Prepared in cooperation with the National Park Service

Assessment of Geochemical and Hydrologic Conditions near Old Yuma Mine in Saguaro National Park, Arizona, 2014–17

Scientific Investigations Report 2018–5019

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By Kimberly R. Beisner and Floyd Gray

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

U.S. Department of the Interior

RYAN K. ZINKE, Secretary

U.S. Geological Survey

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exercising the authority of the Director

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Conversion Factors

U.S. customary units to International System of Units

Multiply	By	To obtain
Length		
inch (in.)	2.54	centimeter (cm)
inch (in.)	25.4	millimeter (mm)
foot (ft)	0.3048	meter (m)
mile (mi)	1.609	kilometer (km)
yard (yd)	0.9144	meter (m)
Area		
acre	4,047	square meter (m ²)
acre	0.4047	hectare (ha)
acre	0.4047	square hectometer (hm ²)
acre	0.004047	square kilometer (km ²)
square mile (mi ²)	259.0	hectare (ha)
square yard (yd ²)	0.836127	square meter (m ²)
Volume		
ounce, fluid (fl. oz)	0.02957	liter (L)
pint (pt)	0.4732	liter (L)
quart (qt)	0.9464	liter (L)
gallon (gal)	3.785	liter (L)
cubic inch (in ³)	0.01639	liter (L)
cubic yard (yd ³)	0.7646	cubic meter (m ³)
Mass		
ounce, avoirdupois (oz)	28.35	gram (g)
pound, avoirdupois (lb)	0.4536	kilogram (kg)

Temperature in degrees Celsius (°C) may be converted to degrees Fahrenheit (°F) as follows:

$$^{\circ}\text{F} = (1.8 \times ^{\circ}\text{C}) + 32.$$

Temperature in degrees Fahrenheit (°F) may be converted to degrees Celsius (°C) as follows:

$$^{\circ}\text{C} = (^{\circ}\text{F} - 32) / 1.8.$$

Datum

Vertical coordinate information is referenced to the North American Vertical Datum of 1988 (NAVD 88).

Horizontal coordinate information is referenced to the North American Datum of 1983 (NAD 83).

Altitude, as used in this report, refers to distance above the vertical datum.

Supplemental Information

Specific conductance is given in microsiemens per centimeter at 25 degrees Celsius ($\mu\text{S}/\text{cm}$ at 25 °C)

Concentrations of chemical constituents in water are given in either milligrams per liter (mg/L) or micrograms per liter ($\mu\text{g}/\text{L}$).

Activities for radioactive constituents in water are given in picocuries per liter (pCi/L).

Results for measurements of stable isotopes of an element (with symbol E) in water, solids, and dissolved constituents commonly are expressed as the relative difference in the ratio of the number of the less abundant isotope (^iE) to the number of the more abundant isotope of a sample with respect to a measurement standard.

Note to USGS users: Use of hectare (ha) as an alternative name for square hectometer (hm^2) is restricted to the measurement of small land or water areas. Use of liter (L) as a special name for cubic decimeter (dm^3) is restricted to the measurement of liquids and gases. No prefix other than milli should be used with liter.

Abbreviations

AZDEQ	Arizona Department of Environmental Quality
BLM	Bureau of Land Management
CERCLA	Comprehensive Environmental Response, Compensation, and Liability Act
EPA	U.S. Environmental Protection Agency
GMWL	global meteoric water line
GPS	global positioning system
ICP-AES	inductively coupled plasma atomic emission spectrometry
ICP-MS	inductively coupled plasma mass spectrometry
LMWL	local meteoric water line
NMDS	non-metric multidimensional scaling
NOSAMS	National Ocean Sciences Accelerator Mass Spectrometry
NWQL	National Water Quality Laboratory
SGS	Société Générale de Surveillance (USGS contract laboratory)
SPLP	synthetic precipitation leaching procedure
TEM	transient electromagnetics
USGS	U.S. Geological Survey

Assessment of Geochemical and Hydrologic Conditions near Old Yuma Mine in Saguaro National Park, Arizona, 2014–17

By Kimberly R. Beisner and Floyd Gray

Abstract

The Old Yuma Mine is an abandoned copper, lead, zinc, silver, and gold mine located within the boundaries of Saguaro National Park, Tucson Mountain District, Arizona. This study analyzed the geochemistry of sediments associated with the Old Yuma mine and assessed hydrologic and geochemical conditions of groundwater to evaluate the area surrounding the Old Yuma Mine. The purpose of the study was to establish the geochemical signature of material associated with the Old Yuma Mine and to compare it with background material and groundwater in the area. Near the mine, groundwater generally flows to the northeast. A locally anomalous steep gradient in groundwater elevation is present beneath alluvial fan deposits in the center of the study area, near the projection of the Old Yuma Fault trend. Few groundwater samples exceeded the EPA drinking water standards. One sample exceeded the EPA primary drinking water standard for arsenic; one sample exceeded the EPA secondary drinking water standard for chloride, iron, and manganese and two other samples exceeded the total dissolved solids secondary drinking water standard. Analysis of groundwater age indicates groundwater with a component of modern water is present on the northwest side of the study area. Groundwater on the southeast side of the study area is primarily older, with a radiocarbon age ranging from approximately 600 to 6,700 years before present.

Concentrations of several elements (As, Bi, Cd, Co, Cu, Fe, Hg, In, Li, Mn, Mo, Pb, Sb, U, V, W, and Zn) were elevated in the waste rock and mine tailings compared with concentrations in sediments collected in background areas. Concentrations of four elements (As, Mo, Pb, and V) in some sediment samples were greater than the EPA regional soil screening levels and (or) Arizona Department of Environmental Quality (AZDEQ) soil screening levels. A subset of 15 sediment samples was leached according to the EPA 1312 leachate method to simulate precipitation interacting with the solid material. The pH of the leachate samples increased following the leaching procedure. Several leachate samples had concentrations that exceeded the EPA drinking water

standards for As, Mn, and Pb. Analysis of leachate samples compared to groundwater samples suggests that groundwater samples collected in this study are similar to each other and distinct from leachate samples associated with mining related material. Results suggest that at this time groundwater samples collected during this investigation are not influenced by elements leached from Old Yuma Mine materials.

Introduction

Saguaro National Park consists of two districts, the Rincon Mountain District and Tucson Mountain District on the far eastern and western sides, respectively, of the city of Tucson, Arizona (fig. 1). The Tucson Mountain District historically experienced gold and silver mining activity from 1880 to the 1970s and in 1994 Saguaro National Park acquired one of these mines, the Old Yuma Mine. The Old Yuma Mine was active from the dawn of the twentieth century through World War I, and produced steel-hardening minerals such as wulfenite, molybdenite, and vanadinite, and the base and precious metals lead, copper, zinc, silver, and gold (National Park Service, 2010).

Mining History

Located on a fault that trends east-northeast and dips steeply to the southeast, the Old Yuma Mine contains a relatively wide lenticular surface expression and a ~300-foot (ft) inclined shaft that dips at an angle of 43° and provides access to its underground workings (Wilson and Schlepp, 2008). Horizontal underground workings occur at the 65-, 100-, 200-, and 300-ft levels off the main incline. Between 1916 and 1947, this underground mine produced 5,700 tons of ore grading 4 percent lead, 1 percent copper, 0.6 percent zinc, 0.3 percent molybdenum, 1 ounce silver per ton, and 0.1 ounce gold per ton. This mine also produced high quality specimens of wulfenite (PbMoO_4), a lead-molybdenum oxide, and vanadinite ($\text{Pb}_5(\text{VO}_4)_3\text{Cl}$), a lead-vanadium mineral.

2 Geochemical and Hydrologic Conditions near Old Yuma Mine in Saguaro National Park

The first claim to the Old Yuma Mine was filed in 1885, and a mill capable of handling 100 tons per day was constructed on site in 1916 for concentrating gold, molybdenum, and vanadium (Wilson and Schlepp, 2008). The mine changed ownership in 1930 and occasionally produced dump ore and surface material, but the mine was primarily used for acquiring mineral specimens. Around 1969, the ceiling of the main mine incline shaft caved in and large slabs of rock fell in single pieces.

In addition to waste rock located around the property, approximately 7,000 cubic yards of tailings remain stockpiled at the Old Yuma Mine site, though this is only part of the original tailings pile. The remainder of the pile was used for road base in the surrounding area (Michael Baker Jr., Inc., 2005). The current mine site includes a large inclined excavation open to the surface, shafts (inclined and vertical), adits (nearly horizontal passageways into the mine), a headframe that was used to hoist the inclined main access shaft, a concrete mill foundation, a solid waste dumping area, and a small leach pad. The leach pad was constructed in 1984 for the purpose of reducing gold ore from the remnant mine tailings, but it was never operational (Michael Baker Jr., Inc., 2005).

A local claimant, Richard A. Bideaux, received a patent on the valid claims from the Bureau of Land Management (BLM) near the time the land transferred from BLM to National Park Service management in 1994 (Comet 1 Lode, Old Yuma #1 Lode, and Old Yuma Placer Mining Claims, which were top-staked on one another and occupied a total of about 22 acres [9 hectares]). Saguaro National Park's primary concern regarding this mine is potential injury owing to onsite hazards (National Park Service, 2010). Old Yuma Mine is currently under a Comprehensive Environmental Response, Compensation, and Liability Act (CERCLA, or "Superfund") investigation (National Park Service, 2010).

Geologic Setting

The Tucson Mountains, in which the Old Yuma Mine is located, are underlain by Late Cretaceous volcanic rocks interpreted as part of the fill of a large ash-flow caldera (Lipman, 1993). Volcanic rocks of the Old Yuma Mine area consist of compositionally diverse lava flows, intrusive dikes, and interleaved sedimentary rocks. Rock units around the Old Yuma Mine are described as aphanitic andesite [Kya] and aphanitic rhyolite and dacite flows [Kyr] (Lipman, 1993). Aphanitic andesite consists of dark-gray, fine-grained andesitic lava flows containing 20–40 percent small phenocrysts of plagioclase, augite, and serpentine pseudomorphs after olivine or orthopyroxene. Aphanitic rhyolite and dacite flows are described as tan to light-gray lava flows containing minor small phenocrysts of sanidine, plagioclase, and recrystallized biotite (Lipman, 1993).

The ore deposit at the Old Yuma Mine consists of a porphyritic andesite or latite dike occupying a dip-slip fault dipping at about 43° through Cretaceous andesite (Wilson and

Schlepp, 2008). The average width of the dike is 8–10 ft, but widens to 20 ft on the 65-ft level. The dike contains scattered pods of silver-rich galena altered to anglesite and cerussite, which released lead for the crystallization of vanadinite and wulfenite. Vanadinite and wulfenite occur in distinct zones and were reported to be milled in separate bins; wulfenite is found primarily on the western part of the fissure and vanadinite on the eastern part (Wilson and Schlepp, 2008).

Hydrologic Setting

No perennial surface water features exist in the study area, but ephemeral washes are present that flow episodically following precipitation events. Groundwater is present in the study area generally as part of fractured bedrock, alluvium, and alluvial fan deposits.

Purpose and Scope

Updated information on groundwater levels in the Old Yuma Mine area and chemistry of mining-related materials and groundwater are needed by Saguaro National Park for a better understanding of the presence and quality of groundwater near the Old Yuma Mine. This report (1) presents a groundwater surface elevation map to estimate the groundwater elevation below the Old Yuma Mine; (2) characterizes the chemistry of mining-related material, background sediment, and water leached from both of these materials; and (3) compares leachate chemistry to groundwater chemistry from the surrounding area. The study area includes groundwater wells located south of Ina Road, north of Camino del Cerro Road, west of Silverbell Road and east of Golden Gate Road (fig. 1).

Methods

Field

Groundwater Elevation

Groundwater-level measurements were collected from 29 sites throughout the study area between December 2014 and February 2017 (table 1; fig. 2). Twenty of the sites were domestic wells that were in use during the study period, and measurements at these sites represent recently pumped water levels. Eight of the sites were wells that are no longer in use, and measurements at these sites represent static water levels. One site was an abandoned mine shaft that has standing water at the bottom.

Figure 1. Map of the study area. *A*, Shaded-relief map showing the location of Old Yuma Mine relative to Saguaro National Park boundary and Tucson, Arizona. *B*, Geologic map of the study area with groundwater sample locations shown.

Figure 1.—Continued

EXPLANATION

Surficial deposits (mostly Holocene and Pleistocene)

- Qal Alluvium
- Qc Colluvium
- Qf Alluvial-fan deposits
- Qfo Older alluvial-fan deposits

Middle Tertiary rocks (mostly Oligocene)

- Tsf Safford Dacite lava flows
- Tss Safford Dacite vent spatter
- Tst Tuffaceous rocks
- Tc Tertiary colluvium

Rocks of the Tucson Mountains caldera cycle (Lower Tertiary and Upper Cretaceous)

- TKai Andesite dikes and irregular intrusions
- Kgp Porphyritic granodiorite dikes and irregular intrusions
- Kad Amole granodiorite pluton
- Kyap Yuma Mine porphyritic andesite
- Kya Yuma Mine aphanitic andesite
- Kyr Yuma Mine aphanitic rhyolite and dacite flow
- Kyrb Yuma Mine aphanitic rhyolite and dacite flow breccia
- TKvs Yuma Mine volcanoclastic sedimentary rocks
- Kow Cat Mountain Tuff densely welded rhyolite
- Kcn Cat Mountain Tuff nonwelded to partly welded rhyolite
- Kcma Cat Mountain Tuff andesite megabreccia
- Kcms Cat Mountain Tuff megabreccia (sedimentary clasts)

- Contact
- Fault—Solid where location is certain, dashed where approximate, dotted where concealed
- Road
- Mine location
- Groundwater sample location—Each labeled site has a unique symbol

Table 1. Groundwater-level measurement site information.

[Map site number refers to number on figure 2; USGS, U.S. Geological Survey; AZ014, Arizona Department of Environmental Quality; GW, groundwater well; SB-TSM, subsurface tunnel, shaft, or mine; NAD83, North American Datum of 1983; NAVD88, North American Vertical Datum of 1988; –, unknown]

Map site number	Agency code	USGS site number	Site name	Site type	Latitude	Longitude	Horizontal datum	Altitude, in feet	Altitude accuracy, in feet	Vertical datum	Well depth, in feet	Screened interval lithology
1	USGS	321713111062401	D-13-12 22CAA1	GW	32.28683	–111.10668	NAD83	2,654.6	0.1	NAVD88	495	Brown and gray volcanic
2	USGS	321749111051001	D-13-12 14DCB	GW	32.29684	–111.08712	NAD83	2,383.9	0.1	NAVD88	405	–
3	USGS	321808111063001	D-13-12 15BDC1	GW	32.30209	–111.10845	NAD83	2,581.7	0.1	NAVD88	500	Shale and lime-stone
4	USGS	321817111061601	D-13-12 15ACB1	GW	32.30475	–111.10442	NAD83	2,480	10	NAVD88	400	Unknown
5	USGS	321817111062501	D-13-12 15BDA1	GW	32.30474	–111.10697	NAD83	2,513.5	0.1	NAVD88	425	Gray basalt
6	USGS	321833111060601	D-13-12 10DCD1	GW	32.30906	–111.10159	NAD83	2,489.8	0.1	NAVD88	500	Volcanics
7	USGS	321836111040601	D-13-12 12DCD1	GW	32.31007	–111.06837	NAD83	2,276.3	0.1	NAVD88	300	Sand and gravel
8	USGS	321836111064801	D-13-12 10CDB1	GW	32.30998	–111.11399	NAD83	2,448	0.1	NAVD88	–	–
9	USGS	321840111055001	D-13-12 10DDA1	GW	32.31103	–111.09729	NAD83	2,449	0.1	NAVD88	–	–
10	USGS	321859111075101	D-13-12 09BCC1	GW	32.31636	–111.13097	NAD83	2,492	0.1	NAVD88	77.5	–
11	USGS	321905111083701	D-13-12 08BDB1	SB-TSM	32.31796	–111.14356	NAD83	2,560.3	0.1	NAVD88	–	–
12	AZ014	321907111065001	D-13-12 09ADA1	GW	32.31925	–111.11444	NAD83	2,371.2	0.1	NAVD88	175	–
13	USGS	321914111063401	D-13-12 10BAC3	GW	32.32068	–111.10949	NAD83	2,372.9	0.1	NAVD88	686	Granite (likely volcanic)
14	USGS	321916111064301	D-13-12 10BBC1	GW	32.32125	–111.11200	NAD83	2,361.9	0.1	NAVD88	405	Sand, black, gray, and red rock
15	USGS	321917111062702	D-13-12 10BAC2	GW	32.32127	–111.10762	NAD83	2,380.3	0.1	NAVD88	687	Brown andesite conglomerate
16	USGS	321917111065101	D-13-12 09AAA1	GW	32.32140	–111.11407	NAD83	2,356.2	0.1	NAVD88	460	Rock with strips of sand rock, rock very hard
17	USGS	321923111062701	D-13-12 10BAA1	GW	32.32302	–111.10743	NAD83	2,362.1	0.1	NAVD88	475	Sandy volcanic
18	USGS	321923111064601	D-13-12 10BBB1	GW	32.32298	–111.11282	NAD83	2,347.9	0.1	NAVD88	400	Sand, black, gray, and red rock

Table 1. Groundwater-level measurement site information.—Continued

[Map site number refers to number on figure 2; USGS, U.S. Geological Survey; AZ014, Arizona Department of Environmental Quality; GW, groundwater well; SB-TSM, subsurface tunnel, shaft, or mine; NAD83, North American Datum of 1983; NAVD88, North American Vertical Datum of 1988; —, unknown]

Map site number	Agency code	USGS site number	Site name	Site type	Latitude	Longitude	Horizontal datum	Altitude, in feet	Altitude accuracy, in feet	Vertical datum	Well depth, in feet	Screened interval lithology
19	USGS	321926111055701	D-13-12 03DDDI	GW	32.32388	−111.09919	NAD83	2,338.7	0.1	NAVD88	365	Alluvial
20	USGS	321928111063301	D-13-12 03CDCI	GW	32.32440	−111.10928	NAD83	2,342.8	0.1	NAVD88	—	—
21	USGS	321929111060501	D-13-12 03DCDI	GW	32.32471	−111.10144	NAD83	2,341.4	0.1	NAVD88	447	Andesite conglomerate
22	USGS	321929111070601	D-13-12 04DCD	GW	32.32500	−111.11902	NAD83	2,385.3	0.1	NAVD88	172	Malapic
23	USGS	321931111061101	D-13-12 03DCAI	GW	32.32523	−111.10316	NAD83	2,336.4	0.1	NAVD88	438	Alluvial
24	USGS	321932111063601	D-13-12 03CCAI	GW	32.32558	−111.11002	NAD83	2,321.9	0.1	NAVD88	390	Sand and gravel
25	USGS	321937111061201	D-13-12 03DBD2	GW	32.32702	−111.10334	NAD83	2,322.1	0.1	NAVD88	465	Alluvial
26	USGS	321941111061201	D-13-12 03DBDI	GW	32.32806	−111.10339	NAD83	2,310	10	NAVD88	370	Cemented boulders
27	USGS	321941111064801	D-13-12 04DADI	GW	32.32799	−111.11419	NAD83	2,320.5	0.1	NAVD88	350	Decomposed granite
28	USGS	321947111060901	D-13-12 03DBAI	GW	32.32975	−111.10237	NAD83	2,292.3	0.1	NAVD88	284	Alluvial
29	USGS	321955111064401	D-13-12 03BCCI	GW	32.33194	−111.11228	NAD83	2,301.3	0.1	NAVD88	390	Sandy gravel, red rock, purple rock, basalt

Figure 2. Groundwater-elevation contour map. Groundwater-level measurement site numbers are given in table 1. Blue arrows represent groundwater flow direction (perpendicular to contour lines and flowing from higher to lower elevations).

EXPLANATION

Surficial deposits (mostly Holocene and Pleistocene)

- Qal** Alluvium
- Qc** Colluvium
- Qf** Alluvial-fan deposits
- Qfo** Older alluvial-fan deposits

Middle Tertiary rocks (mostly Oligocene)

- Tsf** Safford Dacite lava flows
- Tss** Safford Dacite vent spatter
- Tst** Tuffaceous rocks
- Tc** Tertiary colluvium

Rocks of the Tucson Mountains caldera cycle (Lower Tertiary and Upper Cretaceous)

- TKai** Andesite dikes and irregular intrusions
- Kgp** Porphyritic granodiorite dikes and irregular intrusions
- Kad** Amole granodiorite pluton
- Kyap** Yuma Mine porphyritic andesite
- Kya** Yuma Mine aphanitic andesite
- Kyr** Yuma Mine aphanitic rhyolite and dacite flow
- Kyrb** Yuma Mine aphanitic rhyolite and dacite flow breccia
- TKvs** Yuma Mine volcanoclastic sedimentary rocks
- Kcw** Cat Mountain Tuff densely welded rhyolite
- Kcn** Cat Mountain Tuff nonwelded to partly welded rhyolite
- Kcma** Cat Mountain Tuff andesite megabreccia
- Kcms** Cat Mountain Tuff megabreccia (sedimentary clasts)

- Contact**
- Fault**—Solid where location is certain, dashed where approximate, dotted where concealed
- Groundwater elevation contour**
2550
- Mine location**
- Groundwater-level measurement site**

Precise Trimble differential global positioning system (GPS) was used to make the GPS measurements at all sites. Groundwater elevation was calculated by subtracting the water-level measurement below the land surface from the land surface elevation at each site. For sites with more than one water-level measurement, the average of the measurements was used. Groundwater elevation data were contoured in ArcMap 10.5 using natural neighbor interpolation and some minor manual edits.

Water Sampling

Groundwater samples were collected for water-quality analyses from eight wells following standard U.S. Geological Survey (USGS) protocols (U.S. Geological Survey, variously dated; fig. 1B). Before water samples were collected, field parameters including pH, water temperature, specific conductance, dissolved oxygen, and barometric pressure were measured in a flow-through cell during well purging at each well site. Well discharge was measured by field personnel using volumetric techniques or was reported by the owner. Water level, casing dimensions, and pumping rate were used to calculate purge volume and time required to purge three casing volumes prior to sample collection.

Water samples were filtered (0.45 micron, μm) for major cations, trace elements, alkalinity, nutrients, ^{14}C , sulfur isotopes, perchlorate, and lead and strontium isotopes. The major cations, trace elements, and lead and strontium isotope samples were preserved to $\text{pH} < 2$ by adding ultrapure nitric acid. Unfiltered samples were collected for tritium, stable isotopes, and sulfur hexafluoride (SF_6). Alkalinity (field) was computed from titration data using the incremental equivalence method (U.S. Geological Survey, variously dated). Dissolved gases of nitrogen and argon were collected in glass septum bottles, filled, and sealed with a rubber stopper punctured with a needle and removed underwater in a beaker. Two separate samples of dissolved gases and SF_6 were collected and analyzed for each sample.

Sediment Sampling

Sediment samples were collected from 38 sites: 10 from two tailings piles; 10 from a large waste rock pile at the Old Yuma Mine site; 5 along an ephemeral drainage originating at and draining the mine site; and 13 background samples from both sides of the hill to the south of the Old Yuma Mine site. At each site, field technicians collected samples by delineating a 1-square-yard area, and then compositing 10 evenly spaced scoops of soil within each area into a plastic bag. A plastic garden scoop was used to collect soil samples including the top layer and ~2 inches below the surface, and the scoop was cleaned with deionized water between each sample location. Sediment samples were passed through a 2-millimeter sieve

before analysis. Sediment samples use the naming convention OYM-04-T, for example, where the number refers to the sediment sample sequence number and the ending letter refers to the sample type. Samples were numbered sequentially by collection time and assigned one of the following letters: T for tailings, W for waste rock, S for stream sediment, or B for background.

Analytical Water

Water samples were analyzed for major cations, trace elements, and nutrients by the USGS National Water Quality Laboratory (NWQL). Analytical methods from the USGS NWQL included inductively coupled plasma mass spectrometry (ICP-MS) to determine concentrations of Al, Sb, As, Ba, Be, B, Cd, Cr, Co, Cu, Pb, Li, Mo, Ni, Se, Ag, Sr, Tl, W, U, V, and Zn (Garbarino and others, 2006). Inductively coupled plasma atomic emission spectrometry (ICP-AES) was used to analyze for Ca, Fe, K, Mg, Mn, and Na (Fishman, 1993). Anions Cl^- , F^- , and $(\text{SO}_4)^{2-}$ were analyzed by ion chromatography and SiO_2 was analyzed by discrete analyzer colorimetry (Fishman and Friedman, 1989). Nitrate (NO_3^-) plus nitrite (NO_2^-) were analyzed by colorimetry (Patton and Kryskalla, 2011). Perchlorate (ClO_4^-) was analyzed by Weck Laboratories, Inc.

Stable isotope ratios ($\delta^{18}\text{O}$ and $\delta^2\text{H}$) were measured at the USGS Reston Stable Isotope Laboratory following methods by Révész and Coplen (2008a, b). The 2-sigma uncertainties for the stable isotope analyses are 0.2 per mil for $\delta^{18}\text{O}$ and 2 per mil for $\delta^2\text{H}$, reported relative to Vienna standard mean ocean water. The Reston Stable Isotope Laboratory measured $\delta^{34}\text{S}$ of sulfate following methods by Révész and others (2012).

The USGS National Research Program Laboratory in Menlo Park, Calif., measured strontium isotope ratios ($^{87}\text{Sr}/^{86}\text{Sr}$) using methods described in Bullen and others (1996). These methods are precise to 0.00002 or better at the 95-percent confidence level.

^{14}C and $\delta^{13}\text{C}$ ratios were analyzed by the National Ocean Sciences Accelerator Mass Spectrometry (NOSAMS) at Woods Hole Oceanographic Institution. ^{14}C values (reported by NOSAMS as absolute percent modern carbon) were denormalized using equation 5 of Plummer and others (2012) to percent modern carbon (pmc). NetpathXL computed corrected groundwater ages using model 11 “Revised F&G solid ex” (Parkhurst and Charlton, 2008). Groundwater age was computed with ^{14}C values of 0 and 10 pmc for carbonate rock and 100 pmc for soil CO_2 , assuming $\delta^{13}\text{C}$ values of -4.5 per mil for carbonate and -19.1 per mil for soil CO_2 (Kalin, 1994). The University of Miami Tritium Laboratory measured tritium using the electrolytic enrichment and gas counting method, with a reporting limit of 0.3 picocuries per liter (pCi/L).

Dissolved gases of nitrogen and argon were analyzed at the USGS Groundwater Dating Laboratory by methods documented in U.S. Geological Survey (2017a), which are summarized below. The lab analyzed the samples using a Hewlett Packard model 7890B gas chromatograph with helium as the carrier gas. The headspace gas pressure was measured with a pressure transducer. The sample gas was then introduced simultaneously into two sampling loops. One sampling loop was injected into an Alltech CTR-III column at 55 °C to separate argon, nitrogen, and oxygen. These gases were quantified with a thermal conductivity detector. The second sampling loop was injected into an Alltech CTR-I column at 30 °C to separate methane (CH₄) and carbon dioxide (CO₂) from the other constituents. After separation, the gas stream was passed through a nickel methanizer converting CO₂ to CH₄ and the two gases were quantified with a flame ionization detector. The gas chromatograph was calibrated with four gravimetric gas standards and one National Oceanic and Atmospheric Administration air standard at the beginning of each day and checked again at the end of each day. Instrument drift was generally less than 1 percent for argon, nitrogen, and oxygen (Ar, N₂, and O₂) and 1–2 percent for CH₄ and CO₂.

The USGS Groundwater Dating Laboratory also analyzed SF₆ samples, using a purge and trap gas chromatography procedure with an electron capture detector and following the methods summarized below (U.S. Geological Survey, 2017a). The apparatus used for vacuum extraction of SF₆ from groundwater is similar to the system described by Law and others (1994) and Busenberg and Plummer (2000). The apparatus consists of a 950-milliliter glass stripping vessel and various valves that control the vacuum and the flow of gases and water. For water samples, the stripped gas is trapped on a large trap immersed in an isopropyl alcohol-dry ice bath at about -70 °C. The trapped SF₆ is transferred into a small trap cooled in the isopropyl alcohol-dry ice bath by heating the large trap to 96 °C. The small trap is then heated to 96 °C, and opened to inject the SF₆ into the gas chromatograph. The measurement is done by an electron capture detector that is controlled by an integrator and a computer.

Sediment

Sediment samples were analyzed by the USGS Central Region Mineral Resources Laboratory contract laboratory Société Générale de Surveillance (SGS) for major and trace elements following digestion using hydrochloric, nitric, perchloric, and hydrofluoric acids at low temperature. Digested samples were analyzed by ICP-AES and ICP-MS. Calibration on the ICP-AES was performed by standardizing with digested rock reference materials and a series of multi-element solution standards. The ICP-MS was calibrated with aqueous standards, and internal standards were used to compensate for matrix effects and internal drifts. Data were deemed acceptable if recovery for all 42 elements was ±15 percent at five times the lower limit of determination (U.S. Geological Survey,

2013). Total carbon and carbonate carbon were measured for every sediment sample, and the difference between total carbon and carbonate carbon was used to calculate organic carbon. The SGS analyzed total carbon using an automated carbon analyzer, where a weighted sample is combusted in an oxidic atmosphere at 1,370 °C to oxidize carbon to carbon dioxide. Moisture and dust are removed and the carbon dioxide is measured by a solid-state infrared detector. Carbonate carbon is determined as carbon dioxide by coulometric titration. The sample is treated with hot 2-normal perchloric acid and the evolved carbon dioxide is passed into a cell containing a solution of monoethanolamine. The carbon dioxide, quantitatively absorbed by the monoethanolamine, is coulometrically titrated using platinum and silver/potassium-iodide electrodes (U.S. Geological Survey, 2013). The lab determined mercury following digestion using nitric and hydrochloric acids using a FIMS-100 (flow injection mercury system) cold-vapor atomic absorption mercury analyzer (U.S. Geological Survey, 2013).

A split of the sediment samples was analyzed by a partial digestion method by the USGS Central Region Mineral Resources Laboratory using the U.S. Environmental Protection Agency (EPA) 3050 method. The partial digestion samples were analyzed by ICP-MS and ICP-AES.

A separate split of 15 of the sediment samples (4 tailings, 4 waste rock, 2 stream sediment, and 5 background) was used to perform a synthetic precipitation leaching procedure (SPLP) by SGS (EPA SPLP method 1312; U.S. Environmental Protection Agency, 1994). The EPA 1312 method uses a 20:1 liquid to solid ratio. The resulting leachate liquid was analyzed by the same methods as the sediment digestion at SGS using ICP-AES and ICP-MS. Seven leachate samples were filtered with a 0.45-µm filter and acidified to pH<2 with ultrapure nitric acid. The USGS National Research Program Laboratory in Menlo Park, Calif., then measured strontium isotope ratios (⁸⁷Sr/⁸⁶Sr), using methods described in Bullen and others (1996).

Quality Assurance Procedures

Water

One field blank was collected for the groundwater sample set from the site where a Grundfos RediFlo2 portable pump was used to collect the groundwater sample. Certified inorganic blank water was pumped through the pump and sample tubing at the well site prior to sample collection to obtain the field blank. One sequential replicate sample was collected from a different site than the blank sample.

Sediment

Four field replicate samples were analyzed for total and partial digestions. Three laboratory replicate samples were analyzed for the total digestion and one additional lab replicate for total carbon.

Sediment reference materials are homogenized materials that have been analyzed at multiple laboratories to obtain

a common value and distribution statistics. The analysis of reference materials along with environmental samples provides an understanding of how accurate laboratory results are. Four reference materials were obtained through the USGS Geochemical Reference Materials program: andesite, AGV-2 (Wilson, 1998a); granodiorite, GSP-2 (Wilson, 1998b); Cody Shale, SCo-1 (Smith, 1995); and Green River Shale, SGR-1b (Wilson, 2001). Blind reference material samples were submitted along with the sediment samples from this study for both total and partial digestion methods. An additional reference material (granodiorite, GSP-2) was submitted by the USGS Central Region Mineral Resources Laboratory to the contract laboratory for quality control analysis.

Three reference materials specific to the EPA 3050 leachate method—2709 (San Joaquin soil), 2710 (Montana soil), and 2711 (Montana soil)—were analyzed by the USGS Central Region Mineral Resources Laboratory during the analysis of the partial digestion samples.

Sediment Leachate

One replicate was analyzed for sediment leachate samples. The replicate was a separate split from the original sediment sample bag that was leached as a separate sample from the normal sample split and represents variability within a sample as well as lab variability. A separate sample replicate was analyzed for strontium isotopes.

Data Analysis

The data were analyzed using statistical methods to understand similarities and differences of samples within and between groups. The majority of trace elements had one or more values below a laboratory reporting level, with several elements having multiple reporting levels. The statistical methods of Helsel (2012) for data with values below the laboratory reporting level were used to analyze the majority of analytes presented in this study. Several measured trace elements (Al, As, Ba, Mo, Se, Sr, V, and U) did not result in values below the laboratory reporting level for the dataset reported here.

For censored data (in this case, data below the reporting level), boxplots for elements were made using the “cenboxplot” function from the NADA package (Lee, 2015) in R statistical computing environment (R Core Team, 2015). Outlier data points on boxplots were defined for this study as greater than 1.5 times the interquartile range. We analyzed the chemical analysis data from samples for each constituent to determine

the Kaplan-Meier model of the data using “cenfit” from the NADA package in R (Lee, 2015). A p-value threshold of 0.05 (95 percent confidence level) was used to indicate statistical significance for all mentioned statistical tests.

We grouped samples by type of sediment, sediment leachate, and groundwater, then compared sample groups using “cendiff” from the NADA package in R (Lee, 2015). The “cendiff” function uses the Peto-Prentice test (Helsel and Lee, 2006) to determine if there were significant differences between the groups for elements with censored data.

Non-metric multidimensional scaling (NMDS) was used to reduce the complex data structure (many samples and many elements) to represent the pairwise dissimilarity between objects in a low-dimensional space (Buttigieg and Ramette, 2014). We computed U-scores of the data using the “uscore” function for R from Helsel (2016) with default values to calculate the ranks of the scores (Helsel, 2012, 2016). NMDS was performed on the U-scores using “metaMDS” from the vegan package in R (Oksanen and others, 2016) using Euclidean distance, zerodist = add, and autotransform = false (Helsel, 2012). NMDS stress values ≤ 0.1 are considered fair, values ≤ 0.05 indicate good fit, and values ≥ 0.2 are deemed suspect (Buttigieg and Ramette, 2014). For sediment samples, results from all elements were used, and for leachate versus groundwater, a subset of elements (Al, Ag, As, Ba, Cd, Cr, Cu, Mn, Mo, Ni, Pb, Sb, Se, Sr, U, V, and Zn) were used as they were available within both datasets.

A cluster analysis was used to identify similar groups of samples by evaluating minimum differences within groups and maximum differences among groups using the “hclust” function for the elements used in the NMDS analysis. The Calinski criterion was applied with the “cascadeKM” function of the vegan package in R (Oksanen and others, 2016) to determine the number of clusters that maximizes the difference between clusters while minimizing the differences within clusters. The “ANOSIM” function was used to statistically evaluate whether or not groups of samples have significantly different concentration patterns (Helsel, 2012).

Water sample concentrations were compared to the EPA drinking water standards presented in table 2 (U.S. Environmental Protection Agency, 2000, 2017a).

Sediment sample concentrations were compared to the EPA regional and Arizona Department of Environmental Quality (AZDEQ) soil screening levels, presented in table 3 (U.S. Environmental Protection Agency, 2017b; Arizona Department of Environmental Quality, 2009). A comprehensive risk assessment would be needed to understand the screening levels appropriate for the exposure pathways present at the site.

Table 2. U.S. Environmental Protection Agency water-quality standards for drinking water (U.S. Environmental Protection Agency, 2000, 2017a).

[Values presented in units used in this report. NA, not available; MCL, maximum contaminant level; SMCL, secondary maximum contaminant level; TDS, total dissolved solids; $\mu\text{g/L}$, microgram per liter; mg/L , milligram per liter]

Constituent	Units	Primary drinking-water standard	Secondary drinking-water standard
		MCL	SMCL
Al	$\mu\text{g/L}$	NA	50–200
Sb	$\mu\text{g/L}$	6	NA
As	$\mu\text{g/L}$	10	NA
Ba	$\mu\text{g/L}$	2,000	NA
Be	$\mu\text{g/L}$	4	NA
Cd	$\mu\text{g/L}$	5	NA
Cl^-	mg/L	NA	250
Cr	$\mu\text{g/L}$	100	NA
Cu	$\mu\text{g/L}$	1,300	1,000
F^-	mg/L	4	2
Fe	$\mu\text{g/L}$	NA	300
Pb	$\mu\text{g/L}$	15	NA
Mn	$\mu\text{g/L}$	NA	50
NO_3^- , as N	mg/L	10	NA
pH	standard scale	NA	6.5–8.5
Se	$\mu\text{g/L}$	50	NA
Ag	$\mu\text{g/L}$	NA	100
SO_4^{2-}	mg/L	NA	250
TDS	mg/L	NA	500
U	$\mu\text{g/L}$	30	NA
Zn	$\mu\text{g/L}$	NA	5,000

Table 3. U.S. Environmental Protection Agency (EPA) and Arizona Department of Environmental Quality (AZDEQ) soil screening levels (U.S. Environmental Protection Agency, 2017b; Arizona Department of Environmental Quality, 2009).

[Values in milligrams per kilogram (mg/kg); NA, not available]

Element	EPA (composite worker)		AZDEQ (non-residential)
	Carcinogenic target risk	Non-cancer hazard index	
Al	NA	1,100,000	920,000
Sb	NA	470	410
As	3	480	10
Ba	NA	220,000	170,000
Be	6,900	2,300	1,900
Cd	9,300	980	510
Co	1,900	350	13,000
Cu	NA	47,000	41,000
Pb	NA	800	800
Mn	NA	26,000	32,000
Hg	NA	46	310
Mo	NA	5,800	5,100
Ni	NA	NA	20,000
Ag	NA	5,800	5,100
U	NA	230 ^a	200
V	NA	5,800	1,000
Zn	NA	350,000	310,000

^a Uranium soluble salts.

Results

Quality Assessment

Water

Charge balance difference for all samples was less than 5 percent, with a maximum percentage difference of 3.2 percent. All samples were checked and those with values greater than the EPA drinking water standards (U.S. Environmental Protection Agency, 2000, 2017a; table 2) were rerun and verified. Equipment for field parameter measurement and alkalinity titrations was tested during the USGS annual National Field Quality Assurance project, and measurements produced results within the acceptable range.

Nine constituents from the one blank sample collected had values above the laboratory reporting level (Ca, Cl,

ammonia [NH₄], Cr, Cu, Mn, Mo, Ni, and Zn; table 4). Calcium and chloride concentrations in the blank sample were more than 100 times less than those in the environmental samples (table 4). Ammonia, Cr, Cu, Mn, Mo, Ni, and Zn concentrations in the blank samples were less than 10 times some, or all, of each respective constituent concentration in the environmental samples. The blank was collected from the portable pump, which was used at one sample site and likely represents greater potential for contamination compared with other dedicated pump sites. More blank measurements would be needed to statistically understand the potential bias from contamination in the sampling equipment and field conditions.

One groundwater replicate was sampled for this study and the relative percentage differences between the environmental sample and the replicate were less than 10 percent for all constituents, except for lead, perchlorate, and zinc (table 5). More replicate samples would be needed to statistically quantify the variability for each element.

Table 4. Results of field blank analyses.[**Bold** values indicate detection above the laboratory reporting level. mg/L, milligrams per liter; µg/L, micrograms per liter; NA, not available]

Constituent	Units	Detection level	Reporting level	Blank value	Environmental sample range
Ca	mg/L	0.022	0.044	0.153	18.5–364
Mg	mg/L	0.011	0.022	<0.011	0.863–63.7
Na	mg/L	0.06	0.12	<0.06	47–167
K	mg/L	0.03	0.06	0.04	1.39–6.61
Cl ⁻	mg/L	0.02	0.04	0.049	27.7–742
SO ₄ ²⁻	mg/L	0.02	0.04	0.025	20–134
F ⁻	mg/L	0.01	0.02	<0.01	0.15–0.75
NH ₃ , as N	mg/L	0.01	0.02	0.022	<0.01–1.83
NO ₂ , as N	mg/L	0.001	0.002	<0.001	<0.001–0.004
NH ₃ plus NO ₂ , as N	mg/L	0.04	0.08	<0.04	3.86–8.87
PO ₄ ³⁻ , as P	mg/L	0.004	0.008	<0.004	0.009–0.053
Al	µg/L	3	6	<3	5.2–9.2
As	µg/L	0.1	0.2	<0.1	2.8–10.7
Sb	µg/L	0.027	0.054	<0.027	<0.054–1.65
Ba	µg/L	0.6	1.2	0.69	13.4–228
Be	µg/L	0.19	0.38	<0.19	<0.19–<0.76
Cd	µg/L	0.6	1.2	<0.6	<0.6–<2.4
Cr	µg/L	0.3	0.6	0.99	<0.30–6.5
Cu	µg/L	0.8	1.6	3.54	<0.8–2
Fe	µg/L	4	8	6.55	<4–2,020
Pb	µg/L	0.04	0.08	0.048	<0.08–3.68
Mn	µg/L	0.2	0.4	1.4	<0.2–2,200
Mo	µg/L	0.05	0.1	4.58	2.97–14.6
Ni	µg/L	0.2	0.4	2.14	<0.2–2.4
Se	µg/L	0.05	0.1	<0.05	0.64–2.5
Ag	µg/L	0.02	0.04	<0.02	<0.02–<1
U	µg/L	0.014	0.028	<0.014	1.32–11.3
V	µg/L	0.6	1.2	<0.6	6.6–20.4
Zn	µg/L	1.9	3.8	4.62	<1.9–108
ClO ₄ ⁻	µg/L	NA	0.1	<0.1	0.67–1.48

Table 5. Results of groundwater replicate analyses.

[Bold value indicates the percentage difference is greater than 10 percent. mg/L, milligram per liter; µg/L, microgram per liter; pCi/L, picocurie per liter; pM, absolute percent modern carbon (normalized); fg/kg, femtogram per kilogram (1 femtogram = 10⁻¹⁵ grams); µS/cm, microsiemens per centimeter; R, radiochemistry non-detect; NA, not applicable]

Constituent	Units	Environmental result	Replicate result	Concentration difference	Absolute percent difference
Ca	mg/L	18.5	18.7	-0.2	1.08
Mg	mg/L	0.863	0.878	-0.02	1.72
Na	mg/L	103	105	-2	1.92
K	mg/L	1.39	1.43	-0.04	2.84
Cl ⁻	mg/L	39.5	39.6	-0.1	0.18
SO ₄ ²⁻	mg/L	35.7	35.7	0.01	0.02
F ⁻	mg/L	0.752	0.754	-0.002	0.27
Alkalinity, as CaCO ₃ , field	mg/L	166.4	164.4	2.0	1.21
HCO ₃ ⁻ , field	mg/L	201.6	199.2	2.4	1.20
CO ₃ ²⁻ , field	mg/L	0.6	0.6	0.0	0.00
NH ₃ , as N	mg/L	0.0227	<0.01	NA	NA
NO ₂ , as N	mg/L	<0.001	<0.001	NA	NA
NH ₃ plus NO ₂ , as N	mg/L	6.03	5.97	0.06	1.02
PO ₄ ³⁻ , as P	mg/L	0.0116	0.0118	-0.0002	1.71
Al	µg/L	6.62	6.39	0.23	3.49
Sb	µg/L	0.0329	<0.027	NA	NA
As	µg/L	8.06	8.32	-0.26	3.17
Ba	µg/L	20.9	21.2	-0.3	1.43
Be	µg/L	<0.19	<0.19	NA	NA
Cd	µg/L	<0.6	<0.6	NA	NA
Cr	µg/L	6.51	6.75	-0.24	3.62
Cu	µg/L	<0.8	<0.8	NA	NA
Fe	µg/L	<4	<4	NA	NA
Pb	µg/L	0.108	0.205	-0.097	61.98
Mn	µg/L	<0.2	<0.2	NA	NA
Mo	µg/L	14.6	15.2	-0.6	4.03
Ni	µg/L	<0.2	<0.2	NA	NA
Se	µg/L	1.41	1.46	-0.05	3.76
Ag	µg/L	<0.02	<0.02	NA	NA
Sr	µg/L	327	321	6	1.85
U	µg/L	8.66	9.09	-0.43	4.85
V	µg/L	19.2	19.4	-0.2	1.04
Zn	µg/L	6.55	20.8	-14.25	104.20
ClO ₄ ⁻	µg/L	0.93	1.25	-0.32	29.15
³ H	pCi/L	R 0.08	R -0.09	NA	NA
δ ³⁴ S, in SO ₄ ²⁻	per mil	6.86	6.79	0.07	1.03
¹⁴ C	pM	16.96	16.54	0.42	2.51

Table 5. Results of groundwater replicate analyses.—Continued

[Bold value indicates the percentage difference is greater than 10 percent. mg/L, milligram per liter; µg/L, microgram per liter; pCi/L, picocurie per liter; pM, absolute percent modern carbon (normalized); fg/kg, femtogram per kilogram (1 femtogram = 10⁻¹⁵ grams); µS/cm, microsiemens per centimeter; R, radiochemistry non-detect; NA, not applicable]

Constituent	Units	Environmental result	Replicate result	Concentration difference	Absolute percent difference
SF ₆	fg/kga	264.50	261.40	3.10	1.18
TDS ^a	mg/L	366	356	10	2.77
⁸⁷ Sr/ ⁸⁶ Sr		0.70968	0.70971	-0.00003	0.00
δ ¹³ C	per mil	-10.08	-10.07	-0.01	0.10
δ ² H	per mil	-68.50	-68.60	0.10	0.15
δ ¹⁸ O	per mil	-8.86	-8.92	0.06	0.67
pH, lab	standard scale	8.2	8.1	0.1	1.23
SpecCond, lab ^b	µS/cm	559	559	0	0.00
ANC ^c , as CaCO ₃ , lab	mg/L	169	169	0	0.00

^a TDS, total dissolved solids, at 180 °C.

^b SpecCond, specific conductance, at 25 °C.

^c ANC, acid-neutralizing capacity.

Sediment

Four field and three laboratory replicate sediment samples (plus one additional total carbon laboratory replicate) were analyzed with the total digestion method. Relative percentage differences between field environmental samples and replicates were generally less than 20 percent, with the exception of Bi (29 percent) and Hg (40 percent) for OYM-31-B, Mo (29 percent) for OYM-22-S, and Sn (49 percent) for OYM-33-B (table 6). For the laboratory replicates, one sample (OYM-05-T) had three elements whose relative percentage difference was greater than 20 percent: Hg (22 percent), Sn (21 percent), and Te (67 percent) (table 7).

For the 3050 partial-digestion method field replicates, relative percentage difference between field environmental samples and replicates were generally less than 20 percent, with the exception of Fe (26 percent) for OYM-33-B, Nd (22 percent) and Pb (23 percent) for OYM-31-B, Mo (43 percent) for OYM-22-S, and Li (22 percent) for OYM-02-T (table 8).

For standard reference materials analyzed by total digestion, elements Cr in SCo-1 (Smith, 1995), Ti in GSP-2 (Wilson, 1998b), and Fe, Mg, Cr, Mn, Pb, and Sr in AGV-2 (Wilson, 1998a) had laboratory results greater than three standard deviations of the reported value for the associated

reference material (table 9). Relative percentage difference between the reported values and the SGS laboratory values were generally less than 20 percent with the exception of Cr (24 percent) and Y (28 percent) in SCo-1; Ga (29 percent), Nb (62 percent), and Y (22 percent) in SGR-1b; Cr (35 percent) in GSP-2; Be (22 percent), Cr (47 percent), Pb (69 percent) and Sb (27 percent) in AGV-2; and CO₂ (21 percent), carbonate carbon (22 percent), organic carbon (22 percent), and Mn (22 percent) in GSP-2.

Reference materials analyzed by the EPA 3050 method represent a partial digestion and certified values do not exist with which to compare results. Rather, results can be compared with a range from other laboratory results analyzed using the same method (National Institute of Standards and Technology, 2003; table 10). Most values reported here for reference materials analyzed by the EPA 3050 method are within the range reported by other laboratories. Some constituents had values outside of the reported range, which is commonly representative of results from a smaller number of reporting laboratories (2–9), depending on the element. Titanium and vanadium were the only two elements whose laboratory values were outside of the range for all reference materials (2709, 2710, and 2711) (table 10).

Table 6. Field replicate sediment data for total digestion samples collected from Old Yuma Mine study area.

[%, percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-33-B			OYM-31-B			OYM-22-S			OYM-02-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
CO ₂	%	1.09	1.09	0.00	1.81	1.88	3.79	1.39	1.38	0.72	0.33	0.37	11.43
Total C	%	1.47	1.41	4.17	0.89	0.94	5.46	0.64	0.65	1.55	0.28	0.27	3.64
C in CO ₃ ²⁻	%	0.3	0.3	0.00	0.49	0.51	4.00	0.38	0.38	0.00	0.09	0.1	10.53
Organic C	%	1.17	1.11	5.26	0.4	0.43	7.23	0.26	0.27	3.77	0.19	0.17	11.11
Al	%	7.79	7.8	0.13	7.74	7.66	1.04	6.44	6.54	1.54	3.07	3.14	2.25
Ca	%	1.91	1.86	2.65	2.51	2.63	4.67	1.68	1.76	4.65	0.52	0.57	9.17
Fe	%	3.19	3.2	0.31	3.49	3.63	3.93	4.84	4.94	2.04	7.27	7.41	1.91
K	%	3.07	3.09	0.65	3.89	3.67	5.82	3.71	3.81	2.66	1.12	1.15	2.64
Mg	%	0.77	0.76	1.31	0.75	0.79	5.19	0.62	0.62	0.00	0.47	0.48	2.11
Na	%	1.45	1.45	0.00	1.23	1.16	5.86	0.84	0.86	2.35	0.09	0.09	0.00
S	%	0.02	0.02	0.00	<0.01	0.01		0.01	0.01	0.00	0.02	0.02	0.00
Ti	%	0.39	0.41	5.00	0.47	0.48	2.11	0.41	0.44	7.06	0.16	0.16	0.00
Ag	ppm	<1	<1		<1	<1		4	4	0.00	13	15	14.29
As	ppm	10	10	0.00	15	15	0.00	55	57	3.57	276	283	2.50
Ba	ppm	1,010	1,030	1.96	1,090	995	9.11	1,060	1,090	2.79	252	257	1.96
Be	ppm	2	1.8	10.53	1.9	2	5.13	1.6	1.6	0.00	1.3	1.2	8.00
Bi	ppm	0.33	0.35	5.88	0.18	0.24	28.57	6.3	6.65	5.41	118	122	3.33
Cd	ppm	0.4	0.4	0.00	0.5	0.6	18.18	22.9	23.9	4.27	11.1	12.1	8.62
Ce	ppm	81.8	77.2	5.79	74.6	77.1	3.30	62.6	68.7	9.29	20.2	21.3	5.30
Co	ppm	12	11.7	2.53	15.6	15.9	1.90	21.6	24.7	13.39	29.9	31.9	6.47
Cr	ppm	27	27	0.00	40	46	13.95	50	49	2.02	28	24	15.38
Cs	ppm	21	21	0.00	25	26	3.92	15	17	12.50	9	10	10.53
Cu	ppm	30	28.2	6.19	32.6	35	7.10	184	196	6.32	1900	1870	1.59

Table 6. Field replicate sediment data for total digestion samples collected from Old Yuma Mine study area.—Continued

[%; percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-33-B			OYM-31-B			OYM-22-S			OYM-02-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
Ga	ppm	18.4	18.8	2.15	19	18.9	0.53	17.1	19.1	11.05	16	16.3	1.86
Hg	ppm	0.03	0.03	0.00	0.02	0.03	40.00	0.16	0.14	13.33	0.2	0.21	4.88
In	ppm	0.05	0.05	0.00	0.04	0.04	0.00	0.08	0.09	11.76	0.52	0.53	1.90
La	ppm	35.4	34.7	2.00	33.7	33.6	0.30	26.9	30.2	11.56	9.7	10.1	4.04
Li	ppm	82	84	2.41	119	112	6.06	171	171	0.00	228	235	3.02
Mn	ppm	1,330	1,340	0.75	2,510	2,550	1.58	8,240	8,560	3.81	6,070	6,110	0.66
Mo	ppm	1.75	1.85	5.56	1.72	1.75	1.73	532	399	28.57	851	934	9.30
Nb	ppm	12	12.5	4.08	11.6	11.6	0.00	6.4	7.3	13.14	1.7	1.8	5.71
Ni	ppm	15.3	15.3	0.00	26.3	26.7	1.51	35.4	35.3	0.28	18.2	18.3	0.55
P	ppm	570	580	1.74	790	710	10.67	760	790	3.87	390	400	2.53
Pb	ppm	66	61	7.87	219	176	21.77	15,600	16,300	4.39	23,100	24,700	6.69
Rb	ppm	168	172	2.35	222	202	9.43	191	212	10.42	56.2	59.1	5.03
Sb	ppm	2.89	2.97	2.73	5.13	4.78	7.06	18.2	20	9.42	101	103	1.96
Sc	ppm	9.6	9.6	0.00	10.2	10.7	4.78	8.1	9.5	15.91	4	4.2	4.88
Sn	ppm	2	3.3	49.06	94.7	2	191.73	2.9	3.2	9.84	4.1	3.7	10.26
Sr	ppm	339	341	0.59	262	254	3.10	253	255	0.79	112	112	0.00
Te	ppm	<0.1	<0.1		<0.1	<0.1		<0.1	<0.1		0.2	0.2	0.00
Th	ppm	11.9	10.6	11.56	9.1	9.5	4.30	6.5	7.9	19.44	1.4	1.6	13.33
Tl	ppm	1.2	1.2	0.00	2.1	1.9	10.00	1.9	2.1	10.00	1.6	1.6	0.00
U	ppm	2.6	2.6	0.00	2.5	2.6	3.92	2.7	3.1	13.79	3.7	3.9	5.26
V	ppm	83	83	0.00	104	108	3.77	228	244	6.78	339	374	9.82
W	ppm	1.8	2	10.53	2.7	2.5	7.69	24.2	24.9	2.85	100	103	2.96
Y	ppm	21.1	21.2	0.47	20.2	20.7	2.44	16.5	18.7	12.50	4.8	5.2	8.00
Zn	ppm	126	126	0.00	246	239	2.89	12,000	13,100	8.76	34,000	33,800	0.59

Table 7. Laboratory replicate sediment data for total digestion samples collected from Old Yuma Mine study area.

[%, percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-38-B			OYM-22-S			OYM-05-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
CO ₂	%	1.47	1.46	0.68	1.39	1.36	2.18	0.39	0.38	2.60
Total C ^u	%	0.67	0.68	1.48	0.64	0.65	1.55	0.2	0.2	0.00
Al	%	7.8	7.94	1.78	6.44	6.56	1.85	2.81	2.75	2.16
Ca	%	2.71	2.76	1.83	1.68	1.75	4.08	0.56	0.56	0.00
Fe	%	4.56	4.63	1.52	4.84	4.95	2.25	6.75	6.78	0.44
K	%	3.37	3.48	3.21	3.71	3.87	4.22	0.96	0.93	3.17
Mg	%	1.51	1.51	0.00	0.62	0.64	3.17	0.55	0.54	1.83
Na	%	1.21	1.26	4.05	0.84	0.87	3.51	0.08	0.08	0.00
S	%	<0.01	<0.01		0.01	0.01	0.00	0.02	0.02	0.00
Ti	%	0.5	0.53	5.83	0.41	0.44	7.06	0.1	0.12	18.18
Ag	ppm	<1	<1		4	4	0.00	15	14	6.90
As	ppm	13	14	7.41	55	54	1.83	196	191	2.58
Ba	ppm	1,060	1,070	0.94	1,060	1,090	2.79	256	252	1.57
Be	ppm	1.6	1.6	0.00	1.6	1.4	13.33	1.6	1.6	0.00
Bi	ppm	0.43	0.45	4.55	6.3	6.22	1.28	132	128	3.08
Cd	ppm	1.6	1.6	0.00	22.9	22.5	1.76	22	21.5	2.30
Ce	ppm	72.3	74.2	2.59	62.6	62	0.96	20	19.1	4.60
Co	ppm	22.3	22.9	2.65	21.6	21.9	1.38	34.4	33.4	2.95
Cr	ppm	71	71	0.00	50	53	5.83	23	22	4.44
Cs	ppm	30	30	0.00	15	15	0.00	9	8	11.76
Cu	ppm	51.1	52.5	2.70	184	187	1.62	1,100	1,080	1.83
Ga	ppm	19	19.5	2.60	17.1	17	0.59	15.3	15	1.98
Hg	ppm	0.03	0.032	6.45	0.16	0.155	3.17	0.21	0.261	21.66

Table 7. Laboratory replicate sediment data for total digestion samples collected from Old Yuma Mine study area.—Continued

[%; percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-38-B			OYM-22-S			OYM-05-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
In	ppm	0.05	0.05	0.00	0.08	0.08	0.00	0.31	0.3	3.28
La	ppm	33	33.6	1.80	26.9	27.1	0.74	9.1	8.7	4.49
Li	ppm	142	144	1.40	171	176	2.88	202	211	4.36
Mn	ppm	3,060	3,080	0.65	8,240	8,530	3.46	11,100	11,100	0.00
Mo	ppm	21.2	23.8	11.56	532	545	2.41	750	713	5.06
Nb	ppm	10.6	10.2	3.85	6.4	6.6	3.08	1.7	1.6	6.06
Ni	ppm	44.6	44.3	0.67	35.4	36.8	3.88	18.2	18.4	1.09
P	ppm	1,030	1,060	2.87	760	790	3.87	270	260	3.77
Pb	ppm	575	607	5.41	15,600	15,700	0.64	30,400	29,200	4.03
Rb	ppm	175	176	0.57	191	194	1.56	51.3	49.9	2.77
Sb	ppm	3.45	3.55	2.86	18.2	18.1	0.55	56.5	55.4	1.97
Sc	ppm	12.6	12.9	2.35	8.1	8.2	1.23	3.5	3.5	0.00
Sn	ppm	1.7	1.8	5.71	2.9	2.9	0.00	8.6	7	20.51
Sr	ppm	303	309	1.96	253	262	3.50	124	123	0.81
Te	ppm	<0.1	<0.1		<0.1	<0.1		0.2	0.1	66.67
Th	ppm	7.3	7.8	6.62	6.5	6.3	3.13	1.6	1.6	0.00
Tl	ppm	1.4	1.4	0.00	1.9	2	5.13	2.3	2.3	0.00
U	ppm	2	2	0.00	2.7	2.7	0.00	5.8	5.6	3.51
V	ppm	150	152	1.32	228	232	1.74	221	226	2.24
W	ppm	2.3	2.3	0.00	24.2	24.5	1.23	88	87.8	0.23
Y	ppm	20	20	0.00	16.5	16.7	1.20	5.1	4.9	4.00
Zn	ppm	514	504	1.96	12,000	12,400	3.28	47,100	47,200	0.21

^aAn additional sample from OYM-18-W was also analyzed for total carbon. Environmental and replicate results were both 1.08 percent.

Table 8. Field replicate sediment data for U.S. Environmental Protection Agency 3050 method partial digestion samples collected from Old Yuma Mine study area.
[%, percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-33-B			OYM-31-B			OYM-22-S			OYM-02-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
Al	%	2.44	2.41	1.24	2.4	2.73	12.87	1.98	2	1.01	1.25	1.36	8.43
Ca	%	1.35	1.35	0.00	2.2	2.22	0.90	1.45	1.5	3.39	0.527	0.552	4.63
Fe	%	2.19	2.85	26.19	3.37	3.4	0.89	4.62	4.69	1.50	6.95	6.62	4.86
K	%	0.395	0.388	1.79	0.344	0.408	17.02	0.23	0.224	2.64	0.0594	0.0655	9.77
Mg	%	0.725	0.72	0.69	0.761	0.784	2.98	0.632	0.642	1.57	0.522	0.515	1.35
Na	%	0.0224	0.0223	0.45	0.0193	0.02	3.56	0.0263	0.0237	10.40	0.00608	0.00699	13.93
P	%	0.0515	0.0531	3.06	0.0769	0.0673	13.31	0.0698	0.0718	2.82	0.0444	0.0434	2.28
S	%	0.0171	0.0166	2.97	0.0069	< 0.005		0.00627	0.00763	19.57	0.0127	0.0151	17.27
Ti	%	0.0911	0.0919	0.87	0.0542	0.0635	15.80	0.0421	0.0396	6.12	0.0113	0.0119	5.17
Ag	ppm	<2	<2		<2	<2		<2	<2		<2	<2	
As	ppm	<30	<30		<30	<30		58.7	54.8	6.87	340	317	7.00
Ba	ppm	191	191	0.00	228	228	0.00	224	226	0.89	37.2	35.2	5.52
Be	ppm	<1	<1		<1	<1		1.14	1.18	3.45	1.21	1.21	0.00
Cd	ppm	<2	<2		<2	<2		24.5	24.1	1.65	13.9	14.3	2.84
Ce	ppm	54	53.7	0.56	56.8	57.2	0.70	48	50.5	5.08	26.1	25	4.31
Co	ppm	10.2	9.73	4.72	15.3	14.6	4.68	19.4	22.3	13.91	26.7	24.7	7.78
Cr	ppm	24.3	24.1	0.83	41	43.7	6.38	48.2	51.3	6.23	34	33.3	2.08
Cu	ppm	30.3	30.5	0.66	37.8	39.1	3.38	213	222	4.14	2,170	2,090	3.76

Table 8. Field replicate sediment data for U.S. Environmental Protection Agency 3050 method partial digestion samples collected from Old Yuma Mine study area.—Continued

[%; percent; ppm, parts per million; sample identifier: B, background; S, stream sediment; T, tailings]

Constituent	Units	OYM-33-B			OYM-31-B			OYM-22-S			OYM-02-T		
		Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference	Environmental	Replicate	Absolute percent difference
Eu	ppm	<2	<2		<2	<2		<2	<2		<2	<2	
La	ppm	22.2	22.6	1.79	26.3	24.4	7.50	19.2	19.8	3.08	6.42	5.96	7.43
Li	ppm	35.4	35.9	1.40	49.9	54.8	9.36	75.7	73	3.63	44.8	56	22.22
Mn	ppm	1,510	1,530	1.32	3,280	3,150	4.04	16,800	17,000	1.18	9,700	8,600	12.02
Mo	ppm	<2	<2		<2	<2		537	347	42.99	893	927	3.74
Nb	ppm	<4	<4		<4	<4		<4	<4		<4	<4	
Nd	ppm	18.7	21.9	15.76	19.9	16	21.73	<4	<4		<4	<4	
Ni	ppm	13.4	15	11.27	26.9	27.5	2.21	37.2	37.8	1.60	11.6	11.6	0.00
Pb	ppm	60.3	62.5	3.58	217	173	22.56	17,000	16,800	1.18	25,200	24,500	2.82
Sc	ppm	5.09	5.15	1.17	6.21	6.47	4.10	6.48	6.58	1.53	3.28	3.12	5.00
Sr	ppm	96.1	96.4	0.31	79.2	81.5	2.86	93.5	89.4	4.48	34.2	34.7	1.45
Th	ppm	4.31	<4		4.4	4.66	5.74	<4	<4		<4	<4	
V	ppm	64.1	64.3	0.31	88.7	92.3	3.98	204	236	14.55	340	370	8.45
Y	ppm	9.14	9.14	0.00	10.6	10.5	0.95	9.03	9.03	0.00	3.02	2.86	5.44
Yb	ppm	<1	<1		<1	<1		<1	<1		<1	<1	
Zn	ppm	122	124	1.63	254	247	2.79	12,100	13,100	7.94	33,000	31,900	3.39

Table 9A. Total digestion results for sediment reference materials Cody Shale (SCo-1), Green River Shale (SGR-1b), and Granodiorite (GSP-2).

[Accepted reference material concentrations from Smith (1995) and Wilson (1998a, b; 2001). %, percent; ppm, parts per million; SGS, Société Générale de Surveillance; –, not available]

Cody Shale SCo-1						Green River Shale SGR-1b				Granodiorite GSP-2			
		Reported		USGS Contract Lab SGS		Reported		USGS Contract Lab SGS		Reported		USGS Contract Lab SGS	
Constituent	Unit	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference
CO ₂	%	–	–	–	–	–	–	–	–	–	–	–	–
Total C	%	–	–	1	–	–	–	27.9	–	–	–	0.09	–
C in CO ₃ ²⁻	%	–	–	0.69	–	–	–	2.97	–	–	–	0.04	–
Organic C	%	–	–	0.31	–	–	–	24.93	–	–	–	0.05	–
Al	%	–	–	6.98	–	–	–	3.53	–	7.88	0.11	7.82	1
Ca	%	–	–	1.8	–	–	–	5.74	–	1.5	0.04	1.51	1
Fe	%	–	–	3.26	–	–	–	1.88	–	3.43	0.11	3.2	7
K	%	–	–	2.28	–	–	–	1.38	–	4.48	0.12	4.33	3
Mg	%	–	–	1.45	–	–	–	2.46	–	0.58	0.02	0.53	9
Na	%	–	–	0.61	–	–	–	2.15	–	2.06	0.07	1.91	7
S	%	0.063	0.009	0.07	11	1.53	0.11	1.53	0	–	–	0.05	–
Ti	%	–	–	0.29	–	–	–	0.13	–	0.4	0.01	0.34	15
Ag	ppm	–	–	<1	–	–	–	<1	–	–	–	<1	–
As	ppm	12	1	12	0	67	5	64	4	–	–	2	–
Ba	ppm	570	30	541	5	290	40	271	7	1,340	44	1,320	1
Be	ppm	1.8	0.2	2.1	17	–	–	1	–	1.5	0.2	1.4	7
Bi	ppm	0.37	–	0.37	0	–	–	0.79	–	–	–	0.06	–
Cd	ppm	–	–	0.1	–	0.9	–	1	11	–	–	<0.1	–
Ce	ppm	62	6	57.7	7	36	4	36.7	2	410	30	439	7
Co	ppm	11	0.8	10.8	2	12	1.5	11.5	4	7.3	0.8	7.4	1
Cr	ppm	68	5	52	24	30	3	24	20	20	6	13	35
Cs	ppm	7.8	0.7	8	3	5.2	0.3	5	4	1.2	0.1	<5	–
Cu	ppm	29	2	24.9	14	66	9	65.8	0	43	4	42.8	0
Ga	ppm	15	–	17	13	12	–	8.55	29	22	2	23.1	5
Hg	ppm	–	–	0.08	–	0.3	–	0.25	17	–	–	0.02	–
In	ppm	–	–	0.05	–	–	–	0.03	–	–	–	0.04	–
La	ppm	30	1	28.1	6	20	1.8	18.7	7	180	12	183	2
Li	ppm	45	3	43	4	147	26	136	7	36	1	35	3
Mn	ppm	410	30	351	14	267	34	222	17	320	20	293	8
Mo	ppm	1.4	0.2	1.22	13	35	0.9	32.2	8	2.1	0.6	2.38	13
Nb	ppm	11	–	10.3	6	5.2	–	8.4	62	27	2	23.4	13
Ni	ppm	27	4	22.9	15	29	–	25.9	11	17	2	14.8	13
P	ppm	–	–	830	–	–	–	1,170	–	1,300	100	1,240	5
Pb	ppm	31	3	31	0	38	4	43	13	42	3	40	5
Rb	ppm	110	4	112	2	–	–	80.3	–	245	7	244	0
Sb	ppm	2.5	0.1	2.6	4	3.4	0.5	3.81	12	–	–	0.42	–
Sc	ppm	11	1	11.7	6	4.6	0.7	5	9	6.3	0.7	6.1	3

Table 9A. Total digestion results for sediment reference materials Cody Shale (SCo-1), Green River Shale (SGR-1b), and Granodiorite (GSP-2).—Continued

[Accepted reference material concentrations from Smith (1995) and Wilson (1998a, b; 2001). %, percent; ppm, parts per million; SGS, Société Générale de Surveillance; —, not available]

Cody Shale SCo-1						Green River Shale SGR-1b				Granodiorite GSP-2			
		Reported		USGS Contract Lab SGS		Reported		USGS Contract Lab SGS		Reported		USGS Contract Lab SGS	
Constituent	Unit	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference
Sn	ppm	3.7	—	3.4	8	1.9	—	1.7	11	—	—	6.6	—
Sr	ppm	170	16	150	12	420	30	360	14	240	10	224	7
Te	ppm	—	—	<0.1	—	—	—	0.2	—	—	—	<0.1	—
Th	ppm	9.7	0.5	9.9	2	4.8	0.21	4.9	2	105	8	113	8
Tl	ppm	—	—	0.6	—	—	—	0.5	—	—	—	1.3	—
U	ppm	—	—	2.9	—	5.4	0.4	5.4	0	2.4	0.19	2.3	4
V	ppm	130	13	121	7	130	6	112	14	52	4	49	6
W	ppm	1.4	—	1.3	7	2.6	0.06	2.4	8	—	—	0.6	—
Y	ppm	26	4	18.6	28	13	—	10.1	22	28	2	26.5	5
Zn	ppm	100	8	100	0	74	9	70	5	120	10	117	3

Table 9B. Total digestion results for sediment reference materials Andesite (AGV-2) and Granodiorite (GSP).

[Accepted reference material concentrations from Smith (1995) and Wilson (1998a, b; 2001). %, percent; ppm, parts per million; SGS, Société Générale de Surveillance; —, not available]

Andesite AGV-2						Granodiorite GSP			
		Reported		USGS Contract Lab SGS		Reported		USGS Contract Lab SGS	
Constituent	Unit	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference
CO ₂	%	—	—	—	—	0.33	—	0.26	21
Total C	%	—	—	0.04	—	0.18	—	0.18	0
C in CO ₃ ²⁻	%	—	—	0.01	—	0.09	—	0.07	22
Organic C	%	—	—	0.03	—	0.09	—	0.11	22
Al	%	8.95	0.11	8.83	1	7.57	—	7.65	1
Ca	%	3.72	0.09	3.72	0	1.5	—	1.48	1
Fe	%	4.68	0.09	4.32	8	2.77	—	2.7	3
K	%	2.39	0.09	2.35	2	4.25	—	4.09	4
Mg	%	1.08	0.02	0.95	12	0.615	—	0.55	11
Na	%	3.11	0.09	2.93	6	1.87	—	1.74	7
S	%	—	—	<0.01	—	0.074	—	0.07	5
Ti	%	0.63	0.13	0.58	8	0.339	—	0.35	3
Ag	ppm	—	—	<1	—	3.07	—	3	2
As	ppm	—	—	1	—	31.4	—	32	2
Ba	ppm	1,140	32	1,060	7	1,310	—	1,290	2
Be	ppm	2.3	0.4	1.8	22	1.17	—	1.2	3

Table 9B. Total digestion results for sediment reference materials Andesite (AGV-2) and Granodiorite (GSP).—Continued

[Accepted reference material concentrations from Smith (1995) and Wilson (1998a, b; 2001). %, percent; ppm, parts per million; SGS, Société Générale de Surveillance; —, not available]

Andesite AGV-2						Granodiorite GSP			
Reported				USGS Contract Lab SGS		Reported		USGS Contract Lab SGS	
Constituent	Unit	Value	Standard deviation	Lab value	Absolute percent difference	Value	Standard deviation	Lab value	Absolute percent difference
Bi	ppm	—	—	0.06	—	4.28	—	3.88	9
Cd	ppm	—	—	<0.1	—	0.227	—	0.2	12
Ce	ppm	68	3	67	1	405	—	416	3
Co	ppm	16	1	15.3	4	6.36	—	5.9	7
Cr	ppm	17	2	9	47	16.6	—	15	10
Cs	ppm	1.16	0.08	<5	—	<5	—	<5	—
Cu	ppm	53	4	48.3	9	31.3	—	33.3	6
Ga	ppm	20	1	21.4	7	22.2	—	21.9	1
Hg	ppm	—	—	0.01	—	0.26	—	0.26	0
In	ppm	—	—	0.04	—	0.04	—	0.04	0
La	ppm	38	1	38	0	173	—	172	1
Li	ppm	11	—	10	9	34	—	33	3
Mn	ppm	770	20	688	11	281	—	342	22
Mo	ppm	—	—	2.62	—	1.31	—	1.38	5
Nb	ppm	15	1	14.1	6	18.6	—	22.3	20
Ni	ppm	19	3	16.8	12	11.5	—	12.1	5
P	ppm	2,100	100	1,990	5	1230	—	1,140	7
Pb	ppm	13	1	22	69	40.4	—	37	8
Rb	ppm	68.6	2.3	68.2	1	228	—	231	1
Sb	ppm	0.6	—	0.76	27	0.877	—	0.83	5
Sc	ppm	13	1	11.5	12	6.23	—	6	4
Sn	ppm	2.3	0.4	2	13	4.72	—	4.7	0
Sr	ppm	658	17	588	11	226	—	205	9
Te	ppm	—	—	<0.1	—	4.07	—	3.9	4
Th	ppm	6.1	0.6	6.2	2	104	—	109	5
Tl	ppm	0.27	—	0.3	11	2.07	—	1.9	8
U	ppm	1.88	0.16	1.9	1	2.27	—	2.2	3
V	ppm	120	5	112	7	73.3	—	68	7
W	ppm	—	—	0.7	—	6.65	—	6.7	1
Y	ppm	20	1	19.5	3	25.5	—	25.4	0
Zn	ppm	86	8	81	6	117	—	117	0

Table 10. Sediment reference material results for U.S. Environmental Protection Agency 3050 partial digestion; replicate runs listed for each reference material.

[Accepted reference material concentrations from National Institute of Standards and Technology (2003). Blank cell indicates value was not reported. %, percent; ppm, parts per million]

Constituent	Units	2709			2710			2711	
		Laboratory value	Laboratory replicate value	Reported value range	Laboratory value	Laboratory replicate value	Reported value range	Laboratory value	Reported value range
Al	%	3.15	3.33	2–3.1	2.51	2.52	1.2–2.6	2.55	1.2–2.3
Ca	%	1.43	1.54	1.4–1.7	0.43	0.444	0.38–0.48	2.27	2.0–2.5
Fe	%	3.03	3.09	2.5–3.3	2.93	2.19	2.2–3.2	2.03	1.7–2.6
K	%	0.399	0.42	0.26–0.37	0.525	0.525	0.37–0.50	0.507	0.26–0.53
Mg	%	1.35	1.36	1.2–1.5	0.592	0.567	0.43–0.60	0.828	0.72–0.89
Na	%	0.075	0.077	0.063–0.11	0.067	0.066	0.049–0.062	0.037	0.02–0.029
P	%	0.055	0.055	0.05–0.07	0.092	0.088	0.106–0.11	0.077	0.06–0.09
SO ₄ ²⁻	%	0.083	0.087		0.234	0.22		0.039	
Ti	%	0.058	0.061	0.03–0.04	0.131	0.126	0.092–0.11	0.076	0.039–0.048
Ag	ppm	<2	<2		7.77	9.55		<2	
As	ppm	<30	<30	<20	637	615	490–600	121	88–110
Ba	ppm	413	449	392–400	357	373	300–400	223	170–260
Be	ppm	<1	<1		<1	<1		<1	
Cd	ppm	<2	<2	<1	24.1	23.8	13–26	40.7	32–46
Ce	ppm	36.1	41.6		41.2	45.6		59.6	
Co	ppm	11.9	12.3	10.0–15	7.86	7.83	6.3–12	8.08	7–12
Cr	ppm	78.7	83	60–115	22	21.7	15–23	26.8	15–25
Cu	ppm	35	39.1	26–40	3,080	3,010	2,400–3,400	126	91–110
Eu	ppm	<2	<2		<2	<2		<2	
La	ppm	16.8	18		21.5	21.5		30	
Li	ppm	37.7	41		30.6	31.3		19	
Mn	ppm	528	536	360–600	8,650	8,410	6,200–9,000	581	400–620
Mo	ppm	<2	<2	<2	9.59	12.1	13–27	<2	<2
Nb	ppm	<4	<4		<4	<4		<4	
Nd	ppm	14.5	13.3		<4	<4		11.3	
Ni	ppm	78.1	79.4	65–90	10.4	10	8.8–15	17	14–20
Pb	ppm	<20	28.5	12–18	5,440	5,220	4,300–7,000	1,200	930–1,500
Sc	ppm	7.85	8.33		4.74	4.85		4.77	
Sr	ppm	107	114	100–112	110	113	94–110	55.4	48–55
Th	ppm	8.3	8.85		10.5	11.1		10.8	
V	ppm	79.2	81.7	51–70	54	53.3	37–50	57.2	34–50
Y	ppm	11	11.8		13.6	13.4		18.8	
Yb	ppm	1.09	1.2		1.09	1.08		1.87	
Zn	ppm	95.8	157	87–120	6,190	5,960	5,200–6,900	366	290–340

Sediment Leachate

One replicate sample from site OYM-14-W was analyzed for sediment leachate. The replicate analysis indicates that several elements have relative percentage differences between the sample and replicate greater than 20 percent (table 11). The difference may represent variability owing to heterogeneity within a sediment sample. More replicate samples would be needed to statistically quantify the variability for each element.

One replicate sample from the sediment leachate from site OYM-10-T was analyzed for strontium and strontium isotopic value. The samples had the same strontium concentration of 33.4 micrograms per liter ($\mu\text{g/L}$) and strontium isotope ratios of 0.71174 and 0.71178. The difference in strontium isotopes (0.00004 or 0.01 percent) is greater than the 2-sigma uncertainty presented by the USGS National Research Program Laboratory in Menlo Park, Calif. (0.000024).

Table 11. Results of sediment leachate replicate analyses for site OYM-14-W.

[**Bold** values indicate the percentage difference is greater than 20 percent; mg/L, milligrams per liter; $\mu\text{g/L}$, micrograms per liter; NA, not determined]

Constituent	Units	Environmental result	Replicate result	Difference	Absolute percent difference
Initial pH	standard scale	7.57	7.39	0.18	2.41
Final pH	standard scale	9.62	9.51	0.11	1.15
Ca	mg/L	8.16	8.51	-0.35	4.20
Fe	mg/L	0.594	1.03	-0.44	53.69
Mg	mg/L	0.762	0.879	-0.117	14.26
P	mg/L	<0.003	<0.003	NA	NA
K	mg/L	0.890	0.928	-0.038	4.18
Na	mg/L	7.51	7.50	0.01	0.13
S	mg/L	1.6	1.4	0.20	13.33
Al	$\mu\text{g/L}$	457	644	-187	33.97
Sb	$\mu\text{g/L}$	1.1	1.2	-0.1	8.70
As	$\mu\text{g/L}$	22.3	11.3	11.0	65.48
Ba	$\mu\text{g/L}$	14.2	15.4	-1.2	8.11
Be	$\mu\text{g/L}$	0.022	0.026	-0.004	16.67
Bi	$\mu\text{g/L}$	0.35	0.47	-0.12	29.27
Ce	$\mu\text{g/L}$	0.42	0.55	-0.13	26.80
Cd	$\mu\text{g/L}$	0.420	0.628	-0.208	39.69
Co	$\mu\text{g/L}$	0.436	0.622	-0.186	35.16
Cr	$\mu\text{g/L}$	1.38	1.59	-0.21	14.14
Cu	$\mu\text{g/L}$	33.3	48.5	-15.2	37.16
Cs	$\mu\text{g/L}$	0.31	0.42	-0.11	30.14
Ga	$\mu\text{g/L}$	0.26	0.36	-0.10	32.26
In	$\mu\text{g/L}$	<0.01	<0.01	NA	NA
La	$\mu\text{g/L}$	0.34	0.46	-0.12	30.00
Pb	$\mu\text{g/L}$	277	428	-151	42.84
Li	$\mu\text{g/L}$	8.05	9.55	-1.50	17.05
Mn	$\mu\text{g/L}$	94	145	-51	42.68
Hg	$\mu\text{g/L}$	<0.01	<0.01	NA	NA
Mo	$\mu\text{g/L}$	69.9	83.0	-13.1	17.14
Nb	$\mu\text{g/L}$	<0.01	<0.01	NA	NA
Ni	$\mu\text{g/L}$	0.50	0.60	-0.10	18.18
Rb	$\mu\text{g/L}$	0.88	0.90	-0.02	-2.25

Table 11. Results of sediment leachate replicate analyses for site OYM-14-W.—Continued[**Bold** values indicate the percentage difference is greater than 20 percent; mg/L, milligrams per liter; µg/L, micrograms per liter; NA, not determined]

Constituent	Units	Environmental result	Replicate result	Difference	Absolute percent difference
Sc	µg/L	0.52	0.67	−0.15	25.21
Se	µg/L	0.07	0.08	−0.01	13.33
Ag	µg/L	0.024	0.025	−0.001	4.08
Sr	µg/L	40.5	40.9	−0.4	0.98
Te	µg/L	<0.01	<0.01	NA	NA
Th	µg/L	0.05	0.08	−0.03	46.15
Sn	µg/L	0.21	0.07	0.14	100.00
Ti	µg/L	10.9	17.7	−6.8	47.55
Tl	µg/L	0.015	0.011	0.004	30.77
W	µg/L	0.30	0.37	−0.07	20.90
U	µg/L	0.086	0.106	−0.020	20.83
V	µg/L	0.94	1.24	−0.30	27.52
Y	µg/L	0.132	0.185	−0.053	33.44
Zn	µg/L	511	842	−331	48.93

Groundwater Elevation

Spatial

Groundwater elevation was measured between 2014 and 2017 at 29 groundwater sites (fig. 2). Groundwater elevation was generally highest in the southern and western parts of the study area, which also have higher ground surface elevations. Groundwater flow direction is generally toward the northeast in the study area. There was a steep water-level gradient over a small area in the north-central part of the study area where measured groundwater elevation changes by more than 200 ft in less than 0.25 miles. The steep gradient is located beneath surficial Quaternary-fill geologic units, where no structural feature is mapped, but the gradient coincides with the projection of the Old Yuma Fault from Lipman (1993).

Chon and others (2016) measured geophysical data in the area near the large change in groundwater-level elevation using transient electromagnetics (TEM) and reported higher resistivity at depth on the northwestern side of the projected Old Yuma Fault than on the southeastern side. The more conductive area was above the resistive zone at depth on the northwestern side and may relate to the shallow groundwater-level elevations. On the southeastern side, the resistive zone was at the surface, with the more conductive area located below the resistive zone (Chon and others, 2016). The displacement of the conductive zones may be due to the presence of the Old Yuma Fault beneath the basin fill.

Some of the wells show evidence of confined or semi-confined groundwater conditions. Well D-13-12 10BAC3 (well 13 in table 1 and fig. 2) was drilled during this study and

cuttings were collected during drilling (described in table 12). The well was drilled using air rotary, which may break up rock from different intervals to a greater degree, depending on the pressure in that interval; therefore, the descriptions of competency in table 12 are not representative of the original rock material. The well was drilled through 27 ft of unconsolidated material followed by rhyodacite-rhyolite flow material that may be related to the Tertiary Safford Dacite units of Lipman (1993), which are present on the hill to the south and west of the well. A distinct change to obsidian was noted starting around 300 to 440 ft. The drillers encountered a small amount of water above 300 ft during drilling, which dried up quickly, and around 440 ft they had to increase drilling pressure. Below the resistant layer, water was encountered and the total well was drilled to 687 ft. The static water level rose to an elevation of 2,224.6 ft (148.3 feet below land surface, table 13). To the east, well D-13-12 10BAC2 was drilled to the same total depth below the surface and the water level in that well was at an elevation of 1,928.3 ft (452.0 ft below land surface, table 13).

The water-level elevation within the inclined shaft of the Old Yuma Mine was determined to be between 2,400 and 2,450 ft, based on the groundwater elevation map (fig. 2). The elevation of the surface at the top of the mine shaft is 2,575 ft. The shaft is inclined at 43° from horizontal and a total depth of 300 ft was reported when the shaft was created. The 200-ft level is not saturated and the shaft has collapsed below the 200-ft level (National Park Service, 2010). The geometry of the shaft and the groundwater level estimation places the water table within the collapsed portion of the lower 100 ft of the mine shaft. Calculating the vertical distance from the land surface, the water level is below 2,440 ft. There may be

Table 12. Well-cutting descriptions from well D-13-12 10BAC3.

[Samples are numbered such that 1 is at the surface and larger numbers represent progressively deeper samples; mm, millimeter]

Drilling sample number	Description	Comments	Depth, in feet
1	Grayish-colored finely porphyritic rhyodacite-rhyolite with 1 mm phenocrysts of plagioclase and biotite in a granular microcrystalline groundmass; portions of groundmass appear partly aphanitic	Rhyodacite-rhyolite, porphyritic	0–27
2	Light gray to white rhyolitic altered/weathered pumiceous tuff; weathered to clays, devitrified glassy cryptocrystalline matrix, phenocryst altered to kaolin; altered crystal-rich clusters in clayey matrix; rock is friable, crushes in hand	Siliceous weathered tuff	27–63
3	Light tan to pinkish-gray siliceous altered rock (rhyodacitic to rhyolitic); phenocrysts of feldspar, minor biotite; biotite is altered	Rhyodacite-rhyolite, porphyritic	63–72
4	Pinkish-tan to tannish-gray rhyodacitic to rhyolitic rock with 0.5–1 mm phenocrysts of plagioclase, minor scattered biotite	Rhyodacite-rhyolite, porphyritic	
5	Same as above	Rhyodacite-rhyolite, porphyritic	
6	Same as above	Rhyodacite-rhyolite, porphyritic	
7	Same as above; fragments show some secondary white quartz stringers irregularly crosscutting material	Rhyodacite-rhyolite, porphyritic	
8	Same as above; clear intergranular groundmass (microcrystalline)	Rhyodacite-rhyolite, porphyritic	
9	Same as above	Rhyodacite-rhyolite, porphyritic	
10	Same as above; fragments have a planar (flat) habit, are dark tan colored, and interspersed with some light-gray-colored rhyolitic pieces	Rhyodacite-rhyolite, porphyritic	
11	Same as above; some clear quartz crystal fragments	Rhyodacite-rhyolite, porphyritic	
12	Same as above; irregular fine-grained pieces and irregular crystal-rich clusters	Rhyodacite-rhyolite, porphyritic	end 286
13	Same as above; shows some very dark colored aphanitic fragments with phenocrysts in aphanitic matrix	Mixed glassy material, rhyolitic	start 286
14	Mixed obsidian material (dark gray to black in color); glassy fragments mixed with some rhyodacite-rhyolite with phenocrysts	Obsidian mixed with rhyolitic material	
15	Obsidian-predominant fragments; glassy to slightly devitrified (indicating flow margin?)	Obsidian	
16	Same as above; mixed rhyodacite-rhyolite containing plagioclase and biotite phenocrysts in aphanitic glassy matrix	Obsidian mixed with rhyolitic material	
17	Mixed obsidian material; altered glassy fragments mixed with some rhyodacite-rhyolite with phenocrysts	Obsidian mixed with rhyolitic material	
18	Same as above; devitrified, altered rhyolitic material	Obsidian mixed with rhyolitic material	~440
19	Pumiceous tuff; material crystal-rich with altered biotite and plagioclase in altered microcrystalline matrix; crumbles in hand; similar to sample number 2	Tuffaceous lens	
20	Light to medium gray, pinkish-tan rhyodacite-rhyolite with plagioclase and biotite phenocrysts in aphanitic matrix	Rhyodacite-rhyolite, porphyritic	
21	Same as above	Rhyodacite-rhyolite, porphyritic	
22	Same as above	Rhyodacite-rhyolite, porphyritic	
23	Same as above	Rhyodacite-rhyolite, porphyritic	~560
24	Same as above	Rhyodacite-rhyolite, porphyritic	

Table 13. Water-level measurements.

[ASL, above sea level, BLS, below land surface; R, recently pumped prior to measurement; –, not recently pumped]

Map site number	Site name	Date	Time	Water level, in feet BLS	Water level status	Groundwater elevation, in feet ASL
1	D-13-12 22CAA1	1/29/2015	13:00	88.61	R	2,565.99
2	D-13-12 14DCB	3/20/2015	14:00	261.55	R	2,122.35
3	D-13-12 15BDC1	2/24/2015	14:00	267.31	–	2,314.39
4	D-13-12 15ACB1	3/3/2015	14:00	243.25	R	2,236.75
5	D-13-12 15BDA1	3/20/2015	12:30	307.7	–	2,205.80
6	D-13-12 10DCD1	2/17/2015	12:00	417.95	R	2,071.85
7	D-13-12 12DCD1	2/17/2015	10:00	207.36	R	2,068.94
8	D-13-12 10CDB1	12/17/2014	11:40	10.24	R	2,437.76
8	D-13-12 10CDB1	2/25/2016	12:00	11.73	R	2,436.27
8	D-13-12 10CDB1	2/2/2017	16:20	7.99	–	2,440.01
9	D-13-12 10DDA1	2/17/2015	13:00	327.6	R	2,121.40
9	D-13-12 10DDA1	2/9/2016	11:40	327.84	R	2,121.16
10	D-13-12 09BCC1	3/3/2015	11:30	53.7	–	2,438.30
10	D-13-12 09BCC1	5/6/2016	10:58	54.49	–	2,437.51
10	D-13-12 09BCC1	7/14/2016	10:10	55.24	–	2,436.76
10	D-13-12 09BCC1	9/14/2016	9:20	52.98	–	2,439.02
10	D-13-12 09BCC1	11/17/2016	8:10	53.13	–	2,438.87
10	D-13-12 09BCC1	2/2/2017	15:15	53.63	–	2,438.37
11	D-13-12 08BDB1	3/16/2015	10:00	90.92	–	2,460.25
12	D-13-12 09ADA1	2/10/2015	10:00	49.68	–	2,321.52
12	D-13-12 09ADA1	2/25/2016	12:40	56.43	–	2,314.77
13	D-13-12 10BAC3	8/16/2016	9:30	148.3	–	2,224.60
14	D-13-12 10BBC1	2/3/2015	10:50	122.03	R	2,239.87
14	D-13-12 10BBC1	2/25/2016	14:40	122.65	R	2,239.25
15	D-13-12 10BAC2	2/3/2015	10:00	451.98	R	1,928.32
15	D-13-12 10BAC2	2/10/2015	13:30	452.14	R	1,928.16
15	D-13-12 10BAC2	1/22/2016	15:30	429.58	R	1,950.72
16	D-13-12 09AAA1	2/10/2015	9:30	115.29	R	2,240.91
16	D-13-12 09AAA1	1/11/2016	10:10	116.64	R	2,239.56
17	D-13-12 10BAA1	3/16/2015	13:00	421.93	–	1,940.17
18	D-13-12 10BBB1	2/10/2015	11:00	106.97	R	2,240.93
18	D-13-12 10BBB1	2/25/2016	14:30	108.7	R	2,239.20
19	D-13-12 03DDD1	3/3/2015	12:30	236.65	R	2,102.05
20	D-13-12 03CDC1	2/3/2015	14:00	104.39	–	2,238.41

Table 13. Water-level measurements.—Continued

[ASL, above sea level; BLS, below land surface; R, recently pumped prior to measurement; –, not recently pumped]

Map site number	Site name	Date	Time	Water level, in feet BLS	Water level status	Groundwater elevation, in feet ASL
21	D-13-12 03DCD1	3/3/2015	12:00	238.79	R	2,102.61
22	D-13-12 04DCD	1/23/2015	12:00	138.99	–	2,246.31
22	D-13-12 04DCD	2/8/2016	10:38	138.77	–	2,246.53
22	D-13-12 04DCD	7/14/2016	11:10	138.04	–	2,247.26
22	D-13-12 04DCD	9/14/2016	9:35	135.07	–	2,250.23
22	D-13-12 04DCD	11/16/2016	12:50	133.01	–	2,252.29
22	D-13-12 04DCD	2/2/2017	15:45	133.01	–	2,252.29
23	D-13-12 03DCA1	2/24/2015	11:30	228.14	R	2,108.26
24	D-13-12 03CCA1	1/21/2016	9:58	86.81	R	2,235.09
25	D-13-12 03DBD2	2/17/2015	14:00	217.88	R	2,104.22
26	D-13-12 03DBD1	3/16/2015	12:00	207.63	R	2,102.37
27	D-13-12 04DAD1	2/24/2015	10:00	82	–	2,238.50
28	D-13-12 03DBA1	1/22/2016	14:49	190.09	–	2,102.21
28	D-13-12 03DBA1	1/29/2016	11:05	189.87	R	2,102.43
29	D-13-12 03BCC1	2/4/2016	11:00	149.72	–	2,151.58

local anomalies in the groundwater elevation near the mine and the Old Yuma Fault that were not documented during this study because of a lack of groundwater wells near the mine. The TEM data indicate a lower resistivity zone below an elevation of about 2,300 ft to the south and east of the mine and below about 2,100 ft at a station closer to the mine, which may represent either the groundwater elevation or mine workings (Chon and others, 2016). Less resistive zones are shallower (approximately 2,300 ft) at geophysical stations to the north and west of the Old Yuma Mine. Additional data would improve estimates of groundwater elevation near the mine.

Temporal

Groundwater levels were measured multiple times over multiple years in 10 of the 29 wells (table 13). The average water-level difference between the highest and lowest value for the wells with multiple measurements was 4.54 ft, with the greatest change of 22.56 ft at D-13-12 10BAC2 and the smallest change of 0.22 ft at D-13-12 03DBA1. A previous study measured the water level at site D-13-12 09ADA1 (site number 12) in 2011 to be 52.8 ft below the land surface (URS Corporation, 2012), which is between the two water-level measurements made at that well during this study.

Continuous groundwater levels were recorded in two wells: D-13-12 09BCC1 and D-13-12 04DCD. Historical

water-level measurements are available starting in 1949 for D-13-12 04DCD (U.S. Geological Survey, 2017b) and the difference between the highest and lowest water level over the measurement period was 22 ft (fig. 3A). Over the 2-year study period from 2015 to 2017 there was a 6-ft water-level increase at D-13-12 04DCD (fig. 3A). At D-13-12 09BCC1 there was a 2-ft water-level increase over a 3-month period (fig. 3B).

Groundwater Geochemistry

General Chemistry

Eight wells were sampled for a comprehensive geochemical suite of analyses. The temperature of the samples ranged from 25.8 to 30.2 °C, pH ranged from 6.8 to 7.8, specific conductance ranged from 547 to 3,020 microsiemens per centimeter (μS/cm), and dissolved oxygen ranged from 1 to 6.2 milligrams per liter (mg/L). Major ion composition of the groundwater samples varied across the study area. Some wells were dominated by major cation type of calcium (D-13-12 09AAA1, 03CCA1, and 04DCD) and some sodium (D-13-12 10BAC2, 10DDA1, and 10BAC3), and major anion type bicarbonate (D-13-12 09AAA1, 10BAC2, 03DBA1, and 10DDA1) and chloride (D-13-12 04DCD) (figs. 1B and 4). D-13-12 03BCC1 had mixed water type with no dominant ion.

Figure 3. Groundwater level measured with a pressure transducer at two well sites. A, D-13-12 04DCD and B, D-13-12 09BCC1. Blue crosses show discrete water-level measurements. Solid black lines plot hourly measurements; dashed black lines connect discrete measurements.

Figure 4. Piper diagram plotting the major ion proportion of groundwater samples. All axes plot relative concentrations, in percent.

Only one sample (D-13-12 03DBA1) had a value that exceeded the EPA primary drinking water standard for arsenic (10.7 µg/L and 10 µg/L, respectively). Sample D-13-12 04DCD exceeded the EPA secondary drinking water standard for chloride, iron, manganese, and total dissolved solids. However, this well does not have a pump installed and is not currently used to supply drinking water. Two other samples (D-13-12 09AAA1 and 03CCA1) had a total dissolved solids concentration greater than the EPA secondary drinking water standard of 500 mg/L.

Perchlorate (ClO_4^-) can be used as an indicator of the use of explosives (Smith and others, 2015), potentially from nearby mining activities, but is also known to occur naturally (Plummer and others, 2006). Reported concentrations of

perchlorate in groundwater associated with blasting for mining activities at a different mine were variable and had a maximum value of 157 µg/L (Smith and others, 2015). Perchlorate concentrations ranging from 0.12 to 1.8 µg/L were measured in pre-anthropogenic samples from remote parts of the Middle Rio Grande Basin in New Mexico (Plummer and others, 2006). Concentrations reported here ranged from 0.67 to 1.48 µg/L and are similar for samples containing tritium and those without detectable tritium, indicating that the perchlorate measured in the samples is likely naturally occurring. All perchlorate concentrations in this study are less than the EPA interim health advisory value of 15 µg/L (U.S. Environmental Protection Agency, 2008).

Isotopic Analysis

Stable isotopes of oxygen and hydrogen can be used to understand elevation, season, and evaporation effects of water contributing to groundwater. Groundwater samples from this study were compared with the global meteoric water line (GMWL; Craig, 1961) and local meteoric water line (LMWL) derived from precipitation collected near the University of Arizona in Tucson (Eastoe and Dettman, 2016). Stable isotope values from groundwater in the area around the Old Yuma Mine plot to the right of (below) both the GMWL and LMWL (fig. 5). Stable isotope values ranged from -7.1 to -8.86 per mil for $\delta^{18}\text{O}$ and -51.4 to -68.5 per mil for $\delta^2\text{H}$. The $\delta^{18}\text{O}$ values are between the 10-year average value for summer (-6 per mil) and winter (-8.9 per mil), and half of the $\delta^2\text{H}$ values are less than the 10-year average value for $\delta^2\text{H}$ in summer (-42 per mil) and winter (-59 per mil) precipitation for Tucson. This result suggests groundwater in this area is a mix of precipitation recharged during both seasons.

Sulfate precipitation occurs in rain and as dry fallout. Reported sulfur isotopes ($\delta^{34}\text{S}$) in precipitation near Tucson in 1996 and 1997 ranged from 2.1 to 8.5 per mil, with higher values reported in the summer (Kayaci, 1997). Reported sulfur isotope values in dust ranged from 3.6 to 6.9 per mil (Eastoe and others, 2004). The sulfur isotopic ratio from groundwater in this study ranged from 3.25 to 13.96 per mil (fig. 6). Most samples fall within the range of precipitation and dust, with the exception of one sample with a higher value (13.96 per mil). The sample with high $\delta^{34}\text{S}$ was the only sample with a hydrogen sulfide smell noted during sample collection, and it also had the highest concentration of sulfate (134 mg/L). Sulfur isotope values greater than 10 per mil are often attributed to Permian marine gypsum (Eastoe and others, 2004), but may also represent waters that have undergone sulfate reduction (Canfield and Thamdrup, 1994).

Strontium isotopic ratio ($^{87}\text{Sr}/^{86}\text{Sr}$) in water can provide an indication of rock units the water may have interacted with along its flow path. Strontium isotope ratios in this study ranged from 0.70968 to 0.71168. Well D-13-12 10DDA1 had the lowest strontium isotopic ratio (0.70968) and the lowest concentration of strontium (330 mg/L) (fig. 7). That sample was also the farthest south and may represent groundwater moving along a different flow path compared with the other samples. The sample from well D-13-12 03DBA1 had the highest strontium isotopic ratio (0.71168) and the second lowest strontium concentration (480 mg/L) (fig. 7). This sample is the only sample from a well completed in the basin fill. The other samples had higher strontium concentrations and a narrower range of strontium isotope ratios, varying from 0.71027 to 0.71158 (fig. 7).

Figure 5. Stable isotope ratios ($\delta^{18}\text{O}$ and $\delta^2\text{H}$) for groundwater samples. Global meteoric water line (GMWL) from Craig (1961) and local meteoric water line (LMWL) from Eastoe and Dettman (2016).

Figure 6. Stable isotope ratios ($\delta^{34}\text{S}$ and $\delta^{18}\text{O}$) for groundwater samples.

Figure 7. Strontium isotope ratio ($^{87}\text{Sr}/^{86}\text{Sr}$) versus the inverse of strontium concentration ($1/\text{Sr}$) for groundwater samples.

Groundwater Age

Groundwater age is inferred from ^{14}C with corrections based on total dissolved inorganic carbon (the sum of inorganic carbon species carbonic acid, bicarbonate, and carbonate) and $\delta^{13}\text{C}$. Graphs of carbon species were made to understand the potential processes influencing carbon water chemistry at groundwater sample sites before interpretation of groundwater age, similar to Han and others (2012) and Han and Plummer (2016) (fig. 8, table 14). The blue lines on figure 8 represent the “zero-age” lines, which are determined by soil gas and solid carbonate ^{14}C and $\delta^{13}\text{C}$ values. Samples that plot between the zero-age lines on figure 8A do not have a radiocarbon age, and may be explained by geochemical reaction with no radiocarbon decay. Samples that plot above the zero-age area are likely mixtures containing some old recharged water, and samples that plot below the zero-age area may have a radiocarbon age greater than zero, indicating the presence of old water that has undergone radiocarbon decay (Han and Plummer, 2016). Results from NetpathXL are presented in table 14 for the uncorrected age (user defined) and revised Fontes and Garnier model (solid exchange) (Han and Plummer, 2013).

Samples from several sites plot below the zero-age area, indicating they may be old waters that could have undergone ^{14}C decay (fig. 8A) (Han and others, 2012). Sites with a possible radiocarbon age are wells D-13-12 10DDA1 and D-13-12 BAC3. These two sites, as well as D-13-12 03DBA1, also indicate a possible radiocarbon age using 10 pmc for ^{14}C in recharge-zone carbonates from Kalin (1994).

Tritium is a useful tracer for determining if there is a component of water recharged during the period of nuclear bomb testing in the 1950s and 1960s, when tritium in the atmosphere peaked and then decreased over the following decades. Tritium values have stopped decreasing in recent precipitation (after 1992), and average recent values of tritium in precipitation in Tucson are 17 pCi/L (Eastoe and others, 2012). Tritium values of samples from this study ranged from -0.02 pCi/L (which is below the reporting limit of 0.3 pCi/L) to 8.9 pCi/L. Three groundwater samples had tritium above the reporting limit, D-13-12 09AAA1, 04DCD, and 03BCC1

Figure 8. Graphs of carbon data from groundwater samples. A, ^{14}C versus $\delta^{13}\text{C}$, B, ^{14}C versus the inverse of dissolved inorganic carbon concentration ($1/\text{DIC}$), and C, $\delta^{13}\text{C}$ versus $1/\text{DIC}$. Solid gray lines represent Tamers X and (or) Y (Han and Plummer, 2013). Blue lines represent zero-age lines, which are determined by soil gas and solid carbonate ^{14}C and $\delta^{13}\text{C}$ values.

(table 14). The first two of these samples had high tritium (8.9 and 5.4 pCi/L, respectively) and high ^{14}C (103.21 and 101.02 pmc, respectively), and plot in an area on figure 8 indicating interaction with soil CO_2 . The other site, D-13-12 03BCC1, had a low tritium value (0.83 pCi/L) and low ^{14}C (41.83 pmc), and plots within the zero-age area of figure 8, which may indicate some equilibrium condition with respect to carbon that precludes radiocarbon age determination. The tritium data from the other five groundwater samples were below the detection limit, indicating that groundwater at these sites was primarily recharged prior to 1952 (pre-modern), and of these data, two samples (D-13-12 10BAC2 and D-13-12 03CCA1) also plot within the zero-age area of figure 8.

Sulfur hexafluoride (SF_6) is a useful tracer for determining the presence of water recharged since the 1970s and has a high rate of increase in the atmosphere (Busenberg and Plummer,

2000). In addition to atmospheric sources, natural sources of SF_6 are known to be present in rocks and minerals (Busenberg and Plummer, 2000). SF_6 was analyzed in six groundwater samples from the area (table 15). Samples were not analyzed from wells B-13-12 10BAC2 and B-13-12 10BAC3 because they did not contain detectable tritium. In addition, well B-13-12 10BAC3 was drilled using air rotary less than a year before the sample was collected; the air rotary drilling method is known to affect gas concentrations in water samples for several years following drilling (Busenberg and Plummer, 2010).

To determine the influence of excess air on SF_6 , nitrogen and argon gas were measured in groundwater samples to determine the recharge temperature and presence of excess air, which can dissolve in groundwater during recharge and water-table fluctuations. Only one well, D-13-12 04DCD, had excess nitrogen gas (N_2), measured at 3 mg/L. This well had

Table 14. Corrected radiocarbon ages for groundwater.

[pCi/L, picocuries per liter; pmc, percent modern carbon (denormalized); pM, absolute percent modern carbon (normalized); BP, before present; R, radiochemistry non-detect; NA, not applicable]

Site name	Sample date	^3H , in pCi/L	^{14}C , in pmc	^{14}C , in pM	^{14}C error, in pM	$\delta^{13}\text{C}$, in per mil	Minimum corrected age, in years BP ^a	Maximum corrected age, in years BP ^b
D-13-12 09AAA1	1/11/2016	8.9	103.21	101.7	0.22	-13.86	NA	NA
D-13-12 10BAC2	1/11/2016	R 0.20	49.92	48.96	0.13	-11.59	NA	NA
D-13-12 03CCA1	1/21/2016	R 0.03	73.20	71.85	0.23	-11.94	NA	NA
D-13-12 03DBA1	1/29/2016	R 0.10	24.50	23.92	0.09	-9.38	NA	2,074
D-13-12 04DCD	2/8/2016	5.4	101.02	99.7	0.29	-14.65	NA	NA
D-13-12 10DDA1	2/9/2016	R 0.08	17.34	16.96	0.09	-10.08	5,148	6,708
D-13-12 03BCC1	2/29/2016	0.83	41.83	40.86	0.18	-9.49	NA	NA
D-13-12 10BAC3	8/16/2016	R -0.02	43.01	42.23	0.12	-12.08	619	1,520

^aCalculated assuming 0 pmc for ^{14}C in carbonate.

^bCalculated assuming 10 pmc for ^{14}C in carbonate.

Table 15. Dissolved gas values in groundwater samples.

[°C, degrees Celsius; cm^3/L , cubic centimeter per liter; STP, standard temperature and pressure; mg/L, milligrams per liter; fmol/kg, femtomoles per kilogram; one mole is equal to 10^{15} femtomoles; NA, not applicable]

Site name	Date	Recharge temperature, in °C	Excess air, in cm^3/L at STP	Excess N_2 , in mg/L	Bottle headspace, in cm^3	SF_6 , in fmol/kg
D-13-12 09AAA1	1/11/2016	22.2	3.3	NA	1.10	0.33
D-13-12 10BAC2	1/11/2016	19.5	3.1	NA	NA	NA
D-13-12 03CCA1	1/21/2016	19.6	3.1	NA	2.50	0.89
D-13-12 03DBA1	1/29/2016	20.1	3.1	NA	0.80	1.40
D-13-12 04DCD	2/8/2016	23.5	2.5	3.0	1.30	0.43
D-13-12 10DDA1	2/9/2016	18.4	2.8	NA	2.00	1.81
D-13-12 03BCC1	2/29/2016	25.7	3.1	NA	0.90	0.99

a hydrogen sulfide smell and may have some denitrification occurring in the well, which would explain the excess N_2 . Excess air values ranged from 2.5 to 3.3 cubic centimeters per liter (table 15). The calculated apparent age from SF_6 will decrease by approximately 2 years per cubic centimeter of excess air per kilogram of water (U.S. Geological Survey, 2017b). An estimated recharge elevation of 3,000 ft could be used in the SF_6 age calculation because the highest point in the area is Wasson Peak at 4,687 ft and the majority of water would likely recharge at lower elevations than at the peak apex.

All samples collected for SF_6 had concentrations above the reporting limit. Three samples (wells D-13-12 03CCA1, DBA1, and 10DDA1) contained measureable concentrations of SF_6 , but tritium concentration was below the detection limit. These samples had greater concentrations of SF_6 than the samples with measurable tritium. This result may indicate that there is a natural source of SF_6 in the study area; SF_6 has been found to be present in elevated concentrations from volcanic and igneous rocks and certain minerals (Busenberg and Plummer, 2000). Because of evidence of background contributions of SF_6 , no age determinations were made for this dataset.

Sediment Geochemistry

Bulk Sediment

The USGS Central Region Mineral Resources laboratory analyzed sediment samples using total and partial digestion techniques. Both methods were used to provide information about the relative mobility of trace elements associated with the sediment samples. Sediment samples represent the surface and upper few inches of the soil profile.

Total Digestion

Sediment samples were analyzed by total digestion methods (Beisner, 2017; table 1). For the total digestion samples, many elements (As, Bi, Cd, Co, Cu, Fe, Hg, In, Li, Mn, Mo, Pb, Sb, U, V, W, and Zn) were elevated in the waste rock and tailings compared with background sediments; some of these elements (Ag, As, Cu, Mn, Pb, Zn) also were elevated in the stream sediments near the waste rock pile (fig. 9). The concentrations of lead were greatest in the waste rock samples, in stream sediment near the waste rock pile, and in background samples near the Old Yuma Fault (figs. 9, 10, 11). Moving away from the waste rock at the Old Yuma Mine, sediments show a decrease in trace element concentration with distance (fig. 11).

Four elements (As, Mo, Pb, and V) had sample concentrations greater than the EPA regional soil screening levels and (or) the AZDEQ soil screening levels (table 3). For arsenic, all sediment samples were greater than the EPA carcinogenic target risk of 3 milligrams per kilogram (mg/kg), all but one background sample (OYM-32-B) were greater than the AZDEQ soil screening level of 10 milligram per

kilogram (mg/kg), and five waste rock samples (OYM-13, 14, 15, 18, and 19-W) were greater than the EPA non-cancer soil screening level of 480 mg/kg. For molybdenum, two waste rock samples (OYM-14 and 15-W) were greater than the EPA soil screening level of 5,800 mg/kg and the AZDEQ soil screening level of 5,100 mg/kg. For lead, all tailings, waste rock, stream sediments, and two of the background samples (OYM-26 and 36-B) were greater than the EPA and AZDEQ soil screening level of 800 mg/kg. For vanadium, five waste rocks samples (OYM-12, 13, 17, 18, and 19-W) were greater than the AZDEQ soil screening level of 1,000 mg/kg.

Sediment samples were compared with average soil concentrations from the Western United States (Smith and Huyck, 1999) to determine if some elements may be elevated in the background samples collected around the Old Yuma Mine. Often in areas of mining activity, there are naturally elevated trace-element concentrations related to the mineralizing event targeted by mining (Plumlee and Nash, 1996; Church and others, 2007). The background samples were collected to the south of the Old Yuma Mine disturbed area on both sides of the hill (figs. 10, 12). Generally, concentrations of elements in background samples were greatest near the Old Yuma Fault and decreased with distance to the east (fig. 12).

Many elements (Al, Fe, K, Ti, Sb, As, Ba, Be, Cd, Ce, Cs, Co, La, Pb, Li, Mn, Mo, Nb, P, Rb, Sr, Sn, and Zn) had higher values in all background samples near the Old Yuma Mine compared to average soils in the Western United States. Lead concentrations in background samples ranged from 30 to 6,410 parts per million (ppm) with a median value of 219 ppm (table 16, fig. 9D). The average value of lead in soils from the Western United States is 17 ppm. Manganese concentrations were greater in background samples (807–7,280 ppm, median 2,510 ppm; fig. 9C) compared to the average Western United States soil value of 380 ppm. Zinc concentrations also were greater in background samples (88–2,940 ppm, median 246 ppm; fig. 9F) compared to the average Western United States soil value of 55 ppm. These results suggest that some trace element concentrations may be naturally elevated in the sediments associated with the mineralizing event that deposited the ore at the Old Yuma Mine.

Partial Digestion

Sediment samples also were analyzed by the EPA 3050 partial digestion method (Beisner, 2017; table 2). The partial digestion results represent a less aggressive digestion, which identifies constituents that may be more available compared with total digestion results. The concentrations of Al, K, Na, Ti, Ba, and Sr were lower in the partial digestion samples than in the total digestion samples. The concentrations of Ca, Fe, and Co also were lower in the partial digestion samples than in the total digestion samples, with a few exceptions. The concentrations of Cu and Mn, as well as As and Cd (with a few exceptions), were greater in the partial digestion samples than in the total digestion samples.

Figure 9. Boxplots of element concentrations determined for total digestion samples. A, arsenic, B, copper, C, manganese, D, lead, E, silver, and F, zinc. Blue represents background samples, orange represents stream sediments, dark red represents tailings, and red represents waste rock.

Figure 10. Geologic map showing sampling locations and concentrations of lead in sediment samples measured in 2015. Locations from U.S. Geological Survey National Water Information System, 2017.

Figure 11. Graph of lead, zinc, and manganese concentrations in stream sediment samples with increasing downgradient distance from sample site OYM-21-S.

Figure 12. Graph of lead, zinc, and manganese concentrations in background sediment samples versus distance from the Old Yuma Fault.

Table 16. Concentrations of elements from background sediments near the Old Yuma Mine compared with average values from the Western United States.[**Bold** values indicate background samples greater than mean of Western United States values; %, percent; ppm, parts per million]

Element	Units	Mean concentration from Western United States soils ¹	Concentration range of background samples near Old Yuma Mine
Al	%	5.8	6.98–8.52
Ca	%	1.8	0.73–4.78
C	%	1.7	0.67–1.64
Fe	%	2.1	3.08–4.56
Mg	%	0.74	0.38–1.51
K	%	1.8	2.44–4.38
Na	%	0.97	0.84–1.89
S	%	0.13	<0.01–0.02
Ti	%	0.22	0.34–0.54
Sb	ppm	0.47	1.77–25.7
As	ppm	5.5	9–35
Ba	ppm	580	759–1,230
Be	ppm	0.68	1.4–2.2
Cd	ppm	0.06	0.2–5.5
Ce	ppm	65	69.5–88.7
Cs	ppm	6	10–39
Cr	ppm	41	14–71
Co	ppm	7.1	10.5–22.3
Cu	ppm	21	15.6–179
Ga	ppm	16	16–20.4
La	ppm	30	31.1–43.1
Pb	ppm	17	30–6,410
Li	ppm	22	76–152
Mn	ppm	380	807–7,280
Hg	ppm	0.046	0.02–0.06
Mo	ppm	0.85	0.99–29.4
Ni	ppm	15	10–44.6
Nb	ppm	8.7	9–15.6
P	ppm	320	570–1,030
Rb	ppm	69	124–266
Sc	ppm	8.2	8.2–13.9
Ag	ppm	0.05	<1–2
Sr	ppm	200	213–414
Th	ppm	9.1	6.5–11.9
Sn	ppm	0.9	1.7–94.7
U	ppm	2.5	1.8–3.2
V	ppm	70	67–186
Y	ppm	22	18.5–24.1
Zn	ppm	55	88–2,940

¹Average soil data from the Western United States from Smith and Huyck (1999) reported as ppm and converted for some elements in this table.

Sediment Leachate

A subset of 15 sediment samples were leached in accordance with the EPA 1312 leachate method to simulate precipitation interacting with the solid material (Beisner, 2017; table 3). The concentrations of leachate samples, however, do not reflect dilution that leachate waters would undergo in the surrounding environment. The dilution factor would depend on the flow rate and water volume, which was not determined in this study.

The pH of the leachate samples increased following the leaching procedure. The initial pH was lowest for tailings samples (6.53–7.33) and variable for other samples (7.18–9.19), whereas the final pH values following the leachate procedure were generally alkaline (8.67–9.72), indicating that the mine waste has low acid-generating potential.

Several leachate samples exceeded the EPA drinking water standards for arsenic, manganese, and lead. Exceedances occurred for all leachates of tailings and waste rock sediment samples, as well as stream sediments collected in a small drainage near the mine. Some leachates of background samples collected south of the mine had concentrations that exceeded the EPA drinking water standard for lead (OYM-28 and 37-B) and the EPA secondary drinking water standard for manganese (OYM-27-B, 28, and 37-B).

Discussion

Assessment of Groundwater Compared with Sediment and Associated Leachate

Geochemical comparisons can be made between the chemistry of the sediment associated with mining activities compared with sediment of similar geologic origin that has not been mined to understand better what elements are associated with the mining activity. The associated sediment leachate can be compared with the groundwater chemistry to understand if there is a component of the groundwater derived from fluids in contact with mining material. Both of these comparisons provide valuable geochemical fingerprints for mining-related signatures that can be used to assess impacts to the current system and for comparison with future samples.

A NMDS analysis was conducted on the total digestion sediment data to understand dissimilarity between sediment

samples and what constituents may be responsible for the dissimilarity. The NMDS analysis of the sediment data resulted in two convergent solutions after 20 tries with a stress of 0.064 (fig. 13), which implies a fair to good fit (Buttigieg and Ramette, 2014). Many constituents are associated with separation between background samples (P, Sc, Be, organic carbon [OC], K, Y, Ti, Nb, Al, Th, La, Ce, Cs, Rb, Ba, Na, Ga) compared with tailings and waste rock (W, In, Bi, Ag, Fe, Cu, Zn, Sb, Co, As, Li, Mo, Pb, U, Hg, Mn, Cd, V) on the first NMDS axis [NMDS1]. Separation between the tailings (Fe, Ag, In, Bi, W) and waste rock (Cd, V, Mn, Hg, U, Pb, Mo, As, Li) occurs on the second NMDS axis [NMDS2]. The stream sediment samples plot between the tailings, waste rock, and background samples indicating that the stream sediment may have a component of mining-related material (fig. 13).

A similarity analysis (ANOSIM) on the sediment samples relative to the sediment-type group resulted in a test statistic of 0.8559 and a p-value of 0.001, indicating there is a statistical difference between at least two of the groups. A cluster analysis was also run on the same elements used in the NMDS analysis and is shown in figure 14. The Calinski criterion indicates that there are two distinct groups: one group includes all tailings samples and all but one waste rock sample and the other group includes waste rock sample OYM-20-W plus all stream sediment and background samples (fig. 14). The separation of the sediment samples generally indicates the mining-related material is distinct from the stream sediment and background samples.

Leachate samples were compared with groundwater samples using a NMDS analysis. The multivariate results show similar distribution when the analysis used major and trace elements compared with trace elements only; thus, only the trace element analyses are presented here. The NMDS analysis of the leachate data resulted in two convergent solutions after 20 tries with a stress of 0.094 (fig. 15), indicating a fair to good fit (Buttigieg and Ramette, 2014). Several elements were associated with separation between leachate from tailings, waste rock, and stream sediment samples (Ag, As, Cd, Cr, Cu, Mo, Pb, Sb, Zn) compared with groundwater samples (Ba, Se, Sr, U, V) on the first NMDS axis [NMDS1] (fig. 15). Leachates from background sediment samples plotted between the groundwater and mining-related material leachates on the first NMDS axis and separated on the second NMDS axis [NMDS2] based on aluminum (fig. 15). Based on these analyses the groundwater samples do not seem to have been influenced by leachate from mining material at the Old Yuma Mine.

Figure 13. Non-metric multidimensional scaling (NMDS) plot for sediment samples with shaded areas outlining sediment sample type. Blue numbers and colored areas represent background samples, orange represents stream sediment, red represents waste rock, and dark red represents tailings. Numbers refer to the sediment sample sequence number. CC, carbonate carbon, OC, organic carbon, TC, total carbon.

Figure 14. Cluster dendrogram for sediment samples. Numbers refer to the sediment sample sequence number. Blue numbers represent background samples, orange represents stream sediment, red represents waste rock, and dark red represents tailings. Solid lines represent distinct groups as determined by the Calinski criterion and dashed lines show subdivisions considered indistinct.

Figure 15. Non-metric multidimensional scaling (NMDS) plot for leachate and groundwater samples with shaded areas outlining aqueous sample type. Numbers represent the sediment sample sequence number. Letters refer to groundwater samples: A) D-13-12 09AAA1, B) D-13-12 10BAC2, C) D-13-12 03CCA1, D) D-13-12 03DBA1, E) D-13-12 04DCD, F) D-13-12 10DDA1, G) D-13-12 03BCC1, and H) D-13-12 10BAC3. Blue numbers and colored area represents background samples, orange represents stream sediment, red represents waste rock, dark red represents tailings, and gray represents groundwater.

An ANOSIM analysis on the sediment samples relative to the type of sample resulted in a test statistic of 0.761 and a p-value of 0.001, indicating there is a statistical difference between at least two of the groups. A cluster analysis was also run on the same elements used in the NMDS analysis and is presented in figure 16. The Calinski criterion indicates that there are two statistically distinct groups. One group includes all mining-related-material and stream sediment leachates, and the other includes background and groundwater samples (fig. 16). The separation of the samples into two groups indicates the samples associated with mining material and stream sediment are distinct from the samples associated with background as well as the groundwater samples.

Strontium isotope ratio was measured on seven leachate samples. Leachate from tailings, waste rock, and stream sediment samples had higher values of strontium isotope ratios (0.71170–0.71228) compared with leachate from the background sediment samples (0.71060–0.71107) (fig. 17). More sample analyses would be needed to determine if the difference is statistically significant. The groundwater sample strontium isotope ratios (0.70968–0.71168) were less than the ratios for mining-related material and similar to background leachate values. The length of time the leachates were exposed to water was shorter than the travel time of groundwater through the subsurface, so the direct comparison of leachate and groundwater samples cannot be made, but can be made generally.

Figure 16. Cluster dendrogram for leachate and groundwater samples. Numbers represent the sediment sample sequence number. Letters refer to groundwater samples: A) D-13-12 09AAA1, B) D-13-12 10BAC2, C) D-13-12 03CCA1, D) D-13-12 03DBA1, E) D-13-12 04DCD, F) D-13-12 10DDA1, G) D-13-12 03BCC1, and H) D-13-12 10BAC3. Blue numbers and colored area represents background samples, orange represents stream sediment, red represents waste rock, and dark red represents tailings. Solid lines represent distinct groups as determined by the Calinski criterion and dashed lines show subdivisions considered indistinct.

Figure 17. Strontium isotopic value versus strontium concentration for leachate samples. Numbers refer to the sediment sample sequence number. Blue represents background samples, orange represents stream sediment, red represents waste rock, and dark red represents tailings.

Conclusions

On the basis of observed water levels, groundwater is generally moving toward the northeast in the Old Yuma Mine study area. Additionally, there is a locally anomalous steep gradient in the groundwater elevation beneath the Quaternary alluvial fan deposits in the center of the study area near the projected trend of the Old Yuma Fault. Groundwater levels within the study area varied 4.5 ft on average over a 2-year period, with a maximum change of 22 ft from historical water levels. Based on groundwater elevation measurements across the study area, an estimate of groundwater elevation beneath the Old Yuma Mine is between 2,400 and 2,450 ft, suggesting the water table is below the 200-ft level of the mine (which is known to be dry). More groundwater elevations near the mine are needed to refine the local groundwater elevation surface, due to the presence of local anomalies in water table elevation located near the Old Yuma fault.

Few groundwater samples exceeded the EPA drinking water standards. One sample exceeded the EPA primary drinking water standard for arsenic; one sample exceeded the EPA secondary drinking water standard for chloride, iron, and

manganese and two other samples exceeded the total dissolved solids secondary drinking water standard. These results suggest the water sampled in the study area is generally of good quality with localized areas of poor quality water.

Analysis of groundwater age indicates groundwater with a component of modern water, containing tritium above the laboratory reporting level, is present on the northwest side of the study area. Groundwater on the southeast side of the study area is primarily older groundwater with tritium below the laboratory reporting level and radiocarbon age ranging from approximately 600 to 6,700 years before present.

Soil screening levels provide thresholds for comparison with human health risks from the sediment associated with the Old Yuma Mine. Comparison of exceedances of standards from background samples with mining-related samples helps to differentiate hazards related to mining-related material. The geochemistry of sediments associated with the Old Yuma Mine and nearby background samples were analyzed by total and partial digestion methods. The relative similarity of concentrations between the total and partial digestions indicate that many of the trace elements associated with mining activity may be in an easily mobilized state. For total digestion samples, four elements (As, Mo, Pb, and V) had concentrations greater than the EPA regional soil screening levels and (or) the AZDEQ soil screening levels. These elements were elevated in some of the mining-related samples; arsenic and lead were elevated in some background samples. Additionally, many elements (Al, Fe, K, Ti, Sb, As, Ba, Be, Cd, Ce, Cs, Co, La, Pb, Li, Mn, Mo, Nb, P, Rb, Sr, Sn, and Zn) had higher concentrations in all background samples near the Old Yuma Mine compared to other soils in the Western United States, suggesting there may be some naturally elevated trace element concentrations in the sediments associated with the mineralizing event that deposited the ore at the Old Yuma Mine.

A NMDS analysis of the geochemistry of sediment samples indicates the sediment associated with tailings and waste rock have different geochemical signatures than background sediments. Stream sediment samples plotted between the tailings, waste rock, and background samples, indicating they have some component of both mining and background sources.

A subset of 15 sediment samples were leached following the EPA 1312 leachate method to simulate precipitation interacting with the solid material. The pH of the leachate samples increased following the leaching procedure, indicating that waste from the Old Yuma Mine has low acid-generating potential. Several leachate samples exceeded the EPA drinking water standards for arsenic, manganese, and lead. Exceedances occurred for all leachates of mining-related material (tailings and waste rock) as well as stream sediments collected in a small drainage near the mine. Some leachates of background samples collected south of the mine had concentrations that exceeded the EPA drinking water standard for lead and the EPA secondary drinking water standard for manganese. The leachates represent a concentrated solution in contact with mining material and would likely be subject to dilution

as the leachate moves through the groundwater system. A NMDS analysis suggests that groundwater samples collected in this study are similar to each other and distinct from leachate samples associated with mining-related material. Thus, the groundwater samples in this study do not seem to be influenced by the elements associated with leachate from Old Yuma mining material at this time.

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Appendix A. Groundwater sample data from Old Yuma Mine study area.

Appendix A is available as an Excel table and may be downloaded from <https://doi.org/10.3133/sir20185019>.

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