VMS	38.3194	107.0942	6.00	6.00	4.80	1.70	2.30	1240	< 2	13	0.50	< 1	1.2	123	nd	< 2	259	6.0	nd	nd	173
NGT237	GB	PMV	38.6297	106.5811	4.00	0.14	2.20	2.20	0.37	257	15	262	2.20	< 1	5.6	187	nd	5	2620	12.0	nd	nd	886
NGT238	GB	PMV	38.6164	106.5892	5.40	0.16	2.00	2.60	0.44	205	11	992	2.00	< 1	6.9	124	nd	5	2260	9.0	nd	nd	772
NGT239	GB	PMV	38.6111	106.5972	5.80	0.33	2.80	2.40	0.69	409	15	197	1.10	1.0	3.3	98	nd	11	1670	16.0	nd	nd	470
NGT272	TC	PMV	37.9750	107.7222	1.20	2.00	6.90	0.29	1.60	5140	268	389	0.70	6.0	37.2	4570	nd	6	19800	1680	nd	nd	21600
NGT283	GN	TV	38.0172	107.3819	3.10	0.10	1.50	1.50	0.17	951	9	56	< 0.1	2.0	2.0	76	nd	56	1360	11.0	nd	nd	532
NGT284	GN	TN	38.0217	107.3836	4.20	0.29	2.00	2.00	0.23	575	13	67	0.30	4.0	1.3	152	nd	78	2750	20.0	nd	nd	363
NGT302	OU	PMR	38.0622	107.6622	4.20	0.62	3.50	1.40	0.28	2150	250	9	0.40	13.0	199.0	1690	nd	24	45300	454	nd	nd	46100
NGT307	OU	PMR	38.0547	107.6886	2.20	0.07	1.40	1.10	0.14	567	15	53	0.80	4.0	5.1	118	nd	30	1850	16.0	nd	nd	864
NGT308	OU	PMR	38.0408	107.6825	2.40	0.90	4.50	1.00	0.26	1860	43	133	0.90	31.0	6.3	833	nd	43	2660	346	nd	nd	1170
NGT313	IS	TV	37.8561	107.8214	6.00	0.38	4.80	2.50	0.26	154	73	35	0.50	16.0	47.4	447	nd	10	13900	58.0	nd	nd	9030
NGT316	IS	TV	37.8575	107.8492	4.00	0.07	3.20	1.80	0.09	87	186	1560	3.30	66.0	2.3	785	nd	9	5540	1060	nd	nd	458
NGT323	IS-a	TV	37.8850	107.8556	5.90	2.50	4.60	2.40	1.00	7290	16	129	0.70	4.0	4.5	1120	nd	75	1350	37.0	nd	nd	620
NGT325	IS-a	TV	37.8856	107.8572	4.20	0.41	4.20	2.00	0.30	2660	102	265	1.20	71.0	3.9	465	nd	84	4390	203	nd	nd	845
NGT343	RM	BP	37.9183	107.8306	3.30	0.54	3.30	1.70	0.32	2080	10	151	1.00	11.0	5.4	253	nd	76	1350	6.0	nd	nd	1330
NGT345	RM	BP	37.9192	107.8317	3.40	0.23	2.40	1.70	0.30	750	9	137	0.50	7.0	0.3	93	nd	56	987	4.0	nd	nd	129
NGT370	RM	TV	37.9394	107.6708	4.50	0.36	3.40	2.10	0.42	1180	5	60	1.40	4.0	0.3	70	nd	26	567	3.0	nd	nd	139
NGT372	SN	TV	37.9750	107.7222	3.10	4.20	4.00	1.50	0.33	4540	13	16	0.30	23.0	10.4	330	nd	38	986	19.0	nd	nd	1970
NGT375	SN	TV	37.9761	107.7569	5.70	0.27	2.20	2.60	0.32	531	75	223	0.20	5.0	1.3	95	nd	82	1700	72.0	nd	nd	293
NGT376	SN	TV	37.9758	107.7572	3.30	1.80	2.00	1.50	0.21	3420	124	20	0.30	5.0	48.2	418	nd	127	3890	82.0	nd	nd	7920
NGT377	SN	TV	37.9736	107.7242	3.70	3.30	3.50	1.50	0.44	9050	5	14	< 0.1	5.0	14.5	418	nd	49	1290	2.0	nd	nd	2610
NGT386	TE	TV	37.9478	107.8717	3.40	1.50	2.40	1.60	0.33	9680	67	197	0.80	9.0	65.8	2380	nd	94	4270	18.0	nd	nd	12200
NGT387	TE	TV	37.9386	107.7556	7.60	1.50	3.70	3.60	0.52	4260	122	217	3.70	12.0	24.4	514	nd	266	6950	41.0	nd	nd	5220
NGT526	RB	PMV	38.8781	107.0750	2.19	0.02	6.21	0.92	0.08	767	18.2	336	0.03	1.7	0.2	80	0.413	2	565	20.1	13	nd	925
NGT527	RB	PMV	38.8781	107.0750	2.02	0.01	9.84	0.85	0.06	242	20.2	223	0.01	1.9	3.1	620	0.064	5	603	8.58	7.09	nd	483
NGT530	GB	PMV	38.6083	106.5986	5.37	0.42	3.49	2.40	0.83	690	25.3	327	0.90	1.3	4.9	101	0.669	12	1653	16.9	0.802	nd	620
NGT534	TN	PMR	38.7197	106.4972	1.32	1.56	19.09	0.42	0.63	4656	115.8	1326	1.22	35.1	10.0	1075	0.111	12	4116	77.9	1	nd	2131
NGT536	TN	PMR	38.7028	106.4917	1.25	2.37	7.58	0.34	1.68	4862	408.5	308	0.25	5.6	25.5	4015	122	6	31333	1419	1.08	nd	25090
NGT562	TM	PMR	38.5389	106.3961	2.08	7.80	8.82	0.66	2.45	9891	24.8	66	0.14	26.5	52.3	298	0.158	6	6770	19.4	2.37	nd	8628
NGT619	IS	TV	37.8561	107.8206	5.28	1.51	6.52	2.27	0.35	277	55	172	0.20	4.4	49.7	707	0.781	4	6896	37	1.66	nd	10682
NGT625	IS	TV	37.8617	107.8733	4.72	0.31	8.87	1.59	0.28	556	41.6	158	0.63	15.2	1.2	334	0.5	14	1247	95.7	3.6	nd	617
NGT626	IS	TV	37.8614	107.8731	2.46	1.01	9.96	1.02	0.17	107	83.5	10	0.45	109.0	1.9	668	1.13	23	1475	4.44	1.28	nd	278
NGT631	IS-a	TV	37.8567	107.8483	3.43	0.09	2.49	1.49	0.09	97	131.7	669	1.40	38.3	3.5	301	0.706	14	3651	562	1.68	nd	796
NGT639	IS-a	TV	37.8803	107.8497	5.31	0.10	4.76	2.34	0.37	245	6	308	1.10	2.3	0.6	62	3.63	24	346	12	0.518	nd	245
NGT649	MP	TV	37.9606	107.5925	2.72	0.07	3.94	0.90	0.12	125	393.1	338	1.29	32.8	1.3	867	0.132	81	4814	638	2.05	nd	2242
NGT659	SN	TV	37.9525	107.7264	3.41	5.54	3.21	1.62	0.33	7560	27.1	20	2.10	34.1	39.5	1028	1.82	81	1705	3.1	5.88	nd	7548
NGT660	SN	TV	37.9494	107.7297	3.97	0.26	2.09	1.74	0.20	395	170	59	0.56	48.3	50.2	213	0.775	333	33545	152	9.75	nd	9652
NGT661	TE	TV	37.9331	107.7456	1.13	0.17	3.50	0.66	0.05	9069	141.3	40	1.14	0.0	273.0	7608	1.42	60	48712	24.2	3.39	nd	77412
NGT663	TE	TV	37.9339	107.7525	8.06	0.37	4.28	2.34	0.27	895	15.8	144	8.20	7.6	7.1	357	0.525	171	6491	20.9	3.76	nd	1987
NGX363	RM	SLUDGE	37.9275	107.6881	0.42	0.44	25.00	0.03	0.03	963	< 2	287	1.00	13.0	1.4	162	nd	< 2	264	3.0	nd	nd	587

**Table 3.** Analytical results for selected elements in surface-water samples, CWS, Colorado.

[All samples were filtered through 0.45-µm filter and acidified with ultrapure nitric acid. Analyses by ICP-MS, reported in ppb (parts per billion). Number of figures reported in spreadsheet may not be significant. Sample type: DD, dump drainage; DH, artesian flow from drill hole; MD, mine drainage at portal; S, surface water such as a creek; Spr, spring; TPW, pore water in tailings impound (seepage into augered hole); TS, seepage from tailings impound. District: mining districts, Cash, Cashin; CO, Cochetopa; GB, Gold Brick; GN, Galena; IS, Iron Spring; LO, Los Ochos; MP, Mineral Point; OU, Ouray; QC, Quartz, Creek; RB, Ruby; RM, Red Mountain; TE, Telluride; TM, Tomichi; TN, Tincup; UV, Uravan. Watershed: GU, Gunnison River; LF, Lake Fork of Gunnison River; SM, San Miguel River; UN, Uncompahgre River. Samples collected in 1997–98, ICP-MS analysis by commercial laboratory]

Sample ID	pH	Conductivity (µS/cm)	Sample type	District	Watershed	Al (ppb)	As (ppb)	Cd (ppb)	Co (ppb)	Cu (ppb)	Fe (ppb)	Mn (ppb)	Mo (ppb)	Ni (ppb)	Pb (ppb)	Zn (ppb)	Lat (N.) (degrees)	Long (W.) (degrees)	Site name
<b>Tailings pore waters</b>																			
NGTW527	2.5	1,295	TPW	RB	GU	136.6	5.9	4.6	8.1	316.7	2,777	189	0.4	2.4	2,774.6	739.1	38.8678	107.0750	Standard mill
NGTW619	3.4	1830	TPW	IS	SM	18,349.5	36.8	39.9	214.2	330.5	18,413	2,459	1.4	89.5	1,676.4	7,753.1	37.8506	107.8206	Carbonero mill
NGTW620	3.0	1710	TPW	IS	SM	16,226.3	7.0	33.1	168.9	563.3	27,827	2,783	1.0	71.7	1,515.6	8,233.0	37.8506	107.8206	Carbonero mill
NGTW625	2.1	>2000	TPW	IS	SM	190,941.3	5,910.3	567.4	2,673.7	31,734.0	4,731,542	48,494	1,907.9	405.0	2,458.0	58,049.8	37.8512	107.8733	Caribean mill
NGTW629	4.5	660	TPW	IS	SM	<0.2	7.4	0.6	1.3	7.8	3,738	14	2.2	13.2	7.9	100.1	37.8507	107.8492	Caribean mill
NGTW631	5.4	NA	TPW	IS	SM	21.8	3.8	0.9	2.0	4.9	3,188	52	1.2	14.1	23.2	401.2	37.8507	107.8483	Caribean mill
<b>Surface waters</b>																			
NGW105	8.5	440	S	UV	SM	0.0	40.7	0.9	0.1	3.9	200	1	179.6	1.7	<0.1	4.6	38.3361	109.0246	Lion Crk
NGW106	8.3	310	MD	UV	SM	2.7	5.9	0.3	0.0	4.5	64	1	36.9	0.8	0.3	3.6	38.3364	109.0151	
NGW107	8.5	410	MD	UV	SM	0.0	72.9	0.9	0.2	2.1	80	0	174.1	0.9	<0.1	2.2	38.3369	107.0252	
NGW115A	7.4	1740	SP	UV	SM	1.5	1,130.5	7.3	1.5	6.1	1,891	3	1,864.0	15.7	0.8	7.4	38.3089	108.1263	
NGW117	8.1	840	S	UV	SM	3.4	1.5	0.3	0.3	0.8	299	16	68.9	2.6	<0.1	0.8	38.3908	108.0112	Atkinson Crk
NGW205	7.6	410	DP	TM	GU	10.9	0.5	7.3	0.2	7.4	706	32	6.1	2.9	22.5	351.4	38.5372	106.1348	Akron mill
NGW214	7.9	180	MD	TM	GU	2.7	0.4	0.2	0.2	3.6	308	23	3.4	1.9	1.2	179.1	38.5392	106.0156	Akron adit
NGW230	7.8	280	S	VU	GU	9.5	2.6	0.2	0.3	6.1	438	20	2.0	3.0	2.0	36.8	38.3469	107.0051	Camp Crk
NGW231	7.7	310	S	VU	GU	4.8	1.3	0.1	0.3	5.9	522	25	1.6	3.2	0.7	12.3	38.3481	107.0067	Camp Crk
NGW232	7.9	240	MD	GB	GU	5.6	2.3	0.5	0.1	3.5	166	11	5.1	1.1	1.1	38.9	38.6128	106.0188	Carter mine
NGW247	7.4	240	MD	TC	GU	36.5	0.5	0.1	0.3	4.4	237	12	1.8	2.5	0.8	12.1	38.7242	106.0061	
NGW249	7.7	90	SP	MP	GU	12.0	0.2	0.1	0.1	3.8	76	1	1.4	1.7	0.6	4.8	38.3781	106.0111	
NGW259	7.9	680	SP	LO	GU	3.5	5.4	0.1	0.3	2.7	666	1	10.7	4.3	0.6	13.5	38.3764	106.0074	
NGW263	3.7	210	MD	GQ	GU	574.3	0.1	7.7	23.6	1,180.5	1,331	976	3.0	11.6	2.0	410.8	38.6831	106.1422	Bon Ton mine adit
NGW264	5.7	120	MD	GQ	GU	314.8	0.2	7.4	21.5	1,156.3	6,798	1,020	29.4	10.7	1.1	391.3	38.6831	106.1377	Bon Ton mine and dump drain
NGW275	7.6	280	MD	TC	GU	3.1	0.2	0.1	0.4	19.6	611	464	18.1	2.6	2.0	32.0	38.7044	106.0099	Blistered Horn
NGW287	7.4	260	MD	BP	LF	12.8	2.7	1.2	0.2	6.4	334	25	26.0	2.3	8.8	154.2	37.9206	107.0209	American Basin
NGW296	4.7	80	S	MP	UN	655.7	1.2	4.4	1.9	11.4	240	701	0.7	2.5	17.6	625.6	37.9522	107.0849	
NGW298	3.5	200	MD	MP	UN	188.8	0.7	23.2	3.3	76.1	1,327	379	2.8	2.2	147.7	3,112.6	37.9492	107.3879	London mine
NGW299	3.6	410	S	RM	UN	7,071.6	0.2	2.3	16.2	313.0	3,131	765	1.6	16.3	17.8	374.4	37.9492	107.0395	Red Mtn Crk
NGW309	3.4	>2000	SP	OY	UN	12,462.4	2.8	309.2	55.6	2,509.7	12,609	63,766	1.7	128.2	108.2	68,089.8	38.0539	112.1589	
NGW310	3.7	570	SP	IS	SM	3,989.9	0.2	2.3	24.5	178.5	5,376	1,077	2.8	12.6	2.2	298.6	38.0542	107.0432	Iron Spring mine
NGW315	3.0	1660	TD	IS	SM	15,651.6	15.4	205.0	244.8	1,221.9	76,187	60,014	1.3	77.5	911.9	38,183.4	37.8564	110.4192	Carbonero mill
NGW317	7.3	980	TD	IS	SM	17.2	0.8	0.6	9.4	3.1	1,149	2,863	1.1	19.3	2.4	71.4	37.8575	107.0256	Caribou mill
NGW318	7.1	790	MD	IS	SM	2.9	0.6	1.0	10.8	2.5	2,267	2,595	1.0	17.2	0.2	181.9	37.8569	107.0335	Caribou mine
NGW320	6.6	290	MD	IS	SM	411.7	0.2	18.3	10.4	580.6	331	8,161	0.6	8.1	28.9	1,786.7	37.8844	107.3080	Alta mine
NGW328	4.8	250	S	RM	UN	153.2	0.2	0.5	2.7	75.3	365	150	0.5	3.4	2.6	83.7	37.9033	107.0124	National Belle
NGW333	2.9	1340	MD	RM	UN	41,357.3	32.0	6.2	75.9	883.2	60,234	989	0.6	71.4	38.8	840.9	37.9139	107.1132	Champion Gulch
NGW335	2.4	>2000	MD	RM	UN	210,775.5	84.1	44.5	520.1	8,252.9	441,023	7,457	1.1	384.4	51.1	5,471.5	37.9175	107.7535	Guston mine
NGW336	3.1	>2000	MD	RM	UN	194,867.4	31.6	22.1	273.4	765.2	230,955	7,043	0.8	209.7	66.4	3,143.6	37.9172	107.3791	
NGW338	2.6	>2000	DP	RM	UN	42,888.0	293.9	14.3	136.3	2,688.8	260,589	6,767	1.0	148.4	111.9	2,160.3	37.9211	107.2467	
NGW340	2.7	>2000	MD	RM	UN	248,110.7	93.5	47.7	602.9	9,165.3	501,385	8,499	1.0	432.0	57.1	5,978.5	37.9164	107.8033	Guston mine
NGW342	3.5	640	S	RM	UN	10,071.3	2.9	3.4	22.5	838.4	13,220	566	0.2	24.2	48.4	543.6	37.9181	107.0707	Red Mtn Crk
NGW344	2.9	>2000	TS	RM	UN	14,801.0	13.6	78.3	832.1	831.7	184,510	74,094	1.4	102.9	37.7	24,754.6	37.9189	108.3199	Tailings seep
NGW347	3.6	410	S	RM	UN	6,061.7	0.2	2.3	14.9	291.4	2,914	739	0.2	13.7	18.2	363.2	37.9600	107.0508	Red Mtn Crk
NGW348	2.0	1650	MD	RM	UN	43,381.5	1.1	151.6	125.4	435.8	69,504	5,739	2.2	82.9	10.0	10,284.3	37.9372	109.5354	Larson Brothers mine
NGW353	6.3	740	MD	RM	UN	89.3	0.3	3.6	14.9	3.0	2,344	1,480	2.0	11.6	1.4	520.8	37.9278	107.0720	Mt. King mine
NGW357	2.9	1410	MD	RM	UN	38,992.6	22.2	17.1	99.2	665.3	60,250	4,641	0.8	91.1	97.6	3,733.1	37.9242	107.2946	Joker mine
NGW363	5.9	1260	MD	RM	UN	47.9	10.2	0.3	22.0	2.9	20,058	3,512	0.5	22.3	0.8	269.7	37.7442	107.0100	Silver Belle

**Table 3.** Analytical results for selected elements in surface-water samples, CWS, Colorado—Continued.

Sample ID	pH	Conductivity (µS/cm)	Sample type	District	Watershed	Al (ppb)	As (ppb)	Cd (ppb)	Co (ppb)	Cu (ppb)	Fe (ppb)	Mn (ppb)	Mo (ppb)	Ni (ppb)	Pb (ppb)	Zn (ppb)	Lat (N.) (degrees)	Long (W.) (degrees)	Site name
<b>Surface waters—Continued</b>																			
NGW365	3.7	390	S	RM	UN	6,441.1	0.6	3.5	15.0	330.7	6,612	341	0.3	14.1	23.3	596.8	37.9250	107.0680	Red Mtn Crk
NGW367	3.0	290	S	RM	UN	1,823.5	0.8	0.7	6.6	185.0	1,640	52	0.2	5.9	55.4	110.7	37.9175	107.0184	CorksCrkew Gulch
NGW371	6.2	190	S	SN	UN	18.1	0.6	0.6	0.2	1.7	212	134	2.1	1.7	1.1	109.5	38.0028	107.0210	Canyon Crk
NGW373	6.5	190	S	SN	UN	12.9	0.6	0.9	0.3	2.0	193	243	2.0	2.1	0.6	155.4	37.9747	107.0198	Canyon Crk
NGW380	6.9	120	S	MnP	UN	64.0	0.2	1.2	0.2	5.0	137	100	0.9	2.0	0.7	133.3	37.9747	107.0207	Upr. Unconphgre R
NGW382	6.5	480	MD	MnP	UN	38.0	0.8	0.5	0.6	3.5	496	603	5.9	5.7	0.5	115.2	37.9531	107.0175	Old Lout mine
NGW385	7.2	360	S	IS	SM	38.7	0.2	0.5	1.4	2.2	317	209	1.7	4.1	0.1	65.1	37.8661	107.0224	Howard Fk
NGW388	7.3	150	S	TE	SM	11.3	0.5	5.2	0.2	4.7	87	84	1.1	1.4	3.8	750.3	37.9383	107.0915	Savage Crk
NGW391	7.2	110	S	TE	SM	15.8	0.4	1.0	0.1	3.3	77	37	1.0	1.0	4.8	139.2	37.9356	107.0226	San Miguel R
NGW393	3.3	1510	MD	GN	LF	10,534.6	7.8	29.9	168.5	37.2	21,580	57,836	0.7	103.8	71.0	4,784.0	37.9736	107.5084	Palmetto basin
NGW394	2.9	1500?	MD	GN	LF	9,179.0	5.8	30.9	159.7	69.9	17,590	56,318	0.6	97.8	68.2	4,891.6	37.9736	107.5237	Palmetto basin
NGW450	7.1	240	S	GN	LF	9.7	0.1	9.4	9.2	320.2	652	504	1.0	10.9	0.2	1,022.3	37.9772	107.1677	Head Henson Crk
NGW455	2.7	nd	MD	UN?	UN	7,402.5	59.5	165.0	59.0	424.0	23,624	2,191	4.3	37.4	493.8	11,289.6	37.9600	109.7537	San Juan Chief
NGW492	5.9	910	S	SN	UN	18.2	0.8	0.8	0.2	2.6	519	179	9.1	1.7	1.5	163.3	37.9675	107.7211	Canyon Crk
NGW493	5.9	1540	MD	IS	SM	52.3	1.8	1.1	6.1	24.5	6,008	1,445	1.2	11.1	<0.1	148.8	37.8508	107.8494	Caribeu mine drainage
NGW494	5.9	1540	DH	IS	SM	<0.2	1.4	<0.01	9.1	2.2	13,511	850	2.6	28.0	<0.1	68.3	37.9006	107.6675	Drill hole flow
NGW495	2.1	>2000	MD	RM	UN	135,579.7	486.5	7.0	454.1	7,648.3	479,713	228	12.0	326.9	71.9	924.7	37.8508	107.8494	Head Corkscrew
NGW499	3.0	285	MD	RB	GU	1,348.1	<0.03	117.1	17.5	841.8	2,742	5,712	<0.05	16.0	1,348.1	18,910.2	38.8680	107.0739	Standard mine
NGW505	8.2	251	S	VU	GU	21.2	1.5	0.2	0.7	6.5	992	58	1.5	5.9	3.9	132.8	38.3346	107.0014	Cabin Crk
NGW506	8.1	354	S	VU	GU	40.0	2.7	<0.01	0.8	7.7	1,317	201	1.2	10.1	<0.1	16.6	38.3349	107.0053	Cabin Crk
NGW507	3.1	200	MD	RB	GU	3,726.9	<0.03	47.9	10.4	708.2	3,955	3,453	<0.05	5.8	417.0	7,644.9	38.8681	107.0728	Mid-Elk Basin
NGW510	3.1	320	MD	RB	GU	3,929.8	<0.03	54.5	11.0	297.8	3,973	7,351	<0.05	8.2	327.3	7,972.1	38.8834	107.0683	Upper Elk Basin
NGW511	2.9	354	DD	RB	GU	4,084.8	<0.03	80.5	13.7	1,015.5	4,634	4,533	<0.05	10.9	571.1	12,388.2	38.8682	107.0714	Upper Elk Basin
NGW512	3.0	285	MD	RB	GU	1,878.8	<0.03	127.2	18.5	927.8	4,738	7,437	<0.05	17.4	1,373.7	21,470.9	38.8680	107.0739	Standard mine
NGW514	3.3	301	S	RB	GU	3,641.6	<0.03	87.4	13.0	825.3	3,784	4,328	<0.05	7.3	1,229.1	13,117.3	38.8681	107.0739	E Elk Crk
NGW515	5.8	20	S	RB	GU	5.0	0.2	<0.01	0.1	0.3	110	20	0.1	0.7	0.4	2.7	38.8680	107.0742	W Elk Crk
NGW518	3.0	169	DD	RB	GU	9,890.7	<0.03	152.4	51.3	1,402.6	15,890	10,500	<0.05	21.6	613.9	27,647.8	38.8679	107.0744	Standard mine
NGW519	3.1	448	TP	RB	GU	7,893.9	<0.03	153.8	38.6	1,163.9	15,263	7,875	<0.05	16.8	436.9	23,820.3	38.8678	107.0744	Standard mine
NGW520	2.9	765	TPS	RB	GU	4,936.8	<0.03	89.4	32.4	514.8	7,552	6,968	<0.05	11.2	403.4	13,819.3	38.8678	107.0744	Standard mine
NGW521	5.4	62	S	RB	GU	238.4	0.2	14.8	2.2	63.9	379	664	<0.05	2.3	91.7	2,044.2	38.8678	107.0753	Elk Crk
NGW523	5.9	60	S	RB	GU	127.5	1.2	10.0	1.2	37.5	247	398	<0.05	1.9	52.5	1,324.4	38.8672	107.0753	Elk Crk
NGW529	7.2	97	S	GB	GU	11.4	0.6	0.4	0.1	1.8	235	12	1.0	1.2	1.8	42.5	38.6009	106.5989	Gold Crk
NGW531	6.9	90	S	GB	GU	8.0	0.4	0.2	0.1	0.7	216	4	1.1	1.1	0.5	19.9	38.6170	106.5853	Gold Crk
NGW532	6.9	56	S	GB	GU	9.6	0.1	0.0	0.1	0.4	153	2	1.3	0.8	0.2	1.8	38.6501	106.5742	Gold Crk
NGW533	6.7	40	S	GQ	GU	76.7	0.1	0.6	0.9	64.6	95	41	7.4	2.1	0.9	47.7	38.6682	106.4792	Tomichi Crk
NGW537	6.8	66	S	LF	LF	15.8	0.4	0.8	0.7	1.4	154	8	6.1	1.9	2.0	3.4	38.0167	107.3239	Wade Gulch
NGW540	6.9	96	S	LF	LF	68.7	0.2	0.2	0.6	1.4	305	63	1.9	2.0	1.3	8.0	37.9505	107.3064	Red Mtn Gulch
NGW541	6.6	43	S	LF	LF	86.9	0.3	0.1	0.2	0.8	161	58	1.2	1.2	0.2	17.7	37.9003	107.4264	Lake Fk Gunn.
NGW543	6.7	46	S	LF	LF	13.9	0.2	0.0	0.1	0.4	179	2	1.0	1.0	0.1	1.7	37.9000	107.4331	Cottonwood Crk
NGW544	6.8	34	S	LF	LF	13.3	0.3	0.0	0.0	0.3	140	2	0.4	0.7	0.1	0.9	37.9001	107.4342	Cattaract Crk
NGW545	6.9	44	S	BP	LF	12.4	0.6	0.0	0.1	0.4	166	1	1.0	0.9	0.1	0.7	37.9336	107.4631	Grizzly Crk
NGW546	7.1	47	S	BP	LF	41.1	0.1	0.2	0.5	0.8	126	145	0.9	1.4	0.2	30.4	37.9340	107.4581	Silver Crk
NGW547	7.0	46	S	LF	LF	26.2	0.3	0.0	0.1	0.3	140	8	0.6	0.9	0.2	3.0	37.9007	107.3806	Bent Crk
NGW548	6.8	76	S	LF	LF	50.5	0.4	0.0	0.2	0.8	190	50	0.7	1.3	0.3	2.0	37.9847	107.2958	Lake Fk Gunn.
NGW549	6.9	46	S	GN	LF	70.7	0.2	0.0	0.3	0.8	163	23	0.3	1.0	0.2	3.6	38.0007	107.4675	W Fk Henson Crk
NGW550	7.2	42	S	GN	LF	32.0	0.4	0.3	0.2	7.7	134	38	0.2	1.0	0.5	34.9	37.9676	107.5350	Upper Henson Crk
NGW552	7.3	46	S	GN	LF	55.7	0.4	0.1	0.1	1.7	168	11	0.7	1.0	0.9	12.5	38.0171	107.3319	Lower Henson Crk
NGW555	3.0	779	TS	RB	GU	5,199.0	<0.03	86.2	33.6	525.1	5,928	7,160	<0.05	15.3	336.3	13,766.6	38.8678	107.0753	Standard mine
NGW556	6.6	32	MD	RB	GU	89.6	0.1	3.0	0.6	21.5	193	151	<0.05	0.8	104.6	452.6	38.8834	107.0708	Upper Elk Basin
NGW558	7.3	59	S	RB	GU	66.8	2.3	4.0	0.3	13.6	235	108	<0.05	1.4	15.6	541.1	38.8507	107.0600	Lower Elk Crk
NGW559	7.3	37	S	RB	GU	28.3	0.3	0.4	0.1	1.9	143	12	0.6	0.9	3.8	50.0	38.9008	107.0253	Slate R
NGW560	7.3	24	S	RB	GU	43.8	0.2	0.7	0.1	3.3	98	19	<0.05	0.5	11.5	90.6	38.9008	107.0467	Oh<Be<Joyful Crk

**Table 3.** Analytical results for selected elements in surface-water samples, CWS, Colorado—Continued.

Sample ID	pH	Conductivity (µS/cm)	Sample type	District	Watershed	Al (ppb)	As (ppb)	Cd (ppb)	Co (ppb)	Cu (ppb)	Fe (ppb)	Mn (ppb)	Mo (ppb)	Ni (ppb)	Pb (ppb)	Zn (ppb)	Lat (N.) (degrees)	Long (W.) (degrees)	Site name
<b>Surface waters—Continued</b>																			
NGW561	7.0	62	S	TM	GU	30.7	0.1	0.0	0.1	6.7	225	9	3.9	1.0	0.5	4.5	38.5349	106.3892	Tomichi Crk
NGW563	7.1	83	S	TM	GU	20.6	0.1	0.6	0.1	5.0	262	10	4.0	1.2	1.6	104.0	38.5178	106.4061	Tomichi Crk
NGW565	7.2	81	S	TM	GU	19.1	0.2	0.8	0.1	4.8	268	11	4.0	1.3	3.2	124.9	38.5172	106.4147	Tomichi Crk
NGW570	2.7	1025	S	RM	UN	21,555.4	43.6	35.7	79.6	5,861.9	63,499	3,457	<0.05	52.6	189.2	5,505.0	38.8678	107.0750	Red Mtn Crk
NGW571	2.6	1140	MD	RM	UN	98,023.1	345.8	10.9	147.5	5,639.1	163,712	1,829	<0.05	133.2	367.2	2,243.3	37.9004	107.7031	Adit into ferricrete
NGW574	3.3	436	S	RM	UN	13,636.8	37.8	2.4	23.5	1,031.2	22,776	420	<0.05	23.3	51.6	531.5	37.9167	107.6931	Red Mtn Crk
NGW575	2.9	385	S	RM	UN	17,352.3	10.7	2.3	31.3	293.0	26,180	378	<0.05	24.8	29.5	531.5	37.9167	107.6919	Champion Gulch
NGW576	2.5	>2000	MD	RM	UN	469,702.7	101.0	54.8	689.7	8,307.4	680,771	9,422	0.7	454.5	117.9	10,410.1	37.9167	107.6919	Guston mine
NGW577	4.2	122	MD	RM	UN	6,440.0	2.0	1.0	9.7	138.6	9,916	153	<0.05	8.3	5.3	173.6	37.9169	107.6981	Prospect trench
NGW578	2.5	>2000	TS	RM	UN	16,727.7	5.5	50.0	859.5	785.8	185,890	105,535	<0.05	101.3	26.5	30,397.4	37.9175	107.6928	Tailings
NGW580	2.9	885	DD	RM	UN	18,750.1	6.0	5.1	60.4	412.6	12,862	4,122	<0.05	42.1	9.2	1,447.0	37.9174	107.6928	Joker mine
NGW581	3.6	319	S	RM	UN	11,751.9	2.1	3.3	22.3	485.3	14,636	653	<0.05	18.0	34.0	687.7	37.9172	107.6758	Red Mtn Crk
NGW582	3.0	565	S	RM	UN	10,677.9	7.8	0.9	31.1	437.2	16,358	597	<0.05	24.1	78.0	395.9	37.9172	107.6758	Corkscrew Gulch
NGW583	3.3	248	S	RM	UN	4,892.5	0.4	0.1	5.6	18.2	1,396	85	1.0	4.7	2.4	28.5	37.9008	107.6694	Small crk
NGW584	2.5	1035	S	RM	UN	21,270.5	<0.03	1.1	72.4	1,172.8	14,492	86	<0.5	52.9	136.5	153.7	37.9006	107.6675	Corkscrew Gulch
NGW587	2.1	>2000	MD	RM	UN	230,594.5	675.0	7.9	674.8	11,376.3	720,589	335	13.8	485.8	75.9	1,313.1	37.9006	107.6622	Adit, Corkscrew
NGW589	3.9	39	S	RM	UN	732.9	3.4	0.6	2.3	30.5	1,599	17	14.5	3.3	3.6	<5	37.9013	107.6742	Lake at divide
NGW592	3.4	154	S	RM	UN	1,262.0	1.3	0.1	3.3	15.3	1,094	49	2.5	3.1	35.5	8.2	37.9180	107.6794	Small crk
NGW593	3.1	563	S	RM	UN	14,209.1	0.8	0.3	20.7	64.2	4,459	388	0.7	14.6	3.4	99.0	37.9008	107.6744	Small crk
NGW595	2.9	492	S	RM	UN	5,187.8	0.5	0.5	9.9	37.0	1,702	26	<0.05	13.3	103.0	61.5	37.9014	107.6736	Small crk
NGW599	3.2	215	S	RM	UN	2,287.1	1.2	0.7	5.5	107.3	1,617	56	<0.05	4.7	63.7	103.3	37.9337	107.6675	Small crk
NGW605	4.9	278	S	RM	UN	609.9	0.4	0.1	3.2	22.1	1,551	242	<0.05	5.3	1.8	33.5	37.9183	107.6611	Gray Copper Crk
NGW606	6.0	375	MD	RM	UN	58.8	2.2	0.2	2.3	2.0	3,088	1,213	3.8	5.4	<0.1	119.1	38.0834	107.6575	Mountain Silver
NGW609	5.9	74	MD	RM	UN	101.9	<0.03	0.3	0.9	13.2	420	101	<0.05	1.7	3.8	44.0	37.9509	107.6617	Upper Mt Silver
NGW610	3.4	390	S	RM	UN	9,040.2	<0.03	2.2	18.4	363.3	5,584	759	<0.05	17.8	28.4	601.5	37.8511	107.7981	Red Mtn Crk
NGW612	6.2	680	S	IS	SM	178.0	0.4	0.3	7.4	6.3	2,991	538	<0.05	12.0	<0.1	52.3	37.8511	107.7986	Chapman Gulch
NGW613	6.4	620	MD	IS	SM	30.8	1.0	<0.01	12.0	1.5	4,221	833	<0.05	10.6	<0.1	30.0	37.8512	107.8019	Small adit
NGW614	5.9	1540	DH	IS	SM	11.6	1.3	0.2	9.7	3.2	15,058	935	1.3	34.1	<0.1	76.2	37.8508	107.8494	Exploration drill hole
NGW616	5.1	95	S	IS	SM	861.9	<0.03	0.1	2.8	10.5	580	100	<0.05	2.6	56.7	54.2	37.8508	107.8097	Small crk
NGW617	3.4	955	SP	IS	SM	5,730.4	<0.03	1.9	26.0	248.0	3,056	1,056	<0.05	14.8	5.3	432.0	37.8507	107.8089	Iron Spring
NGW618	5.5	365	S	IS	SM	177.9	<0.03	<0.01	1.8	4.2	1,249	102	<0.05	6.2	<0.1	20.4	37.8508	107.8144	E Howard Fk
NGW621	5.5	890	S	IS	SM	56.8	<0.03	4.7	14.2	1.3	3,251	1,607	<0.05	18.4	<0.1	701.9	37.8839	107.9072	Side creek
NGW627	6.1	330	S	IS	SM	49.0	<0.03	0.3	1.5	2.2	1,064	137	0.7	7.1	<0.1	57.7	37.8506	107.8467	S Fork San Miguel
NGW632	5.9	910	MD	IS	SM	71.6	1.9	1.1	8.8	30.4	8,924	2,169	<0.05	20.7	<0.1	219.2	37.8508	107.8494	Caribou mine
NGW633	5.9	820	TD	IS	SM	16.1	<0.03	0.5	7.2	1.7	3,448	1,953	<0.05	21.8	<0.1	67.2	37.8508	107.8497	Flow on Caribou tails
NGW634	5.5	348	S	IS	SM	41.9	<0.03	0.6	2.1	2.0	1,369	194	<0.05	7.4	<0.1	117.6	37.8510	107.8561	Howard Fork
NGW635	6.9	135	S	IS	SM	<0.2	0.4	<0.01	0.2	<1	519	2	0.8	2.1	<0.1	<5	37.8343	107.8858	Lake Fk San Miguel;
NGW636	4.7	450	MD	IS	SM	3,107.2	4.3	3.2	31.7	221.2	7,740	3,644	<0.05	24.6	3.6	702.7	37.8514	107.8811	Adit W of Campgrnd
NGW637	6.7	348	S	IS	SM	91.7	<0.03	0.5	2.1	5.2	1,370	272	<0.05	5.5	<0.1	89.1	37.8514	107.8811	Howard Fork
NGW641	6.2	118	S	MP	UN	51.9	<0.03	0.9	0.3	3.9	363	88	<0.05	2.3	<0.1	151.9	37.9677	107.6375	Upr Unconpahgre R
NGW642	6.5	472	MD	MP	UN	30.1	<0.03	35.9	3.1	19.9	1,266	7,484	<0.05	7.7	<0.1	7,258.7	37.9515	107.6250	M Breen adit
NGW644	7.0	83	S	MP	UN	46.1	<0.03	1.2	0.2	5.4	284	161	<0.05	1.3	1.3	237.0	37.9514	107.6269	Mineral Crk
NGW645	6.9	128	S	MP	UN	304.0	<0.03	1.2	0.5	11.7	445	174	<0.05	2.4	3.4	211.9	37.9510	107.5864	Upr Unconpahgre R
NGW651	3.0	707	S	MP	UN	13,284.8	55.7	137.7	55.3	288.9	45,195	4,192	1.7	34.6	781.2	15,629.0	37.9176	107.6214	Mineral Crk
NGW652	3.6	170	S	MP	UN	3,044.3	0.9	16.9	5.9	62.0	1,389	2,913	<0.05	7.1	34.5	2,213.3	37.9014	107.6539	Upr Unconpahgre R
NGW654	3.1	443	MD	RM	UN	2,246.2	1.9	1.1	16.5	47.7	10,184	457	<0.05	23.0	3.3	191.4	37.0339	107.6806	Adit drainage, small mine
NGW656	5.8	202	S	OU	UN	71.0	<0.03	0.9	4.1	9.0	1,105	267	<0.05	5.9	<0.1	164.6	37.9672	107.7464	Unconpahgre R
NGW657	5.9	69	S	SN	UN	31.4	<0.03	1.0	0.5	2.7	296	360	0.9	1.3	1.2	180.6	37.9516	107.7278	Sneffels Crk
NGW658	6.9	48	S	SN	UN	20.9	<0.03	0.7	0.1	3.3	217	11	<0.05	0.6	<0.1	154.5	37.9183	107.7486	Imogene Crk
NGW662	6.5	48	S	TE	SM	<0.2	0.7	<0.01	0.1	<1	335	5	<0.05	0.6	<0.1	20.6	37.9339	107.7586	Savage Crk
NGW670	6.7	90	S	TE	SM	29.6	<0.03	7.5	0.3	16.4	327	103	0.8	2.9	8.3	1,563.2	37.9344	107.7672	Savage Crk
NGW671	7.3	48	S	TE	SM	56.7	3.6	0.7	0.2	3.7	244	110	<0.05	2.2	1.3	155.3	37.9173	107.7706	Marshall Crk

**Table 3.** Analytical results for selected elements in surface-water samples, CWS, Colorado—Continued.

Sample ID	pH	Conductivity (µS/cm)	Sample type	District	Watershed	Al (ppb)	As (ppb)	Cd (ppb)	Co (ppb)	Cu (ppb)	Fe (ppb)	Mn (ppb)	Mo (ppb)	Ni (ppb)	Pb (ppb)	Zn (ppb)	Lat (N.) (degrees)	Long (W.) (degrees)	Site name
<b>Surface waters—Continued</b>																			
NGW673	6.9	53	S	TE	SM	11.3	0.7	2.4	0.1	9.2	245	8	<0.05	1.2	<0.1	510.7	37.9337	107.8036	Ingrahm Crk
NGW677	7.2	73	S	TE	SM	15.7	0.3	<0.01	0.1	1.2	328	1	1.0	1.0	<0.1	<5	37.9675	107.7467	Bear Crk
NGW678	5.6	66	S	SN	UN	29.0	0.5	0.8	0.4	2.4	270	377	<0.05	1.5	1.3	185.3	37.9676	107.7472	Sneffels Crk
NGW679	6.0	94	MD	SN	UN	13.5	1.5	<0.01	0.2	<1	338	13	6.5	1.7	<0.1	10.2	37.9681	107.7611	Eclipse mine
NGW680	6.1	44	S	SN	UN	18.1	<0.03	<0.01	0.1	<1	239	3	0.6	0.8	<0.1	<5	37.9675	107.7211	Yankee Boy Crk
NGW681	5.7	121	S	SN	UN	21.6	0.4	0.9	0.3	2.1	510	179	1.2	2.2	<0.1	163.6	37.8846	107.7161	Canyon Crk
NGW683	7.1	12	S	RM	UN	17.6	<0.03	<0.01	0.1	<1	168	5	<0.05	0.8	<0.1	<5	37.9346	107.6644	Mineral Crk ditch
NGW685	6.6	150	S	RM	UN	229.3	<0.03	1.5	1.7	110.4	477	501	<0.05	4.5	10.2	284.6	38.8846	107.7161	Albany Crk
<b>Samples collected in 1999, ICP-MS analysis by USGS laboratory</b>																			
NGW820	2.9	130	SP	RM	UN	770.0	0.4	0.2	4.1	180.0	2,400	25	0.1	2.8	4.6	22.0	37.8956	107.6533	Spring near mine/vein, red altered
NGW823	4.6	15	S	RM	UN	750.0	<0.4	5.0	2.8	11.0	150	1,100	0.2	2.3	12.0	860.0	37.9128	107.6539	Head Gray Copper creek
NGW827	2.8	189	MD	RM	UN	3,200.0	11.0	0.4	6.4	360.0	6,100	68	0.2	5.6	8.1	66.0	37.9178	107.6558	Vernon adit, lower
NGW828	3.5	33	S	RM	UN	600.0	<0.4	0.1	1.8	6.0	220	59	0.0	1.3	0.4	7.0	37.9206	107.6556	Side creek from red altn
NGW829	3.0	121	S	RM	UN	1,900.0	3.9	0.3	4.5	150.0	5,400	65	0.1	4.0	4.1	52.0	37.9194	107.6561	Gray Copper crk west of Vernon
NGW830	3.0	92	S	RM	UN	1,200.0	0.8	0.1	3.8	41.0	3,100	38	0.0	3.3	3.2	20.0	37.9172	107.6550	Gray Copper crk east of Vernon
NGW831	2.0	>2000	MD	RM	UN	>400,000	1,000.0	9.6	1,000.0	28,000.0	1,500,000	720	19.0	780.0	61.0	2,000.0	37.9053	107.6675	Corkscrew headwall adit
NGW834	6.8	172	S	IS	SM	82.0	<0.4	0.6	1.9	4.0	240	370	1.0	1.2	0.2	110.0	37.8652	107.8670	Howard Fk at Ames
NGW835	7.7	121	S	IS	SM	52.0	<0.4	0.3	0.7	3.0	120	120	1.1	1.0	0.3	43.0	37.8722	107.8933	South Fork San Miguel
NGW837	3.2	312	MD	RM	UN	24,000.0	300.0	7.0	49.0	2,000.0	65,000	1,200	0.1	41.0	170.0	1,800.0	37.8997	107.7019	Hudson mine drain
NGW839	3.0	204	S	RM	UN	14,000.0	100.0	3.1	25.0	950.0	29,000	790	0.5	22.0	83.0	880.0	37.9000	107.7061	Headwat Red Mtn Crk, no mines
NGW840	2.5	>2000	MD	RM	UN	>400,000	91.0	44.0	700.0	10,000.0	810,000	9,100	0.6	450.0	88.0	8,900.0	37.9164	107.6917	Guston mine drain
NGW841	2.9	284	S	RM	UN	33,000.0	20.0	6.6	45.0	1,800.0	50,000	1,200	0.1	44.0	71.0	1,700.0	37.9186	107.6969	Red mtn creek middle
NGW842	2.9	762	MD	RM	UN	70,000.0	29.0	23.0	120.0	910.0	140,000	5,600	0.1	100.0	140.0	8,000.0	37.9233	107.6928	Joker tunnel drainage
NGW844	5.5	415	MD	IS	SM	1,600.0	<0.4	1.4	22.0	34.0	26,000	1,200	0.4	4.7	1.8	360.0	37.8608	107.8299	Mine drain makes Fe floc
NGW845	3.7	325	S	IS	SM	5,100.0	<0.4	1.1	16.0	92.0	4,400	700	0.1	3.7	3.9	280.0	37.8600	107.8299	Surface flow, much Fe floc
NGW850	7.3	51	S	TM	GU	23.0	<0.4	0.7	0.0	4.0	170	13	2.9	2.2	5.9	140.0	38.5169	106.4232	Tomichi creek near smelter
NGW852	7.3	131	MD	TM	GU	2.0	<0.4	0.4	0.0	1.0	130	120	1.7	2.6	0.5	430.0	38.5394	106.3953	Akron tunnel drainage
NGW853	7.8	50	S	TM	GU	20.0	<0.4	0.2	0.1	4.0	160	15	3.2	2.3	1.1	38.0	38.5378	106.3972	Tomichi crk 200 m below Akron
NGW856	8.7	45	S	GB	GU	40.0	0.7	0.2	0.1	1.0	130	10	0.7	1.8	1.8	20.0	38.6064	106.5994	Gold Crk east of mines and tails
NGW857	8.2	48	S	OU	UN	110.0	<0.4	1.3	0.1	110.0	100	12	0.6	2.8	4.1	190.0	38.0655	107.6718	Dexter crk below mines and mill
NGW860	6.9	184	S	na	UN	26.0	<0.4	1.6	6.1	60.0	360	310	1.1	7.9	1.6	260.0	38.0483	107.6862	Uncompahgre north side Ouray
NGW862	7.7	240	S	na	UN	100.0	<0.4	0.7	1.9	38.0	160	150	1.8	4.9	1.1	52.0	38.1889	107.7472	Uncompahgre north of Ridgeway
NGW864	6.4	42	S	SN	UN	40.0	0.5	0.4	0.1	10.0	85	79	0.7	2.0	1.4	84.0	37.9769	107.7589	Sneffels crk above Atlas tails
NGW865	6.8	55	S	SN	UN	<2	<0.4	0.2	<0.02	0.5	52	0	1.1	0.2	<0.05	98.0	37.9753	107.7461	Sneffels crk below Atlas tails
NGW866	7.0	64	S	SN	UN	39.0	<0.4	1.2	0.0	20.0	84	6	1.1	2.4	1.8	280.0	37.9658	107.7275	Imogene crk SW Camp Bird
NGW869	7.2	51	MD	SN	UN	11.0	<0.4	0.9	<0.02	13.0	72	2	2.0	2.3	1.0	270.0	37.9483	107.7311	Adit drain old CB#3
NGW870	5.8	55	MD	SN	UN	200.0	<0.4	22.0	4.5	320.0	2,000	1,000	0.2	2.9	9.8	4,600.0	37.9500	107.7339	Adit drain Hidden Treasure, red floc
NGW871	6.6	124	S	TE	SM	53.0	<0.4	26.0	0.4	36.0	140	400	0.6	3.5	10.0	8,200.0	37.9347	107.7528	Surface flow at Argentine dump
NGW873	7.1	190	S	SN	UN	16.0	<0.4	0.9	0.1	11.0	160	110	3.0	2.0	0.7	200.0	37.9758	107.7195	Canyon Crk below Camp Bird