

# Engineering Evaluation/Cost Analysis of a Removal Action at the Johnny M Mine and Adjacent Properties

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prepared for:

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# Acronyms and Abbreviations

AOC	Settlement Agreement and Administrative Order on Consent for Removal Action
ARARs	Applicable or Relevant and Appropriate Requirements
ASTM	American Society of Testing and Materials
bgs	below ground surface
BRA	background reference area
°C	degrees Celsius
CERCLA	Comprehensive Environmental Response, Compensation, and Liability Act
cm s <sup>-1</sup>	centimeters per second
COPC	constituents of potential concern
DMR	discharge monitoring report
EE/CA	Engineering Evaluation/Cost Analysis
EPA	U.S. Environmental Protection Agency
ERG	Environmental Restoration Group, Inc.
ET	evapotranspiration
°F	degrees Fahrenheit
FRTR	Federal Remediation Technologies Roundtable
ft	foot/feet
g min <sup>-1</sup>	gallons per minute
HASP	health and safety plan
Hecla	Hecla Limited
IC	institutional control
ITRC	Interstate Technology & Regulatory Council
License	radioactive materials license
М	million
m	meter
$m^2$	square meters
m <sup>3</sup>	cubic meters
$m^3 yr^{-1}$	cubic meters per year
mg kg <sup>-1</sup>	milligrams per kilogram
$\mu R h^{-1}$	microRoentgens per hour
NMEID	New Mexico Environment Improvement Division
NMED	New Mexico Environment Department

# Acronyms and Abbreviations

NORM	naturally occurring radioactive material
NRC	U.S. Nuclear Regulatory Commission
pCi g <sup>-1</sup>	picocuries per gram
pCi L <sup>-1</sup>	picocuries per liter
pcf	pounds per cubic foot
PRSC	post-removal site control
RAO	remedial action objective
Ranchers	Ranchers Exploration and Development Corporation
RSL	regional screening level
SIR	Site Investigation Report (ERG, 2013)
TENORM	technologically-enhanced, naturally occurring radioactive material
UMTRCA	Uranium Mill Tailings Radiation Control Act
USACE	U.S. Army Corps of Engineers
WCS	Waste Control Specialists
yd <sup>3</sup>	cubic yards

# **Executive Summary**

This Engineering Evaluation/Cost Analysis (EE/CA) evaluates alternatives for a non-time-critical removal action at the Johnny M Mine and adjacent properties (project area) located in McKinley County, New Mexico (NM). The project area contains remnants of the surface deposition of mine-related material containing naturally occurring radionuclides in the uranium-238 decay series and certain indicator metals (arsenic, barium, lead, molybdenum, selenium, and vanadium).

This EE/CA was prepared on behalf of Hecla Limited (Hecla) and New Mexico Land, LLC by Environmental Restoration Group, Inc. (ERG) and Alan Kuhn Associates, LLC. It was prepared 1) to satisfy the requirements of paragraph 38 of the Settlement Agreement and Administrative Order on Consent for Removal Action (AOC), dated August 16, 2012, between Hecla and New Mexico Land, LLC; and the U.S. Environmental Protection Agency (EPA, 2012a) and 2) in accordance with "Guidance on Conducting Non-Time-Critical Removal Actions under CERCLA" (EPA, 1993).

Sufficient data to prepare this EE/CA were collected during a site investigation in 2012 at the project area and subsequent work (ITASCA, 2013). This investigation was required by the AOC. The Site Investigation Report (SIR; ERG, 2013) that documents the majority of this work was prepared by ERG and Alan Kuhn and Associates, LLC; submitted to the EPA in 2013, and approved by the EPA in February 2014.

The following are provided in this EE/CA:

- a description of the physical, demographic, and other characteristics of the project and surrounding areas;
- a streamlined risk evaluation focusing on human health and based on current conditions and potential future land use;
- an identification of removal action objectives (RAOs);
- an identification and analysis of removal action alternatives based on effectiveness, implementability, and cost;
- a comparative analysis of removal action alternatives; and
- a recommended removal action alternative.

The streamlined risk evaluation identifies radium-226 and uranium, which is co-located with radium-226, as the constituents of potential concern (COPCs) in the project area.

The RAOs identified and addressed in this EE/CA are:

- reduce soil concentrations of COPCs below a level resulting in an excess human cancer risk of 1x10<sup>-4</sup>;
- reduce soil concentrations of COPCs below a Total Hazard Quotient of 1; and
- minimize or eliminate the migration of mine-related material containing elevated soil concentrations of COPCs to surface water, air, and land.

The removal action alternatives evaluated are: 1) no action, 2) on-site disposal, and 3) off-site disposal.

The recommended removal action alternative for the project area is on-site disposal. This removal action alternative meets all of the RAOs and satisfies Applicable or Relevant and Appropriate Requirements and is the most effective, implementable, and cost effective of the alternatives evaluated. Potential exposure of workers and the public to mine-related material can be effectively mitigated through use of common engineering and administrative controls. Potential environmental and safety impacts associated with off-site transportation of mine-related material are avoided. Access controls associated with the repository would be implemented and maintained. An enforceable, restrictive covenant would be recorded to control future use of the land owned by New Mexico Land, LLC, which is where the repository would be located. Land use within the project area would not otherwise be restricted.

# **Section 1.0 - Introduction**

This Engineering Evaluation/Cost Analysis (EE/CA) evaluates removal action alternatives for the Johnny M Mine and adjacent properties (project area), located in McKinley County, New Mexico (NM). Sufficient data to prepare this EE/CA were collected during a site investigation in 2012 at the project area and subsequent work (ITASCA, 2013). This investigation was required by the Settlement Agreement and Administrative Order on Consent for Removal Action (AOC), dated August 16, 2012, between Hecla and New Mexico Land, LLC; and the U.S. Environmental Protection Agency (EPA, 2012a). The Site Investigation Report (SIR; Environmental Restoration Group [ERG], 2013) that documents the majority of this work was prepared by ERG and Alan Kuhn and Associates, LLC; submitted to the EPA in 2013, and approved by the EPA in February 2014.

The project area contains remnants of the surface deposition of mine-related material containing naturally occurring radionuclides in the uranium-238 decay series and indicator metals. Long-lived radionuclides in the uranium-238 series include naturally occurring isotopes of uranium (uranium-238, uranium-235, and uranium-234) in naturally occurring isotopic ratios, radium-226, and thorium-230.

Mine-related material includes soil and rock from the historic mining operations that is elevated in uranium decay series radionuclides, including water treatment residuals in the historic mine water treatment area and along the discharge path. Mine-related material does not include background concentrations of naturally occurring radioactive materials or stable elements. This definition is consistent with the definition in the AOC. Indicator metals are defined for the purposes of this EE/CA as the metals (arsenic, barium, lead, molybdenum, selenium, and vanadium) that are indicative of mine-related material (ERG, 2013).

Indicator metals and uranium are sufficiently co-located with radium-226 such that radium-226 concentrations in soil and surrogate measurements can be used to guide removal of mine-related material (ERG, 2013).

This EE/CA was prepared on behalf of Hecla Limited (Hecla) and New Mexico Land, LLC of Coeur D'Alene, Idaho by ERG and Alan Kuhn Associates, LLC, both from Albuquerque, NM. It was prepared to satisfy the requirements of paragraph 38 of the AOC.

# 1.1 Purpose

The purpose of this EE/CA is to describe the objectives for a non-time-critical removal action (removal action); identify and evaluate available removal action alternatives, and recommend a removal action. It was prepared in accordance with EPA's "Guidance on Conducting Non-Time-Critical Removal Actions under CERCLA" (EPA, 1993). The EE/CA is organized as follows:

- Sections 1 and 2: Introduction and Site Characterization summarize the data used to characterize the nature and extent of mine-related material in the project area and evaluate potential risks to human health.
- Section 3: Streamlined Risk Evaluation presents a streamlined human health risk evaluation for radionuclides and indicator metals, based on the existing nature and extent of mine-related material and potential future land uses, and identifies the constituents of potential concern (COPCs) for the project area.
- Section 4: Identification of Removal Action Scope, Goals, and Objectives identifies the scope, goals, and removal action objectives (RAOs) and summarizes applicable or relevant and appropriate requirements (ARARs).

- Section 5: Identification and Evaluation of Removal Action Alternatives identifies applicable technologies and develops alternatives for removal actions at the project area. This section also evaluates each of the alternatives considered based on effectiveness, implementability, and cost.
- Section 6: Comparative Analysis of the Removal Action Alternatives provides a comparative analysis of alternatives considered to identify trade-offs.
- Section 7: Recommended Removal Action Alternative identifies the alternative that best satisfies the criteria used in the evaluation and meets RAOs.

# **1.2** Location

The project area, shown on Figure 1, is located on private land within the Ambrosia Lake mining district in McKinley County, NM, just north of New Mexico Highway 605 and 4.4 miles west of the village of San Mateo. It lies within portions of Sections 7 and 13; and all of Section 18 in Township 13 North, Range 8 West divided into the three areas described in Section 2.1. The geographic location of the project area is Latitude 35.361959 and Longitude -107.7211956, as identified in the AOC.

# **1.3 History and Current Site Conditions**

Development of the Johnny M Mine was initiated by Ranchers Exploration and Development Corporation (Ranchers), a lessee, in 1972. The first ore was produced in 1976. Ore production ended in 1982. All ore was shipped off-site for the milling and recovery of uranium. Ore was hauled within the project area from the mine area and along Marcus Road for approximately one mile southwest to New Mexico Highway 605.

Ranchers merged with Hecla Mining Company (now Hecla Limited) in 1984. The mine property was reclaimed over time, starting in 1982. The radioactive materials license (License) was terminated in 1993 (NRC, 1993).

Approximately 286,000 tons of tailings sands from the Kerr McGee Mill were used as underground structural support material (backfill material) as part of the mining operation, an activity requiring a License, which was issued by the New Mexico Environmental Improvement Division (NMEID) on June 21, 1977. Backfill operations at the mine started upon receipt of the License and continued until January 1982. Two small surface locations, totaling approximately two acres, were used to store the backfill material. NMEID relinquished oversight of the uranium recovery licensing program, and therefore the License, to the U.S. Nuclear Regulatory Commission (NRC) in 1986.

The NRC approved reclamations plans for licensed materials and oversaw reclamation activities from 1987 to 1992. The reclamation plans addressed the two backfill storage areas mentioned above and other areas where radiation levels in soil from licensed material exceeded NRC standards. The NRC terminated the License on May 21, 1993, following the satisfactory completion of these reclamation activities.

Water from mine dewatering and operations was contained in on-site settling ponds. The locations of these ponds are shown on Figure 2, along with other mine site features. Pond 1 was constructed in 1973 followed by Pond 2 in 1974. Each pond was approximately 100 by 400 by 15 feet (ft) deep and constructed in subgrade native materials consisting of the Mancos Shale (New Mexico Environment Department [NMED], 2010).

The flow rate of water from the pond(s) varied over time but averaged approximately 700 gallons per minute (g min<sup>-1</sup>) based upon limited available information (Ranchers, 1978). This flow rate was much lower during mine development (EPA, 1975).

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Starting in 1973, discharge from the settling ponds was by gravity flow through an unlined ditch. A 12-inch diameter pipeline replaced the ditch in March of 1978. The location of the ditch and pipeline are shown on Figure 2. Mine water was treated in the two settling ponds, including the use of a coagulant and a barium chloride solution. This occurred until 1982 when mine production ceased.

From a review of the few discharge monitoring reports (DMRs) available, it appears the Johnny M Mine initially had a single flow monitoring device, which recorded the total flow volume pumped from the underground workings to the two settling ponds. Recycle water was drawn from the settling pond water for reuse in the underground mining activities, thus creating a recirculating flow rate. A specific flow meter was installed in October 1981 to monitor the discharge from Pond 2. Mine closure began in January 1982, thus only November and December 1981 had both total flow rate pumped out of the mine and total DMR flow rate from Pond 2, representative of normal production operations. Flow rates from the mine for these two months were 862 and 823 g min<sup>-1</sup> (842.5 g min<sup>-1</sup> average). DMR flow rates from Pond 2, for these same two months, were 527 and 580 g min<sup>-1</sup> (553.5 g min<sup>-1</sup> average). These are the only estimates available for average DMR flow rates from Pond 2 for normal operations. The average recirculating flow rate for underground use for these two months is estimated at 289 g min<sup>-1</sup>.

Site investigations and/or remediation activities, addressed in Section 2.5, have been conducted by Ranchers, EPA, NMED, and Hecla.

The project area is currently unoccupied and described in Section 2.1.

# **1.4 Regulatory History**

This EE/CA was prepared in accordance with the AOC and the Comprehensive Environmental Response, Compensation, and Liability Act of 1980 (CERCLA) to support a non-time-critical removal action.

The mine backfilling operation required a License under the Atomic Energy Act, as amended by the Uranium Mill Tailings Radiation Control Act (UMTRCA). A License was in place from 1977 to 1993.

EPA issued NPDES Permit No. NM0026573 to the Johnny M Mine. This permit was terminated in 1982 when site operations ceased. The Johnny M Mine addressed state groundwater requirements under the NMEID-approved groundwater discharge plan DP-20.

EPA Region 6 conducted aerial surveys in October 2009 of residential and near-residential areas in the Grants and Cebolleta Land Grant Areas in New Mexico to identify anomalous surface expressions of uranium concentrations. The Johnny M Mine was identified in one of these aerial surveys (EPA, 2010b).

In November 2010, EPA Region 6 received a request for assistance from the NMED to evaluate the project area for potential removal action. Subsequently, EPA performed two investigations of the project area, as outlined in Section 2.5.

# 2.1 Site Description

The project area includes the historic Johnny M Mine (Area A) and adjacent properties, specifically the properties west of Area A within the western half of Section 18 (Area C); within both the eastern half of Section 18 and southern half of Section 7 (Area B); and drainage pathways to the west of Area C. A background reference area (BRA) was established in the southwest corner of Section 7 (Township 13 North, Range 8 West) as described in the SIR (ERG, 2013).

Figure 3 shows the project area and BRA referred to in this EE/CA.

# 2.2 Physical Setting

# 2.2.1 Geography

The geography of the project area is representative of northwestern New Mexico and, specifically, the San Juan Basin. San Juan Basin topography is characterized by the combination of two land forms: mesas which dip gently to the north and broad valleys with intermittent streams. Arroyos have incised the mesas by headward erosion, forming steep-sided canyons.

Area A is relatively flat and bordered on three sides by mesas extending to approximately one hundred ft above the mine. The project area broadens to the south, east, and west of the mine. An east-west trending drainage (the primary arroyo) crosses the southern edge of Area A and extends to the western edge of Area C. A small mesa outcrops in the east-center of Area C. The mesa that curves around three sides of Area A also crosses the middle of Area B, occupying approximately 40 percent of the latter.

The project area is proximal to the San Mateo Mine, which has similar physical and environmental conditions and mine-related material.

# 2.2.2 Land Use

Livestock grazing is the predominant land use within one kilometer of the project area.

# 2.2.3 Population

The vicinity of the project area is sparsely populated. The village of San Mateo is approximately 4.4 miles to the southeast. Small residences, on large areas of land supporting mostly livestock grazing, occur in the vicinity of the project area. The nearest residence is across State Highway 605, approximately 1,000 ft from the southwest corner of the land in Area C owned by New Mexico Land, LLC.

# 2.2.4 Man-made Features

Man-man features in Area A include a Quonset type structure and several concrete pads; two of which contain circular concrete caps, one over the shaft and the other over a vent rise. There are power lines and poles in the project area running east-west near the Section 7/18 boundary. The waste rock piles and drainages along the edge of the Area A contain concrete debris, pipes, and exposed wires (mine-related debris). There are two former settling ponds in Area A, partially backfilled, within historically reclaimed areas.

There are no man-made features on Area B, other than fences, electricity poles, and roads.

Area C contains mine-related debris in the north-south trending drainage near the northeast corner of the area. Aboveground structures were removed from Area C in 2013. Segments and pieces of transite and steel pipe occur along the former path of a discharge pipeline. Pop-ups for telephone wires are located along Marcus Road. There are two historic monitoring wells located within the southern portion of Area C, near Highway 605.

## 2.2.5 Climate

The description of the climate in the project area presented here is adopted largely from "Baseline Data Report for the Roca Honda Mine, Revision 1" (Roca Honda Resources, 2011).

The regional climate may be classified as arid to semiarid continental, characterized by cool, dry winters, and warm, dry summers. Abundant sunshine, low relative humidity, and large annual and diurnal ranges in temperature are characteristic of the area.

Temperature extremes, measured at the San Mateo weather station, have ranged from -35.0 degrees Fahrenheit (°F: -37.2 degrees Celsius [°C]) in January 1971 to 102.9 °F (39.4 °C) in June 1962. Average high and low temperatures are 40.6 °F (4.78 °C) and 15.4 °F (-9.22 °C) for the coldest (January) and 82.9 °F (28.3 °C) and 55.0 °F (12.8 °C) for the warmest month (July). The average diurnal variation in temperature throughout the year at San Mateo is generally 25 to 30 °F (approximately 15 °C).

Precipitation data obtained at the weather station in San Mateo, NM from 1918 to 1988 indicate that the project area has an average annual rainfall of less than 9 inches (ITASCA, 2014). The wettest period is in late summer and early fall. Winter is the driest season. Approximately half of the annual precipitation in the region occurs in July through September, which averages more than 50 days of brief thunderstorms per year. The storms are sometimes heavy and can be accompanied by hail and strong, gusty winds. These storms may bring several inches of rain to small areas in a short time, and runoff frequently causes local flash floods. In addition, precipitation events lasting several days may occur occasionally in September and October when the remains of tropical cyclones move into the area from the Gulf of Mexico or Gulf of California. Snow falls from November through March and is light on valley floors, but increases at the higher elevations of the nearby mesas and mountains. The estimated average annual snowfall is 26 inches for the San Juan Basin.

Relative humidity is highest in the early morning when it is approximately 70 percent in the winter and 45 percent in the summer. Relative humidity typically falls to approximately 40 percent in the winter and 15 to 20 percent in the summer, as the day progresses and temperatures rise. June is usually the driest month: mid-afternoon relative humidity is typically less than 15 percent. The sun shines approximately 80 to 85 percent of the time in June; approximately 75 to 80 percent for the rest of the summer; and 65 to 70 percent in winter.

The annual rate of evaporation in the region is approximately 75 to 80 inches. The net annual lake evaporation for the region is 30 to 40 inches. Pan evaporation rates for two measuring stations near the project area indicate yearly evaporation rates of about 63 inches per year.

Large-scale (or synoptic) winds in the region are most frequently from the southwest and west and are strongest between March and June, with the highest average speeds in March. Strong winds can accompany frontal activity associated with late winter and spring low pressure systems and thunderstorms. The strong spring winds often bring considerable dust into the area.

# 2.3 Geology and Geomorphology

# 2.3.1 Geology

The description of the geology of the region and project area presented here is adopted from "Analysis of Groundwater Conditions at the Former Johnny M Mine, McKinley County, New Mexico" (ITASCA, 2013), which is provided as Appendix A. Excerpts of the report are presented below and focus on the relationship of geology to hydrogeology.

# <u>Regional</u>

Three structural features associated with the San Juan Basin (the Zuni uplift, Chaco slope, and Rio Grande Rift) are particularly important to the hydrogeology of the region. The Zuni uplift is located approximately 25 to 30 miles southwest of the project area. This uplift is an important regional structural feature that exposes rocks as old as Precambrian in age and is an important location of regional recharge to groundwater. The area of transition from the Zuni uplift to the central part of the San Juan Basin is the Chaco slope, where regional sedimentary strata of mainly Mesozoic age dip gently to the northeast, into the central part of the basin. The dip of the rock units varies between 4 to 8 degrees. The Rio Grande Rift is located on the southeast margin of the San Juan Basin and groundwater flow in the southeastern portion of the basin is generally directed toward this regional structural feature.

The stratigraphic column of geologic units encountered regionally includes several units, such as the Menefee Formation, Point Lookout Sandstone, and Mount Taylor volcanics, that are not present in the project area due to erosion.

Soils that are classified as alluvium or colluvium can comprise up to approximately the upper 80 ft of geologic materials in the region; e.g., the alluvium in the San Mateo Creek drainage. These soils typically are unsaturated near the mesas and become saturated near the San Mateo Creek drainage, which flows intermittently.

# Project Area

The main body of the Mancos Shale lies below the surficial soils in the project area, forming a widespread regional aquitard that locally is approximately 600 to 1,000 ft thick. The Mancos Shale represents the interplay of transgressive and regressive episodes of the epicontinental Western Interior Seaway. Shale, mudstone, claystone, and limestone were deposited during transgressions, and sandstones were deposited during regressions. The Twowells Sandstone Tongue, an interbed of Dakota Sandstone, occurs between the main body of the Mancos Shale and the Whitewater Arroyo Tongue of the Mancos Shale. Other localized sandstone lenses are also present within the main body of the Mancos Shale. Two such lenses exist in the Mancos Shale in Area C.

The Dakota Sandstone is located below the Mancos Shale and was deposited during the initial transgression of the seaway, although, as previously noted, there is some interbedding between these formations. The Twowells Sandstone Tongue is the uppermost unit of the Dakota Sandstone with thickness of about 30 to 120 ft, averaging approximately 70 ft. This is the uppermost water bearing zone in the project area and, based on the depth of one of the wells installed at the former (b) (6) residence (GMD-04: depth to water at 624 ft below top of casing and a total depth of 715 ft below ground surface [bgs]), also appears to be the unit in which GMD-04 is screened. Below the Twowells Sandstone is another approximately 50 to 150 ft of Mancos Shale (the Whitewater Arroyo Shale Tongue), and below that is the 20 to 80 ft thick main body of the Dakota Sandstone. The historic Johnny M Mine potable water well used during mine operations (no number available) was apparently screened in the main body of the Dakota Sandstone (water level at a depth of 673 ft below top of casing and a total depth of 1,084 ft bgs).

The Morrison Formation is located below the main body of the Dakota Sandstone. The uppermost portion of the Morrison Formation is the Brushy Basin Member. Excluding the sandstone Poison Canyon Tongue at its base, the Brushy Basin Member is green shale with very low hydraulic conductivity. The Brushy Basin Member averages about 100 ft thick in the local area. The Johnny M Mine recovered ore from sandstones in the Morrison Formation, namely the Poison Canyon Tongue, at the base of the Brushy Basin Member, and the subjacent (approximately 25 ft below) Westwater Canyon Member of the Morrison Formation, at depths of approximately 1,300 to 1,400 ft bgs.

## 2.3.2 Geomorphology

The geomorphology of the project area is typical of the mesa-and-valley terrain of the Colorado Plateau. Mesas capped by Gallup Sandstone are separated by pediments with shallow alluvial, colluvial, and eolian soils. Mesa slopes are retreating gradually, over geological time, as the Gallup Sandstone caprock is undermined by erosion of the Mancos Shale below it, forming talus slopes and colluvial fans that cover the bases of the mesas. Mancos Shale underlies the project area, either outcropping or covered by colluvial/alluvial soils derived from the mesa slopes. Most of the colluvial fans are geomorphologically active with deeply incised arroyos that are actively headcutting (degrading).

Arroyos in the project area display the following characteristics:

- An upstream reach of channel erosion and headcutting into the talus and colluvial slopes. Channels are bare of deep-rooted vegetation; and side slopes are oversteepened and undercut. Erosion occurs in this section during all runoff events.
- A midstream reach (10s to 100s of feet) over which the arroyo channel loses definition; and the sides of channels diverge and become shorter and flatter. Channel vegetation in this section is large enough to deflect most of the flow, and lower plant stems are covered by soil. Erosion and deposition in this section is in a general equilibrium, depending on the amount of flow in each runoff event.
- A downstream reach of deposition with low, rounded channel side slopes. Channel vegetation in this reach is relatively substantial in size and density, similar to the surrounding ground. Deposition occurs therein, during most runoff events.

Arroyos disappear once they reach the valley floors (pediments) due to active aggradation, or deposition of soil eroded from the colluvial fans, with the exception of the major watershed channels. No major watershed channels cross the project area. In general, each arroyo system originating on the mesas north and east of Areas A and C has an upstream reach and a downstream reach separated by a short midstream reach. Figure 4 shows the actively eroding (upstream), transitional (midstream), and actively aggrading (downstream) sections of the arroyos.

The primary arroyo (shown on Figure 4) running east to west across Area B, south of area A, has an actively eroding (degrading) reach east of Area B, in the headwater canyons. The midstream reach of the primary arroyo, starting east of the former pipeline crossing and ending 100-200 ft west of the property fence in Area C, appears to be in equilibrium with the current hydrologic regime. The secondary arroyos, tributaries of the primary arroyo that extend north toward Area A, display upstream characteristics with headcutting into mine-related material and underlying native soils. To the west, the downstream reach of the primary arroyo has an aggrading channel through most of Area C (see Figure 4). The northeastern portion of the land owned by New Mexico Land, LLC appears not to be contributing much sediment to the primary arroyo. In addition, only the downstream reaches of arroyos from the north cross the vicinity of the northeastern portion of the land owned by New Mexico Land, LLC.

The lower elevations of the ground surface in Area A to the toes of the mesa slopes have been substantially altered from its natural condition by mine-related activities. Arroyos have apparently been filled or displaced; talus and colluvial deposits have been excavated and placed as fill in the mine area, and the two mine water settling ponds have been partially backfilled from mine development through operations and subsequent reclamation. The present geomorphological features of this area are consequently recent and do not reflect either the original or natural conditions.

Considering the active headcutting that is occurring in the tributaries extending to the north from the primary arroyo in Area A, it is likely that without significant engineering controls additional headcutting would eventually occur to the north into and through the mine waste rock and historic settling pond area. Figure 4 depicts these tributaries.

In addition to the mine-related impacts, drainage features in the project area have been altered from natural conditions by 1) pre-mining diversions to bring runoff closer to the (former) Marcus Ranch and 2) culverts under Highway 605. Subsequent alterations include installation of the access road (Marcus Road) and the diversion along Highway 605, east of Marcus Road. Marcus Road acts as a surface flow barrier in the south half of Area C. These are shown on Figure 5.

Other pre-mining drainage features west of the project area that are unrelated to natural conditions are depicted in Figure 5, which also depicts current directions of surface water runoff.

Vegetation patterns and surficial soils on ground surfaces outside of arroyos, Area A, and the mesa slopes appear to be geomorphologically stable. Although a substantially thick Quaternary eolian sand is interbedded with alluvium in the southern part of Area A and along the primary arroyo, no dune fields or deflation basins were identified on the ground surface of Areas A, B or C. Therefore, wind scour and deposition do not appear to be active on the project area.

# 2.4 Hydrology

The description of the hydrology of the region and project area presented here are largely adopted from "Analysis of Groundwater Conditions at the Former Johnny M Mine, McKinley County, New Mexico," (ITASCA, 2013, see Appendix A). Excerpts of the report are presented in the following sections.

#### 2.4.1 Regional

In the San Juan Basin (including the project area), there are several thick, very low-permeability shale layers (e.g., the Mancos Shale, Brushy Basin Member of the Morrison Formation, and the Recapture Shale) that hydraulically separate the formations that serve as water resources in the region. These shale layers separate the deeper water-bearing units (i.e., the Gallup Formation, Dakota Sandstone Formation, and Westwater Canyon Member of the Morrison Formation) from each other, the shallow water-bearing formations, and the much shallower alluvial groundwater systems. For the purposes of this EE/CA, the separation between the deeper water-bearing units and the shallower water-bearing units is the top of the Gallup Formation, which overlies the Mancos Formation that outcrops within the project area. Only the deeper water-bearing units and shallow, unsaturated alluvium exist in the project area.

In general, groundwater recharge enters the deeper water-bearing units as precipitation on permeable formations that crop out along the southern margin of the San Juan Basin and on the flanks of the Zuni, Chuska, and San Mateo mountains. Groundwater then flows downgradient, either northwestward to discharge along the San Juan River, or, in the southeast portion of the basin (where the project area is located), northeastward, eastward, and southeastward toward the Rio Grande Rift, to discharge to tributaries of the Rio Grande River including the Rio Salado, Rio Puerco, and Rio San Jose rivers.

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The pattern of regional groundwater movement within the deeper units in the southeastern part of the San Juan Basin is greatly influenced by the Zuni uplift, San Mateo Dome, Rio Grande Rift, and McCartys syncline.

The movement of groundwater through the alluvial valleys is influenced by topography and surface water drainages and is independent of—and sometimes flows in directions opposing—groundwater movement in the deep water-bearing units. Volcanic rocks of the Mt. Taylor volcanic field exist less than five miles to the east and south of the project area. This is an area of local and regional groundwater recharge for shallower rocks of the Tertiary and Upper Cretaceous age. However, these younger, shallower waterbearing units in the region (e.g., the Menefee Formation and Point Lookout Sandstone) are not present in the project area. Where present regionally, these units occur higher in the stratigraphic sequence and are hydraulically separated from the deeper water-bearing resources bearing units (i.e., Dakota Sandstone and Morrison Formation) by the Mancos Shale aquitard.

Important water bearing units, such as the Dakota Sandstone, are substantially deeper below land surface (approximately 350 to 700 ft deeper per mile down dip) to the northeast of the project area than they are beneath the project area because of the dip associated with the Chaco slope. Accordingly, the geologic units in the project area that could be considered water resources, such as the Dakota Sandstone, are less desirable as a source of groundwater downgradient of the project area due to high costs of drilling deep wells. Groundwater flow in the deep Dakota Sandstone and Morrison formations is to the east-southeast in the region of the project area.

The nearest domestic wells in the general topographical downgradient direction from the project area are screened in the much shallower Menefee Formation or Point Lookout Sandstone. Both units are absent in the project area and are located above the Mancos Shale aquitard, which is the uppermost bedrock unit present in the project area. These wells are at least four miles east of the project area; furthermore, the hydraulic gradient in the project area is downward, away from the units in which these wells are screened.

# 2.4.2 Project Area

#### Shallow Groundwater (Surficial Soils)

The investigations reported in the SIR (ERG, 2013) found no saturated zones in alluvium within the project area. Although shallow groundwater can be found in the San Mateo Creek drainage alluvium, the limited watersheds of San Mateo tributaries within and upstream of the project area, relatively steep gradients of these tributaries, and low precipitation of the semi-arid climate should control and limit alluvial groundwater to intermittent, seasonal flow.

#### Deep Groundwater (Dakota Sandstone and Morrison Formation)

The Mancos Formation, dominated by the Mancos Shale, separates the deep groundwater in the project area from both direct infiltration of rainwater and hydraulic connection with shallow alluvium deposits. The hydraulic conductivity in the Mancos Shale is generally very low, on the order of  $5 \times 10^{-8}$  centimeters per second (cm s<sup>-1</sup>). To put this value into context, a compacted clay liner for a municipal landfill typically has a permeability (hydraulic conductivity) of approximately  $1 \times 10^{-6}$  cm s<sup>-1</sup>. Isolated sandstone lenses typically occur within the Mancos Shale and have been noted in drill logs from the project area.

The hydraulic conductivities of the Dakota Sandstone are  $9 \times 10^{-5}$  to  $5 \times 10^{-4}$  cm s<sup>-1</sup>, suggesting it is capable of transmitting low to moderate volumes of water depending on its thickness.

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The hydraulic conductivity of the Westwater Canyon Member varies from  $7 \times 10^{-6}$  to  $6 \times 10^{-4}$  centimeters per second (cm s<sup>-1</sup>). The direction of flow for groundwater in the project area in the Westwater Canyon member is towards the north-northeast. The hydraulic gradient for the Westwater Canyon Member is approximately 0.03 ft/ft to the northeast. Groundwater velocities are 2 to 160 ft per year in the Westwater Canyon Member, assuming an effective porosity of 0.1. It would take groundwater approximately 3,330 to 2,640,600 years to travel one mile, based upon this range of values.

The direction of flow in the Dakota Sandstone and Morrison formations is downward and eastward/northeastward from the project area, away both vertically and laterally, from the New Mexico Land, LLC property and vertically away from the topographically downgradient domestic/stock wells in the Menefee Formation and Point Lookout Sandstone. In addition to the vertically downward gradients in these deep formations, the thick shales of the Mancos Formation further separate the Menefee/ Point Lookout water-bearing units to the north and east of the project area from hydraulic connection with the underlying Dakota and Morrison water-bearing units.

# 2.5 Site Investigations

The EPA five year plan for the Grants Mineral Belt (EPA, 2010a) led to several regional investigations conducted by the EPA. The first investigation was an aerial radiological assessment (EPA, 2010b) of the area, which was followed by two site-specific investigations (EPA, 2011 and 2012b).

The AOC focuses on mine-related material in the project area. The field activities documented in the SIR (ERG, 2013) were conducted at the project area in 2012.

## 2.5.1 EPA Investigations

The aerial radiological assessment was conducted over the nominal Grants and Cebolleta Land Grant Areas in October 2009 (EPA, 2010b). Additional investigations within the project area, conducted as a response to the findings of the aerial survey and land use, occurred from 2010 through 2012; and consisted of GPS-based gamma surveys; indoor and outdoor exposure rate measurements, static (integrated) and down-hole gamma measurements; and soil and groundwater sampling (EPA, 2011 and 2012b).

The down-hole measurements were made in two phases. Phase 1 consisted of 136 measurements made in the center of 136, 100-ft by 100-ft grids placed over approximately 31 acres that encompassed the former **(b) (6)** residence and horse stables in Area C. Phase 2 consisted of 209 and 97 measurements made in the center of 100-ft by 100-ft and 200-ft by 200-ft grids, respectively, established over the remainder of the former **(b) (6)** property (now owned by New Mexico Land, LLC) exhibiting elevated surface gamma readings.

The nodes of each grid were logged to depths of up to 36 inches with a Ludlum Model 44-10 2-inch by 2-inch sodium iodide detector coupled to a ratemeter.

The second site-specific investigation conducted by the EPA consisted of soil sampling and analysis, a GPS-based gamma walkover survey and static (integrated) measurements in Area A.

Relevant results of the previous EPA investigations are as follows:

- Radium-226 concentrations in surface (0 to 3 or 0 to 6 inches bgs) and near surface (collected at 6-inch intervals to 30 inches bgs) soil samples collected in the vicinity of the former Jackson residence were 0.662 to 370 picocuries per gram (pCi g<sup>-1</sup>).
- Thorium-232 concentrations in these soil samples were -0.396 to 6.972 pCi g<sup>-1</sup>.

- Exposure rates measurements made in and around the former (b) (6) residence were 14.8 to 103.5 microRoentgens per hour ( $\mu$ R hr<sup>-1</sup>).
- Elevated gamma levels were observed in 40 of the 136 Phase 1 borings. The depth of contamination was not defined in ten of the borings.
- Elevated gamma levels were observed in 27 of the 306 Phase 2 borings. The depth of contamination was not defined in nine of the borings.
- Elevated gamma levels were observed at 83 of the 99 locations where static (integrated) gamma count rates were measured.
- The concentrations of radium-226 in 12 surface soil samples collected from Area A and a background area were 2.64 (background) to 317 pCi g<sup>-1</sup>.

# **2.6 Investigations Conducted Under the AOC**

The field activities reported in the SIR (ERG, 2013) included geomorphological field and GPS-based gamma walkover surveys; exposure and static gamma count rate measurements at fixed points; and soil sampling for geotechnical parameters, radionuclides and indicator metals. Soil sampling for radionuclides, and indicator metals due to their expected co-location, was guided by down-hole gamma measurements.

Exposure rates, predicted from project area-wide gamma count rate measurements, are 10.4 to 401.1, averaging 17.4  $\mu$ R h<sup>-1</sup>. There are 57,226 gamma count rate measurements in the data set, with a standard deviation of 15.2  $\mu$ R h<sup>-1</sup>. The distribution of predicted exposure rates is best described by the median (14.0  $\mu$ R h<sup>-1</sup>) and quartiles, given that it is non-parametric. The first and third quartiles are 13.1 and 15.2  $\mu$ R h<sup>-1</sup>, respectively.

The estimated depth of mine-related material in the Area A borings (see Figure 6), based on down-hole logging, is summarized as follows:

- The range is 0 (Borings AA-02 and AA-05) to greater than 7 m (Boring AA-09).
- The maximum is 3 m in borings located outside of the historical ponds (AA-01 through AA-05, AA-11, and AA-12).
- The maximum is greater than 7 m in borings located inside the historical ponds (AA-06 through AA-10).

The estimated depth of mine-related material in the Area C borings (see Figure 7), based on down-hole logging, is summarized as follows:

- The range is 0.1 (Boring AC-09) to 1.2 m (Boring AC-07) in the cluster of Borings AC-07 through AC-11.
- The deepest level of mine-related material was 1.7 m, in Boring AC-06. This boring is located in the primary arroyo.
- The range is 0.3 to 0.6 m in the borings (AC-03 through AC-05; AC-12 through AC-14, AC-20 through AC-22) located in the arroyo on the western edge of Area C and extending onto Section 13.

• The range is 0 to 1.4 m in the borings (AC-01 and AC-02; AC-15 through AC-19; and AC-23) associated with the pipeline. The maximum estimated depth was observed in Boring AC-17.

The areal extent of the mine-related material was estimated using the results of gamma surveys. It is consistent with the areal extent reported in previous investigations and the vertical extent was further delineated. Surface elevations were estimated by merging data from historical and current topographic surveys. The depths of the mine-related material were estimated using results of down-hole logging, soil sampling, and geotechnical properties.

The findings of the investigation are:

- The horizontal and vertical extents of potential mine-related material were delineated sufficiently to support remedy selection and design.
- A representative BRA was established in Area B. The BRA is isolated from mine-related material and its soil types are representative of the majority of low-lying portions of Area C.
- The estimated volume of mine-related material is 457,000 cubic meters (m<sup>3</sup>), delineating to the project area background concentration (0.9 pCi g<sup>-1</sup>) of radium-226 in soil. The estimated volume of mine-related material is 314,000 m<sup>3</sup>, delineating to 3.5 pCi g<sup>-1</sup> radium-226 in soil<sup>1</sup>. The estimated volume of mine-related material using the 5 pCi g<sup>-1</sup> radium-226 plus background in soil standard<sup>2</sup> applied at the San Mateo Mine is 272,000 m<sup>3</sup>. These volumes represent in situ, unexcavated (bank) volumes. The volume associated with delineating to 3.5 pCi g<sup>-1</sup> radium-226 in soil (314,000 m<sup>3</sup>) is equivalent to about 413,000 cubic yards (yd<sup>3</sup>), which was rounded up to 500,000 yd<sup>3</sup>, assuming a 20 percent increase to account for the swelling of the mine-related material upon excavation. The latter volume is assumed for the purposes of this EE/CA.
- Indicator metals and uranium are sufficiently co-located with radium-226 such that radium-226 concentrations in soil and surrogate measurements can be used to guide removal of mine-related material (ERG, 2013).
- Any transport of mine-related material is primarily limited to soil erosion runoff from Area A to arroyos.
- On-site sources of soil cover materials are adequate for use in a potential on-site repository.
- The project area has geotechnical and geomorphological attributes that are suitable for siting a repository for mine-related material.

Elevated count rates in the project area are associated with mine-related material. Count rates do not increase near rock outcrops in any portion of the project area, indicating that there is no significant radiological mineralization therein.

<sup>&</sup>lt;sup>1</sup> A radium-226 concentration of 3.5 pCi g<sup>-1</sup> would be the cleanup level established for residential land use. It is the background concentration of radium-226 (approximately 1 pCi g<sup>-1</sup>) plus the concentration of radium-226 in soil (2.5 pCi g<sup>-1</sup>) that results in an excess cancer risk of  $1 \times 10^{-4}$ .

 $<sup>^{2}</sup>$  A radium-226 concentration of 5 pCi g<sup>-1</sup> plus background is the UMTRCA standard for the cleanup of radium-226 in surface soil.

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# 2.7 Previous Removal Actions

There have been no previous removal actions, as defined by CERCLA (42 U.S.C. §9601, et seq.) in the project area.

# 2.8 Source, Nature, and Extent of Mine-related material

The nature and extent of mine-related material within the project area are defined in the SIR (ERG, 2013) and EPA investigation reports (EPA, 2010 and 2012), as described in Section 2.5.

This section presents a streamlined human health risk assessment, including a conceptual site model, for the project area. Risk screening is performed for radionuclides and indicator metals present within the project area, using potential future land use scenarios.

# 3.1.1 Conceptual Site Model

A conceptual site model for risks in the project area is shown in Figure 8. The sources of mine-related material and their release mechanisms, exposure routes or pathways, and potential receptors are discussed below.

## 3.1.2 Sources of Mine-related material

The primary sources of mine-related material at the project area are the waste rock pile and two settling ponds used historically for both water storage for recycling and water treatment. These primary sources are located in Area A. Field investigations indicate secondary sources are soils in and around the project area where mine-related material was moved by wind and water erosion, human re-purposing of the materials and spills. The secondary sources are located in Areas A and C. The mine-related material contains naturally occurring radionuclides in the uranium-238 decay series, as discussed in Section 1.0. Long-lived radionuclides in this series include naturally occurring isotopes of uranium (uranium-238, uranium-235, and uranium-234) in naturally occurring isotopic ratios, radium-226, and thorium-230. The indicator metals in the mine-related material are arsenic, barium, lead, molybdenum, selenium, and vanadium.

## 3.1.3 Release Mechanisms

Figure 8 identifies the possible release mechanisms for the primary sources of mine- related materials, including:

- wind erosion of the waste rock pile (re-suspension);
- emanation of radon-222 from the waste rock material into air;
- water erosion (including sheet flow, rill and gully erosion) of the waste rock pile;
- human re-purposing (e.g., use of mine-related material as structural fill in the mine area);
- seepage from the settling ponds; and
- discharges from the settling ponds.

The release mechanisms of mine-related material from soil are re-suspension and emanation to air; and leaching and infiltration into the soil as shown on Figure 8. Infiltration and percolation of mine-related material from soil to groundwater is not a complete pathway because 1) groundwater is intermittent within the alluvium in the project area and 2) the presence and thickness of the Mancos Shale and depth to the nearest aquifer at the project area, as discussed in Section 2.4.2, preclude constituents in the mine-related material from reaching groundwater. The areal extent and thickness of the Mancos Shale is well known by its extensive outcrop, forming the Ambrosia Lake and San Mateo valleys, and the many mine shafts advanced through the Mancos Shale in the area. Because of this, the groundwater exposure pathway is not further discussed or evaluated.

## 3.1.4 Exposure Pathways

Potential exposure pathways at the project area (see Table 1) include exposure to mine-related material in waste rock, soil, and air. These pathways include direct contact, inhalation, ingestion (both directly and directly via uptake by biota), and external radiation exposures. Each pathway is described in more detail below. The drinking water pathway was eliminated, considered unrealistic because of the small watershed, dry climate, and lack of shallow groundwater. A domestic water well would need to be screened very deep, below the highly impermeable Mancos Shale for access to drinking water.

## **Direct Exposure Pathway**

The primary exposure pathway at the project area is direct exposure to waste rock and surface soil containing mine-related material. The predominant exposure route for humans is direct exposure to gamma radiation from gamma-emitting radionuclides; e.g., radium-226 and its short-lived decay products. Other pathways, including incidental soil ingestion and fugitive dust inhalation, are less important to human radiological risk. The direct exposure pathway is important to all potential project area receptors.

#### Air Exposure Pathway

Mine-related material in Areas A and C is susceptible to wind erosion due to their low cohesion and soil moisture content; and the sparse vegetative cover that are characteristic of the region. Additionally, radon-222 can readily emanate from mine-related material into the air. The radon-222 pathway is particularly important when considering exposures of residential and commercial receptors to indoor concentrations of radon-222 and its short-lived decay products. Inhalation of metals, given an appropriate particle size/composition and long-lived radionuclides derived from mine-related material, is important in outdoor exposure scenarios for all receptors.

#### **3.1.5 Potential Receptors**

The project area consists entirely of private land in a sparsely populated area. New Mexico Land, LLC owns most of Area C. Areas A and B are owned by others. The predominant land use in the area is livestock grazing with occasional use of the project area by a rancher to support the livestock. Additionally, recreational use; e.g., big game hunting and hiking; of the land surrounding the project area, is known to occur.

For the purposes of this evaluation, the following human receptors were evaluated based on potential future land uses:

- a resident rancher;
- a worker (indoor and outdoor); and
- a recreational visitor.

# 3.1.6 Streamlined Human Risk Evaluation

A streamlined evaluation of risks to human health was performed, based on 1) applicable EPA guidance (EPA, 1993) and 2) the estimated current extent of mine-related material within the project area and associated concentrations of radionuclides and indicator metals identified in the SIR (ERG, 2013). This section summarizes the methods used to evaluate risks and presents and discusses the results. The purpose of this evaluation is to identify exposure pathways and estimate health risks based on current conditions

and potential future land uses; information that will be used for further evaluation of whether potential removal actions are warranted (EPA, 1993). Risk-based preliminary remediation goals (PRGs) for impacted soils were also developed at the EPA's request.

#### **<u>Risks from Radioactive Constituents (Radionuclides)</u>**

Radiological health risks (expressed as lifetime attributable carcinogenesis) were modeled using RESRAD Version 7.0 (Argonne National Laboratory [ANL], 2014) for three hypothetical receptor scenarios involving conservative exposures to mine-related material within Areas A and C, as identified and delineated in the SIR (ERG, 2013). The three receptor scenarios were: a resident rancher, a worker routinely working on-site in a commercial facility (worker), and a recreational visitor such as a hunter occasionally camping on-site. The land use scenarios used here are conservative and with the exception of the recreational visitor, do not represent the current or expected future land use within the project area. The radionuclides considered were those identified in the SIR as being elevated above background concentrations in Areas A and C: natural uranium, thorium-230, and radium-226. The modeled exposure pathways and parameter selections for each area and receptor scenario, to the extent possible, were based on site-specific data, relevant studies of nearby sites, and/or pertinent regulatory or RESRAD guidance. This information is provided in Tables 1 and 2. RESRAD default parameters were used in cases where applicable information was unavailable. The default exposure duration in RESRAD (30 years) is consistent with EPA guidance (EPA, 1991).

The RESRAD modeling results are provided in Figures 9 (Area A) and 10 (Area C), each of which depict model-predicted excess cancer risks over time by dose, pathway, and radionuclide. The maximum total risk values in Area A were  $3.3 \times 10^{-2}$ ,  $1.4 \times 10^{-2}$ , and  $3.2 \times 10^{-4}$  for the resident rancher, worker, and recreational visitor, respectively. Maximum total risk values in Area C were  $3.3 \times 10^{-3}$ ,  $1.9 \times 10^{-3}$ , and  $3.6 \times 10^{-5}$  for the resident rancher, worker, and recreational visitor scenarios, respectively. Radon-222 is the primary risk pathway for the resident rancher and worker scenarios due to indoor occupancy, followed by external gamma radiation. External gamma radiation is the primary risk pathway for the recreational visitor scenario as there is no exposure to indoor radon or indoor shielding of gamma radiation. Radon risks for a resident rancher scenario in Area A slightly exceed the estimated risk to the general population due to long term exposure to indoor radon concentrations at the EPA's 4 picocuries per liter (pCi L<sup>-1</sup>) action level for radon (Figure 9). Radon risks for other receptor scenarios in Area A, and all scenarios in Area C, are lower than those associated with the 4 pCi L<sup>-1</sup> action level (Figures 9 and 10).

Radium-226 is the dominant source of health risks (due to emanation of radon gas and emission of gamma radiation) in all cases. Thorium-230 begins to contribute slightly to the total risk over time, due to the ingrowth of radium-226 (Figures 9 and 10). Excess risk from all radionuclides and pathways declines rapidly after about 800 years as radioactive contaminants approach complete depletion from surface soils within the modeled area due to soil erosion, leaching and radioactive decay. The maximum total risk for each receptor scenario and Area (A or C) exceeds the EPA's target risk range under CERCLA (10<sup>-6</sup> to 10<sup>-4</sup>), with the exception of the recreational visitor scenario for Area C. Removal actions that would reduce concentrations of radium-226 to acceptable levels are expected to also reduce concentrations of other COPCs to acceptable levels, given that radionuclides and indicator metals associated with mine-related material are co-located with radium-226 (ERG, 2013).

# **Chemical Risks from Indicator Metals and Uranium**

Human health risks associated with uranium and indicator metals were evaluated by comparing measured values (ERG, 2013) to their respective EPA Regional Screening Levels (RSLs) for soils under residential or industrial receptor scenarios (EPA, 2014a), both of which are conservative for the site. The metals evaluated were arsenic, barium, lead, molybdenum, selenium, and vanadium. The RSLs used for these

comparisons are applicable to sites in the southwestern U.S. (EPA Region 6) and are based on carcinogenesis or other health hazards such as chemical toxicity (EPA, 2014a). Respective RSL values were developed using a total cancer risk =  $10^{-6}$  or a Total Hazard Quotient = 0.1 (EPA, 2014a).

The results for each area studied in the SIR (Areas A, B, C and the BRA) are shown as box plots for each indicator metal and uranium against residential and industrial RSLs in Figure 11 (concentrations reported in milligrams per kilogram (mg kg<sup>-1</sup>). The results indicate three general conditions in soil:

- Background concentrations of certain indicator metals exceed RSLs;
- Concentrations of certain indicator metals in the project area are less than RSLs; and
- Concentrations of certain indicator metals in the project area exceed RSLs.

None of the indicator metals are COPCs. The results indicate that arsenic concentrations in the BRA exceed the residential RSL. Vanadium concentrations in 22 samples at 10 of the 15 background locations exceed the residential RSL (Figure 11). EPA policy with respect to background at CERCLA sites is that cleanup levels are not set at concentrations below natural background levels. In addition, the CERCLA program does not remediate sites to concentrations below natural or anthropogenic background levels (EPA, 2002).

Aside from slight exceedances of residential RSLs for molybdenum in one sample, none of the reported values of barium, lead, or molybdenum exceed their respective residential RSLs (Figure 11). Five samples of selenium at three locations, all in Area A, exceed the residential RSL. However, the mean of the selenium concentrations is below the RSL.

A number of samples in Areas A and C exceeded the residential RSL for uranium, while 5 samples in Area A exceed the industrial RSL (Figure 11). The mean concentration of uranium in Area A exceeds the residential RSL. Therefore, it is included as a COPC for the entire project area.

These results indicate that, in general, radium-226 and uranium are the only COPCs with respect to human health risks from mine-related material in the project area. Again, the concentrations of radium-226 dominate current health risks at the project area, and removal of materials to reduce radium-226 concentrations to acceptable levels, is expected to reduce the levels of all other COPCs to acceptable levels.

#### **Preliminary Remediation Goals**

Preliminary Remediation Goals (PRGs) for soils within Areas A and C were calculated for each receptor scenario using the EPA's PRG Calculator for Radionuclides (EPA, 2014b). Consistent with the ratioanale and criteria used to determine PRGs for soil in the EPA's Five-year Plan for remediation of uranium properties within the Grants Mineral Belt (EPA, 2010a; Weston, 2009), a bounding risk value of 3 x10<sup>-4</sup> was adopted for determination of PRGs for soil within the Johnny M Project Area<sup>3</sup>. Results of this modeling (Table 3) indicate that PRGs for the most conservative receptor scenario (resident) and most limiting (restrictive) radionuclide (radium-226) are consistent with the 2.5 pCi/g above background soil cleanup level adopted by the EPA for remediation of uranium properties within the Grants Mineral Belt (Weston, 2009). Hecla plans to use the 2.5 pCi/g of radium-226 above background for areas outside of those addressed by ICs. For areas addressed by ICs, Hecla plans to meet or exceed one of the alternative PRGs in Table 3.

<sup>&</sup>lt;sup>3</sup> EPA guidance indicates that an annual total effective dose equivalent of 12 mrem y<sup>-1</sup> is equivalent to 3 x10<sup>-4</sup> excess cancer risk (EPA, 2014c). This risk level is considered by EPA to be essentially equivalent to a presumptively safe risk level of 1 x10<sup>-4</sup> (EPA 1997, OSWER 9200.4-18).

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# Section 4.0 - Identification of Removal Action Scope, Goals, and Objectives

#### 4.1 Determination of Removal Action Scope and Objectives

The scope of this EE/CA is limited to mine-related material within the project area. The RAOs for the project area are intended to mitigate, reduce, or eliminate the potential for exposure of human receptors to project area COPCs. The RAOs apply to soil containing mine-related material in Areas A, B, and C as identified in the SIR.

The RAOs for the project area are to:

- reduce soil concentrations of COPCs below a level resulting in an excess human cancer risk of 1x10<sup>-4</sup>;
- reduce soil concentrations of COPCs below a Total Hazard Quotient (THQ) of 1; and
- minimize or eliminate the release of mine-related material with unacceptable concentrations of COPCs to surface water, air, and land.

The soil concentrations of COPCs that meet the RAOs are based on probable future land use. Land use in Areas A, B, and portions of C not owned by New Mexico Land, LLC is assumed to be resident rancher. A combination of industrial/commercial and livestock grazing is assumed with institutional controls (ICs) to be established for the land in Area C owned by New Mexico Land, LLC (the western half of Section 18 shown in Figure 3).

Appendix B addresses federal and state ARARs germane to the removal action. Removal action alternatives are evaluated for compliance with ARARs in Section 6.0.

#### 4.2 Determination of Removal Schedule

It is anticipated that, upon approval of the EE/CA, planning and implementation of a removal action in the project area could take two years or more to complete, depending on weather and other factors.

# **5.1 Identification and Evaluation of Removal Technologies**

Removal action alternatives use one or more technologies, which can be grouped into several categories:

- <u>Access Controls and ICs</u>: Access controls include measures that prevent or reduce receptor exposure by limiting access or use of impacted areas. ICs are non-engineered instruments such as government and/or proprietary controls that reduce the potential for human exposure to contamination by limiting land or resource use.
- <u>Engineering Controls</u>: Measures such as caps and drainage controls implemented to mitigate contaminant mobility and the potential for receptor exposure.
- <u>Excavation and On-Site Disposal</u>: Removal of contaminated material by conventional means and disposal in an on-site repository.
- <u>Excavation and Off-Site Disposal</u>: Removal of contaminated material by conventional means and off-site disposal.
- <u>Treatment</u>: Contamination mitigation by treatment to destroy, immobilize, or extract COPCs.

Each technology is preliminarily screened in this section to determine if it should be retained for further evaluation.

#### 5.1.1 Access and Institutional Controls

Access controls and ICs include a combination of physical and legal measures to preclude 1) trespass on the project area or 2) use of the project area for activities that could result in unintended non-radiological and radiological exposures on or off the project area. Access controls are the legal and physical barriers to unauthorized entry to the project area that include signage, fences, and locked gates. ICs are legal devices that make unlawful any use of the project area that is deemed incompatible with the radiological and chemical characteristics of the project area and involve the perpetual custody or oversight by an entity that can maintain such control. ICs, according to EPA, 2000a, are:

- "non-engineered instruments such as administrative and/or legal controls that minimize the potential for human exposure to contamination by limiting land or resource use; [and]
- generally to be used in conjunction with, rather than in lieu of, engineering measures such as waste treatment or containment; can be used during all stages of the cleanup process to accomplish various cleanup-related objectives; should be "layered" (i.e., use multiple ICs) or implemented in a series to provide overlapping assurances of protection from contamination".

ICs can include government controls (e.g.; zoning restrictions, ordinances) and/or proprietary controls (e.g.; restrictive covenant, easement).

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#### 5.1.2 On-site Disposal

The technologies to consolidate and dispose of mine-related material within the project area, and specifically within the land owned by New Mexico Land, LLC, include standard excavation techniques and equipment readily available to the mining and construction industries, as well as technologies for consolidation and disposal of wastes and radiation control that have been used for many years under the UMTRCA, Resource Conservation and Recovery Act, and CERCLA programs. Standard earthworking equipment such as dozers, scrapers, excavators, and loaders can excavate the mine-related material identified with the radiological survey methods and technologies used in the site investigations described in Section 2.5. The excavated material can be transported by scraper or truck to the on-site repository location for final placement. The same equipment can be used for earthwork to construct the repository. Sources of soil for a repository liner and cover materials are available in the project area, as described in the SIR (ERG, 2013). Preliminary engineering analyses indicate that the available soil and rock would have the physical properties needed to construct the low permeability and erosion-resistant elements of the repository. A conceptualized cross section of a potential repository is shown in Figure 15.

On-site disposal of mine-related material can meet both the project RAOs and ARARs and has extensive precedent, having been used successfully at a large number of sites including the nearby San Mateo Mine approximately one mile from the project area, Northeast Churchrock Mine near Gallup, NM; and many other hard-rock mine sites in New Mexico and elsewhere.

## 5.1.3 Off-site Disposal

The technologies to consolidate the mine-related material for off-site disposal include the same standard excavation techniques and equipment that would be used for on-site disposal. Standard earthworking equipment such as dozers, scrapers, and loaders can excavate the mine-related material identified with the radiological survey methods and technologies used in the site investigations described in Section 2.5. The excavated mine-related material would be loaded onto trucks with top covers for transport either directly to a licensed disposal facility or to the BNSF rail siding in Milan, NM. The mine-related material would be loaded into lined rail gondola cars and fully enclosed, if it is shipped by rail. Rail transportation, where mentioned below, assumes trucking to the BNSF rail siding and then by rail to the disposal facility.

The mine-related material could be shipped to the Energy Solutions waste disposal facility in Clive, Utah or the Waste Control Specialists waste disposal facility in Andrews County, Texas. Both of these facilities have the equipment, technical capabilities, capacities, and permits to accept the mine-related material. The road and rail transportation routes from the project area to these facilities are shown in Figure 12.

Off-site disposal has been used successfully elsewhere, resulting in both long-term and short-term effectiveness and elimination of any exposure pathways to constituents in mine-related material exceeding cleanup criteria, once disposal has occurred at the receiving facility. This technology can meet the project RAOs and ARARs. However, this technology is typically used for smaller volumes than those estimated for this removal action.

# 5.1.4 Treatment Methods

Several methods developed for the removal of inorganic elements or compounds (i.e., metals and radionuclides that include the COPCs) from soil were evaluated. Each would involve the consumption of large amounts of water, introduction of chemicals or cementitious materials to the project area, and manipulation of mine-related material.

#### Soil Washing

Soil washing is a process that uses physical and/or chemical techniques to separate metals from soil. Constituents are concentrated into a much smaller volume of residue, which is either recycled or disposed. Wash water can consist solely of water or can include additives such as acids, bases, surfactants, solvents, chelating or sequestering agents which are utilized to enhance the separation of constituents from soils (Interstate Technology & Regulatory Council [ITRC], 1997). Soil washing has not been used on soils with concentrations of constituents as low as those at the project area, even though it has been successful in treating soils with higher concentrations of uranium (ITRC, 2010).

Soil washing would use the tendency of the COPCs to concentrate in silt and clay, if applied at the project area or sites with similar soils (Misra et al, 2001; Federal Remediation Technologies Roundtable [FRTR], 2014). The soil washing process would separate the fine soil (silt and clay), which contains the majority of the COPCs, from the coarse soil (sand and gravel). The smaller volume of fine soil can then be disposed of either on-site or off-site, leaving the clean coarser soil to be used as cover or backfill on-site provided it meets the cleanup criteria.

Soil washing has not been used widely for removing metals from soil. There have been only six recorded applications of soil washing in the U.S. through 2011 (EPA, 2013). The reasons for this include the need for large volumes of water, introduction of chemicals into the wash water, need for disposal or treatment of the radium-enriched wash water (a newly created waste stream), need for specialized washing equipment, difficulty of containment of water during and after washing, and increased risk of worker exposure.

Given this discussion, the RAOs and ARARs could be met with this technology, but for the reasons just noted it is not practical.

Ex Situ Source Control – Solidification/ Stabilization. Solidification typically refers to processes that encapsulate a liquid waste, or one with both solid and liquid phases, to form a solid material that restricts the migration of soluble COPCs by reducing the surface area exposed to leaching or by coating the waste with low-permeability materials. Solidification requires the addition of a solidifying agent that causes a chemical reaction within the mine-related material. Solidifying inorganic binders include cement, fly ash, lime, soluble silicates, and sulfur-based binders. The addition of the solidifying agent would substantially increase the volume of the mine-related material. Solidification is often used together with stabilization, which is defined below.

Because the mine-related material is mostly dry, solidification would require substantial quantities of water derived from a source(s) on-site or near the project area for mixing the solidifying agent with the mine-related. The water requirement, mixing equipment and need for a solidifying agent make the solidification technology a poor option in terms of both implementability and cost.

Solidification would substantially reduce the mobility of the COPCs in the mine-related material but would not eliminate or significantly reduce the magnitude of the direct radiation pathway to human receptors. Solidification as a single treatment option would not meet the project RAOs.

Stabilization, in the context of an EE/CA, refers to any process that uses chemical reactions to reduce the leachability of COPCs in the mine-related material by immobilizing them using chemical reactions (EPA, 2000b). Stabilization has been used on 217 sites through 2007 (EPA, 2007). Stabilization is often combined with solidification when the same chemical treatment can accomplish both.

Stabilization of COPCs in the mine-related material is related to solubilization of COPCs by rain water, either as runoff or infiltration. Only barium significantly increases dissolution of radium from soil, with somewhat lesser increases in radium solubility from ammonium acetate, ammonium nitrate, and chloride (Markose et al, 1985). Rain water lacks these constituents and ineffectively solubilizes radium (Shearer and Lee, 1964). The primary uranium minerals mined at the Johnny M Mine; coffinite and uraninite are

insoluble in water (Z´avodsk´a et al, 2008). Uranium as a salt or carbonate is soluble in water (Barthelmy, 2014). Therefore, the benefit of using stabilization for the mine-related material would be only to minimize the liquid-to-solid ratio of the materials, essentially reducing the rate of infiltration therein.

Stabilization alone would have poor long-term effectiveness because it would not substantially reduce the mobility of COPCs in the mine-related material. The mine-related material is dispersed: it would have to be concentrated into a smaller area for stabilization treatment and utilize special equipment, rendering poor both the implementability and cost for this method. Stabilization would take at least two construction seasons plus planning time and not substantially affect radium leaching in that time frame, making it a poor short-term measure. Stabilization reduces the liquid-to-solid ratio only slightly in either the long or short term, making it only fair in reducing radium mobility.

There are no unique benefits to this method compared to the others being considered. Stabilization would not eliminate or significantly reduce the magnitude of the direct radiation pathway to human receptors and therefore would not meet project RAOs.

No other chemical treatment alternative has been included in the screening of technologies because research has shown that soil-cleaning technologies using a combination of chloride washing and flotation, washing with distilled water and humic acid, and other technologies are still in development, require large quantities of water, produce a substantial chemical and radiological waste stream, and are prohibitively expensive.

# **5.2** Technology Screening

The treatment technologies –soil washing and ex situ solidification and stabilization– have complicating factors that eliminate them from further evaluation as removal alternatives. These factors include the 1) increased volume of mine-related material subject to removal action, associated with solidification and stabilization compounds, 2) creation of multiple waste streams (soil washing), 3) inability to meet the RAOs without an engineered cover (all three), 4) use of substantial quantities of water (all three), 5) need for the consolidation and multiple handling of mine-related material associated with treatment options and increased potential for worker exposures (all three), and 6) the technology alone would not meet the RAOs (all three).

The dispersal of mine-related material in the project area makes it necessary to bring the materials to one location for the effective application of any of these three treatments. All three would require substantial quantities of water (a limited resource in the area), either to wash the soil or distribute a chemical through the material. Solidification and stabilization both require the addition of treatment compounds, which increases the final volume of the mine-related material. Solidification alone would reduce the mobility of constituents by encapsulating them in a solid matrix. Stabilization alone would reduce their solubility by chemical means. However, in each case the COPCs would remain in the treated mine-related material, which would still require an engineered cover to eliminate or significantly reduce the magnitude of the direct radiation pathway to human receptors to meet the RAOs. The waste stream from soil washing would have to be removed from the project area or stabilized/solidified and covered on-site to meet the RAOs.

Access controls and ICs are insufficient alone to satisfy the RAOs and therefore are not retained as a standalone alternative. Technologies that utilize earthwork methods are well-established, flexible, and capable of achieving the RAOs. Therefore, on-site and off-site disposal are retained for the evaluation of removal alternatives.

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# 5.3 Identification of Removal Action Alternatives

# 5.3.1 No action

No action would be taken in this alternative. The mine-related material would remain where it is, without additional measures to stabilize or isolate it. It would remain as accessible as it presently is to wildlife, livestock, and humans, restricted by existing fences, signage and gates. The mine-related material would continue to be exposed to erosion by wind and water. The no-action alternative provides the baseline for comparison with the other removal action alternatives.

# 5.3.2 On-site Disposal with Access and Institutional Controls

In this removal alternative, all mine-related material in which the concentration of radium-226 exceeds a provisional clean-up criterion of 3.5 pCi g<sup>-1</sup> (the extent of which is adopted from the SIR and depicted as Figure 13), would be excavated, hauled, consolidated, and isolated in a project area repository within land owned by New Mexico Land, LLC.

The potential locations of the repository are illustrated on Figure 14. They were chosen for the following reasons:

- The Mancos Shale would serve as an effective, natural low permeability liner.
- The shallow alluvial or eolian soils that blanket the surface of most of Area C are underlain by the low permeability Mancos Shale or a sandstone interbed of the Mancos Formation.
- There are on-site sources of materials that could be used to construct a repository cover and low permeability liner, if needed.
- The repository can be isolated from arroyos.
- There is no groundwater within several hundred feet of the ground surface.

The repository would be sited above the 100-year floodplain; in an area that is readily accessible to construction equipment and underlain by either bedrock or low-permeability soil (natural or emplaced).

A conceptualized cross section of the on-site repository is shown in Figure 15. The cross section depicts the mine-related material enveloped by a liner (natural or emplaced) and cover, as described below.

#### Excavation of Basin

Excavation of a basin into the existing soil or rock would produce the material to be used to construct the repository cover. Note that the Mancos Shale would be left in place to act as a natural liner where it underlies the excavated repository basin. Other eolian, alluvial, and residual soils within the project area also would be used to construct the repository cover and low permeability liner, as needed.

#### <u>Liner</u>

Observations from test borings advanced during the investigations reported in the SIR (ERG, 2013) confirm that the Mancos Shale serves as an effective, natural liner under the area of the historic settling ponds and mine-related material that have been in place since the 1970s. No additional liner would be needed to isolate the mine-related material placed therein, if the repository location is also underlain by shale. A compacted clay soil liner would be placed across the footprint of the repository to preclude the migration of any

moisture that might be capable of draining from the mine-related material, if the selected repository location is not directly underlain by shale.

The Mancos Shale has been used successfully as both an in-place natural liner; and a source for constructed earthen liners and covers at other uranium mine and mill sites. These include the heap leach tanks and raffinate ponds at Hecla's Durita Project in Colorado, the L-bar Uranium Operations tailings impoundment in New Mexico, and the Department of Energy Crescent Junction disposal site for the uranium tailings relocated from the Atlas Moab mill site in Utah.

#### Placement of the Mine-related Material

The mine-related material in the project area exceeding cleanup criteria would be excavated and placed in the repository. The materials having the highest concentrations of radium-226 would be placed first, in the deepest level of the repository. These would be followed by materials with increasingly lower concentrations of radium-226 with the shallowest level of the repository containing materials with the lowest concentrations of radium-226. This layered segregation of mine-related material would support the attenuation of radon within the mine-related material and thereby minimize the potential for radon to migrate upward through the repository cover.

The mine-related material would be placed dry, with moisture applied only to the extent needed to control dust. The mine-related material would be placed over any liner as soon as possible after liner construction.

# <u>Cover</u>

An ET cover design would be developed if the on-site disposal alternative is selected. The mine-related material would be capped by an evapotranspiration (ET) cover consisting of on-site soils that include weathered Mancos Shale. The ET cover would serve four functions: 1) physically contain the mine-related material, 2) minimize the potential for infiltration of water into the mine-related material, 3) limit radon flux at the surface of the cover to the ARAR, and 4) provide a growth medium for vegetation.

Conventional covers typically consist of a single monolithic layer of soil that is thick enough to serve the barrier functions (functions 1-3 above) but not the fourth function (growth medium). An ET cover is an alternative to conventional cap and cover systems:

"ET cover systems are designed to rely on the ability of a soil layer to store the precipitation until it is naturally evaporated or is transpired by the vegetative cover. In this respect they differ from more conventional cover designs in that they rely on obtaining an appropriate water storage capacity in the soil rather than...engineered low hydraulic conductivity [barrier components]. ET cover system designs are based on using the hydrological processes (water balance components) at a site, which include the water storage capacity of the soil, precipitation, surface runoff, evapotranspiration, and infiltration. The greater the storage capacity and evapotranspirative properties are, the lower the potential for percolation through the cover system." (EPA, 2013)

An ET cover would have two or more layers. The top layer would be appropriately (relatively loose) compacted soil (sandy and silty sand) thick enough to store sufficient moisture to support vegetation. The bottom layer would be a clay-rich soil that would act as a hydraulic barrier against infiltration to the underlying mine-related material. An intermediate layer, consisting of a filter layer of graded sand over a capillary break layer of free-draining gravel, would be included if needed for the hydrologic (water balance) functions of the cover.

The cover layer thicknesses would be determined by both radon flux and infiltration modeling using the RADON model; and HYDRUS<sup>®</sup> software or an equivalent program, respectively. It is expected that the cover thickness would be driven by infiltration, and the thickness modeled using HYDRUS would be

sufficient to attenuate radon to an acceptable level. The RADON model would be used during the cover design phase to demonstrate this.

A conceptual design was assumed for the purposes of this EE/CA to consist of 1.0 ft of shale clay compacted to not less than 95 percent maximum dry density (MDD) with an optimum moisture content (in accordance with American Society of Testing and Materials (ASTM) Standard D-698 [ASTM, 2012]) covered by 3.0 feet of silty sand (SM, in accordance with the United Soil Classification System) compacted to 90 percent MDD. The specifications for thicknesses and compaction described here are consistent with those of radon covers used at UMTRCA sites to attenuate radon to an acceptable level.

ITASCA modeled this conceptual design of the ET cover to evaluate the potential for infiltration of water into the mine-related material (ITASCA, 2014) using HYDRUS<sup>®</sup> and relevant geotechnical and hydraulic properties of the project area soils listed in Table 4. Conservative values for saturated hydraulic conductivity listed in that table;  $1.10 \times 10^{-3}$  and  $1.3 \times 10^{-4}$  cm s<sup>-1</sup> for the sand and clay layers, respectively, were used in the model. These are two orders of magnitude higher that the respective values in Table 4 for the same soil compacted to 95 percent MMD at up to four percent above optimum moisture. Using the more conservative properties, the model predicts that water would not infiltrate into the mine-related material but instead residual moisture would migrate *upward*, *out* of the mine-related material into the cover. The flux of moisture reaches a maximum in approximately 1,000 days. The results of the model indicate that this cover would be protective of groundwater in the project area.

The specifics of the cover design would be determined during the planning phase of the removal action.

# **Erosion Protection**

Erosion protection would be achieved primarily by diverting runoff to the approximate original (premining) drainage courses and vegetating the repository cover. Rock mulch also may be incorporated into the top soil lift of the cover to enhance seed nesting and erosion resistance. Larger riprap would be applied to drainage courses that are adjacent to the repository. Sandstone of the Mancos Formation interbeds was tested in 2012 and found to be durable enough for use as riprap, if needed for erosion control.

Disturbed ground outside of the repository footprint would be graded to a stable, erosion-resistant surface and then re-vegetated at the same time the repository cover is vegetated. Post-removal site controls (PRSCs) would include fencing and signs around the repository to control entry. The repository would be inspected annually for at least 12 years after the last year of augmented seeding, during which the establishment of vegetation would be evaluated. The standards for the establishment of vegetation are established in the ARARs.

This alternative also would include access controls and ICs. The repository, located on land owned by New Mexico Land, LLC, would be surrounded by reinforced fencing to preclude grazing animals. A restrictive covenant would be put in place to limit and control future land use on property owned by New Mexico Land, LLC.

# 5.3.3 Off-site Disposal

In this alternative, the mine-related material exceeding cleanup criteria would be eliminated from the project area; i.e., excavated and removed from the site for off-site disposal at a facility permitted to receive and dispose of naturally occurring radioactive material (NORM) or technologically-enhanced, naturally occurring radioactive material (TENORM). The off-site disposal alternative would require the identified mine-related material to be excavated, loaded into trucks, and either transported directly to a licensed off-site disposal facility or hauled to Milan, NM for loading onto rail cars for transport by rail to a disposal

facility. There are no permitted facilities in New Mexico: the mine-related material would be transported to either the Energy Solutions or WCS facility.

Disturbed ground would be restored to approximate pre-mining topography and re-vegetated after the minerelated material has been removed from the project area. PRSCs would be minimal, consisting of 12 years of annual inspection and evaluating the establishment of vegetation after the last year of augmented seeding. The standards for the establishment of vegetation are established in the ARARs.

#### **5.4 Evaluation of Removal Action Alternatives**

This section presents an evaluation based on EPA EE/CA guidance (EPA, 1993) for each alternative identified in Section 5.3. Each alternative was evaluated using three general criteria: 1) effectiveness, 2) implementability, and 3) cost, including their subcomponents.

#### 5.4.1 Effectiveness

The effectiveness of an alternative refers to its ability to meet the RAOs within the scope of the removal action, including the final disposition of the mine-related material and soil cleanup level, if any.

#### **Overall Protection of Public Health and the Environment**

Each alternative was evaluated as to how well it would protect public health and the environment. This evaluation drew on assessments of long-term effectiveness and permanence, short term effectiveness, compliance with ARARs, and whether the alternative would meet the RAOs.

#### No Action Alternative

This alternative would not meet the RAOs. Sources of the COPCs and other metals and radionuclides would remain in the project area in their existing configuration. The human health risks associated with the mine-related material would remain unchanged.

#### **On-site** Disposal

This alternative as described in Section 5.3 would meet the RAOs. Mine-related material exceeding cleanup criteria would be consolidated into a stable, permanent configuration and capped with an ET cover that minimizes the infiltration of water and radon flux. The excavated areas would be graded and re-vegetated to provide a stable soil surface. Access to and land use around the repository would be controlled by fencing and ICs, respectively. This alternative would minimize the potential for 1) direct exposure of human receptors to COPCs in mine-related material) and 2) the release of mine-related material to air, water and land.

#### Off-site Disposal

This alternative as described in Section 5.3 would meet the RAOs. Excavation of mine-related material exceeding cleanup criteria would be removed from the project area and disposed of at an appropriately permitted facility. The excavated areas would be graded and re-vegetated to provide a stable soil surface. This alternative would eliminate 1) the potential for direct exposure of human receptors to COPCs in soil at levels exceeding the cleanup criteria and 2) the release of mine-related material to air, water, and land.

#### **Compliance with ARARs**

Potential ARARs for the project area are detailed in Appendix B.

#### No Action Alternative

This alternative would not trigger or satisfy any ARARs.

#### **On-site** Disposal

Implementation of this alternative would satisfy all ARARs.

#### **Off-site Disposal**

Implementation of this alternative would satisfy all ARARs.

#### Long Term Effectiveness and Permanence

This evaluation pertains to the extent and effectiveness of the removal action in achieving the durability and permanence RAOs, including controls that may be required to manage the mine-related material remaining at the project area at the conclusion of a removal action.

#### No Action Alternative

This alternative would have no long term controls regarding mine-related material at the project area.

#### **On-site** Disposal

This alternative would isolate the mine-related material exceeding cleanup criteria in a repository within the project area. The long-term effectiveness of the repository would depend on the design, construction, and inspection/maintenance of the cover following the removal action; however, substantial precedent exists from which to design and construct the repository's long-term effectiveness and permanence. Access controls, in the form of a fence and signs around the repository, would be installed to preclude entry by deer, elk, grazing animals, and humans; and establish and maintain the vegetative cover. Inspection and maintenance of the fence and signs would be required.

PRSCs would be periodic inspections of the repository cover and access controls, with follow-up maintenance as needed.

Disturbed ground outside of the repository footprint would be graded and then re-vegetated after the minerelated material has been removed. Existing fencing would be sufficient to limit access by livestock while vegetation is being re-established. A minimum of 12 years of annual inspection after the last year of augmented seeding would be conducted to evaluate the establishment of vegetation and erosion controls.

ICs in the form of a restrictive covenant would be in place for the lands in the project area owned by New Mexico Land, LLC. The mechanism and authority for enforcing these controls would be defined in the covenant.

#### Off-site Disposal

Disturbed ground would be restored to approximate pre-mining contours and re-vegetated, after the minerelated material has been removed from the project area. PRSCs would consist of up to 12 years of annual inspections after the last year of augmented seeding and evaluations of the re-establishment of vegetation. Existing fencing would be sufficient to limit access by livestock while vegetation is being re-established.

#### **Short-Term Effectiveness**

This section addresses the impacts of each alternative during implementation before the RAOs have been met. The alternatives are evaluated with respect to their potential effects on human health and the environment during and immediately after implementation.

#### No Action Alternative

The no-action alternative would produce no public health, worker or environmental concerns during implementation.

#### **On-site** Disposal

This alternative requires disturbance and movement of mine-related material within and limited to the project area, with consolidation in the repository. The mine-related material would be only briefly exposed to release by natural forces and construction activities at this time. This alternative would minimize the handling steps to move the mine-related material from the existing locations to the repository, depending on the equipment used. Thus, worker exposures would be minimized. These steps and the related hazards are:

- excavate with scraper, haul and place in the repository, resulting in worker inhalation of, and skin contact with, dust;
- excavate with loader or excavator, load into trucks, resulting in worker inhalation of, and skin contact with, dust; and
- spread and compact mine-related material in the repository, resulting in worker inhalation of, and skin contact with, dust.

A health and safety plan (HASP), incorporating controls for occupation exposures to workplace hazards, including radiation, would be in effect during implementation of the removal action.

On-site disposal in a repository has only local limited potential for release of mine-related material and minimal risk of exposure to public receptors. Environmental and public impacts from the excavation of mine-related material at the project area and consolidation into the repository can be effectively managed with engineering controls such as dust suppression and barriers to sediment transport in case of rain events.

#### Off-site Disposal

This alternative requires disturbance and movement of mine-related material within the project area. The mine-related material would potentially be exposed to release by natural forces and multiple handling during excavation, loading, hauling, transfer and disposal.

A HASP incorporating controls for occupational exposures to workplace hazards, including radiation, would be in effect during implementation of the removal action.

Environmental and public impacts from the excavation of mine-related material at the project area can be effectively managed with engineering controls such as dust suppression and barriers to sediment transport in case of rain events.

Implementation of off-site disposal would require multiple handling of the mine-related material and transport over long distances, during which there would be the potential for worker and public exposure and accidental spillage of materials along transportation corridors. These include:

- Excavation of mine-related material from existing locations resulting in worker inhalation of, and skin contact with, dust.
- Loading of mine-related material, either at the point of excavation or at a load-out location resulting in worker inhalation of, and skin contact with, dust.
- Transport by truck either to a rail siding or directly to a disposal facility with spillage due to accident or to incomplete enclosure in the truck resulting in soil, water, and vegetation contamination along the transport route; worker and public inhalation of, and skin contact with, dust.

- If transported by rail, offloading and transfer of mine-related material to gondola cars at the rail siding, resulting in worker and public inhalation of, and skin contact with, dust; soil, water, and vegetation contamination in the vicinity of the siding.
- If transported by rail, derailment and bulk spills or leakage of mine-related material resulting in worker and public inhalation of, and skin contact with, dust; soil, water, and vegetation contamination in the vicinity of the derailment or leakage.
- Off-loading and transfer to the disposal location at the disposal facility, resulting in worker inhalation of, and skin contact with, dust.

The transportation of the mine-related material over long distances on public highways also adds the potential risk of injury to workers (the drivers) and the public due to traffic accidents.

Off-site disposal also would consume large quantities of fossil fuel, resulting in large carbon emissions.

## 5.4.2 Implementability

The section evaluates the removal action alternatives based on the technical and administrative feasibility of their implementation.

## **Technical Feasibility**

The evaluation of technical feasibility assesses the reasonableness of putting the alternative in place; i.e., whether the methods and equipment are proven and appropriate for application.

## No Action Alternative

The no action alternative would require no implementation, thus its technical feasibility is not applicable.

#### **On-site** Disposal

The technology (construction equipment) used to implement this alternative is common and readily available. There are few technical difficulties in implementing this alternative. Construction materials needed to implement this alternative, such as riprap and cover material for the repository, are readily available at the project area.

Monitoring the effectiveness of this alternative is technically feasible by using soil sampling to confirm soil cleanup levels have been met and performing periodic inspections of the site to evaluate erosion and vegetation in excavated areas, as discussed in Section 5.3.

This alternative has been used at many sites; e.g., the San Mateo and Northeast Churchrock mines, and the Ambrosia Lake (Rio Algom), Phillips, and Homestake mills; and proven to be reliable. The effectiveness of the Mancos Shale as both a natural in-place liner and source for very effective liner and cover material has been demonstrated on a number of similar projects, as discussed above.

# Off-site Disposal

The technologies involved to implement this alternative; e.g., construction equipment, highway legal haul trucks or rail cars, are common and readily available. There are few technical difficulties in implementing this alternative.

Monitoring the effectiveness of this alternative is technically feasible by using soil sampling to confirm soil cleanup levels have been met and performing periodic inspections of the project area to evaluate erosion and vegetation in excavated areas, as discussed in Section 5.3.

This alternative has been used at many sites and proven to be reliable for volumes typically smaller than those estimated for this removal action.

### Administrative Feasibility

The evaluation of administrative feasibility assesses the activities needed as part of the coordination with regulatory offices and agencies, other than the EPA. Permits and waivers, including ICs and the availability of services and materials; and support agencies in the State of New Mexico are evaluated for each alternative.

## No Action Alternative

This alternative would involve no activities requiring permits or coordination with agencies. The availability of services and materials is not applicable. The acceptance of this alternative by the State of New Mexico is unlikely.

# **On-site Disposal**

This alternative would require excavation in a primary arroyo. This activity could require consultation with the U.S. Army Corp of Engineers (USACE). Otherwise, the project area is entirely on private land; therefore, consultation with federal agencies other than the EPA would be unlikely. Community members and private land owners near the project area might have concerns regarding this alternative. These concerns could be communicated in the formal public comment process and would be addressed prior to the selection of the removal action.

The ICs proposed in this alternative to control future land use around the repository would be in the form of a restrictive covenant that is enforceable in New Mexico.

New Mexico state agencies would likely prefer this alternative since it would reduce potential radiological exposures more than the other alternatives and is technically similar to reclamation practices of the New Mexico's Abandoned Mine Land Program.

# **Off-site Disposal**

This alternative would require excavation in a primary arroyo. This activity could require consultation with the USACE. Otherwise, the project area is entirely on private land; therefore, consultation with federal agencies other than the EPA would be unlikely. Community members and private land owners near the project area and along the transportation routes might have concerns regarding this alternative. These concerns could be communicated in the formal public comment process and would be addressed prior to the selection of the removal action.

Both of the disposal facilities named in this alternative have the required licenses, permits, and capacities to dispose of the mine-related material from the project area. However, legal and contractual arrangements would have to be negotiated with the carriers and the receiving disposal facility, and it would be necessary to meet the regulatory requirements for transport of the mine-related material. Uncertainty associated with scheduling off-site shipments of mine-related material to the disposal facility and the ability of the disposal facility to receive mine-related material within the necessary time-frame proposed in Section 4.2 could be problematic.

#### Engineering Evaluation/Cost Analysis for the Johnny M Mine and Adjacent Properties Hecla Limited/New Mexico Land, LLC

## 5.4.3 Cost

Each removal action alternative was evaluated to determine its projected costs. Capital and PRSC costs are presented in Table 5. The costs were estimated using volume estimates, vendor quotes, available literature and other sources deemed appropriate. Detailed cost estimates are provided in Appendix C.

#### No Action Alternative

There are no capital or PRSC costs associated with this alternative.

#### **On-site** Disposal

The capital cost for this remedial action alternative is estimated at \$5.6 million (M). The annual PRSC cost for this alternative is estimated to be \$56,000 for inspection and maintenance.

#### **Off-site** Disposal

The cost estimate for this alternative is dependent on the disposal site and the transportation mode for the mine-related material to the disposal site. The capital costs are as follows:

- Disposal at Energy Solutions and transporting by truck and rail is \$85.6M
- Disposal at Energy Solutions and transporting by truck is \$191M
- Disposal at WCS and transporting by truck and rail is \$151M
- Disposal at WCS and transporting by truck is \$221M
- The PRSC cost for this alternative is minimal compared to the capital cost and should be no more than \$21,000 per year for 12 years.

# Section 6.0 - Comparative Analysis of the Removal Action Alternatives

Several removal action alternatives were identified and evaluated for potential selection for the Johnny M mine-related material. The removal action alternatives are compared in this section on the basis of effectiveness, implementability and cost. Effectiveness and implementability are rated qualitatively and costs quantitatively. For the purposes of the qualitative comparison, the rating scale is defined as:

- Poor: Unable to adequately address the RAOs and ARARs
- Fair: Able to adequately address some of the ARARs, RAOs, and other selection criteria
- Good: Able to adequately address all of the ARARs, RAOs, and selection criteria

Costs also were evaluated with these criteria with the basis of the rating being the relative magnitude of the cost: good as low, fair as medium, and poor as high cost, respectively.

The relative rating of and comparison between alternatives is evaluated below and summarized in Table 6.

# 6.1 No Action Alternative

This alternative represents the baseline conditions for the project area and would provide no mitigation for the protection of human health and the environment. Risks associated with radionuclides and indicator metals to human receptors would remain unchanged. The cost is estimated to be low. The effectiveness is ranked as poor, implementability as good since nothing would be done, and costs as good.

# 6.2 On-site Disposal

This alternative would consolidate all mine-related material with concentrations of COPCs exceeding soil cleanup criteria into a repository located on land in the project area owned by New Mexico Land, LLC. The estimated volume of excavated mine-related material is 500,000 yd<sup>3</sup>. This alternative would achieve the project RAOs and ARARs, while minimizing materials handling and potential exposure to workers and the public. Therefore, both short-term and long-term (permanence) effectiveness of this alternative is ranked as good. Access controls and ICs would be used to limit access to and land use near the repository. No technical or administrative issues were identified with this alternative; thus, implementability is ranked as good.

The estimated cost for this alternative is \$5.6M (about \$11.28/yd<sup>3</sup>) which includes \$1.85M to construct a repository. This cost is more than offset by the savings gained from not incurring the off-site transport and disposal costs in the Off-site Disposal alternative, which would be prohibitively high. The cost for this option was ranked as good.

# 6.3 Off-site Disposal

This alternative would remove all mine-related material containing COPCs with concentrations exceeding soil cleanup criteria from the project area and dispose of it off-site. The estimated volume of excavated mine-related material is 500,000 yd<sup>3</sup>. Risks from accidents involving transportation of the materials to the disposal facility are high and the consumption of fossil fuels due to this transportation is very large, even though this alternative would achieve the project RAOs. Additionally, mine-related material would have to be handled at multiple points, increasing the potential for environmental releases and worker exposure. For these reasons the long-term effectiveness is ranked as good but the short-term effectiveness is ranked as fair.

There are significant issues regarding administrative feasibility: there would be substantial effort required to plan off-site removal. Planning the logistics and safety measures for long-distance transport of large volumes of material over at least two years would be complex. In addition, contracts with haulers and the disposal facility would be needed and the regulatory requirements for transportation would have to be met; thus complicating administrative feasibility. The logistical challenges and administrative complexity negatively impact the administrative feasibility of off-site disposal, so the overall implementability is ranked as poor.

The cost associated with this alternative are prohibitive at 85.6M ( $172/yd^3$ ) to 221M ( $442/yd^3$ ), depending on the disposal facility and mode of transportation. The costs for this alternative are ranked as poor, because they are prohibitive.

The recommended removal action for the project area is "On-site Disposal." This removal action meets all of the RAOs and ARARs, is the most cost effective, and has been used extensively at other CERCLA and mining sites in the Grants Mineral Belt in NM. Potential exposures to workers and the public to mine-related material could be effectively mitigated through use of common engineering and administrative controls. Potential environmental and safety impacts associated with off-site transportation of mine-related material would be eliminated. Access controls associated with the repository would be implemented and maintained. An enforceable, restrictive covenant would be recorded to control future use of the land owned by New Mexico Land, LLC, which is where the repository would be located. Land use within the project area would not otherwise be restricted.

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Tables

Pathway <sup>a</sup>	Resident Rancher	Worker Scenario	Recreational Visitor Scenario
External Gamma	•	•	•
Inhalation	٠	•	•
Plant Ingestion	•		
Meat Ingestion	•		
Milk Ingestion			
Aquatic Foods			
Soil Ingestion	•	•	•
Radon	•	•	•

# Table 1. Potential exposure pathways used for RESRAD modeling.

Notes:

<sup>a</sup>Drinking water pathway not realistic due to small watershed, dry climate, and lack of shallow groundwater (domestic water well would likely need to be screened very deep, below highly impermeable Mancos Shale).

Table 2. Site-specific RESRAD parameters used to model risk for three difference receptor scenarios within Areas A and C.

Application	Model Parameter	Value	Rationale/Comments	Source/Reference	
Occupancy (Resident	Fraction on-site indoor occupancy	0.5	Assumes resident rancher scenario, occupancy similar to resident farmer but unfavorable climate/conditions for farming.		
Rancher Scenario)	Fraction on-site outdoor 0.25 H		Assumes resident rancher scenario, about 42 hours per week working outdoors on-site (similar to resident farmer).		
Occupancy (Worker	Fraction on-site indoor occupancy	0.17	Assumes industrial worker, 6 hours per day indoors, 250 days per year (1,500 hour s per year)	Table 2.3, RESRAD Version 6 User's Manual (Yu et al., 2001)	
Scenario)	Fraction on-site outdoor occupancy	0.06	Assumes industrial worker, 2 hours per day outdoors, 250 days per year (500 hours per year).		
	Fraction o-site indoor occupancy	-	Indoor occupancy not applicable for recreationist scenario.		
Occupancy (Recreational Visitor Scenario)	Fraction on-site outdoor occupancy	0.038	Conservatively assumes a recreationist spends two weeks camping on-site (336 hours per year outdoors on-site).	SAIC, 2010 (EE/CA for the nearby San Mateo Mine)	

Application	Model Parameter	Value	Rationale/Comments	Source/Reference
	Resident Rancher Inhalation Rate (m <sup>3</sup> y <sup>-1</sup> )	8,400	RESRAD default (residential scenario)	Table 2.3, RESRAD Version 6 User's Manual (Yu et al., 2001)
Inhalation Rate by Receptor Scenario	Worker Inhalation Rate (m <sup>3</sup> y <sup>-1</sup> )	11,400	RESRAD guidance for worker scenario	
	Recreational Visitor Inhalation Rate 14,000 (m <sup>3</sup> y <sup>-1</sup> )		RESRAD guidance for recreationist scenario	
	Area (m <sup>2</sup> )	85,800	Approximate total areal extent of mine related materials in Area A ( $\approx 21$ acres)	
Contomination Zono for	Thickness (m)	0.77	Calculated estimate of median depth of mine materials from polygon shapes used to estimate total volume in Area A.	SIR (ERG, 2013)
Contamination Zone for Area A	Median above-background radionuclide concentration (pCi/g)	Uranium = 35.3 Thorium -230 = 35.7 Radium-226 = 32	Median of measured values minus background (background = mean $+ 2\sigma$ ). Natural uranium isotopes partitioned based on approximate natural radiological abundance for each (48.9% each for Uranium-238 and -234, 2.2% for Uranium- 235).	SIR (ERG, 2013) NUREG-1620, Appendix H (NRC, 2003)

Table 2. Site-specific RESRAD parameters used to model risk for three difference receptor scenarios within Areas A and C (continued).

Application	Model Parameter	Value	Rationale/Comments	Source/Reference	
	Area (m <sup>2</sup> )	475,782	Approximate total areal extent of mine related materials in Area C ( $\approx$ 118 acres).	SIR (ERG, 2013)	
Contamination Zone for	Thickness (m)	0.51	Calculated estimate of median depth of mine materials from polygon shapes used to estimate respective volume in Area C.	SIR (ERG, 2013)	
Area C	Median above-background radionuclide concentration (pCi g <sup>-1</sup> )	Uranium = 4.1 Thorium-230 = 1.6 Radium-226 = 3.4	Median of measured values minus background (background = mean + $2\sigma$ ). Natural uranium isotopes partitioned based on approximate natural radiological abundance for each (48.9% each for Uranium-238 and -234, 2.2% for Uranium-235).	SIR (ERG, 2013) NUREG-1620, Appendix H (NRC, 2003)	
Gamma Shielding	Gamma penetration factor	0.4	Value recommended by EPA (NRC, 2003) and used in EPA's 1996 generic RESRAD risk/dose assessment for uranium and thorium.	Appendix H, NUREG-1620 (NRC, 2003) EPA, 1996	

Table 2. Site-specific RESRAD parameters used to model risk for three difference receptor scenarios within Areas A and C (continued).

Application	Model Parameter	Value	Rationale/Comments	Source/Reference	
Contaminated	Livestock water	0.3	Conservatively assumes 30% supplied by small on-site surface water stock pond. Deep groundwater well provides remainder (screened in confined regional aquifer below very low permeability Mancos Shale).	Groundwater assessment report (Itasca, 2013) Roca Honda Baseline Data Report (RHR, 2011)	
Fractions	Plant food and meat (resident scenario only)	0.3Conserve on-site s groundwin in confri permeat0.1 (plant) 1 (meat)Small g scenaric (consist) though livestoc2.24Local da (5.01 m0.21Local da (8.45 in sites.ent0.8Mean of sites.	Small garden possible for a resident rancher scenario, similar to urban resident scenario (consistent with RESRAD guidance). Possible (though unlikely) that all meat consumed from livestock could be raised in subject area.	Table 2.3, RESRAD Version 6 User's Manual (Yu et al., 2001)	
	Wind Speed (m s <sup>-1</sup> )	2.24	Local data for nearby Roca Honda Mine site (5.01 mph annual average).	Roca Honda Baseline Data Report (RHR, 2011)	
Meteorological Data	Annual Precipitation (m)	0.21	Local data for nearby Roca Honda Mine site (8.45 inch annual average).	Roca Honda Baseline Data Report (RHR, 2011)	
	Evapotranspiration Coefficient	0.8	Mean of cited range for semi-arid uranium mill sites.	NUREG-1620, Appendix H (NRC, 2003)	
Unsaturated Zone	Thickness (m)	185	Minimum based reported thicknesses of Mancos Shale in this area (600-1000 feet). The Mancos Shale is effectively unsaturated.	Groundwater assessment report (Itasca, 2013)	

Table 2. Site-specific RESRAD parameters used to model risk for three difference receptor scenarios within Areas A and C (continued).

Notes:

Isotope	Area A Resident PRG (pCi/g)	Area A Worker PRG (pCi/g)	Area A Recreator PRG (pCi/g)	Area C Resident PRG (pCi/g)	Area C Worker PRG (pCi/g)	Area C Recreator PRG (pCi/g)
Ra-226+D	2.9E+00	1.6E+01	3.2E+01	2.8E+00	1.6E+01	3.1E+01
Th-230	4.0E+01	1.1E+04	2.9E+04	4.0E+01	1.1E+04	2.9E+04
U-234	4.9E+01	1.8E+04	3.5E+04	4.9E+01	1.8E+04	3.5E+04
U-235+D	2.7E+01	2.4E+02	4.7E+02	2.7E+01	2.3E+02	4.6E+02
U-238	5.4E+01	2.0E+04	3.9E+04	5.4E+01	2.0E+04	3.9E+04

Table 3: Calculated total PRGs for soil by project area, receptor scenario and radionuclide.

	Coord	linate							Percent Passing - U.S. Sieve Numbers									
Sample ID	Northing	Easting	Depth (ft bgs)	USCS Classification	Soil Type	Moisture Content	Liquid Limit (%)	Plasticity Index	1-1/2"	3/4''	3/8''	#4	#8	#16	#30	#50	#100	#200
CN-01 sand	1586931	2754194	0-2	SM	Silty Sand	2.8	NP			98	91	86	81	79	77	74	35	23.4
CN-02 sand	1587168	2754237	0-6	SM	Silty Sand	2.4	NP							99	97	76	25	16.4
CN-03 sand	1587180	2754381	0-10	SM	Silty Sand	2.8	NP						99	99	98	84	23	16.0
CN-04 sand	1587229	2754567	0-10	SM	Silty Sand	3.3	NP							99	98	85	34	24.8
CN 05 sand	1587102	2754569	0-12	SM	Silty Sand	2.6	NP				98	98	97	97	97	88	26	18.5
CN-06 sand	1587111	2754399	0-8	SM	Silty Sand	2.4	NP							98	80	23	16.7	
CN-07 sand	1586728	2754434	0-12	SP-SM	Poorly Graded Sand w/Silt	1.6	NP				99	96	94	92	91	78	21	10.0
CN-01 shale	1586931	2754194	12-18				34	19										
CN-02 shale	1587168	2754237	6.5-8	CL	Lean Clay	7.0	30	16					99	98	96	87	67	57.2
CN-03 shale	1587180	2754381	10-16			6.0	30	16						99	97	88	59	48.9
CN-04 shale	1587229	2754567	10-14	CL	Lean Clay	8.5	31	15							99	95	81	69.6
CN 05 shale	1587102	2754569	12-15				21	7										
CN-06 shale	1587111	2754399	8-12	CL	Lean Clay	6.7	28	13			97	95	93	92	91	86	69	58.2
CN-07 shale	1586728	2754434	12-20			5.4	25	10			99	96	93	90	88	84	70	40.1

Table 4. Geotechnical and hydraulic properties of potential cover and liner soils.

	AS'	TM D 698 Data		temold Para )% Compa		90% C and	cm s <sup>-1</sup> at Compaction Optimum oisture	K <sub>sat</sub> , cm s <sup>-1</sup> , at 95% Compaction									
	Opt. Moist. Cont.	Max. Dry Density	Moist. Cont.	Dry Bulk Density	% of Max. Density			Dry Bulk Density	% of Max. Density	Moist. Cont.	K	K <sub>sat</sub> Dry Bulk Density Density Density Cont.		K	K <sub>sat</sub>		
Sample ID	(%, g/g)	(pcf)	(%, g/g)	(pcf)	(%)		Oversize Corrected	(pcf)	(%)	(%, g/g)		Oversize Corrected	(pcf)	(%)	(%, g/g)		Oversize Corrected
CN-01 sand	10.7	117.8	10.7	106.1	90.1	1.40E-03	1.20E-03										
CN-02 shale	11.0	116.3	11.0	104.9	90.2	8.35E-05	-	109.2	95	11.0	1.49E-06	-	110.4	95	15.0	1.76E-07	-
CN-04 shale	12.0	114.0	11.8	102.8	90.1	2.56E-05	-	107.9	95	12.0	1.06E-06	-	108.6	95	16.0	2.69E-07	-
CN-06 shale	10.8	117.8	11.0	106.1	90.0	2.80E-04	2.65E-04	111.1	95	10.8	8.24E-05	7.81E-05	111.7	95	14.8	9.56E-07	9.05E-07
CN-07 sand	11.8	108.6	12.0	97.7	90.0	8.38E-04	-										
CN-07 shale	10.5	118.0	10.6	106.1	90.0	1.34E-04	-	111.1	94	10.5	5.88E-05	-	111.7	95	14.5	4.46E-06	-

#### Table 4. Geotechnical and hydraulic properties of potential cover and liner soils (concluded).

Notes:

 $cm s^{-1} = centimeters per second$ 

ft bgs = feet below ground surface

g/g = weight of the water/the weight of the dry soil matrix, both in grams  $K_{\text{sat}}$  = saturated hydraulic conductivity

pcf = pounds per cubic feet

		Removal Alternatives							
		Off-site Disposal							
Task	Energy Solutions Rail Transport	Energy Solutions Truck Transport	WCS Rail transport	WCS Truck Transport	On-site Disposal				
Mobilization/ Demobilization	92,000	92,000	92,000	92,000	148,000				
Worker Health and Safety	442,000	442,000	442,000	442,000	207,000				
Radiological Remedial Support Services	430,000	430,000	430,000	430,000	319,000				
Construction Management	894,000	894,000	894,000	894,000	583,000				
Site Preparation	28,000	28,000	28,000	28,000	32,000				
Removal of Mine-related material <sup>b</sup>	61,002,000	166,133,000	49,140,000	118,772,000	2,438,000				
Disposal At Licensed Facility	22,497,000	22,497,000	99,985,000	99,985,000	NA				
Repository Construction	NA	NA	NA	NA	816,000				
Erosion Protection	NA	NA	NA	NA	517,000				
Site Restoration	191,000	191,000	191,000	191,000	524,000				
Post-Removal Site Controls	21,000	21,000	21,000	21,000	56,000				
TOTAL COST	85,597,000	190,728,000	151,223,000	220,855,000	5,640,000				

#### Table 5. Estimated costs of evaluated alternatives<sup>a</sup>.

Notes:

<sup>a</sup>All costs in U.S. dollars, rounded up to nearest 1,000.

<sup>b</sup>Includes excavation, loading, and hauling to point of disposal.

NA = not applicable

# Table 6. Comparison of alternatives against selection criteria.

Alternative	Long-term effectiveness and permanence	Short-term effectiveness	Implementability	Cost
No Action	Poor	Poor	Good	Good
Off-site Disposal	Good	Fair	Poor	Poor
On-site Disposal	Good	Good	Good	Good

Notes:

Poor: Unable to adequately address the ARARs and selection criteria Fair: Able to adequately address some of the ARARs and selection criteria Good: Able to adequately address all of the ARARs and selection criteria ARARs = Applicable or Relevant and Appropriate Requirements Figures

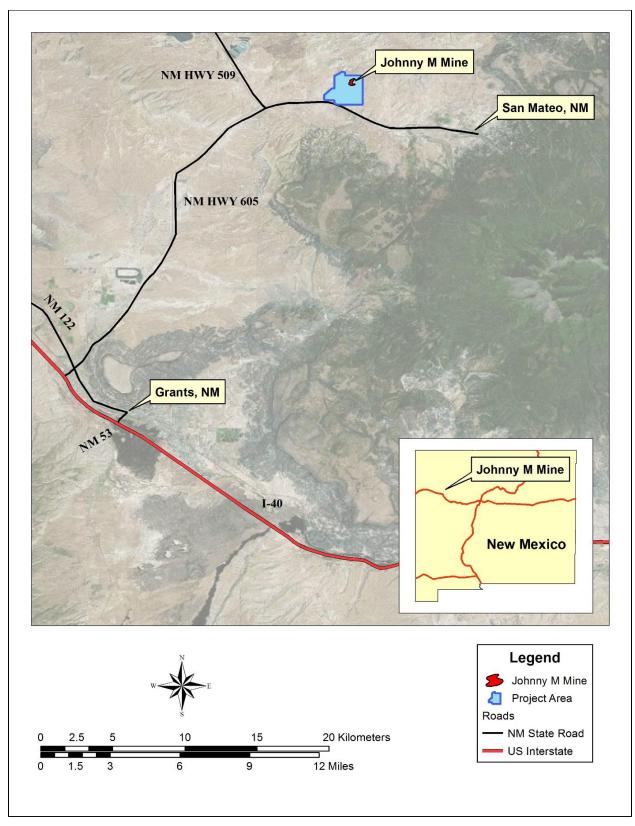


Figure 1. Location of the project area.

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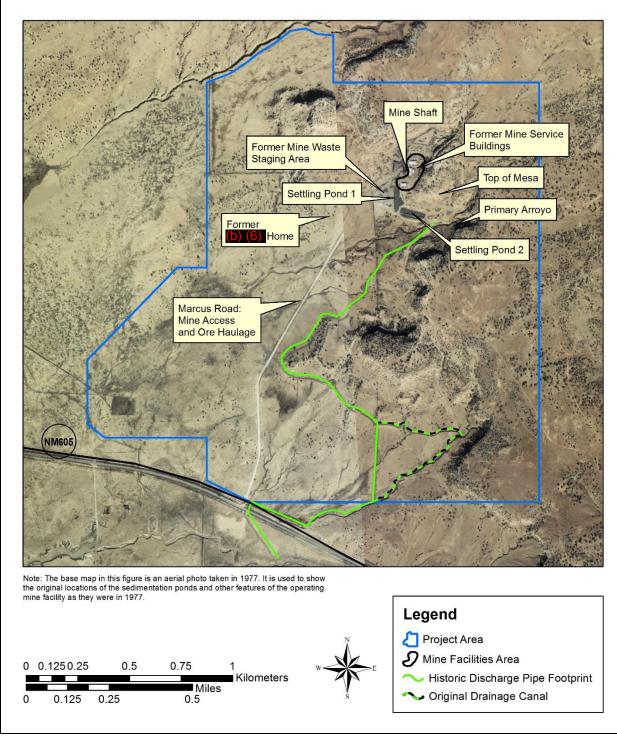


Figure 2. 1977 Historical aerial photo with project area features.

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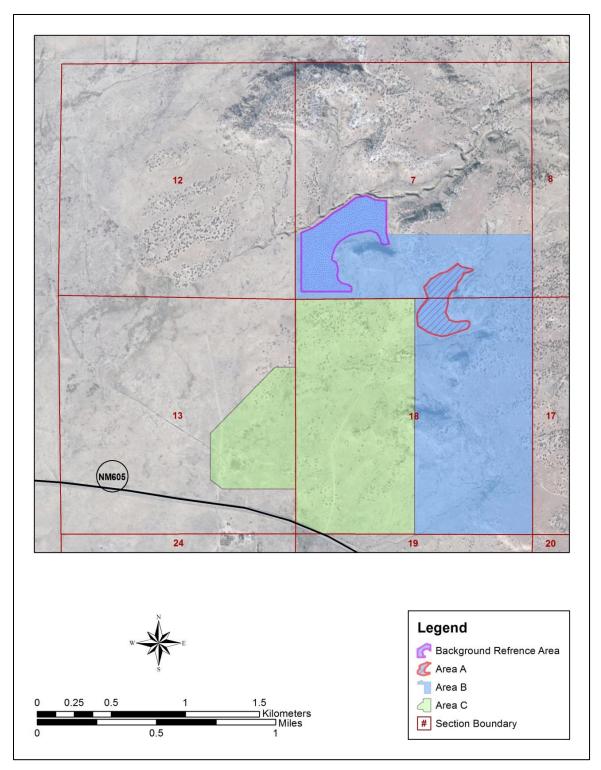
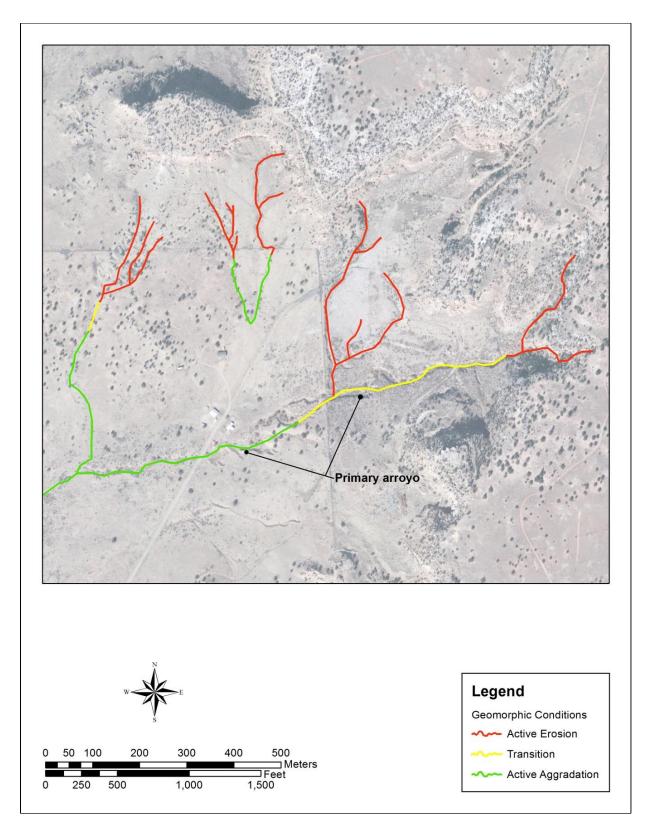
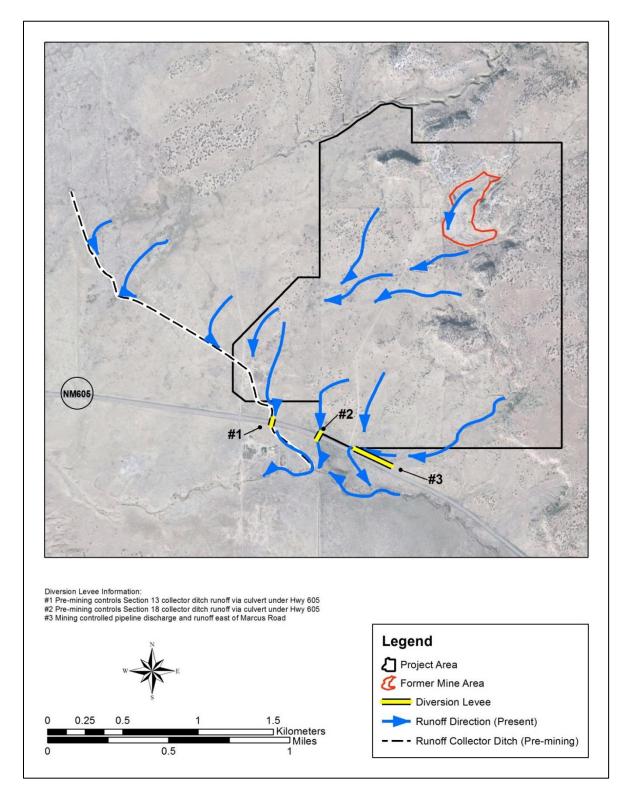


Figure 3. Areas A, B, and C with Background Reference Area [adopted from the SIR (ERG, 2013)].



# Figure 4. Geomorphological characterization of arroyos in the project area [adopted from the SIR (ERG, 2013)].

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# Figure 5. Additional geomorphological features and runoff patterns in the project area [adopted from the SIR (ERG, 2013)].

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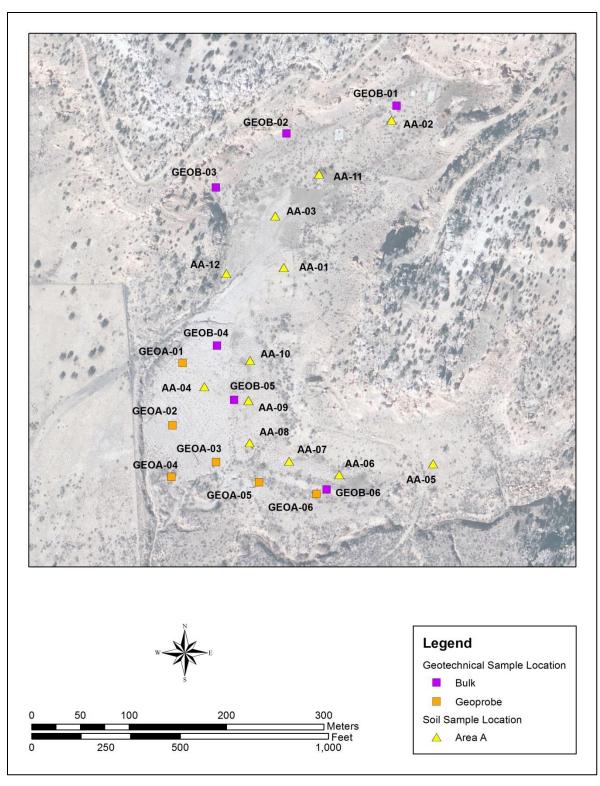


Figure 6. Area A soil sample locations [adopted from the SIR (ERG, 2013)].

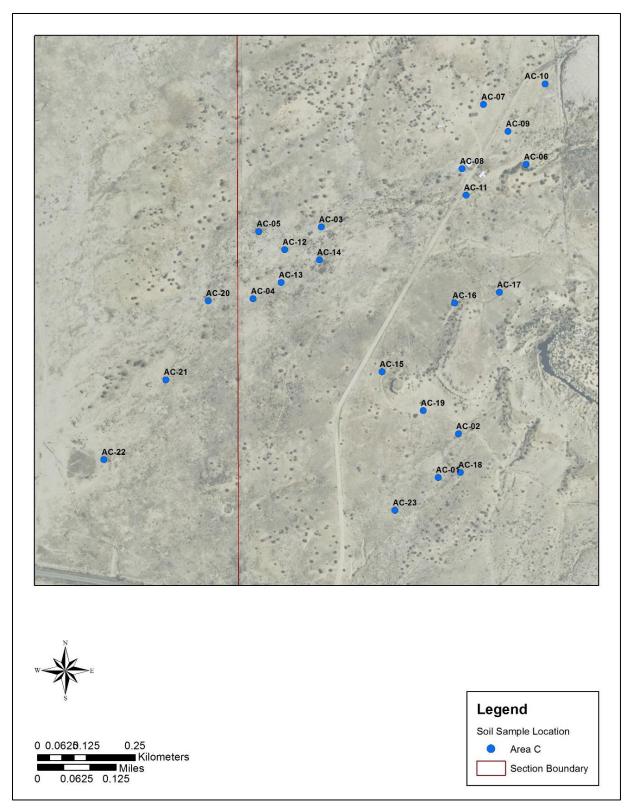


Figure 7. Area C soil sample locations.

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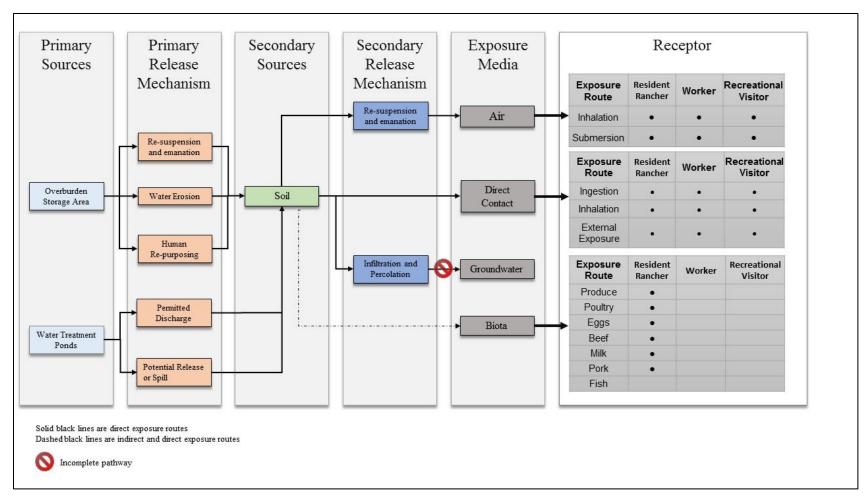


Figure 8. Sources, release mechanisms, and potential exposure pathways at the project area.

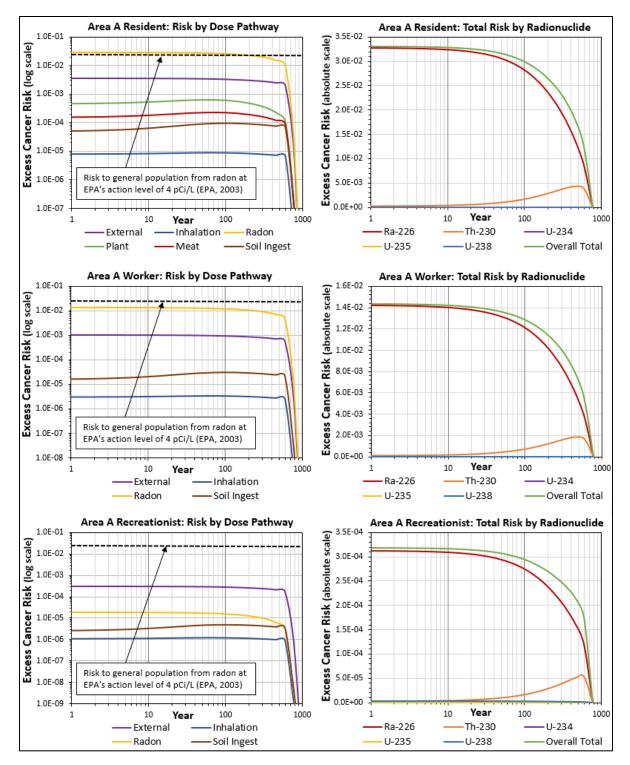


Figure 9. RESRAD modeling results: lifetime excess cancer risk by receptor scenario for Area A.

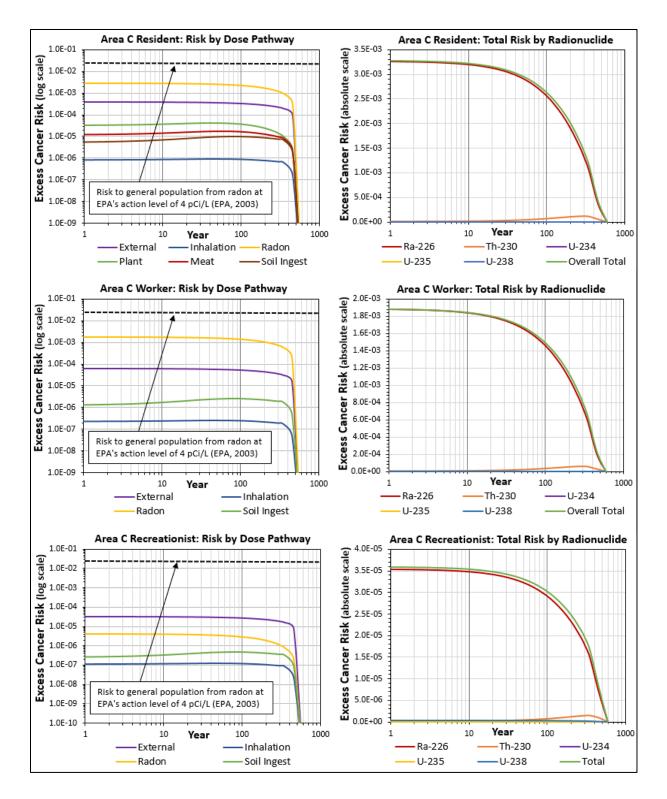


Figure 10. RESRAD modeling results: lifetime excess cancer risk by receptor scenario for Area C.

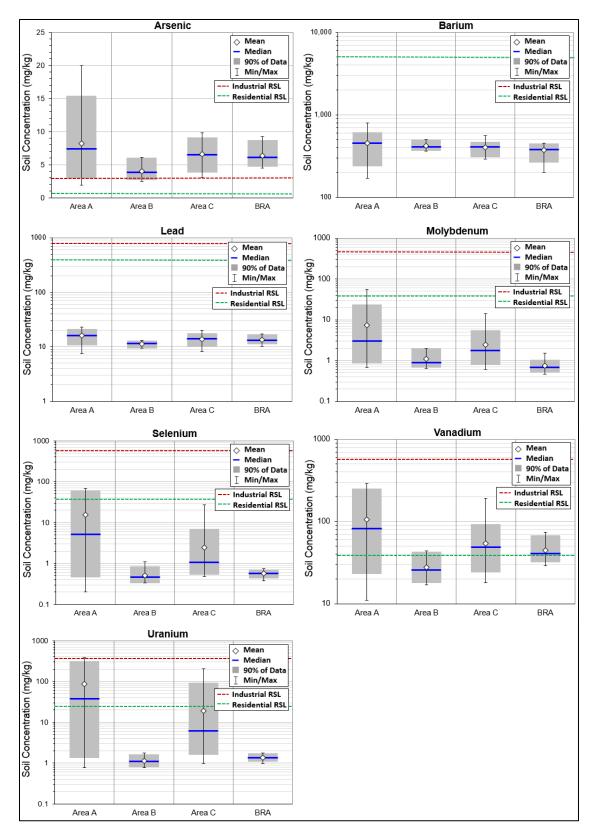
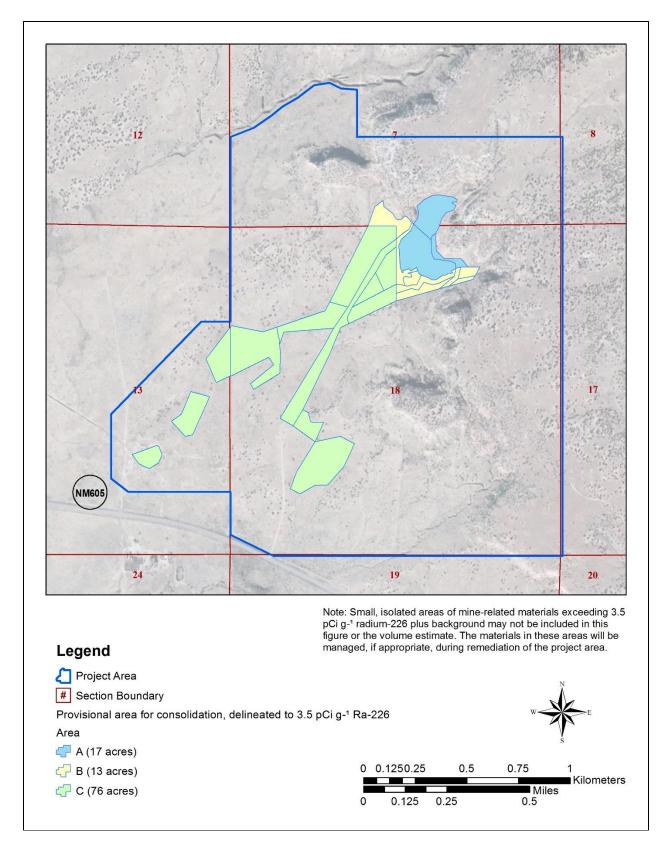


Figure 11. Comparisons of measured uranium and indicator metals against EPA regional screening levels for soils.

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Figure 12. Road and rail transportation routes from the project area to disposal facilities.



# Figure 13. Extent of area considered for consolidation used to estimate costs.

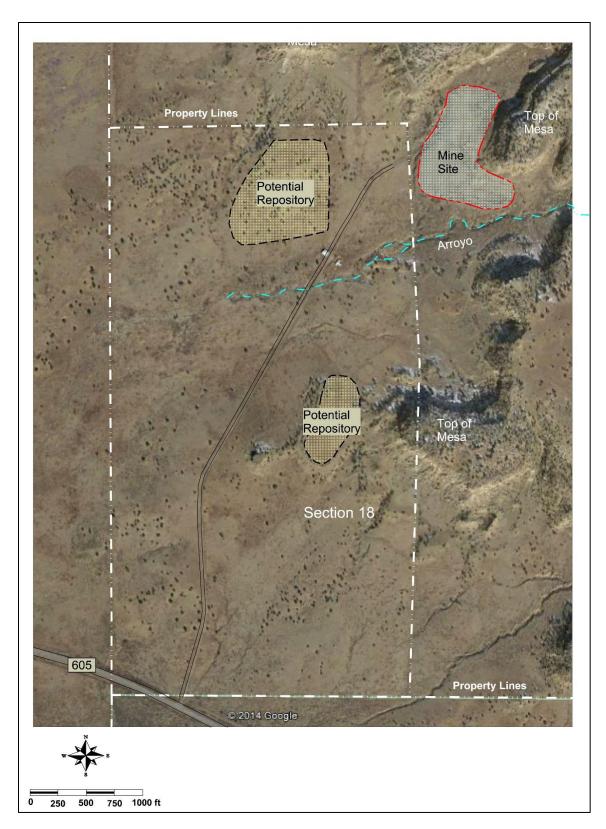


Figure 14. Potential repository locations in the project area.

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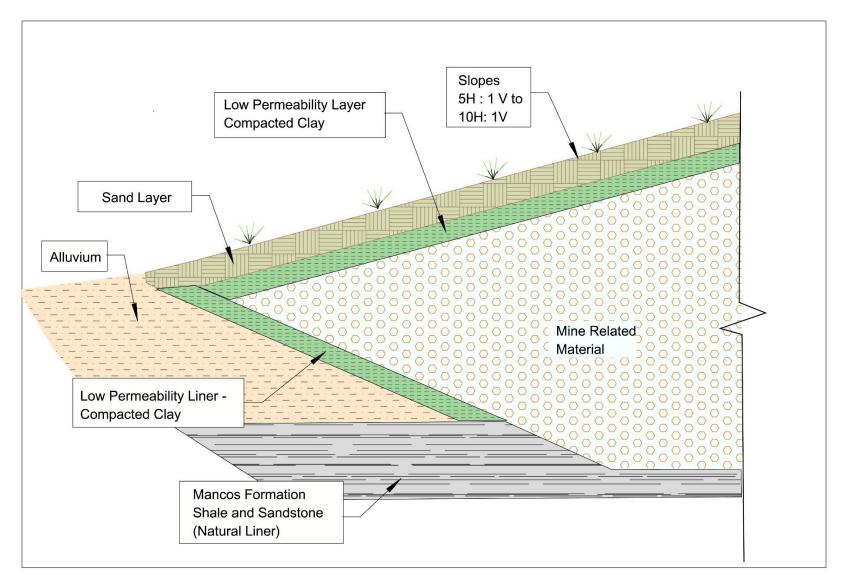


Figure 15. Conceptualized cross section of a potential repository.

Appendix A

Analysis of Groundwater Conditions at the Former Johnny M Mine, McKinley County, New Mexico (ITASCA, 2013)

# ANALYSIS OF GROUNDWATER CONDITIONS AT THE FORMER JOHNNY M MINE MCKINLEY COUNTY, NEW MEXICO

Prepared for Hecla Limited Coeur d'Alene, Idaho

Prepared by Itasca Denver, Inc.

March 2013





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## 1.0 INTRODUCTION

Itasca Denver, Inc., (Itasca) was asked by Hecla Limited (Hecla) to review existing geologic, hydrogeologic, and geochemical data related to the former Johnny M Mine (Project Area) and surrounding area, located in McKinley County, New Mexico (Figure 1). Itasca was requested to address the following questions regarding potential groundwater and surface-water quality impacts associated with the Project Area:

- Is shallow groundwater quality in the Project Area affected by mine water during operations or leaching from mined materials?
- Has the groundwater quality of the former domestic wells in the Project Area (GMD-04 and/or GMD-05) or other groundwater resources been affected by mining-related activity in the Project Area?
- Is the quality of groundwater in the Project area affected by the presence of backfilled tailing sand in the underground workings?

The following sections provide a description of the geology, hydrogeology, and geochemistry of the Project Area and discuss Itasca's evaluation of these water-quality questions.

#### 1.1 BACKGROUND

The Johnny M uranium deposit was discovered in November 1968 and work began on sinking a shaft in late 1972. Ore in the Johnny M Mine came from the Poison Canyon tongue of the Morrison Formation Brushy Basin Member and from a zone near the top of the Morrison Formation Westwater Canyon Member, at depths of between 1,300 and 1,400 feet below ground surface (ft bgs). Production appears to have started in 1976 with the shipment of low-grade ore to Kerr McGee's uranium mill located at Ambrosia Lake. No milling occurred on site; all ore was shipped off site for processing. Production at the mine ended in 1982.

The ore-bearing zone originally was saturated, and was dewatered to facilitate mining. Starting in August 1977, backfilling was performed to enhance the geomechanical stability of the stopes (areas of the mine from which ore had been produced). Approximately 286,000 tons of tailings sand were obtained from the Ambrosia Lake mill and placed within the mine to backfill stopes. Backfilling occurred using a mixture of mine-supplied water and sand, which was slurried into the stopes.



Initially mine discharges consisted of water resulting from dewatering and mine operations (e.g., drilling). Later, the slurry water was collected within the mine and pumped to land surface as part of the ongoing mine dewatering operations. (Mine water for purposes of this report includes water derived from groundwater dewatering, drill water, and slurry water.) Pumping from the mine averaged approximately 800 gallons per minute (gpm), and the recovered water was discharged to two treatment ponds that were excavated into native materials (Ponds 1 and 2 are shown in Figure 2). The recovered water was treated in the ponds by the addition of a coagulant and barium chloride, and then discharged to the San Mateo Creek drainage channel via a one-mile open ditch that was later replaced by a 12-inch diameter pipe (Figure 2).

The mine-water discharge plan, as described above, was approved by the New Mexico Environmental Improvement Board. The area to which treated-water discharge occurred is underlain by up to 80 ft of alluvium/colluvium on top of the Mancos Shale. During and after mining, water samples were collected from groundwater monitoring wells and surface-water locations. The locations of these sampling points are indicated in Figure 2. Figure 2 was constructed from three different hand-drawn maps; the locations of several sampling points are deemed approximate. Water samples collected at MW-1 represent the quality of the treated discharge waters from the surface-treatment ponds. A summary of the sampling points is provided in Table 1.

Upon completion of mining, the mine shaft was sealed with a reinforced four-foot-thick engineered concrete plug. The plug was set between the Dakota Formation and the Westwater Canyon Member of the Morrison Formation.

#### 1.2 NEARBY PROJECTS AND OTHER DATA

In addition to the data that were available for the Project Area, the assessment provided in this report also considered data that were available from several nearby projects. A significant amount of geologic, hydrogeologic, and geochemical data were available from the Baseline Data Report (BDR) (Roca Honda Resources 2011) for the proposed Roca Honda (RH) mine that is located approximately one mile directly east of the Project Area (Figure 1). Data were also available from studies associated with the USDA Forest Service Non-Time Critical Removal Action (Science Applications International Corporation 1994) at the former San Mateo Mine that is located approximately two miles south of the Project Area. Geologic information in the Project Area was



supplemented from a geologic log for one of the former domestic wells in the Project Area (GMD-04). Lastly, water-quality samples were collected from the two former domestic wells in the Project Area by the New Mexico Environmental Department (NMED) on behalf of the United States Environmental Protection Agency (USEPA) (NMED2011). The data available to Itasca were sufficient to answer the questions posed by Hecla.



## 2.0 <u>GEOLOGY</u>

#### 2.1 REGIONAL GEOLOGY

The regional geology of the San Juan Basin is shown in Figure 1. The Project Area, the proposed RH permit area, and the former San Mateo Mine are also shown relative to the area depicted. Three structural features associated with the San Juan Basin (the Zuni uplift, Chaco slope, and Rio Grande Rift) are particularly important to the hydrogeology of the Project Area, as discussed below. The Zuni uplift is located approximately 25 to 30 miles southwest of the Project Area. This uplift is an important regional structural feature that exposes rocks as old as Precambrian in age and is an important location of regional recharge to groundwater. The area of transition from the Zuni uplift to the central part of the San Juan Basin is the Chaco slope (Figure 3), where regional sedimentary strata of mainly Mesozoic age dip gently to the northeast, into the central part of the basin. The dip of the rock units varies between four to eight degrees. The Rio Grande Rift is located on the southeast margin of the San Juan Basin and groundwater flow in the southeastern portion of the basin is generally directed toward this regional structural feature (Figure 3).

The stratigraphic column of geologic units encountered regionally is shown in Figure 4 and includes several units, such as the Menefee Formation, Point Lookout Sandstone, and Mount Taylor volcanics, that are not present in the Project Area due to an erosional unconformity.

#### 2.2 SPECIFIC GEOLOGY OF PROJECT AREA

Understanding the geology and stratigraphy of the Project Area in relation to groundwater sampling activities, is critical to evaluating potential water-quality impacts to groundwater in the Project Area. Figure 5 is a stratigraphic column for the Project Area. It is particularly significant for answering two of the questions posed that some of the uppermost formations present regionally are not present in the Project Area, as they have been removed by erosion. Figure 5 also includes the stratigraphic locations of the screened zones for monitoring and domestic wells sampled for groundwater-quality investigations. The Gallup Sandstone is present in the Project Area, and it caps the mesas that occur within and in the vicinity of the Project Area; however, this sandstone is generally not saturated in the Project Area.



## 2.2.1 Surficial Sediments

In the Project Area, surficial sediments that are classified as alluvium or colluvium range in thickness from 0 to 80 ft. These sediments are typically thin and unsaturated near the mesas and become thicker and saturated near the San Mateo Creek drainage channel (Figure 2), a stream that flows intermittently.

## 2.2.2 Mancos Shale

As mentioned previously, the Menefee Formation and Point Lookout Sandstone do not exist in the Project Area because they have been eroded. As a result, the main body of the Mancos Shale is below the surficial sediments or the Gallup Sandstone. The Mancos Shale forms a widespread regional aquitard that is approximately 600 to 1,000 ft thick locally. The Mancos Shale represents the interplay of transgressive and regressive episodes of the epicontinental Western Interior Seaway. Shale, mudstone, claystone, and limestone were deposited during transgressions, and sandstones were deposited during regressions (Environmental Sciences Laboratory 2011). The Twowells Sandstone Tongue, an interbed of the Dakota Sandstone, occurs between the main body of the Mancos Shale and the Whitewater Arroyo Tongue of the Mancos Shale. One of the former domestic wells in the Project Area, GMD-04, which is located upgradient of the mine, appears to have been screened in this interbed within the Mancos Shale. Other localized sandstone lenses are also present within the main body of the Mancos Shale. As will be discussed later, the other former domestic well in the Project Area (GMD-05) is probably screened within the Mancos Shale.

#### 2.2.3 Dakota Sandstone

The Dakota Sandstone is located below the Mancos Shale and was deposited during the initial transgression of the seaway, although, as previously noted, there is some interbedding between these formations. The Johnny M Mine potable groundwater-supply well (WW in Tables 1 and 2) was screened in the Dakota Sandstone (Figure 5). The Twowells Sandstone Tongue is the uppermost unit of the Dakota Sandstone and ranges in thickness from about 30 to 120 ft (Roca Honda Resources 2011), with an average thickness of approximately 70 ft. This is the uppermost bedrock water-bearing zone in the Project Area and, based on the depth of GMD-04 (depth to groundwater at 624 ft below top of casing and a total depth of 715 ft bgs), also appears to be the



unit in which GMD-04 is screened. Below the Twowells Sandstone is another approximately 50 to 150 ft of Mancos Shale (the Whitewater Arroyo Shale Tongue), and below that is the 20 to 80 ft thick main body of the Dakota Sandstone. Based on the drilling log, well WW appears to be screened in the main body of the Dakota Sandstone (water level at a depth of 673 ft below top of casing and a total depth of 1,084 ft bgs).

## 2.2.4 Morrison Formation

The Morrison Formation is located below the main body of the Dakota Sandstone. The uppermost portion of the Morrison Formation is the Brushy Basin Member. Excluding the sandstone Poison Canyon Tongue at its base, the Brushy Basin Member is green shale with very low hydraulic conductivity (as evidenced by very slow draindown from the overlying Dakota Sandstone following dewatering of the Morrison Formation sandstones during mining (Rosel 1979)). The Brushy Basin Member averages about 100 ft thick in the local area. As previously mentioned, the Johnny M Mine recovered ore from sandstones in the Morrison Formation, namely the Poison Canyon Tongue, at the base of the Brushy Basin Member, and the subjacent (approximately 25 ft below) Westwater Canyon Member of the Morrison Formation, at depths of approximately 1,300 to 1,400 ft bgs. The mine was backfilled with tailings sand that was slurried into the mine workings in the Morrison Formation, and several water-quality sampling locations discussed below are within this zone (i.e., well 15, well 143, the North Vent pipe, UG4, UG5, UG6, DS2, and DN1; see Table 1; see also Figure 2 for locations of well 15, well 143, and the North Vent pipe).



#### 3.0 HYDROGEOLOGY

#### 3.1 REGIONAL HYDROGEOLOGY

In the San Juan Basin (including the Project Area), there are several thick, very low-permeability shale layers (e.g., the Mancos Shale, Brushy Basin Member of the Morrison Formation, and the Recapture Shale) that hydraulically separate the formations that serve as groundwater resources in the region. These shale layers separate the deeper groundwater resources (i.e., the Gallup Formation, Dakota Sandstone Formation, and Westwater Canyon Member of the Morrison Formation) from each other, as well as from the much shallower alluvial groundwater systems and shallow groundwater resource units (i.e., Point Lookout Sandstone and Menefee Formation) that are present regionally (INTERA 2012). Thus, recharge and discharge associated with these deeper units are a function of their outcrop exposures.

In general, groundwater recharge enters the groundwater-flow system as precipitation on permeable formations that crop out along the southern margin of the San Juan Basin and on the flanks of the Zuni, Chuska, and San Mateo mountains. Groundwater then flows downgradient, either northwestward to discharge along the San Juan River, or in the southeast portion of the basin (where the Project Area is located), northeastward, eastward, and southeastward (see Figure 3) toward the Rio Grande Rift, to discharge to tributaries of the Rio Grande, including the Rio Salado, Rio Puerco, and Rio San Jose. Potentiometric surface maps indicate that the pattern of regional groundwater movement within the deeper units in the southeastern part of the San Juan Basin is greatly influenced by the Zuni uplift, the Chaco slope, and the Rio Grande Rift (Roca Honda Resources 2011).

The movement of groundwater through the alluvial valleys is influenced by topography and surface-water drainages and is independent of—and sometimes flows in directions opposing—groundwater movement in the deep water-bearing units. Volcanic rocks of the Mt. Taylor volcanic field are present less than five miles to the east and south of the Project Area. This is an area of local and regional groundwater recharge for shallower rocks of the Tertiary and Upper Cretaceous age. The younger, shallower groundwater-bearing units in the region (e.g., the Menefee Formation and Point Lookout Sandstone) are not present in the Project Area. Where present regionally, these units occur higher in the stratigraphic sequence. The direction of groundwater flow for the shallow



water-bearing unit in the region, the Menefee Formation (Figure 6), is to the northwest. The elevations of the water table (in the Menefee Formation) are approximately 600 to 700 ft above the potentiometric surface of the Westwater Canyon Member (cf. Menefee Formation and Westwater Canyon Member potentiometric contours in Figure 6). The higher potentiometric surface in the Menefee Formation indicates that there is a downward vertical gradient, and the vertical hydraulic gradient may be due, at least in part, to the low permeability of the Mancos Shale that separates alluvium and shallow water-bearing bedrock units from the deeper water-bearing units in the Project Area.

Other important regional water-bearing units, such as the Dakota Sandstone, are substantially deeper, moving away from the Project Area to the northeast. The Dakota Sandstone dips downward at an angle of 350 to 700 ft per mile to the northeast of the Project Area because of the dip associated with the Chaco slope. Accordingly, the geologic units present in the Project Area that could be considered groundwater resources, such as the Dakota Sandstone, are less desirable as a source of groundwater downgradient of the Project Area due to depth and the associated high costs of drilling and pumping water from deep wells. There are no identified domestic or stock wells completed in the Morrison Formation or Dakota Sandstone to the northeast of the Project Area. The distance of this well search is over ten miles from the mine. The nearest domestic wells in the general downgradient direction of the Project Area (wells 4, 7, 132, and 133 in Figure 7) are screened in the much shallower Menefee Formation or Point Lookout Sandstone. These wells are at least four miles northeast of the Project Area (Figure 7); furthermore, the hydraulic gradient in the vicinity of the Project Area is downward, away from the units in which these wells are screened. Figure 3 shows the basin-wide general regional pattern of deep groundwater flow in the Jurassic (Morrison Formation) and Cretaceous (Dakota Sandstone) waterbearing units (relevant to the Project Area) and Figure 8 shows the potentiometric surface and groundwater flow directions specific for the Westwater Canyon Member of the Morrison Formation in the southeastern portion of the San Juan Basin. As noted in Figures 3 and 8, groundwater flow in the deep Dakota Sandstone and Morrison formations is to the east-southeast based upon a regional analysis. Figure 6 shows that in the vicinity of the Project Area, deep groundwater flows to the northeast.



## 3.2 SPECIFIC HYDROGEOLOGY OF THE PROJECT AREA

#### 3.2.1 Shallow Groundwater (Surficial Sediments)

Groundwater flow within the surficial sediments (alluvium and colluvium) that are located on the slopes and within the alluvial valleys follows the local topography (flow in the alluvium within the Project Area is generally to the west/southwest) in the opposite direction of groundwater flow in the bedrock (to the east/northeast). The alluvium is a source of groundwater to wells that are located near the San Mateo Creek drainage channel. The creek is also a source of groundwater recharge.

During mining operations, treated mine water was discharged from the ponds to a ditch and later to a pipe that eventually emptied into the San Mateo Creek drainage channel. A portion of this water, along with precipitation runoff, infiltrated these alluvial sediments and flowed to the San Mateo Creek drainage channel. Later, the pond water was piped farther down the slope, discharging at or near the San Mateo Creek drainage channel. Discharged water that infiltrated the surficial sediments would have perched on top of the Mancos Shale forming a saturated zone within the shallow surficial sediments; monitoring wells GW7, GW8, GW8A, and GW9 were installed and screened at the contact between the surficial sediments and the Mancos Shale (Figure 5) to monitor groundwater quality at this contact in response to discharges from the surface treatment ponds.

#### 3.2.2 Intermediate Groundwater (Mancos Shale)

The hydraulic conductivity in the Mancos Shale is generally very low, on the order of 5 x  $10^{-8}$  cm/s (Roca Honda Resources 2011). To put this value into context, a compacted clay liner for a municipal landfill typically has a permeability (hydraulic conductivity) of approximately  $1 \times 10^{-7}$  cm/s (Benson and Trast 1995). Isolated sandstone lenses typically occur within the Mancos Shale (Environmental Sciences Laboratory 2011) and have been noted in drill logs from the Project Area. For example, 'gray sandstone' was noted in the geologic log at 115 to 130 ft bgs from a former domestic well within the Project Area located upgradient of the mine (OSE#B-01544, subsequently identified as GMD-04) within the 615 ft thick Mancos Shale interval. The well log for the other well at the former residence located upgradient of the mine, GMD-05, was not available for evaluation. It was noted though that GMD-04 was drilled as a replacement because



GMD-05 failed to produce water at sufficient rates (b) (6), pers. comm. with M. Schierman of ERG). As discussed below, the quality of groundwater from GMD-05 also appears similar to that reported elsewhere for the natural groundwater quality associated with the Mancos Shale.

#### 3.2.3 Deep Groundwater (Dakota Sandstone and Morrison Formation)

The hydraulic conductivity of the Dakota Sandstone ranges from  $9 \times 10^{-5}$  to  $5 \times 10^{-4}$  cm/s (INTERA 2012). The hydraulic conductivity values for the Dakota Sandstone suggest that it is capable of transmitting low to moderate volumes of water depending on its thickness. Wells producing from the Dakota Sandstone yield in the range of 1 to 75 gpm with a median value of 12 gpm (Roca Honda Resources 2011).

The hydraulic conductivity of the Westwater Canyon Member varies from 7 x  $10^{-6}$  to 6 x  $10^{-4}$  cm/s (INTERA 2012). These values suggest that the Westwater Canyon Member transmits low to relatively high quantities of water, again, depending on its thickness. Wells completed in the Westwater Canyon Member have been pumped at rates between 10 and 560 gpm with typical values around 100 gpm (Roca Honda Resources 2011). As noted in Figure 6, the direction of flow for groundwater in the Project Area in the Westwater Canyon Member is towards the north-northeast.

The hydraulic gradient calculated from the potentiometric surface map for the Westwater Canyon Member shown in Figure 6 is approximately 0.024 ft/ft to the northeast. Assuming an effective porosity of 0.1 (Roca Honda Resources 2011) yields a range of groundwater velocities of 2 to 150 ft per year in the Westwater Canyon Member. Based upon this range of values, it would take groundwater approximately 35 to 2,600 years to travel one mile. Assuming the hydraulic gradient of 0.024 ft/ft, the elevation of the potentiometric surface would be at an elevation of approximately 5,350 ft above mean sea level in the vicinity of well 133 (Figure 7), a well screened in the Menefee Formation. The elevation of the bottom of this well is approximately 6,760 ft (Roca Honda Resources 2011). This means that there is approximately 1,200 ft of separation between groundwater in the Morrison Formation and the Menefee Formation.



#### 4.0 GROUNDWATER QUALITY

The available groundwater-quality data related to the Project Area were compiled by Itasca and are provided in Table 2 and discussed below.

The assessment of potential past and future groundwater impacts resulting from historical mining operations hinges on the potential migration of uranium (U) in groundwater. The following paragraphs provide an overview of the geochemistry of U and its potential to migrate in groundwater.

Uranium movement in groundwater is dependent upon the geochemical conditions of the environment, particularly with respect to pH and oxidation state (i.e., Eh). Uranium in an oxidizing environment is capable of migrating with groundwater, unlike in reducing conditions such as those found in groundwater in the Dakota Sandstone and Morrison Formation in the Project Area. Figure 9 shows an Eh-pH diagram for the simplified geochemical system composed of U, silica, and water at 25°C. Minerals such as coffinite (USiO<sub>4</sub>) illustrated in Figure 9 and uraninite  $[UO_2(a)]$  which occupies a similar but smaller stability range to that illustrated for coffinite in Figure 9 contain U in its reduced form, the U(IV) valence state, and are relatively insoluble and stable under reducing conditions. Whereas, the mineral schoepite [UO<sub>2</sub>(OH)<sub>2</sub>•H<sub>2</sub>O] (Figure 9) contains U in the U(VI) valence state. The U(VI) valence state is predominant in more oxidizing conditions, such as those frequently associated with surface water and shallow groundwater. It is often present as a  $UO_2^{+2}$  ion or associated hydroxide and/or carbonate complexes. Unlike the U(IV) valence state that is predominant in more reducing conditions, the uranyl hydroxide and/or carbonate complexes can increase U migration in groundwater relative to U(IV). Furthermore, the mineral schoepite, which forms in more oxidizing conditions, is more soluble than the minerals coffinite and uraninite that form under more reducing conditions. Accordingly, the solubility of U minerals also contributes to the ability of U to migrate in oxidizing conditions typically associated with surface water and shallow groundwater.

Radium is generally not of concern in the Project Area based upon work conducted by the NMED (NMED 2010). The NMED indicated in a review of geochemistry in the San Mateo Creek (SMC) area that:



[Radium] Ra does not appear to be a contaminant of concern in the ground water system of the SMC study area because it is relatively insoluble, does not tend to form soluble complexes with other ions, was easily precipitated out of acidic mill tailings by the addition of BaSO<sub>4</sub>, and has a strong tendency to adsorb onto various mineral surfaces such as clays and other silicate minerals (Landa, 1980). Based on the water sample results from EPA, 1975, and the results from this investigation, Ra does not appear to be a radiochemical of concern or a reliable indicator of legacy U mining and milling impacts.

In contrast, U concentrations from this investigation indicate that this radionuclide is elevated in the ground water, and the geochemical conditions support transport of this metal in the aqueous environment.<sup>1, 2</sup>

The NMED (2010) concluded that the estimated average U concentration in groundwater samples that are assumed not to be impacted by mining or milling discharges is less than 5  $\mu$ g/L.

## 4.1 SHALLOW GROUNDWATER QUALITY

#### 4.1.1 Shallow Groundwater-Quality Observations

Shallow groundwater, when present, is separated from the deeper Dakota Sandstone Formation and Morrison Formation water-bearing zones by more than 600 ft of relatively impermeable (Mancos) shale. Hence, the mining activities in the Project Area with the potential to have affected the quality of shallow groundwater were the surface activities associated with discharging mine water into the ponds/ditch and the potential for leaching of stockpiled mine-related materials on the land surface.

During infiltration events in the Project Area, surface water infiltrates downward and perches temporarily on the bedrock (Mancos Shale) surface before it moves downgradient.

Potential degradation of shallow groundwater from the above activities is evaluated below by comparison of the water quality associated with dewatering water, sand slurry, shallow

<sup>&</sup>lt;sup>1</sup> Note that  $BaCl_2$  was used for water-quality treatment at the Johnny M Mine. This forms an insoluble  $BaSO_4$  coprecipitate that quantitatively removes radium.

<sup>&</sup>lt;sup>2</sup> The NMED text cited here is in reference to surficial, oxidizing conditions. Uranium is much less soluble and mobile under reducing conditions, such as those in the Johnny M Mine following inundation by groundwater at the end of mining.



groundwater monitoring wells, and well(s) located on a nearby ranch. As illustrated in Figure 5, the shallow groundwater monitoring wells located within the Project Area were typically constructed to collect water from the contact between the surficial sediments and the top (weathered surface) of the Mancos Shale. Weathered zones of the Mancos Shale have been noted as being naturally affected by geochemical processes including pyrite oxidation, carbonate dissolution, gypsum precipitation, release of nitrate from weathering of organic material, and solubilization of U (Environmental Sciences Laboratory 2011). Consistent with these processes, Figure 10 illustrates that the sulfate concentrations observed in the shallow groundwater monitoring wells (GW7, GW8A, and GW9) are actually higher, in most cases, than the concentrations observed in sand slurry (MWS-3), dewatering discharge, and MW-1 (which was the monitoring location for discharge from the pipeline or ponds prior to entry into the San Mateo Creek drainage channel), or any of the water-quality samples collected from within the underground mine (e.g., DN1, DS2, UG4, UG5, UG6, North Vent pipe).

Similarly, Figure 11 illustrates the nitrate and U concentrations from shallow groundwater wells, the upgradient wells (former domestic wells) within the Project Area, sand slurry, and various mine-water samples. The shallow groundwater-well samples generally cluster around the U and nitrate geometric mean for the Mancos Shale, but with slightly lower U concentrations. In contrast, U concentrations in mine waters are typically an order of magnitude higher than those observed in the shallow groundwater monitoring wells, and the nitrate concentrations are typically one to three orders of magnitude lower in mine waters than in the shallow groundwater monitoring wells.

Water chemistry was measured in 1993 in a groundwater sample from a well located on the Marcus Ranch, which is a shallow groundwater well located on the north side of the San Mateo Creek drainage channel and downgradient of the former Johnny M Mine discharge location (Figure 2). The gross alpha concentration (activity) was  $6\pm15$  pCi/L, the <sup>226</sup>Ra concentration was 0.20±0.28 pCi/L, the gross beta concentration (activity) was  $7\pm29$  pCi/L, the dissolved U was 3.5 µg/L, arsenic was less than 0.005 µg/L, lead was less than 0.01 µg/L, molybdenum was less than 0.02 µg/L, selenium was less than 0.01 µg/L, and vanadium was less than 0.01 µg/L. In summary, concentrations of U, <sup>226</sup>Ra, arsenic, lead, molybdenum, selenium, and vanadium were either below detection limits or below drinking water-quality standards (Science Applications International Corporation 1994).



## 4.1.2 Conclusions Regarding Potential Impacts to Shallow Groundwater Quality

Mining-related discharge water that infiltrated the shallow surficial sediments and perched on the surface of the Mancos Shale more than 25 years ago as a result of mining activities would not now be contributing seepage to the San Mateo Creek drainage system. Subsequent overland runoff and infiltration over the past 25 years would have concentrated in drainage features and tended to 'flush' sediments. Given that the runoff waters would be rich in dissolved oxygen, this oxygenated water would have mobilized any U, or 'flushed' any U from the sediments.

The groundwater quality measured in a water sample from the Marcus Ranch well indicated that the water-bearing surficial sediments have not been impacted by the historical discharges from the mine or by the current conditions within the Project Area. Whereas the other radionuclides (alpha and gross beta) had large errors surrounding the measured concentrations, the reported concentrations do not indicate impacts, particularly when considered together with the low concentrations of U, <sup>226</sup> Ra, and other metals typically associated with mine water.

#### 4.2 INTERMEDIATE GROUNDWATER QUALITY

#### 4.2.1 Intermediate Groundwater-Quality Observations

The groundwater-quality compositions of the two upgradient wells (former domestic wells) within the Project Area (GMD-04 and GMD-05) are quite different from one another. The results of groundwater analyses for wells GMD-04 and GMD-05 are shown in Table 2. The quality of groundwater from GMD-04 can be summarized as follows:

- a mixed calcium/sodium-bicarbonate/sulfate water type;
- at or near the USEPA human health-based maximum contaminant limit (MCL)<sup>3</sup> for gross alpha (17.3±4.01 picocuries per liter (pCi/L) vs. MCL of 15 pCi/L); there is no applicable State standard<sup>4</sup> for gross alpha for groundwater;
- at or near the MCL for radium radioactivity (3.33±1.15 pCi/L for <sup>226</sup>Ra plus 2.67±0.75 pCi/L for <sup>228</sup>Ra vs. MCL of 6.0 pCi/L combined), although this is substantially less than the applicable State standard of 30 pCi/L for radium in groundwater;

<sup>&</sup>lt;sup>3</sup> USEPA primary MCL (includes both safety factors and lifetime exposure scenarios) and secondary MCL (addressing aesthetic quality) values.

<sup>&</sup>lt;sup>4</sup> State standards for groundwater are the New Mexico Water Quality Control Commission (NMWQCC) standards, which are applicable to domestic water supply.



- exceeds the secondary MCL for manganese (68.1 micrograms per liter (μg/L) vs. secondary MCL of 50 μg/L); there is no applicable State standard for manganese in groundwater;
- exceeds the secondary MCL for sulfate (270 milligram per liter (mg/L) vs. secondary MCL of 250 mg/L); although this is substantially less than the applicable State standard of 600 mg/L for sulfate in groundwater; and
- exceeds the secondary MCL for total dissolved solids (TDS) (709 mg/L vs. secondary MCL of 500 mg/L); although this is substantially less than the applicable State standard of 1,000 mg/L for TDS in groundwater.

In summary, GMD-04 exceeds secondary MCL values for manganese, sulfate, and TDS and the primary standard for gross alpha.

In comparison, the groundwater quality from well GMD-05 can be summarized as follows:

- a sodium-chloride water type;
- exceeds the secondary MCL and State standard for chloride in groundwater (1,500 mg/L vs. secondary MCL and State standard of 250 mg/L);
- exceeds the secondary MCL for sulfate (280 mg/L vs. secondary MCL of 250 mg/L), although this is substantially less than the State standard of 600 mg/L for sulfate in groundwater; and
- exceeds the secondary MCL and State standard for TDS in groundwater (3,070 mg/L vs. secondary MCL of 500 mg/L and State standard of 1,000 mg/L).

In summary, GMD-05 exceeds applicable secondary MCL and State groundwater-quality standards for chloride and TDS. As previously mentioned, this well does not produce sufficient rates of water flow for use as a domestic well.

As noted previously, well GMD-04 appears to be screened in the upper portion of the Dakota Sandstone (the Twowells Sandstone Tongue). The water quality of the Dakota Sandstone was characterized in the Marquez, New Mexico area by Daniel B. Stephens and Associates, Inc. (DBSA 2008), who provided the following description:

The Dakota Sandstone is a sodium-bicarbonate water type near recharge areas with increasing sulfate concentrations downgradient (Dam 1995). Water quality in the Dakota Sandstone is variable and generally acceptable for domestic, livestock, and industrial use (Dam 1995). In some areas the



groundwater has elevated TDS and sulfate concentrations that exceed standards (Table 4). Trace elements that were detected at concentrations above standards include iron and manganese (Table 5).

The TDS, sulfate, and manganese water quality exceedances reported for the Dakota Sandstone are consistent with the groundwater quality observed in GMD-04. Gross alpha and radium radioactivity were not reported by DBSA for the Dakota Sandstone; however, the Dakota Sandstone has been reported as a host for low grade U deposits in the Grants Mineral Belt (Green 1980). In fact, U concentrations in the Dakota Sandstone measured in the Johnny M Mine water well in January 1973, prior to the initial mine shaft reaching the ore zone, were 340  $\mu$ g/L, which would typically equate to a gross alpha of more than 200 pCi/L.

The chemistry of groundwater samples from GMD-05 is generally consistent with background groundwater quality in the Mancos Shale. Figure 10 illustrates the chloride and sulfate concentrations for groundwater samples from GMD-04 and GMD-05, the ranges (minimum, maximum, and geometric mean) observed in water samples from the Mancos Shale regionally (Environmental Sciences Laboratory 2011), and from groundwater, surface water (e.g., MW-1 in Table 2), and mine water collected in the Project Area. Note that the chloride concentrations (which, together with sodium comprise the majority of the dissolved constituents) in GMD-05 are higher than for any of the other waters in the Project Area and this well is located vertically and laterally upgradient of the former Johnny M Mine. Of the water-quality samples included in Figure 10, only groundwater from the Mancos Shale (regionally) has chloride concentrations as high as those observed in GMD-05. There is a lower proportion of sulfate relative to chloride observed in GMD-05 (in comparison with the Mancos-Shale trend), which could be an artifact of locally reducing conditions (that would also account for the low dissolved U and metals in water from this well), or could be a result of limited availability of deeper-water samples from the Mancos Shale (because groundwater wells are not typically completed in the Mancos Shale). However, it has been noted that groundwater from deep (greater than 27 m below ground surface) wells in the Mancos Shale have "a sodium chloride composition, in stark contrast to the sulfate-dominated water in shallow, more weathered horizons" (Morrison et al. 2012).



## 4.2.2 <u>Conclusions Regarding Potential Impacts to Intermediate Groundwater Quality</u>

Uranium concentrations in GMD-04 and GMD-05 are 3  $\mu$ g/L and <2  $\mu$ g/L, respectively, as presented in Section 4.0, which, as noted previously, are not indicative of mining-related impacts (NMED 2010).

The horizontal hydraulic gradients within the Dakota Sandstone and Morrison Formation are northeastward/eastward, away from the Project Area so that potential water-quality impacts within the Dakota Sandstone and lower units would migrate away from the former domestic wells in the Project Area. Lastly, approximately 600 ft of relatively impermeable shale (Mancos Shale) separates former surface operations from the screened interval of GMD-04. The groundwater quality observed in GMD-04 is consistent with naturally occurring conditions in the Dakota Sandstone and is not indicative of legacy U mining impacts.

Impacts from mine water cannot account for the groundwater quality observed in well GMD-05 because this well is upgradient of the Morrison Formation and Dakota Sandstone groundwaters in the Project Area, and water from this well has higher concentrations of chloride than any of the mine waters. Well GMD-05 appears to be representative of naturally occurring poor groundwater quality in a geologic unit of low transmissivity, most likely the Mancos Shale. The groundwater quality of these upgradient wells (former domestic wells) within the Project Area is unrelated to mining activity; therefore, the water-quality analysis from these wells should not be used for evaluating the question of whether shallow groundwater quality in the Project Area is impacted as a result of past mining activities.

#### 4.3 DEEP GROUNDWATER QUALITY

## 4.3.1 Deep Groundwater-Quality Observations

Water quality in the underground workings at the Johnny M Mine was monitored prior to and during backfilling with a sand slurry that started in August of 1977 and was completed sometime prior to cessation of mining activity in 1982.

Figure 12 illustrates the water quality of various groundwater, mine-water, and surface-water samples compiled from various sources (see also Table 2). The actual water-quality parameters analyzed differ somewhat between sampling events due to the differing objectives of the various



sampling events. The earliest data illustrated in Figure 12 are groundwater samples from the Dakota Sandstone during development of the initial mine shaft prior to any ore mining. There are numerous sampling data during backfill placement. In terms of subsequent monitoring, a sample was collected from the North Vent pipe in 1985, and the NMED conducted sampling of the upgradient former domestic wells in the Project Area. The North Vent pipe is a sampling point at the former ventilation shaft of the Johnny M Mine used to sample groundwater quality in the backfilled mine (Westwater Canyon Member). Additional sampling of the groundwater quality in the Morrison Formation that hosts the backfilled underground workings (e.g., wells 15 and 143) has recently been conducted as part of baseline water-quality evaluations being conducted for the proposed Roca Honda Project (Roca Honda Resources 2011). The North Vent pipe, well 15, and well 143 draw groundwater from the Westwater Canyon Member Canyon Member of the Morrison Formation.

As illustrated in Figure 12, the sand-slurry water (MSW-3) had elevated concentrations of various constituents (i.e., arsenic, nitrate, molybdenum, selenium, vanadium, gross alpha, radium, thorium, U, chloride, sulfate, and TDS) as compared to the other water samples. Water samples were collected 26 times at location MSW-3 from September 1977 to December 1978 (Table 2). Although the sand-slurry water contained notably elevated concentrations of some water-quality constituents, that water was removed from the mine after the backfill was deposited. Mine-water samples illustrated a much smaller and temporary increase in some constituents during and/or immediately following backfill placement, but the subsequent analyses of water quality within and near the mine (North Vent pipe, well 15, and well 143) all indicate that the groundwater quality in the Morrison Formation has since returned to background concentrations as represented by analyses of groundwater samples from the aforementioned locations. The concentrations of these constituents observed in the underground workings were much lower as a result of immobilization under the circumneutral and reducing conditions of ambient groundwater. On the other hand, the slurry water was initially oxidizing and in some cases mildly acidic. In addition, the slurry water was pumped from the mine, treated, and discharged.

The 1985 sample from the North Vent pipe indicated the following concentrations in groundwater at a depth within the backfilled mine: arsenic was 0.011 mg/L; molybdenum was 0.3 mg/L; selenium was <0.005 mg/L; vanadium was <0.1 mg/L; chloride was 11.9 mg/L; sulfate was 205 mg/L, TDS was 495 mg/L; and nitrate, gross alpha, radium, thorium, and U were not reported. The



water-quality parameters measured either met water-quality standards that existed at the time or reflected natural groundwater conditions.

In the most recent sampling event for well 143 (September 23, 2010), all constituents either meet USEPA public drinking water system standards or are similar to background concentrations. Specifically, arsenic, nitrate, selenium, molybdenum, and vanadium are all below limits of detection; gross alpha radiation is 6 pCi/L, radium (226 plus 228) is 4.9 pCi/L, thorium (230) is 0.5 pCi/L, and U is 3.2 µg/L, chloride is 18 mg/L, sulfate is 276 mg/L, and TDS is 737 mg/L. For comparison, the geometric mean of sulfate and TDS concentrations in the Westwater Canyon Member of the Morrison Formation (based on 48 samples from the nearby area) is 425 and 1,047 mg/L, respectively (Roca Honda Resources 2011).

Similarly, recent sampling events for well 15 indicated that all of these constituents meet USEPA public drinking water system standards. Specifically, arsenic, nitrate (with the exception of one sample reported at the detection limit of 0.1 mg/L), selenium, molybdenum, vanadium, and U are all below detection; gross alpha radiation is less than or equal to 3.4 pCi/L, radium (226 plus 228) is less than or equal to 1.62 pCi/L, thorium (230) is less than 0.2 pCi/L, chloride is less than or equal to 9 mg/L, sulfate is less than or equal to 181 mg/L, and TDS is less than or equal to 591 mg/L. Well 15 is located east of the former Johnny M Mine and, for deeper groundwater, could represent a downgradient sampling point from the mine.

## 4.3.2 <u>Conclusions Regarding Potential Impacts to Deep Groundwater Quality</u>

Backfilled sand that was placed into the Johnny M Mine are unlikely to impact deep groundwater quality because

- 1) the slurry water was removed from the mine immediately following placement of the backfill, and
- 2) the materials used (or considered for use) in backfilling operations in the Grants Mineral Belt (Thomson and Heggen 1982; Thomson et al. 1986), and at Johnny M Mine specifically (Gamble 1992), were largely devoid of the finer particles (e.g., clays) that carry the majority of the leachable/reactive metal content (Thomson and Heggen 1982; Thomson et al. 1986).



Removal of the fine material (e.g., clay) was an important consideration in the use of the sand for placement. Sand was used as backfill material because it was easier to transport and to handle within the mine. Also, the use of sand as backfill was driven by safety concerns, i.e., 'unsized tailings' material would not drain properly and could cause a potentially dangerous 'muck rush' condition within the mine. The finer particles (e.g., clays), less than 200 mesh, contained the majority of the leachable reactive load (Thomson et al. 1986), and this material was retained at the Ambrosia Lake mill and tailings facility. These facilities were not located in the Project Area. Thus, the removal of the fines (e.g., clay) substantially reduced the amount of leachable constituents, including metals, the metalloids such as arsenic (As) and selenium (Se), and sulfate (SO<sub>4</sub>). Analyses described in Thomson et al. (1986), and summarized in Thomson's Tables 3 and 4 (see below), indicate the difference in compositions for the sand and clay fractions from undisclosed operations in the Grants Mineral Belt. The tables show that large enrichment factors are present, with the fine (e.g., clay) fraction always showing enrichment relative to the sand fraction.

		,		• '
Element	Sand (ppm)	Clay (ppm)	Water* (mg/l)	Enrichment Clay/Sand
Al	35633.3	66800.0	3.3	1.9
As	3.16	18.47	0.12	5.8
Ba	695.8	945.8	0.0	1.4
Ca	2362.2	32887.8	323.9	13.9
Cr	10.4	317.9	0.2	30.7
Fe	2081.6	25994.4	7.8	12.5
ĸ	22366.7	25855.6	228.5	1.2
Мg	0.0	3173.3	0.0	17.5
Mõ	10.2	178.6	0.3	17.5
Se	8.24	80.31	0.90	9.8
Гh-223	1.828	7.513	0.002	4.1
U-239	29.27	226.41	0.70	7.7
V	82.4	928.8	0.6	11.3

Table 3. Average concentrations in each fraction of backfilled sands, determined by INAA (9 samples)

Table 4. Average concentrations in each fraction of acid-
leach uranium, mill tailings by INAA (2 samples)

Element	Sand (ppm)	Clay (ppm)	Water* (mg/l)	Enrichment Clay/Sand
Al	37850.0	60050.0	0.0	1.6
As	2.51	20.44	0.17	8.2
Ba	778.5	642.0	0.0	0.8
Ca	2775.0	41150.0	470.1	14.8
Cr	9.88	157.30	0.0	15.9
Fe	2387.5	30370.00	11.5	12.7
К	23 500.0	25400.0	64.9	1.1
Mg	0.0	2650.0	0.0	
Mo	5.39	216.00	0.0	40.1
Se	8.17	124.10	0.13	15.2
Th-223	1.172	2.488	0.0	2.1
U-239	19.04	118.35	0.10	6.2
V	169.45	954.00	0.69	5.6

 Concentrations in 250 ml of solution containing 100 g of backfilled sands. \* Concentrations in 250 ml of solution containing 100 g of acid-leach uranium mill tailings.

Note: Tables 3 and 4 were copied directly from the original peer-reviewed paper. However, the Th-223 values reported may actually represent Th-233, which is derived from neutron activation of Th-232 (used in instrumental neutron activation analysis [INAA]).

In addition to the fact that sand, rather than fines, was utilized in backfilling, the geochemical conditions in the backfilled mine act to limit solubility and thus the potential for metals migration.



Over the last 15 to 20 years, subaqueous disposal of tailings has been employed at numerous mining operations to limit the formation of acid-rock drainage and the subsequent leaching of metals. The mitigation of acid and metals leaching by subaqueous disposal is due to the slow rate of diffusion of oxygen through water relative to air; a water cover is used primarily to halt pyrite oxidation and subsequent acid-rock drainage in the near surface (MEND 2001). Molecular oxygen is the primary driver for oxidation reactions involving pyrite (FeS<sub>2</sub>). Upon the cessation of dewatering, groundwater collecting in the Johnny M Mine would behave similarly, with backfilled sand and other minerals in the mine environment stabilized by the reducing conditions. Limited access of oxygen and the presence of organic matter (mainly humic materials present in the Westwater Canyon Member), which would consume any small amounts of residual oxygen, produces a reducing environment (low Eh) that would stabilize U and other constituents as mineral solids, immobilizing them in the deep groundwater system.

For example, the insoluble minerals uraninite and coffinite are stable under reducing conditions (Figure 9), where the presence of electrons donors, such as humic substances, result in low Eh values. Furthermore, the  $U^{+4}$  ion that is predominant in these conditions does not have a strong tendency to form aqueous complexes that could increase the concentration of U in solution; rather, the  $U^{+4}$  ion tends to form mineral precipitates such as uraninite or coffinite.

The geochemical conditions in the mine after mining was completed and the mine became resaturated are expected to have returned to conditions similar to those that were present prior to mining (i.e., reducing conditions that were responsible for the precipitation of the U(IV) minerals that formed the original ore deposit). These reducing conditions have re-stabilized elements, such as U, that were associated with the ore deposit or backfilled sand. The Johnny M deposit formed under reducing conditions in sediments that were rich in humic materials (derived from plant matter) that allowed for the precipitation of U, which was introduced by periodic volcanic episodes (Falkowski 1980). Thus the distribution of U was influenced directly by the volcanic episodes. It is expected that without exposure to atmospheric oxygen, humic material and other organic matter still present in the geologic materials near the mine, together with reduced minerals such as authigenic pyrite (FeS<sub>2</sub>) and jordisite (MoS<sub>2</sub>), will continue to support a low Eh environment in and near the mine workings. Dissolved U will precipitate either as coffinite or as uraninite, thus limiting both concentrations and mobility in groundwater.



Redox conditions that limit U mobility are consistent with studies in the general area and with groundwater quality observations in the Project Area. Thomson and Heggen (1982) discuss several redox related processes and estimate the native groundwater to be within a pH range of 6 to 8, with a maximum Eh of approximately 0.17 volts (at a pH of 6), and a minimum Eh of approximately -0.12 volts (pH 8). In Thomson et al. (1986) the authors showed an ore zone region that suggested even lower Eh conditions are possible. These ore zone conditions have Eh values as low as -0.28 volts. These lower conditions are near the boundary between 'organic' carbon and inorganic carbon. This region is also within the boundary between sulfide and sulfate. In addition to the abundant humic materials, the ore zones also contain authigenic (formed in place) pyrite (Falkowski 1980), which will drive redox conditions toward an ore-formation or pre-mining Eh. As discussed below, these organic and sulfide rich, low Eh conditions in the backfilled and saturated Johnny M Mine are apparent from the notable decreases in concentrations of elements such as arsenic, radium, selenium, thorium, U, and vanadium that have occurred subsequent to cessation of dewatering activities in 1982 (Figure 12). The assumptions used by Thomson et al. (1986) to define the Eh-pH region of the Morrison Formation subsequent to mining appear to be a reasonable and appropriate representation for groundwater in the ore zones in the Project Area. Accordingly, under reducing conditions, such as in the deep groundwater in the Dakota Sandstone and Morrison Formation at the Johnny M Mine, U is relatively immobile and has low aqueous concentrations.

In summary, there is no indication that the mine activities have negatively affected groundwater quality in or downgradient of the underground workings at the Johnny M Mine. Water quality at Johnny M Mine and other backfilled underground U mines in the Grants Mineral Belt are similar to their mine water quality prior to backfilling. At the Johnny M Mine, the nearby groundwater wells that provided water samples from depths similar to the mine workings (e.g., wells 15 and 143, and the North Vent pipe) have solute concentrations similar to background conditions. These results are consistent with expected U geochemistry; U is mobile in surficial, oxidizing conditions, but is immobile in the reducing conditions present within the Dakota Sandstone and Morrison Formation in the Project Area.



#### 5.0 SUMMARY AND CONCLUSIONS

Itasca was asked by Hecla Limited to analyze existing hydrogeologic data relevant to conditions at the former Johnny M uranium mine that is located in McKinley County, in northwestern New Mexico. The former mine is located within a historic uranium mining district referenced as the Ambrosia Lake uranium mining district. Mining of ore occurred at the Johnny M Mine from approximately 1976 until 1982 when operations ceased.

Hecla Limited had the following questions that Itasca was to address in its analysis:

- Question 1: Is shallow groundwater quality in the Project Area affected by mine water during operations or leaching from mined materials?
- Question 2: Has the groundwater quality of the former domestic wells in the Project Area (GMD-04 and/or GMD-05) or other groundwater resources been affected by mining-related activity in the Project Area?
- Question 3: Is the quality of groundwater in the Project Area affected by the presence of backfilled tailing sand in the underground workings?

Itasca reviewed the analyses of mine-water, groundwater, and surface-water samples collected during and immediately after mining, as well as samples collected recently by NMED and other contractors working in the nearby area. In addition, a significant amount of data regarding the geology, hydrogeology, and geochemistry of the area surrounding the former Johnny M Mine are available from the Baseline Data Report and other reports generated by or for the proposed Roca Honda Mine that is located approximately one mile east of the former Johnny M Mine.

Based upon Itasca's review and analysis of the existing data, Itasca offers the following conclusions.

#### Answer to Question 1

Shallow groundwater quality was measured historically in three former groundwater monitoring wells that were located downgradient of the mine-water discharge pathway through the ditch and upgradient of the San Mateo Creek drainage channel. The shallow groundwater monitoring wells were located and screened to collect groundwater samples from the contact between the surficial sediments and the top (weathered surface) of the Mancos Shale. Weathered zones of the Mancos



Shale have been noted as being naturally affected by geochemical processes, including pyrite oxidation, carbonate dissolution, gypsum precipitation, release of nitrate from weathering of organic material, and solubilization of U. Consistent with these processes, the sulfate concentrations observed in the shallow groundwater monitoring wells (GW7, GW8A, and GW9) were actually higher than the concentrations observed in sand slurry water, dewatering discharge or any of the water-quality samples collected from within the underground mine. The shallow groundwater well samples generally cluster around the U and nitrate geometric mean for the Mancos Shale, but with slightly lower U concentrations. In contrast, U concentrations in discharged (treated) mine waters were typically an order of magnitude higher than those observed in the shallow groundwater monitoring wells, and the nitrate concentrations were typically one to three orders of magnitude lower in discharged (treated) mine waters than in the shallow groundwater monitoring wells.

In addition to having groundwater quality that was substantially poorer due to natural conditions (e.g., higher concentrations of sulfate, nitrate, chloride, and TDS) than that associated with mine activities, the shallow groundwater system is transient. During infiltration events, surface water will infiltrate downward and perch on the bedrock (Mancos Shale) surface temporarily. However, the surficial sediments are typically unsaturated and the Mancos Shale is an aquitard. There is no indication that any water that infiltrated the shallow surficial sediments and temporarily ponded on the surface of the Mancos Shale as a result of mining activities more than 25 years ago is contributing seepage to the San Mateo Creek drainage channel or to underlying water-bearing units today. Subsequent overland runoff over the past 25 years would have concentrated in the ditch and tended to have 'flushed' any surficial sediments. Given that the runoff waters would probably be rich in dissolved oxygen, this oxygenated water would have mobilized U, or 'flushed' any U from the surficial sediments.

Water chemistry measured in a shallow groundwater well located on the north side of the San Mateo Creek drainage channel and downgradient of the former Johnny M Mine discharge location (Marcus Ranch well) indicated that the water-bearing surficial sediments have not been impacted by the historical discharges from the mine or by the current conditions within the Project Area. Concentrations of U, <sup>226</sup>Ra, arsenic, lead, molybdenum, selenium, and vanadium were either below detection limits or below drinking water standards. Whereas the other radionuclides (alpha and gross beta) had large errors surrounding the measured concentrations, the reported



concentrations do not indicate impacts, particularly when considered together with the low concentrations of U, <sup>226</sup> Ra, and other metals typically associated with mine water.

#### Answer to Question 2

The former domestic wells in the Project Area are located upgradient of the former Johnny M Mine (and associated mining activities) and are screened at intermediate depths, either within an upper interbed of the Dakota Sandstone Formation (GMD-04) or within the Mancos Shale (GMD-05). The quality of the groundwater samples collected from these wells is reflective of natural background conditions encountered in these two formations. The groundwater quality observed in well GMD-04 is consistent with naturally occurring conditions in the Dakota Sandstone and is inconsistent with legacy U mining impacts. The groundwater quality observed in well GMD-05 appears to be representative of naturally occurring poor groundwater quality in a low transmissivity geologic unit, most likely the Mancos Shale. Impacts from mine water cannot account for the groundwater quality observed in wells GMD-05 as this well is upgradient of the mine (both vertically and laterally), and the water in this well has higher chloride concentrations than any of the mine waters.

These wells are also hydraulically separated from shallow groundwater that may have been impacted in the past from mine water discharges to the land surface via a ditch or from a pipe by more than 600 ft of Mancos Shale that has a hydraulic conductivity of approximately 5 x  $10^{-8}$  cm/sec, a value lower than compacted clay liners used in landfills.

Other geologic units used as potential groundwater resources such as the Menefee Formation and Point Lookout Sandstone do not exist in the Project Area because they have been eroded. The nearest wells that are screened in these two formations are more than four miles to the northeast which is downgradient of the Project Area, due to the regional dip of the geologic units. The potentiometric surface of the Morrison Formation is estimated to be more than 1,200 ft below a well screened in the Menefee Formation. Given this large vertical separation and the fact that there is a downward gradient, water quality in the Morrison Formation or the Dakota Sandstone is not expected to impact these shallower geologic units.

The regional dip of the geologic units and the groundwater flow direction within the Dakota Sandstone and the Morrison Formation are towards the northeast. There are no domestic or stock



wells completed in these formations to the northeast of the mine. These two formations are not used for groundwater supply northeast of the mine because the depths of these formations increase due to the regional dip, thus making drilling to these units and pumping groundwater uneconomical.

In summary, there is no evidence that the former mine is currently having an impact on groundwater quality—either at the upgradient former domestic wells in the Project Area or elsewhere.

#### Answer to Question 3

Subaqueous disposal of tailings has been employed at numerous mining operations to limit the formation of acid-rock drainage and the subsequent leaching of metals. This is because of the slow rate of diffusion of oxygen through water. A water cover is used primarily to halt pyrite oxidation and subsequent acid-rock drainage in the near surface. Molecular oxygen is the primary driver for oxidation reactions involving pyrite (FeS<sub>2</sub>). Similarly, limiting oxygen to U(IV) bearing minerals such as coffinite (USiO<sub>4</sub>) and uraninite [UO<sub>2</sub>(cr) or UO<sub>2</sub>(a)] will also hinder their dissolution. Upon the cessation of dewatering, the Johnny M Mine would behave similarly to a saturated tailings deposit. Limited access of oxygen and the presence of organic matter (mainly humic materials present in the Westwater Canyon Member) would consume any small amounts of residual oxygen and produce a reducing environment (low Eh) that would stabilize and immobilize U and other constituents as mineral solids.

The overall water quality in the underground workings at the Johnny M Mine was monitored prior to and during backfilling with a sand slurry (and subsequent removal and treatment of the slurry water), that started in August of 1977 and was completed prior to cessation of mining activity in 1982. Subsequently, a sample was collected from the North Vent pipe in 1985 and some additional sampling of the groundwater quality in the Morrison Formation that hosts the backfilled underground workings has recently been conducted as part of the baseline water-quality evaluations being conducted for the proposed Roca Honda Project. The tailings slurry water had elevated concentrations of various constituents (i.e., arsenic, nitrate, molybdenum, selenium, vanadium, gross alpha, radium, thorium, U, chloride, sulfate, and TDS). However, the concentrations of these constituents observed in the underground workings were much lower as a



result of immobilization under reducing geochemical conditions (the slurry water was initially oxidizing and in some cases mildly acidic), and because the slurry water was pumped from the mine, treated, and discharged. In the most recent sampling event for well 143, a well screened at a depth approximately coincident with the ore zone, all constituents either meet USEPA public drinking water system standards or are similar to background concentrations.

The historical data, coupled with the voluminous amount of data collected by the proposed Roca Honda project, allows for the assessment of probable impacts to surface and groundwater quality from the former Johnny M Mine.

Itasca is of the opinion that additional investigation of groundwater and surface-water quality in the vicinity of the Project Area is not technically warranted, as sufficient information exists to assess the probable impacts to surface and groundwater quality from the former Johnny M Mine, as discussed in this report.

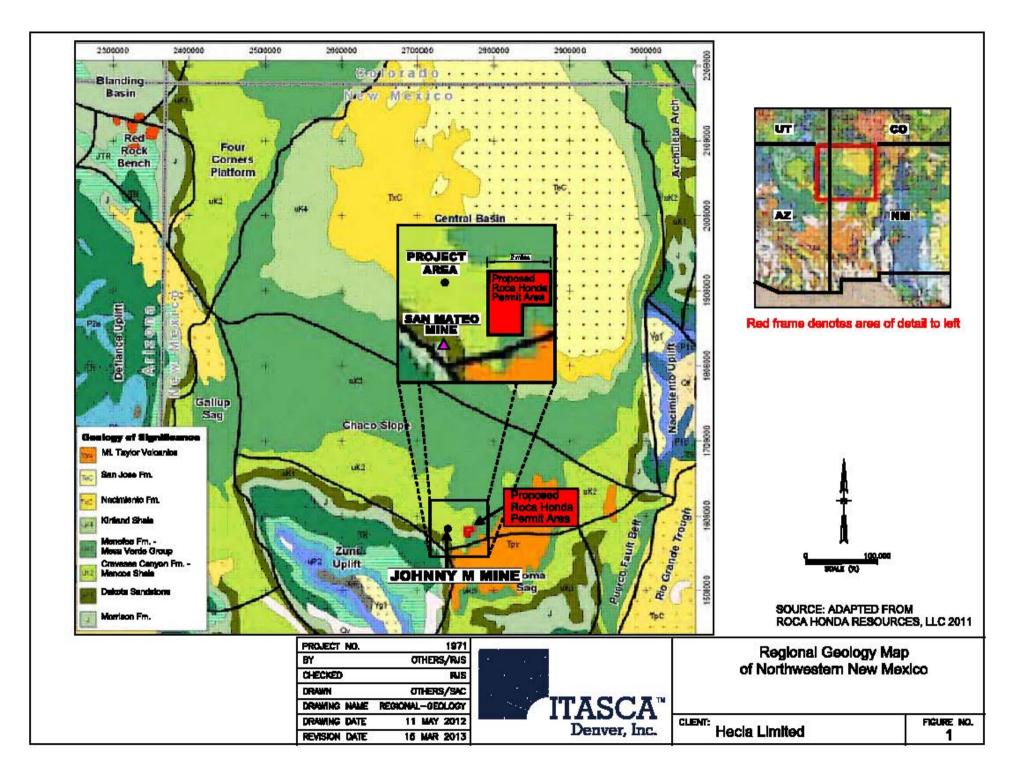


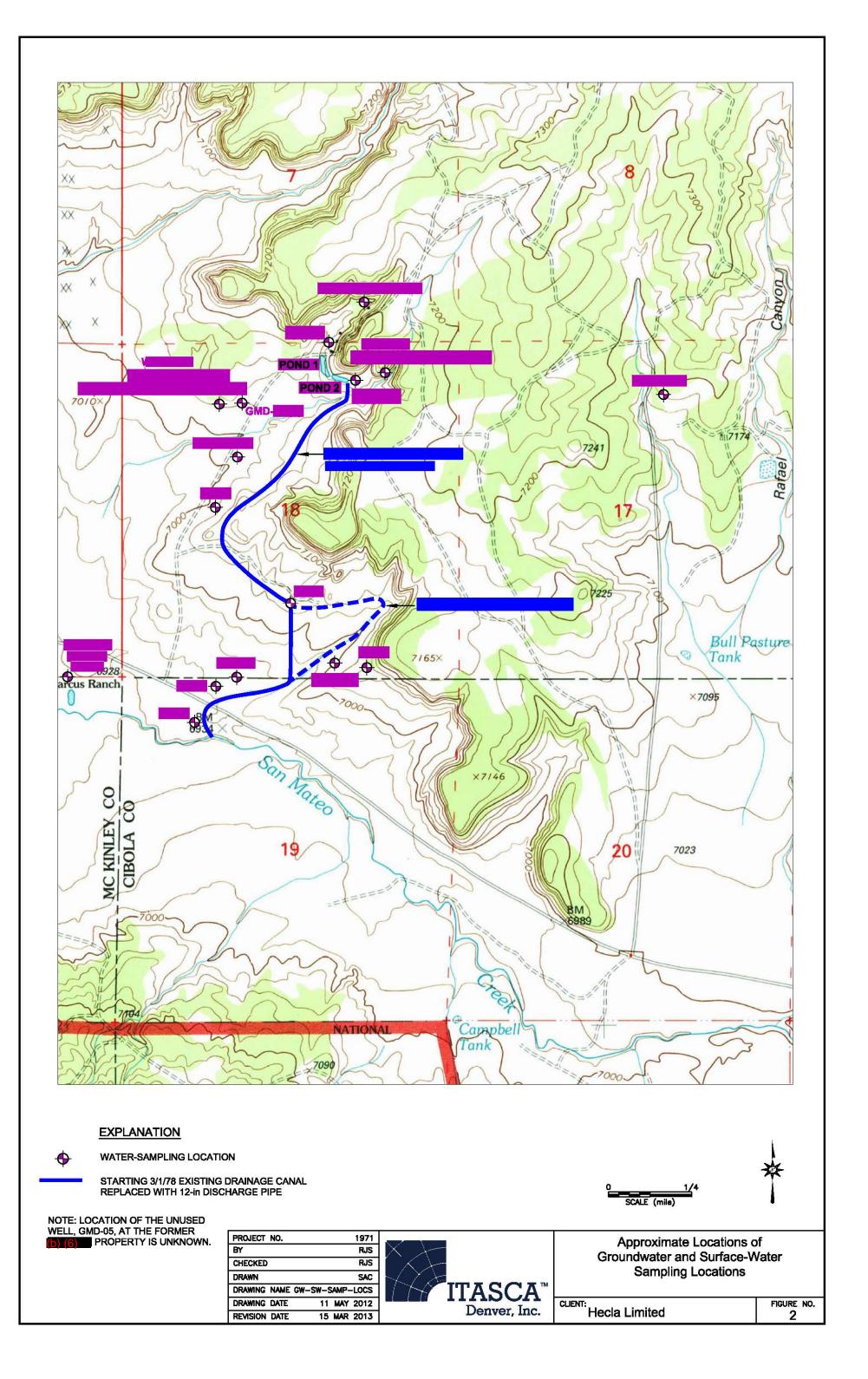
#### 6.0 <u>REFERENCES</u>

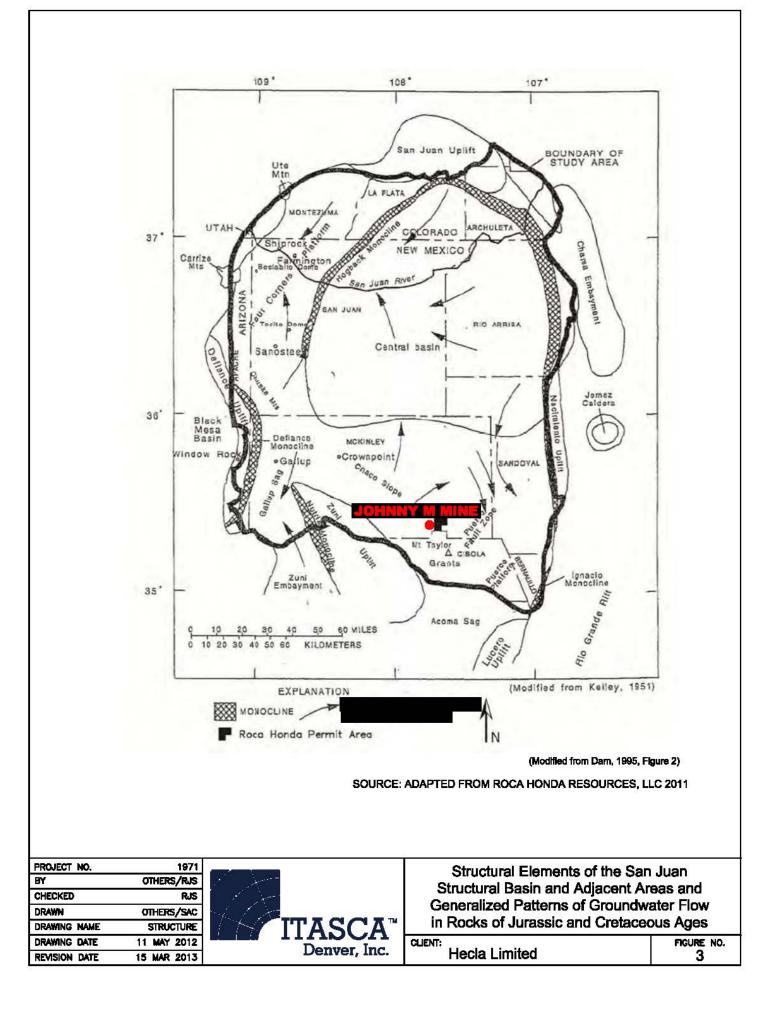
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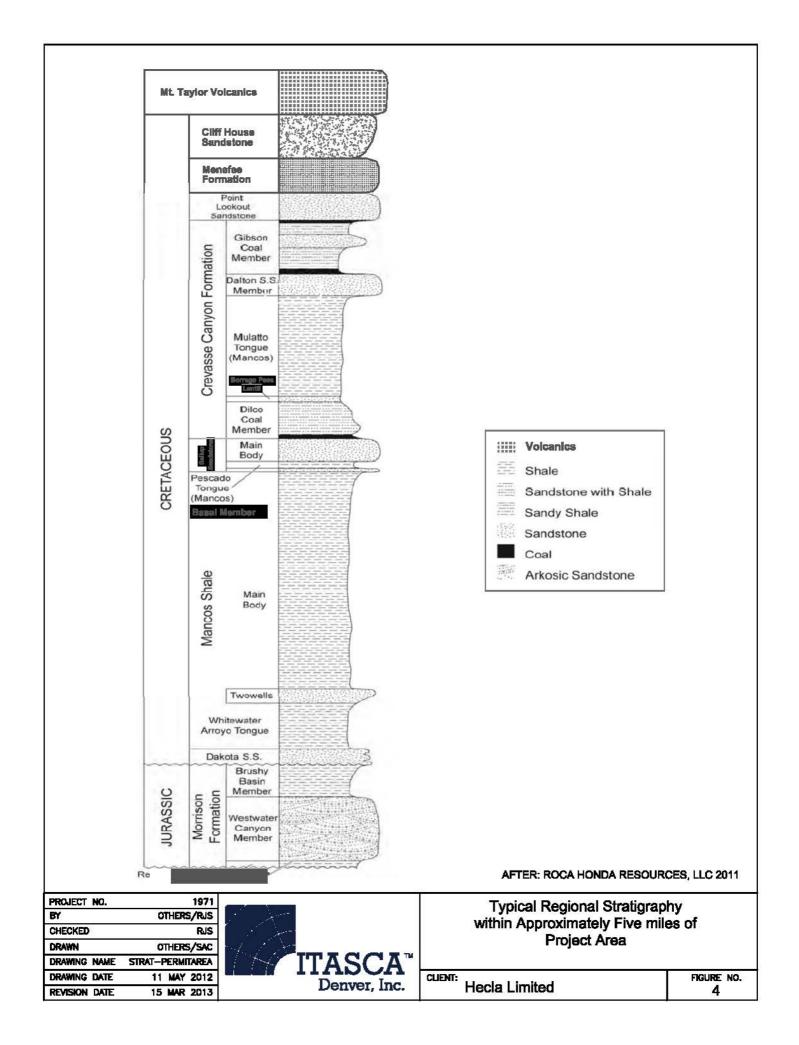


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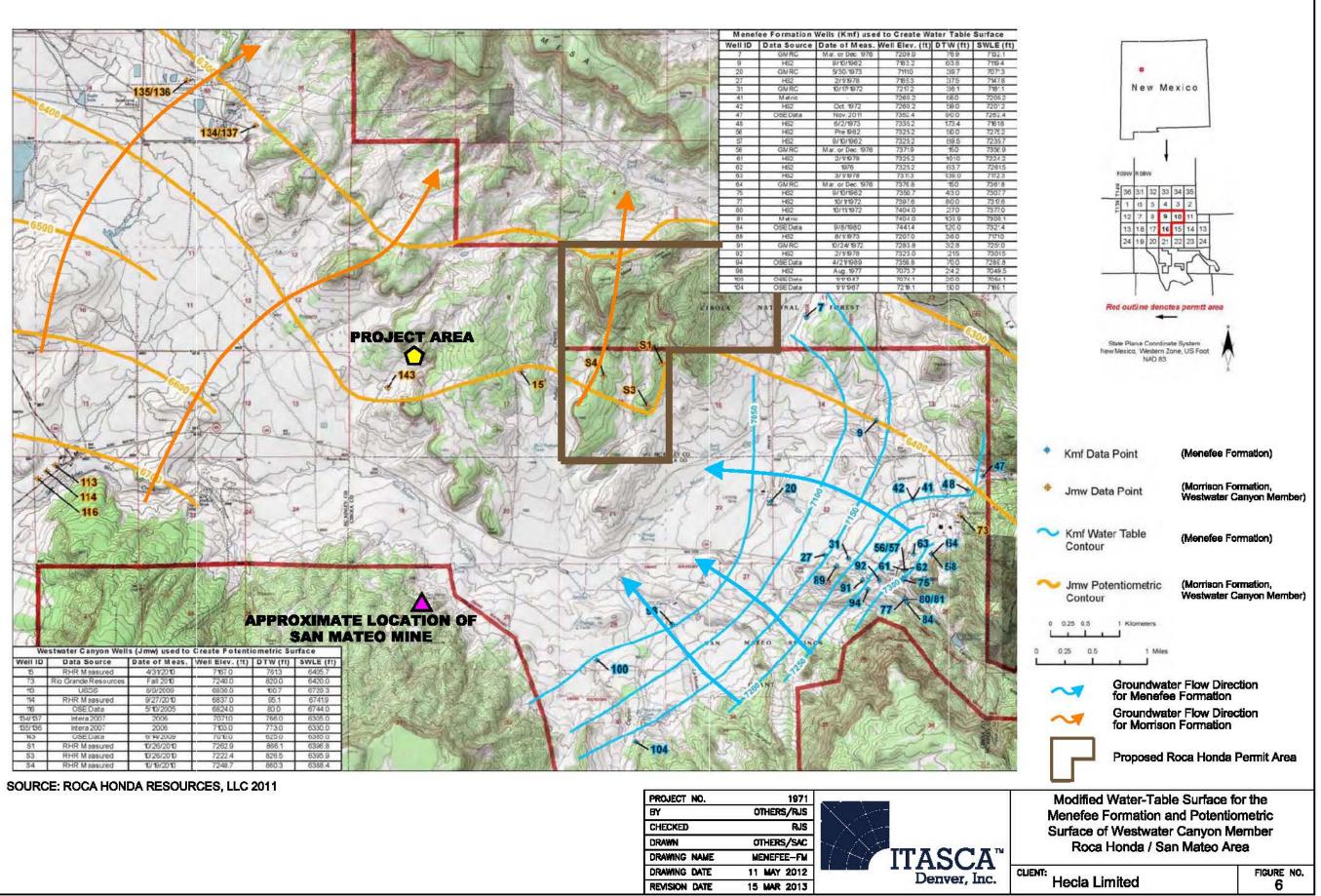




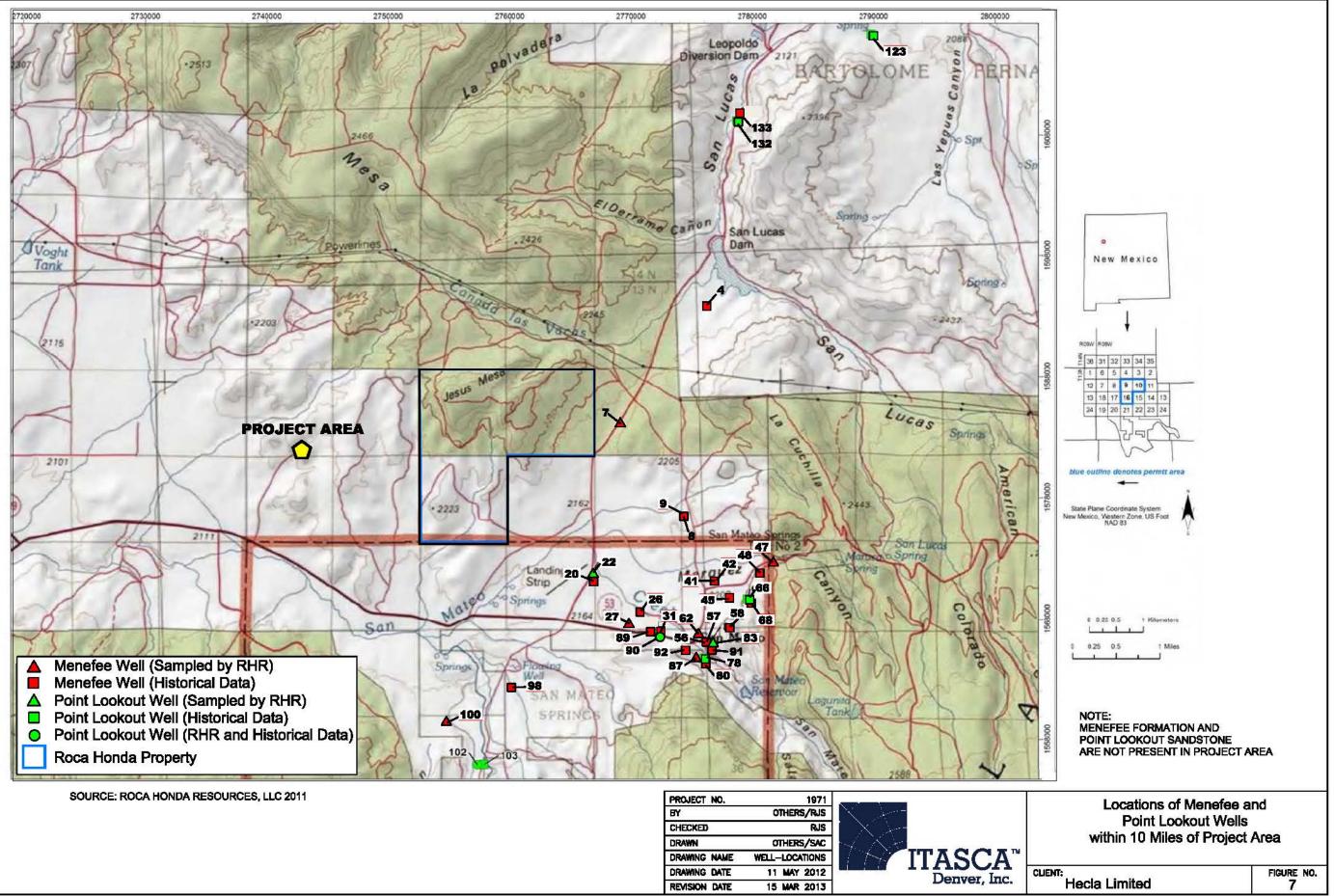


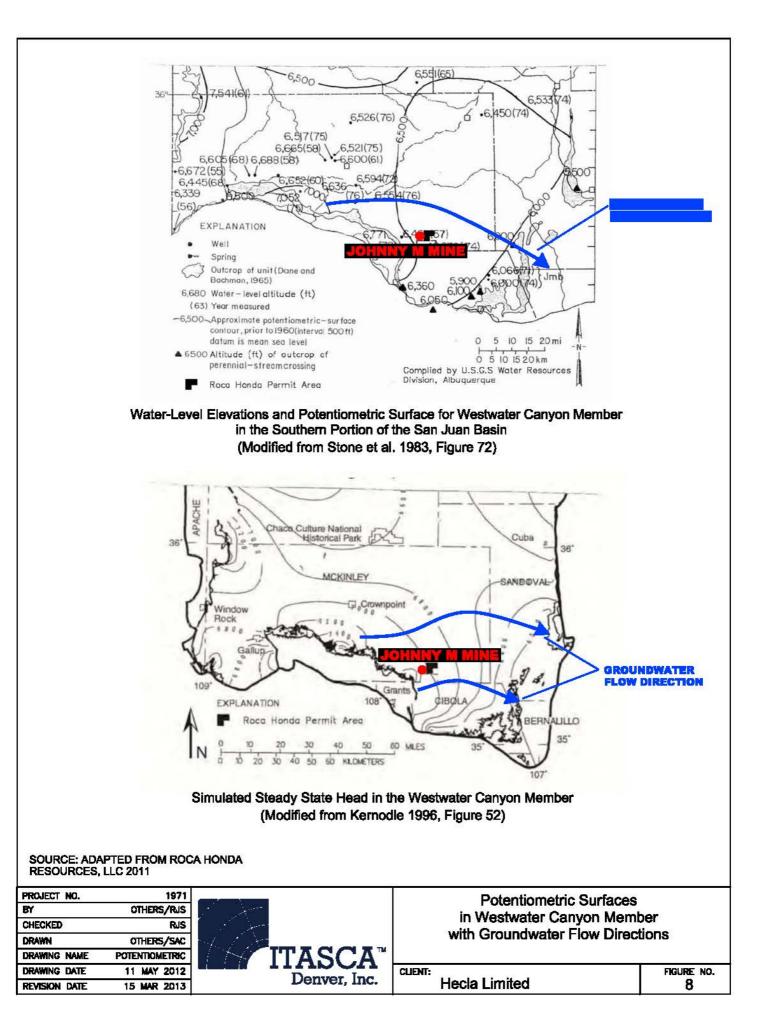


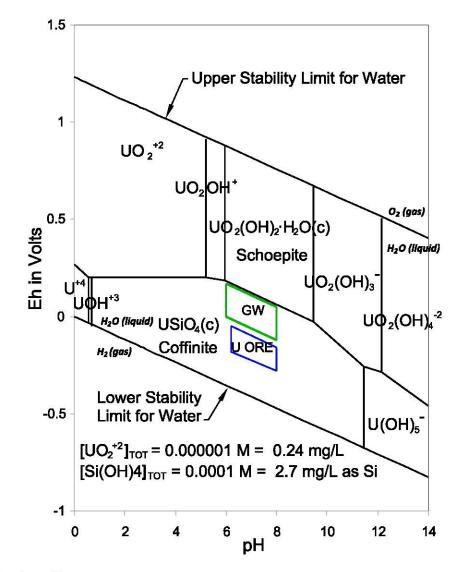
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			RA	NGE OF	GW8 GW8A						UG	5, UG6, 2, DN1	
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	DAKO	TA SANDSTONE (MAIN BODY)	2	20-80									
	BRUS	HY BASIN MEMBER	3	0-190									
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### Explanation

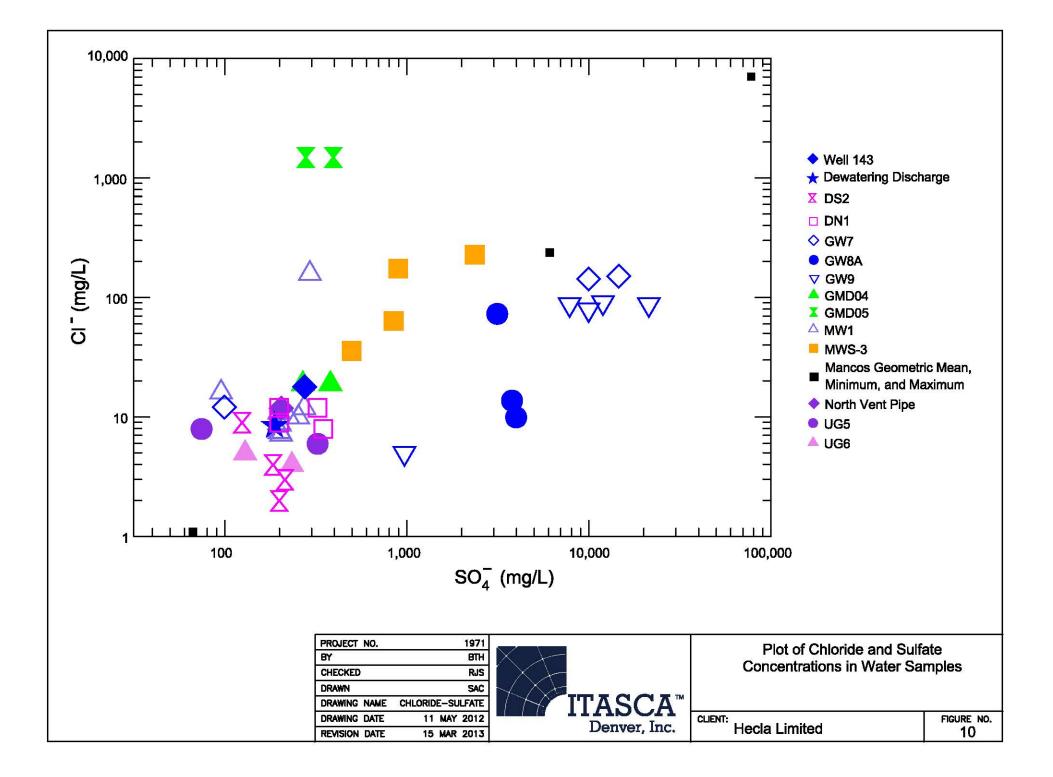


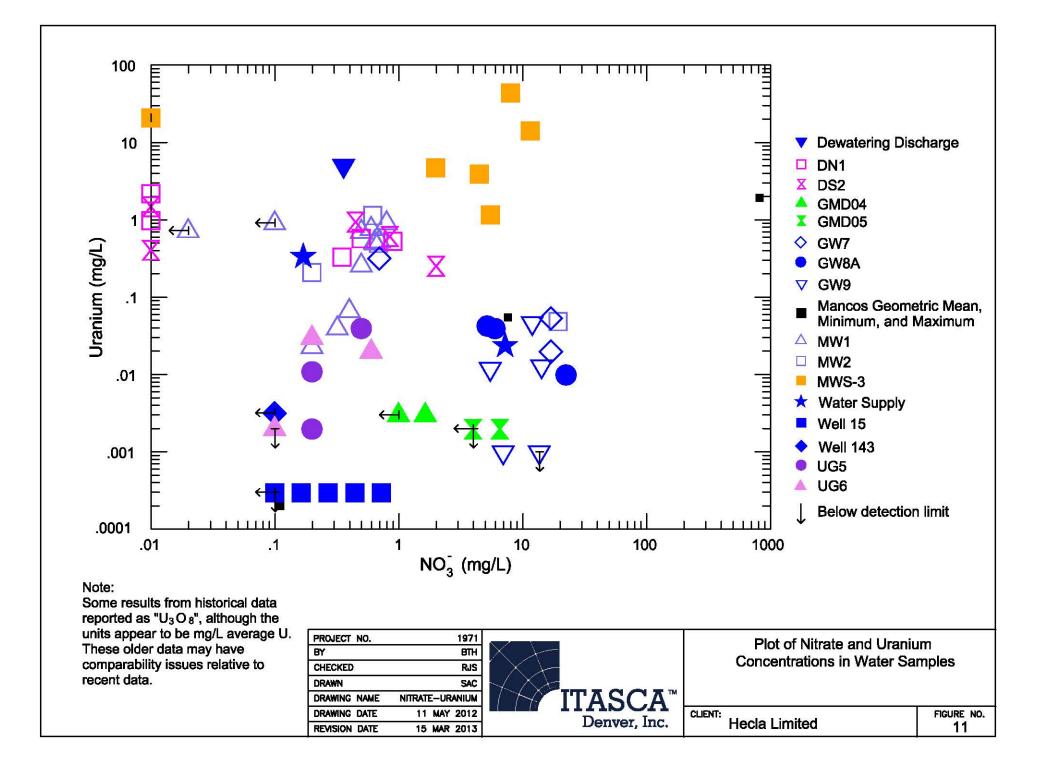
Approximate Range of Regional Groundwater Discussed by Thomson et al. (1986)

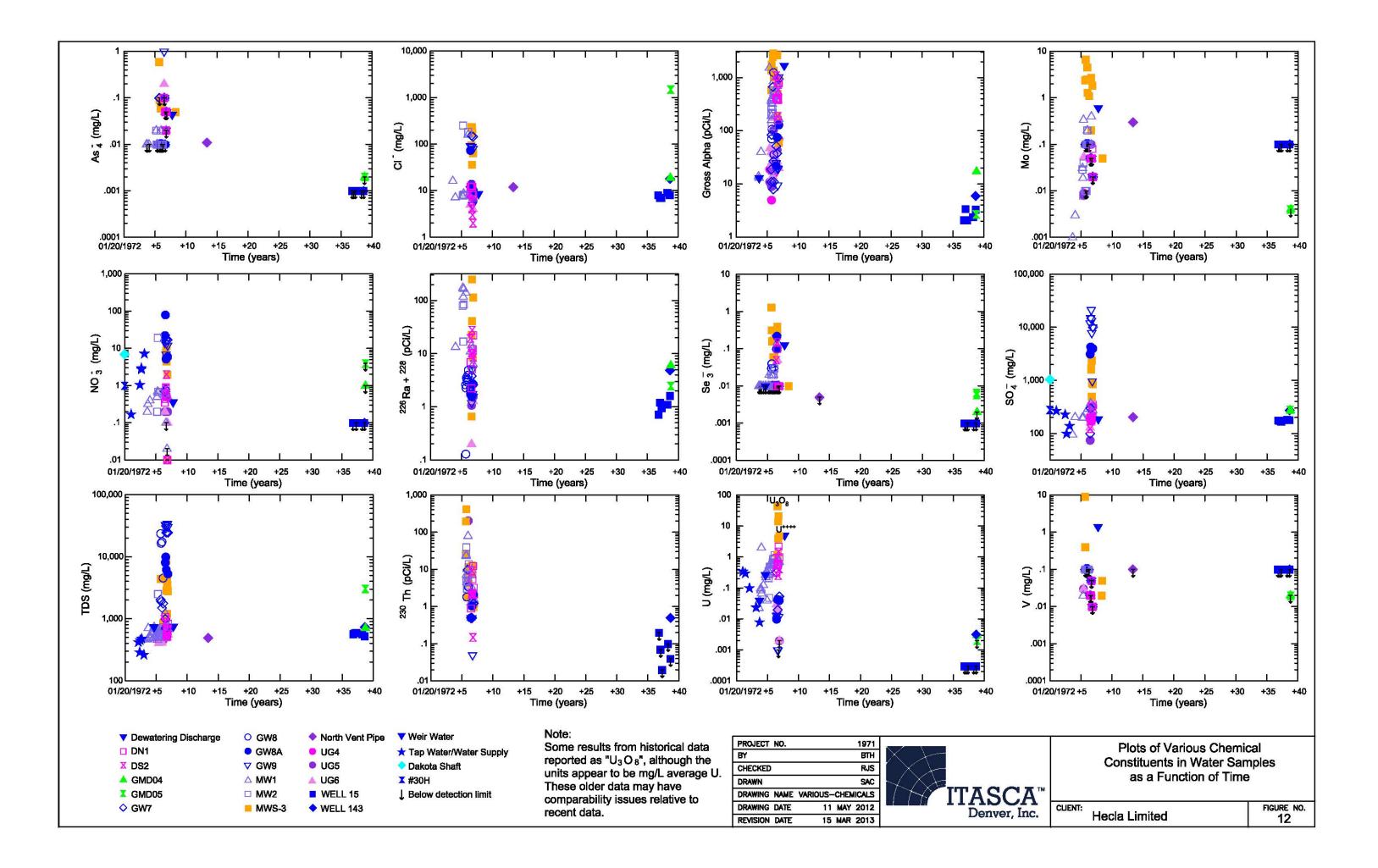
Approximate Range of Groundwater within Ore Zones by Thomson et al. (1986)

Note: The stability field for uraninite occupies a similar but smaller stability range to that illustrated for coffinite

PROJECT NO.	1971		Eh - pH Diagram for Uraniu	Im
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# TABLE 1 Water-Sampling Location Summary

Location ID	Sample Location Type	Formation	Notes
Well 15	Groundwater	Westwater Canyon Member (Morrison Formation)	
Well 143	Groundwater	Westwater Canyon Member (Morrison Formation)	
#30H	?	?	Not used in analysis; data from Johnny M Mine records; location unknown
Dakota Shaft	Groundwater	Dakota Sandstone	
Discharged Water	Mine Water - Discharged	Not Applicable	Mine dewatering samples, collection point unknown
DN-1	Mine Water	Westwater Canyon Member (Morrison Formation)	Mine drainage ditch on north side of shaft prior to intersection of main underground sump
DS-2	Mine Water	Westwater Canyon Member (Morrison Formation)	Mine drainage ditch on south side of shaft prior to intersection of main underground sump
GMD-00	?	?	Not used in analysis; data from NMED/EPA Grants Mining District sampling; location unknown
GMD-01	?	?	Not used in analysis; data from NMED/EPA Grants Mining District sampling; location unknown
GMD-02	?	?	Not used in analysis; data from NMED/EPA Grants Mining District sampling; location unknown
GMD-03	?	?	Not used in analysis; data from NMED/EPA Grants Mining District sampling; location unknown
GMD-04 (Well 17)	Groundwater	Twowells Sandstone Tongue (Dakota Sandstone Interbed within Mancos Shale)	Former domestic well in Project Area; recently used as residential well
GMD-05	Groundwater	Sandstone Lens within Mancos Shale ?	Former domestic well in Project Area; recently unused
GW-7	Groundwater	Alluvium/Mancos Shale Contact	Monitoring well near mine discharge canal
GW-8	Groundwater	Alluvium/Mancos Shale Contact	Monitoring well near mine discharge canal
GW-8A	Groundwater	Alluvium/Mancos Shale Contact	Monitoring well near mine discharge canal
GW-9	Groundwater	Alluvium/Mancos Shale Contact	Monitoring well south of Tailings Pond #2
Mancos Geometric Mean	Groundwater	Mancos Shale	Data from Environmental Sciences Laboratory, 2011
Mancos Maximum	Groundwater	Mancos Shale	Data from Environmental Sciences Laboratory, 2011
Mancos Minimum	Groundwater	Mancos Shale	Data from Environmental Sciences Laboratory, 2011
Marcus Ranch Well	Groundwater	Alluvium	Data from Science Applications International Corporation, 1994
MW-1	Surface Water	Not Applicable	Sampling location at discharge of second of two settling ponds
MW-2	Surface Water	Not Applicable	Sampling location at discharge of drainage canal prior to entry to San Mateo Creek
MWS-3	Tailings Slurry Decant	Not Applicable	Water used to slurry backfill sands into mine
North Vent Pipe	Groundwater	Westwater Canyon Member (Morrison Formation)	Pipe inserted in the north vent pipe shaft to sample backfilled mine water
Tap Water	Groundwater	Dakota Sandstone	Water from Johnny M Mine potable well, collected at the tap
UG-4	Mine Water	Westwater Canyon Member (Morrison Formation)	Sampling location midway between north and south ore bodies
UG-5	Mine Water	Westwater Canyon Member (Morrison Formation)	Sampling location in northrern ore body
UG-6	Mine Water	Westwater Canyon Member (Morrison Formation)	Sampling location in southern ore body
Water Supply (WW)	Groundwater	Dakota Sandstone	Water from Johnny M Mine potable well, collection point unknown
Weir Water	Mine Water - Discharged	Not Applicable	Mine dewatering samples at discharge weir

# TABLE 2 Water-Quality Results Compilation (page 1 of 3)

Location ID	Sample	Arsenic	Selenium	TDS	Molybdenum	Vanadium	Radium 226	Radium 228	Radium (226 & 228)	Thorium 230	Lead 210	Lead	Gross Alpha	Zinc	Uranium	Uranium-234	Uranium-235	Uranium-238	Calcium	Magnesium	Potassium	Sodium	Chloride	рН	Sulfate Alk	linity U3O	8 Nitrate (NO <sub>3</sub> as N)
	Date	mg/l	mg/l	mg/l	mg/l	mg/l	pCi/L	pCi/L	pCi/L	pCi/L	pCi/L	mg/l	pCi/L	mg/l	mg/l	pCi/L	pCi/L	pCi/L	mg/l	mg/l	mg/l	mg/I	mg/l	log[H]	mg/l	mg/l	l mg/l
15	11/13/2008 2/12/2009	< 0.001 < 0.001	< 0.001 < 0.001	567 588	< 0.1 < 0.1	< 0.1 < 0.1	0.25 +or - 0.13 0.43 +or - 0.18	0.47 +or - 0.73 0.78 +or - 0.73		< 0.2 +or - 0.10 < 0.07 +or - 0.08		< 0.05 < 0.05	2.1 +or - 1.6 3.4 +or - 1.6	0.1	< 0.0003				51 46	15 13	4 3	143 127	8	7.9 7.52		26 18	< 0.1
15	5/18/2009 4/28/2010	< 0.001	< 0.001	591	< 0.1	< 0.1	<0.1 +or - 0.11	0.85 +or - 0.8		< 0.02 +or - 0.10 < 0.1 +or - 0.10		< 0.05	2.1 +or - 2 2.4 +or - 2.3	0.08	< 0.0003				48 47	14	4	136	7	7.81	169	31 47	< 0.1
15 15	9/20/2010	< 0.001 < 0.001	< 0.001 < 0.001	568 523	< 0.1 < 0.1	< 0.1 < 0.1	0.5 +or - 0.17 0.32 +or - 0.15	0.61 +or - 0.72 1.3 +or - 0.59		< 0.04 +or - 0.10		< 0.05 < 0.05	3.3 +or - 3.3	0.08	< 0.0003 < 0.0003				45	13 12	4	135 139	8	7.63 7.85	180	56	< 0.1 < 0.1
143	9/23/2010 26318	< nd	< nd	737	< nd	< nd	3.2	1.7 +or -		0.5		< nd	6	0.4	0.0032				105	21	4	104	18	7.89		95	< nd
#30H Dakota Shaft	26318																		122 5.2	23	6.8	160 600			280 1050	nil	-1
Discharged Water	Unknown		< 0.01	737			65.5 +or - 1.3							0.1	0.266												
DN-1 DN-1	7/18/1978 8/13/1978	< 0.1	< 0.01	1076	< 0.05	< 0.02			7.06 +or - 0.32	0.9 +or - 0.8	7.6 +or - 4.2	< 0.02	416 +or - 21	0.15									12	7.39	200	0.33	
DN-1	8/14/1978	-	-	-	-				-	-	-	-	-	-									-	-	-	-	-
DN-1	9/13/1978	< 0.05	0.1	528	0.05	< 0.02			2.26 +or - 0.09	2.36 +or - 0.61	11.4 +or - 1.9	< 0.02	405 +or - 16	0.28									13	6.5	-	0.53	5 0.9
DN-1 DN-1	9/14/1978 9/20/1978	-	-	-	-				-	-	-	-	-	-									-	-	- 190		-
DN-1	10/19/1978	< 0.05	< 0.01	692	< 0.05	< 0.05			5.11 +or - 1.4	5.5 +or - 0.8	3.1 +or - 2	< 0.02	161 +or - 20	< 0.01									9	8.2	200	0.58	
DN-1 DN-1	11/16/1978 12/21/1978	< 0.02	< 0.01	572 836	0.08	< 0.01			12.38 +or - 5.9 22.4 +or - 7.9	12.4 3.24 +or - 1.1	7.1 +or - 4.5 8.3 +or - 5	< 0.02	387 +or - 46 790 +or - 90	0.6									12	9.01 7.97	325 350	2.2	
DS-2	7/18/1978	< 0.1	0.05	900	< 0.05	< 0.02			19.4 +or - 2.48	2.1 +or - 1.9	29.8 +or - 6.4	< 0.02	1092 +or - 95	0.5									9	7.33	125	0.94	
DS-2	8/13/1978	-		-	-	•			-	-	-	-	-	-									-	-	-	•	-
DS-2 DS-2	8/14/1978 9/13/1978	- < 0.05	- 0.15	- 560	- 0.05	< 0.02			- 10.46 +or - 1.3	- 9.4 +or - 1.2	- 22.6 +or - 2.4	- < 0.02	- 907 +or - 33	- 0.23									- 6	- 6.42	-	- 0.61	
DS-2	9/14/1978	-		-	-	-			-	-	-	-		-									-	-	180	-	
DS-2 DS-2	9/20/1978 10/19/1978	- < 0.05	- < 0.01	- 693	- < 0.05	- < 0.05			- 27.98 +or - 25	- 2.1 +or - 0.2	- 3.5 +or - 2	- < 0.02	- 190 +or - 25	- < 0.01									- 4	- 8.12	- 185	- 0.25	
DS-2 DS-2	11/16/1978	< 0.05	< 0.01	560	0.05	< 0.05			27.98 +or - 25 10.09 +or - 9.8	0.15	3.5 +or - 2 5.3 +or - 4	< 0.02	190 +or - 25 591 +or - 60	0.21									3	8.45	215	1.5	
DS-2	12/21/1978	< 0.05	< 0.01	524	< 0.02	< 0.01			8.5 +or - 0.64	2.16 +or - 0.93	5.8 +or - 6	< 0.02	410 +or - 45	0.01									2	8.77	200	0.41	. 0.01
GMD-00 GMD-00	11/8/2010 11/8/2010	< 0.002	< 0.002 < 0.002		< 0.004	< 0.02						< 0.002		0.0488	< 0.002				3.12 3.01	1.04 0.997	< 1	13.7 13.4		7			
GMD-00	11/8/2010						<0.752	<0.962					< 0.768			<0.414	< 0.127	< 0.213									
GMD-01	11/8/2010 11/8/2010	0.0028	< 0.002 < 0.002		0.0051 0.0047	< 0.02						< 0.002 < 0.002		< 0.02	< 0.002 < 0.002				46.6 45.6	13.6 13.3	4.31 4.32	23.8 23.4		7			_
GMD-01 GMD-01	11/8/2010	0.0026	< 0.002		0.0047	< 0.02	<0.65	<0.979				< 0.002	1.28 +or - 0.767		< 0.002	1.598 +or - 0.552	0.147 +or - 0.168	0.864 +or - 0.387	45.0	13.3	4.32	23.4		/			
GMD-02	11/9/2010	0.0063	0.0294		0.0096	0.0232						< 0.002		< 0.02	0.0293				3.28	0.344	< 1	387					
GMD-02 GMD-03	11/9/2010 11/9/2010	0.0066	0.0301 0.0299		0.0091 0.0093	0.0229						< 0.002		< 0.02	0.0292				3.27	0.345	< 1	385		8.7			_
GMD-03	11/9/2010	0.0063	0.0299		0.0088	0.0216						< 0.002		< 0.02	0.0279				3.17	0.338	< 1	370		8.7			
GMD-04	11/8/2010	< 0.002	< 0.002		< 0.004	< 0.02						< 0.002		0.474	0.003				104	21.3	4.11	104					
GMD-04 GMD-04	11/8/2010 11/8/2010	< 0.002	< 0.002	709	< 0.004	< 0.02	3.33 +or - 1.15	2.67 +or - 0.75				< 0.002	17.3 +or - 4.01	0.432	0.003	1.255 +or - 0.476	< 0.238	0.877 +or - 0.385	100	20.5	3.92	99.8	19	7.4	270	70	< 1
GMD-05	11/8/2010	< 0.002	0.0058		< 0.004	< 0.02						< 0.002		< 0.02	< 0.002				16.6	4.61	4.54	1080					
GMD-05 GMD-05	11/8/2010 11/8/2010	< 0.002	0.0066	3070	< 0.004	< 0.02	1.05 +or - 0.71	1.39 +or - 0.57				< 0.002	< 2.67	< 0.02	< 0.002	<0.248	< 0.272	< 0.278	17.6	4.89	7.22	1160	1500	7.8	280	90	< 4
GW-7	7/18/1978	0.1	< 0.01	992	0.05	< 0.02		101 101	2.3 +or - 0.8	< 0.5 +or - 0.1	< 1 +or - 0.5	< 0.02	52 +or - 5	0.3									12.2	6.6	100	0.32	
GW-7	8/13/1978	-	-	-	-				-	-	-	-		-									-	-	300	-	-
GW-7 GW-7	8/14/1978 9/13/1978	< 0.1	- 0.01	708	-				-	-	-	< 0.02	-	-									-	-			-
GW-7	9/14/1978	-	-	-	-				-			-		-									-	-	-	-	-
GW-7 GW-7	9/20/1978 10/19/1978	< 0.01	0.01	24740	< 0.05	< 0.05			3.81 +or - 0.78	< 0.5 +or - 0.1	< 1 +or - 0.5	0.4	38 +or - 2	0.02									152	6.56	14634	0.02	17
GW-7	11/16/1978	-	-	-	-				-	-	-		-	-									-	-	-	-	-
GW-7 GW-7	12/21/1978 9/30/1977	< 0.05	< 0.01 < 0.01	24592	< 0.02	< 0.01	121		2.62 +or - 5.2	1.23 +or - 0.8	7 +or - 5 8.2 +or - 6.6	0.04	960 +or - 100 19 +or - 8	0.14									144	7	10000	0.05	4 17
GW-7	10/14/1977	< 0.1	0.01	2030	< 0.1 < 0.1	< 0.1	1.21 +or - 0.2 0.47 +or - 12				8.2 +or - 6.6 50 +or - 11		19 +or - 8 11 +or - 4	-					-	-			-		-	-	
GW-7	10/24/1977	< 0.1	< 0.01	-	< 0.1	< 0.1	0.81 +or - 0.2						85 +or - 33	-	-				-	-	-	-	-	-	-	-	-
GW-7 GW-7	11/14/1977 12/7/1977	< 0.01	< 0.01 < 0.01	1910	< 0.1 < 0.1	< 0.1 < 0.1	2.04 +or - 0.48 0.72 +or - 0.3						0 +or - 5 0 +or - 5	-					-	-	•	-	-	-	-	-	
GW-7	1/6/1978	< 0.01	< 0.01	1506	< 0.1	< 0.1	138 +or - 6			9.83 +or - 3.06	15 +or - 9		678 +or - 121	-	-				-	-	-	-	-	-	-	-	-
GW-7	1/25/1978 2/8/1978	< 0.01 < 0.01	0.01	-	< 0.1	< 0.1	3.85 +or - 0.45				 6 +or - 2	<u> </u>	8 +or - 6							-	-	-	<u> </u>	<u> </u>		-	· ·
GW-7 GW-7	3/9/1978	< 0.01	0.03	- 1770	< 0.1 < 0.1	< 0.1 < 0.1	1.23 +or - 0.31 1.68 +or - 0.28				6 +or - 2	1	27 +or - 7 11 +or - 5				1			-	-	-	-	-	-	-	
GW-7	4/3/1978		0.01	-	< 0.1	< 0.1								-					-	-	-			-	-	-	· ·
GW-7 GW-8	5/8/1978 9/30/1977	< 0.01	< 0.01		< 0.1 < 0.1	< 0.1			2.61 +or - 0.31				 0 +or - 2.5							-					-	-	
GW-8	10/14/1977	< 0.01	0.04	23722	< 0.1	< 0.1			0.13 +or - 0.09	-	-		0 +or - 4														
GW-8	10/24/1977 11/14/1977	< 0.01	0.04	-	< 0.1	< 0.1			2.27 +or - 0.36	- 1.83 +or - 0.89	-		71 +or - 34 108 +or - 30														
GW-8 GW-8	12/7/1977	< 0.01	< 0.01	- 16600	< 0.1	< 0.1			3.53 +or - 0.54 3.33 +or - 0.49	-	-		0 +or - 26														
GW-8	1/6/1978	< 0.01	< 0.01	17500		< 0.1			2.93 +or - 0.43	-	-		0 +or - 29														
GW-8 GW-8	1/25/1978 2/8/1978	< 0.01 < 0.01	< 0.01 < 0.01	-	< 0.1 < 0.1	< 0.1			3.19 +or - 0.42 3.97 +or - 0.55	- 3.41 +or - 1.57	- 7 +or - 2		0 +or - 5 1260 +or - 365														
GW-8	3/9/1978	< 0.01	< 0.01	4560	< 0.1	< 0.1			4.93 +or - 0.64	-	-		36 +or - 36														
GW-8	4/3/1978 5/8/1978	< 0.01	< 0.01 < 0.01	-	< 0.1	< 0.1			-				-														
GW-8 GW-8A	7/18/1978	< 0.01	0.1	- 8080	< 0.1	< 0.1			- 1.75 +or - 0.6	- < 0.5 +or - 0.1	< 1 +or - 0.5	0.02	- 22 +or - 3	0.5									73.6	8.54	3150	0.01	22.5
GW-8A	8/13/1978	-	-	-	-	-			-	-	-	-	-	-									-	-	4300	-	-
GW-8A GW-8A	8/14/1978 9/13/1978	< 0.1	- 0.22	9968	-	-	-	+	-	-	-	0.35	-	-			1						-	-	-	-	-
GW-8A	9/14/1978	-	-	-	-	-			-	-	-	-		-							1		-	-	-	-	· ·
GW-8A GW-8A	9/20/1978 10/19/1978	< 0.01	0.01	6212	< 0.05	< 0.05			1.56 +or - 0.54	1.05 +or - 0.5	< 1 +or - 0.5	0.05	76 +or - 2	0.14	<u> </u>								13.8	8.05	3800	0.04	- 5.2
GW-8A GW-8A	11/16/1978	-	-	-	-	-	1		-	-	-	-	-	-			1						-	-	-	-	
GW-8A	12/21/1978	< 0.05	< 0.01	5310	< 0.02	< 0.01			2.69 +or - 5.3	2.07 +or - 0.9	2.7 +or - 5	0.06	130 +or - 20	< 0.01									10	7.69	4000	0.04	6



# TABLE 2 Water-Quality Results Compilation (page 2 of 3)

Location ID         Date         mg/l         mg/l         mg/l           6W-9         7/18/1978         1         0           GW-9         8/13/1978         -         -           GW-9         8/13/1978         -         -           GW-9         8/13/1978         -         -           GW-9         9/13/1978         -         -           GW-9         9/14/1978         0.1         0           GW-9         9/14/1978         <         0.1           GW-9         10/15/1978         < 0.01         0           GW-9         11/16/1978         < 0.02         < 0           GW-9         12/21/1978         < 0.05         < 0           Mancos Geometric Mean         0.0021         0           Mancos Minimum         0.0229         7           Marcus Ranch Well         7/1/1993         < 0.001         < 0           MW-1         10/20/1975         < 0.01         < 0           MW-1         12/21/1976         <<         < 0           MW-1         12/21/1976         <<<0         < 0           MW-1         12/21/1976         <<<0         < 0           MW-1         12/21/1976	mg/l         mg/l           0.2         33088            0.2         32080            0.2         29280            0.2         29280            0.1         31880            0.01         31380            0.01         34252            <001         30432            0.001         34252            0.001         34252            0.0021         30432            0.0001         468            0.0001         468            <0.01         468            <0.01         7.557            <0.01         468            <0.01         536            <0.01         536            <0.01         536            <0.01         536            <0.01         536            <0.01         537            <0.01         536            <0.01         537            <0	bdenum         Vanadium         Radium 22           ng/l         pc/l            005         < 0.02            0.05         -            0.05         -            0.05         -            0.06         -            0.07             0.08             0.09             0.000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0000             0.0010	pCi/L         pCi/L           pCi/L         pCi/L           1.81         +or - 0.6           -         -           <	- - - < 0.5 +0r - 0.1 < < 0.05 +0r - 0.01 < 0 <	Lead 210 pCi/L : 1 +or - 0.5 - : : +or - 0.5 : 1 +or - 0.5 : 1 +or - 0.5 : 1 +or - 0.5 : 3 +or - 5.5 - - - - - - - - - - - - -	Lead mg/l 0.25 - 21.5 - 21.5 - 0.45 0.45 0.45 0.45 0.45 0.45 0.45 0.45	Gross Alpha pCi/L 25 +or - 4 20 +or - 1 9.5 +or - 5 19.4 +or - 10 1040 +or - 101 1040 +or - 101 1044 +or - 7 13 +or - 9	Zinc    mg/l 2.6    	mg/l	Uranium-234 Uranium pCi/L pCi 		Calcium mg/l 336 48 600	Magnesium mg/l 392 6.2 7000	Potassium mg/l	Sodium mg/l 	Chloride mg/l 91.4 - - - - 888 88 88 88 88 80 238 238 1.1 7098 1.2 7098	log[H] 4.66 - - 5.22 4.83 4.24 4.75 7.4 4.2	Sulfate         Alkalini           mg/l         -           12000         -           15000         -           15000         -           21425         -           21425         -           7869         -           975         -           10000         -           6114         -           67         0           78003         17266           -         -           -         -	mg/l 0.013	(NO <sub>3</sub> as N) mg/i 14.3 - 20 - - 7 7 13.7 5.5 12 7.6 0.1 816 - 0.2 0.32
GW-9         7/18/1978         1         0           GW-9         8/13/1978         -         -           GW-9         8/14/1978         0.1         0           GW-9         9/13/1978         -         -           GW-9         9/13/1978         -         -           GW-9         9/14/1978         -         -           GW-9         9/20/1978         < 0.0         0           GW-9         10/19/1978         < 0.05         < 0           GW-9         11/16/1978         < 0.02         < 0           GW-9         11/16/1978         < 0.02         0           GW-9         11/16/1978         < 0.022         7           Mancos Maximum         0.0229         7           Marcus Ranch Well         7/1/1993         < 0.005         < 0           MW-1         11/20/1975         < 0.01         < 0           MW-1         11/20/1975         < 0.01         < 0           MW-1         11/21/1976         -<<<0         < 0           MW-1         11/21/1976         -<<<0         < 0           MW-1         12/21/1976         -<<<0         < 0           MW-1         12/21/1976 </th <th>0.2 <math>33088</math> <math>&lt;</math> <math>0.2</math> <math>29280</math> <math>                       0.01</math> <math>34252</math> <math>&lt;</math> <math>&lt;</math> <math>0.01</math> <math>34252</math> <math>&lt;</math> <math>&lt;</math> <math>0.01</math> <math>30432</math> <math>&lt;</math> <math>0.0027</math> <math>0.0001</math> <math>  0.0027</math> <math>5757</math> <math>  &lt;</math> <math>0.01</math> <math>468</math> <math>&lt;</math> <math>&lt;</math> <math>0.01</math> <math>536</math> <math> &lt;</math> <math>0.01</math> <math>536</math> <math> &lt;</math> <math>0.01</math> <math>535</math> <math> &lt;</math> <math>0.01</math> <math>571</math> <math> &lt;</math> <math>0.01</math> <math>571</math> <math> &lt;</math> <math>0.01</math> <math>535</math>         -</th> <th>105     &lt; 0.02       -     -       -     -       -     -       -     -       0.05     -       0.05        0.05        0.02     &lt; 0.05       0.02     &lt; 0.01       0.0006        0.0006        0.0100        0.0100     3.85       0.001     2.92       -     10.3       93.2     -       10.3     93.2       -     17.5       1003     13.5       -     65.5       -     70.7       -     65.5       -     6.8       -     7.1       -     6.5       -     7.1       -     65.5       1031     172</th> <th>181     +or     0.6       1     -   <th>&lt; 0.5 +or 0.1 &lt;</th><th><pre>&lt; 1 +or - 0.5 &lt; 1 +or - 0.5</pre></th><th>0.25 - - - - 0.45 0.2 0.5 0.72 - - - - - - - - - - - - -</th><th>25 +or - 4 - - - 20 +or - 1 9.5 +or - 5 19.4 +or - 10 1040 +or - 101 - - - - - - - - - - - - -</th><th>2.6  2.1 1.2 1.86 1.6</th><th>0.0553 224 0.0002 (0 1.922 4 0.0035 0.023 0.04</th><th>24.3 0.2 0</th><th>11.9 0.1</th><th>336</th><th>392 6.2</th><th>14.3</th><th>1692 9.3</th><th>91.4 - - - - - - - - - - - - -</th><th>4.66 - - - 5.22 4.83 4.24 4.75 7.4 4.2 8.5</th><th>2000 12000 - - 21425 7869 975 10000 6114 67 0 78003 1726</th><th>0.013           -           -           -           0.001           &lt; 0.001           &lt; 0.001           0.012           0.047</th><th>14.3 - 20 - 7 13.7 5.5 12 7.6 0.1 816 - 0.2</th></th>	0.2 $33088$ $<$ $0.2$ $29280$ $                       0.01$ $34252$ $<$ $<$ $0.01$ $34252$ $<$ $<$ $0.01$ $30432$ $<$ $0.0027$ $0.0001$ $  0.0027$ $5757$ $  <$ $0.01$ $468$ $<$ $<$ $0.01$ $536$ $ <$ $0.01$ $536$ $ <$ $0.01$ $535$ $ <$ $0.01$ $571$ $ <$ $0.01$ $571$ $ <$ $0.01$ $535$ -	105     < 0.02       -     -       -     -       -     -       -     -       0.05     -       0.05        0.05        0.02     < 0.05       0.02     < 0.01       0.0006        0.0006        0.0100        0.0100     3.85       0.001     2.92       -     10.3       93.2     -       10.3     93.2       -     17.5       1003     13.5       -     65.5       -     70.7       -     65.5       -     6.8       -     7.1       -     6.5       -     7.1       -     65.5       1031     172	181     +or     0.6       1     - <th>&lt; 0.5 +or 0.1 &lt;</th> <th><pre>&lt; 1 +or - 0.5 &lt; 1 +or - 0.5</pre></th> <th>0.25 - - - - 0.45 0.2 0.5 0.72 - - - - - - - - - - - - -</th> <th>25 +or - 4 - - - 20 +or - 1 9.5 +or - 5 19.4 +or - 10 1040 +or - 101 - - - - - - - - - - - - -</th> <th>2.6  2.1 1.2 1.86 1.6</th> <th>0.0553 224 0.0002 (0 1.922 4 0.0035 0.023 0.04</th> <th>24.3 0.2 0</th> <th>11.9 0.1</th> <th>336</th> <th>392 6.2</th> <th>14.3</th> <th>1692 9.3</th> <th>91.4 - - - - - - - - - - - - -</th> <th>4.66 - - - 5.22 4.83 4.24 4.75 7.4 4.2 8.5</th> <th>2000 12000 - - 21425 7869 975 10000 6114 67 0 78003 1726</th> <th>0.013           -           -           -           0.001           &lt; 0.001           &lt; 0.001           0.012           0.047</th> <th>14.3 - 20 - 7 13.7 5.5 12 7.6 0.1 816 - 0.2</th>	< 0.5 +or 0.1 <	<pre>&lt; 1 +or - 0.5 &lt; 1 +or - 0.5</pre>	0.25 - - - - 0.45 0.2 0.5 0.72 - - - - - - - - - - - - -	25 +or - 4 - - - 20 +or - 1 9.5 +or - 5 19.4 +or - 10 1040 +or - 101 - - - - - - - - - - - - -	2.6 2.1 1.2 1.86 1.6	0.0553 224 0.0002 (0 1.922 4 0.0035 0.023 0.04	24.3 0.2 0	11.9 0.1	336	392 6.2	14.3	1692 9.3	91.4 - - - - - - - - - - - - -	4.66 - - - 5.22 4.83 4.24 4.75 7.4 4.2 8.5	2000 12000 - - 21425 7869 975 10000 6114 67 0 78003 1726	0.013           -           -           -           0.001           < 0.001           < 0.001           0.012           0.047	14.3 - 20 - 7 13.7 5.5 12 7.6 0.1 816 - 0.2
GW-9         8/13/1978         -         -           GW-9         8/14/1578         0.1         0           GW-9         9/13/1378         -         -           GW-9         9/14/1378         -         -           GW-9         9/14/1978         -         -           GW-9         9/12/1978          0.01         0           GW-9         10/16/1978         < 0.05         < 0           GW-9         11/16/1978         < 0.02         < 0           GW-9         11/16/1978         < 0.02         < 0           Mancos Geometric Mean         0.0011         0           Mancos Maxinum         0.0229         7           Marcus Ranch Well         7/1/1933         < 0.05         < 0           MW-1         10/20/1975         < 0.01         < 0           MW-1         12/21/1976         -         < 0           MW-1         12/21/1976	.         .           0.2         22280           -         -           0.01         31880           <         0.01           31880         <           <         0.01           31880            <         0.01           34252            <         0.01           0.0827            0.0001         -           0.0001         -            0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01           <         0.01            0.01            0.01            0.01            0.01           0.01	·         ·           ·         ·           ·         ·           ·         ·           0.05            0.05            0.05            0.02            0.0006            0.0006            0.0190            0.0190            0.02            0.0190         3.85           0.02         0.01           0.03         3.85           0.03         3.85           0.03         13.5           0.03         11.5           0.03         40.9           65.5         70.7           102         68           7.1         102           6.8         7.1           0.031         1172	Image: constraint of the second se	- - - < 0.5 +0r - 0.1 < < 0.05 +0r - 0.01 < 0 <		- 21.5 - 0.45 0.5 0.72 - 0.5 0.72 - - - - - - - -	- - - 20 +or - 1 9.5 +or - 5 19.4 +or - 10 1040 +or - 101 - 		0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	- - - 88 88 5 80 238 1.1 7098		15000 - - 21425 7869 975 10000 6114 67 0 78003 1726		- 20 - 7 13.7 5.5 12 7.6 0.1 816 - 0.2
GW-9         9/13/1978         .         .           GW-9         9/14/1978         .         .           GW-9         9/20/1978         <	.         .         .           .         .         .         .           .         .         .         .           .         .         .         .         .           .         .         .         .         .           .         .         .         .         .         .           .         .         .         .         .         .         .           .	$\begin{array}{c c c c c c c c c c c c c c c c c c c $	5.05         +or - 0.89           1.77         +or - 0.75           1.58         +or - 6.1           1.58         +or - 6.1           0.28         -           -         -           0.28         -           -         -           -         -           -         -           0         -           -         - <td><pre>&lt; 0.05 +or - 0.01 &lt;</pre></td> <td><pre>1 +or - 0.5 1 +or - 0.5</pre></td> <td>- 0.45 0.2 0.5 0.72 - - - - - - -</td> <td>9.5 +or - 5 19.4 +or - 10 1040 +or - 101 </td> <td>1.2 1.86 1.6</td> <td>0.0002 (0 1.922 4 0.0035 0.023 0.04</td> <td>0.2 0</td> <td>0.1</td> <td>48</td> <td>6.2</td> <td>0.9</td> <td>9.3</td> <td>88 5 80 238 1.1 7098</td> <td>4.83 4.24 4.75 7.4 4.2 8.5</td> <td>7869           975           10000           6114           67         0           78003         1726</td> <td>&lt; 0.001 0.012 0.047</td> <td>- 7 13.7 5.5 12 7.6 0.1 816 0.2</td>	<pre>&lt; 0.05 +or - 0.01 &lt;</pre>	<pre>1 +or - 0.5 1 +or - 0.5</pre>	- 0.45 0.2 0.5 0.72 - - - - - - -	9.5 +or - 5 19.4 +or - 10 1040 +or - 101 	1.2 1.86 1.6	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	88 5 80 238 1.1 7098	4.83 4.24 4.75 7.4 4.2 8.5	7869           975           10000           6114           67         0           78003         1726	< 0.001 0.012 0.047	- 7 13.7 5.5 12 7.6 0.1 816 0.2
GW-9         9/14/1978         ·         ·           GW-9         9/20/1978         < 0.01	$\begin{array}{c c c c c c c } < 0.01 & 32032 & < < \\ < 0.01 & 34252 & < \\ < 0.01 & 30432 & < \\ 0.0827 & & & \\ 0.0001 & & & \\ < 0.01 & 468 & < \\ < 0.01 & 468 & < \\ < 0.01 & 468 & < \\ < 0.01 & 545 & \\ < 0.01 & 536 & \\ < 0.01 & 511 & & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ < 0.01 & 545 & \\ \\ < 0.01 & 545 & \\ \\ < 0.01 & 545 & \\ \\ < 0.01 & 545 & \\ \\ < 0.01 & 545 & \\ \\ < 0.01 & 545 & \\ \\ \\ < 0.01 & 545 & \\ \\ \\ < 0.01 & 545 & \\ \\ \\ \\ \hline \\ 0.01 & 545 & \\ \\ \\ \hline \\ 0.02 & 503 & \\ \\ 0.01 & < \\ \\ \\ \hline \\ 0.02 & 503 & \\ \\ \hline \\ 0.01 & \\ \\ \hline \\ \\ \end{array}$	$\begin{array}{c c c c c c c c c c c c c c c c c c c $	5.05         +or - 0.89           1.77         +or - 0.75           1.58         +or - 6.1           1.58         +or - 6.1           0.28         -           -         -           0.28         -           -         -           -         -           -         -           0         -           -         -	<pre>&lt; 0.05 +or - 0.01 &lt;</pre>	<pre>1 +or - 0.5 1 +or - 0.5</pre>	0.2 0.5 0.72 < 0.01 < 0.001 - - -	9.5 +or - 5 19.4 +or - 10 1040 +or - 101 	1.2 1.86 1.6	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	88 5 80 238 1.1 7098	4.83 4.24 4.75 7.4 4.2 8.5	7869           975           10000           6114           67         0           78003         1726	< 0.001 0.012 0.047	13.7           5.5           12           7.6           0.1           816           0.2
$ \begin{array}{c c c c c c c c c c c c c c c c c c c $	$\begin{array}{c c c c c c c } < 0.01 & 32032 & < < \\ < 0.01 & 34252 & < \\ < 0.01 & 30432 & < \\ 0.0827 & & & \\ 0.0001 & & & \\ < 0.01 & 468 & < \\ < 0.01 & 468 & < \\ < 0.01 & 468 & < \\ < 0.01 & 545 & \\ < 0.01 & 536 & & \\ < 0.01 & 536 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ < 0.01 & 545 & & \\ \hline \end{array}$	$\begin{array}{c c c c c c c c c c c c c c c c c c c $	5.05         +or - 0.89           1.77         +or - 0.75           1.58         +or - 6.1           1.58         +or - 6.1           0.28         -           -         -           0.28         -           -         -           -         -           -         -           0         -           -         - <td><pre>&lt; 0.05 +or - 0.01 &lt;</pre></td> <td><pre>1 +or - 0.5 1 +or - 0.5</pre></td> <td>0.2 0.5 0.72 &lt; 0.01 &lt; 0.001 - - -</td> <td>9.5 +or - 5 19.4 +or - 10 1040 +or - 101 </td> <td>1.2 1.86 1.6</td> <td>0.0002 (0 1.922 4 0.0035 0.023 0.04</td> <td>0.2 0</td> <td>0.1</td> <td>48</td> <td>6.2</td> <td>0.9</td> <td>9.3</td> <td>88 5 80 238 1.1 7098</td> <td>4.83 4.24 4.75 7.4 4.2 8.5</td> <td>7869           975           10000           6114           67         0           78003         1726</td> <td>&lt; 0.001 0.012 0.047</td> <td>13.7           5.5           12           7.6           0.1           816           0.2</td>	<pre>&lt; 0.05 +or - 0.01 &lt;</pre>	<pre>1 +or - 0.5 1 +or - 0.5</pre>	0.2 0.5 0.72 < 0.01 < 0.001 - - -	9.5 +or - 5 19.4 +or - 10 1040 +or - 101 	1.2 1.86 1.6	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	88 5 80 238 1.1 7098	4.83 4.24 4.75 7.4 4.2 8.5	7869           975           10000           6114           67         0           78003         1726	< 0.001 0.012 0.047	13.7           5.5           12           7.6           0.1           816           0.2
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	< 0.01	102     < 0.01	1.77     +or - 0.75       1.58     +or - 6.1       1.58     +or - 6.1       0.28     -       -     -	0 <	1 +or - 0.5	0.5 0.72 < 0.01 < 0.001 - - -	19.4 +or - 10 1040 +or - 101 6 +or - 15 14 +or - 7	1.86	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	5 80 238 1.1 7098	4.24 4.75 7.4 4.2 8.5	975           10000           6114           67         0           78003         1726	0.012	5.5 12 7.6 0.1 816 0.2
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	< 0.01	$\begin{array}{c c c c c c c c c } &< 0.01 & & & \\ \hline 0.0000 & & & \\ \hline 0.0000 & & & \\ \hline 0.0100 & & & \\ 0.02 & < 0.01 & & 0.2 & + or & \\ \hline 0.02 & < 0.01 & & 0.2 & + or & \\ \hline 0.02 & & & 0.2 & & \\ \hline 0.02 & & & 0.2 & & \\ \hline 0.02 & & & 0.2 & & \\ \hline 0.03 & & & 0.2 & & \\ \hline 0.03 & & & & 0.2 &$	1.58     +or - 6.1       0.28     -       -     -			0.72 < 0.01 < 0.001 < 0.001 - -	1040 +or - 101 6 +or - 15 14 +or - 7	1.6	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	80 238 1.1 7098	4.75 7.4 4.2 8.5	10000           6114           67         0           78003         1726	0.047	12 7.6 0.1 816 0.2
Mancos Minimum         0.0002         0           Mancos Maximum         0.0229         7           Marcus Ranch Well         7/1/1993         < 0.005	0.0001            7.557            < 0.01	$ \begin{array}{c c c c c c c c c c c c c c c c c c c $				< 0.001 < 0.001 - - -	6 +or - 15 14 +or - 7	0.517	0.0002 (0 1.922 4 0.0035 0.023 0.04	0.2 0	0.1	48	6.2	0.9	9.3	1.1 7098	4.2 8.5	67 0 78003 1726		0.1 816 0.2
Mancos Maximum         0.0229         7           Marcus Ranch Well         7/1/1993         < 0.005	7.557            < 0.01	0.0190           0.02         < 0.01				< 0.001 < 0.001 - - -	14 +or - 7	0.517	1.922     4       0.0035     0       0.023     0							7098	8.5	78003 1726		816 0.2
Marcus Ranch Well         7/1/1993         < C.005	< 0.01         <           < 0.01         468         <           < 0.01         709            < 0.01         536            < 0.01         536            < 0.01         531            < 0.01         541            < 0.01         545            < 0.01         737            < 0.01         545            .         0.01         545           .         .         .         .           < 0.01         545         .           .         .         .         .           .         .         .         .           .         .         .         .           .         .         .         .           .         .         .         .           .         .         .         .         .           .         .         .         .         .           .         .         .         .         .           .         .         .         .         .           .         .	0.02         < 0.01         0.2         +or-           3.85             0.001         3.85            10.3         93.2            0.03         13.5            0.03         13.5            0.03         65.5            0.03         70.7            0.04         6.8            0.05         7.1            0.031         172				< 0.001 < 0.001 - - -	14 +or - 7	0.517	0.0035 0.023 0.04		280		7000	/1.5	23000					0.2
$\begin{array}{ c c c c c c c c c c c c c c c c c c c$	< 0.01         709           < 0.01         491           < 0.01         536           < 0.01         511           < 0.01         511           < 0.01         541           < 0.01         541           < 0.01         541           -         -           -         -           -         -           < 0.01         571           < 0.01         571           < 0.01         571           < 0.01         571           < 0.01         515           0.01         515           0.01            0.02         503           0.01             0.02            0.01	2.92           -         10.3           93.2         17.5           40.9         65.5           70.7         102           6.8         7.1           95.5         17.2				< 0.001 - - -		0.517	0.04							- 16.2		96		
$\begin{array}{ c c c c c c c c c c c c c c c c c c c$	< 0.01	- 10.3 93.2 17.5 1003 13.5 65.5 65.5 70.7 102 6.8 7.1 6.8 7.1 6.5 5.5 1031 172				- -	13 +or - 9	- -								-		-		0.32
MW-1 $2/2/1976$ .         < 0           MW-1 $2/1/1976$ .         < 0	< 0.01	17.5           40.9           65.5           70.7           102           6.8           7.1           65.5           102           103				-		-								-		-		
NW-1 $2/11/1976$ < 0.01         < 0           MW-1 $6/2/1976$ -         < 0	< 0.01	13.5           40.9           65.5           70.7           102           6.8           7.1           65.5           1031				-			0.085							-		-		
$\begin{array}{c c c c c c c c c c c c c c c c c c c $	< 0.01	40.9 65.5 102 6.8 70.7 102 6.8 7.1 6.5 5 1031 172					40.5	- 0.016	2.005 0.0672			-				- 7.2		- 205		0.4
$\begin{tabular}{ c c c c c c c c c c c c c c c c c c c$	< 0.01 541 	70.7 102 6.8 7.1 6.5.5 1.031 172					40.5	-	0.125							-		-		0.4
MW-1         12/13/1976         -         -           MW-1         1//1977         -         -           MW-1         2/7/1937         -         -           MW-1         3/16/1977         -         <		102 6.8 7.1 65.5 1.031 172				-		-	0.266							-		-		
$\begin{tabular}{ c c c c c c c c c c c c c c c c c c c$	< 0.01         460           0.02         503           0.01         515           0.01         <	6.8 7.1 65.5 1.031 172	· · · · · · · · · · · · · · · · · · ·			-		-	0.227							-		-		<b></b>
MW-1         3/16/1977         -         < 0           MW-1         4/15/1977         0.02         < 0	< 0.01         460           0.02         503           0.01         515           0.01         <	.031 65.5 172				-			0.0403							-				
NW-1         4/15/1977         0.02         < 0           MW-1         5/16/1977         < 0.01	< 0.01         460           0.02         503           0.01         515           0.01         <	0.031 172				-		-	0.395							-		-		
MW-1         5/16/1977         < 0.01         0           MW-1         6/15/1977         < 0.01	0.02         503           0.01         515           0.01         <		<1			- < 0.001		- < 0.01	0.278							- 7.6		- 200		0.5
MW-1 9/30/1977 0.01 0	0.01         <		<1			< 0.001		0.007	0.531							7.5		202		0.7
	0.02 504 < 0.01	0.34 0.02 165	1	22.2	05	< 0.001	1540 +or - 40	< 0.01	0.516							8.7		202		0.64
	0.01				96 +or - 10 : 3.3	-	370 +or - 30 192 +or - 14	-								-		-		
MW-1 10/24/1977 0.01 0		0.1 < 0.1 135 +or -	4 -	7.26 +or - 2.64		-	161 +or - 23	-								-		-		
	0.01 528 <				0 +or - 2	-	417 +or - 33	-	0.58							-		-		<b></b>
		0.1 < 0.1 30.4 +or - 0.1 < 0.1 2.13 +or -			14 +or - 6 7.2 +or - 5.1	-	321 +or - 34 17 +or - 11	-	0.85									-		<u> </u>
MW-1 1/25/1978 0.02 0		0.2 0.1 90.6 +or -	3.9 -	79.5 +or - 4.9		-	410 +or - 67	-	0.65							-		-		
MW-1 2/8/1978 0.01 - MW-1 3/9/1978		0.2 < 0.1 - 0.1 < 0.1 -	-		0 +or - 2 8 +or - 2	-	861 +or - 102 515 +or - 80	- 0.01	•							- 160		- 295		<b></b>
MW-1 4/3/1978		0.1 < 0.1 -	· ·	3.43 101 2.42	8 101 - 2	-	515 101- 80	-								-		-		
MW-1 5/8/1978		.1 < 0.1 -	•			-		-	-							-		-		
MW-1         7/18/1978         < 0.1         < 0           MW-1         8/13/1978         -         -         -	< 0.01 1236 <	.05 < 0.02	11.1 +or - 2.4	1.3 +or - 1.9	9.7 +or - 5.1	< 0.02	618 +or - 32	0.3								9	7.33	200 375	0.502	0.65
	< 0.01 488	· ·	-	-	-	< 0.02	-	-								-	-	-		0.7
	0.15 524	0.1 < 0.02	14.33 +or - 1.02	9.76 +or - 1.1	12.6 +or - 2.1	< 0.02	485 +or - 17	0.04								11	6.58	-	0.915	0.8
MW-1 9/14/1978 MW-1 9/20/1978 < 0.01 < 0	< 0.01 632	.4 < 0.05		7.1 +or - 1	- 9.8 +or - 5	- 0.08	- 706 +or - 11	- 0.06								10	7.1	175 255	0.91	< 0.1
		0.05 < 0.05	12.27 +or - 2.5	3 +or - 0.2	5.2 +or - 3	< 0.02	759 +or - 65	< 0.01								11	8.25	200	0.775	0.6
		0.02 < 0.01 0.02 < 0.01	3 +or - 6 6.36 +or - 8.6		4.7 +or - 3 5 +or - 6	< 0.02 < 0.02	155 +or - 20 170 +or - 10	0.18								10 12	9.2 8.54	225 275	0.725	< 0.02
		0.029 - 79.4	<1	1.02 +01 - 0.0	-	< 0.001	-	< 0.01	0.211							7.9	8.3	275	0.71	0.2
		.008 - 9.9	7.2		-	< 0.001	-	0.026	0.049							252	8.09			19.4
		0.039 - 84 0.01 < 0.1 115 +or -	3 -	25.8 +or - 0.5	- 12 +or - 7	< 0.001	- 300 +or - 30	0.01	0.502					-		8.3	8.14			0.69
		0.01 < 0.1 120 +or -			0 +or - 2	-	211 +or - 14	-								-	-			· ·
		0.1 < 0.1 66.7 +or - 0.01 < 0.1 156 +or -		4.82 +or - 1.82 3.1 +or - 1.76	- 16 +or - 6	-	123 +or - 20 354 +or - 30	-	- 0.65							-	-		!	· ·
		0.01 < 0.1 150 +01 =			0 +or - 2	-	265 +or - 31	-	0.825							-	-			-
		0.1 < 0.1 7.49 +or -		0 +or - 0.2	0 +or - 2	-	221 +or - 51	-	0.891							-	-			-
		0.2 < 0.1 16.2 +or - 0.1 < 0.1 110 +or -		6.02 +or - 1.91 7.21 +or - 2.77	- 8.8 +or - 54	-	385 +or - 64 583 +or - 59	-	- 0.695					-		-	-			-
MW-2 3/9/1978 0.01 0		0.1 < 0.1 51 +or -				< 0.05	471 +or - 63	0.01								185	8.35			0.62
		0.1 < 0.1 - 0.1 < 0.1 -			-	-	-		0.68			_			+	-			<u> </u>	+
		2.4 9 177 +or -	3	- 197 +or - 6	160 +or - 13		- 1380 +or - 60	-								-	-			i i
		0.1 0.4 24.5 +or -	1	26.7 +or - 0.5			589 +or - 84													
		5.7 < 0.1 640 +or - 2.5 < 0.1 192 +or -		417 +or - 16 1.8 +or - 1.58	- 50 +or - 15		1807 +or - 163 2484 +or - 173													
MWS-3 1/25/1978 < 0.01 0	0.06 -	l.5 < 0.1 132 +or -	4	6.72 +or - 0.9	-		2894 +or - 199													
					36 +or - 2		1383 +or - 141													
		0.1 < 0.1 19.9 +or - 1.1 < 0.1 -	1.5	4.19 +or - 2.37 -	10 +or - 1		1014 +or - 114 -													
MWS-3 7/18/1978		•		•	-	-		-								-	-	•	· · ·	· · ·
MWS-3 8/13/1978 MWS-3 8/14/1978		· · ·		-	-			-								-	-			
	0.3 2876	2.7 < 0.02	22.9 +or - 2.48	13 +or - 2.4	36.4 +or - 7.8	< 0.02	2686 +or - 90	0.32								94.4	6.25		44.2	- 8
MWS-3 9/14/1978		•		•	-	-		-								-	-	1600		·
MWS-3 9/20/1978 MWS-3 10/19/1978 < 0.05 < 0	 < 0.01 4288 <		 252.1 +or - 6	- 12 +or - 1	- 27.3 +or - 7	- < 0.02	- 929 +or - 90	- < 0.01								- 230	- 7.55	2375	- 14.25	- 11.6
MWS-3 11/16/1978 < 0.02 < 0	< 0.01 3536	.84 < 0.01	8.68 +or - 6.5	0	21.8 +or - 8	< 0.02	862 +or - 80	0.92								176	7.8	900	21	< 0.01
MWS-3         12/21/1978         < 0.05         < 0           MWS-3         7/18/1978         -         -         -	< 0.01 2792 <	0.02 < 0.01	115 +or - 19	0.99 +or - 0.7	15.8 +or - 7	< 0.02	60 +or - 80	0.05								64	7.83	850	4.8	2
MWS-3 //18/19/8 MWS-3 8/13/1978						-		-								-				
MWS-3 8/14/1978	-		-	-	-	-	-	-								-	-			· ·
MWS-3         9/13/1978         < 0.05         0           MWS-3         9/14/1978         -         -         -	0.4 8548		0.66 +or - 0.08	2.46 +or - 0.71	13.2 +or - 4	< 0.02	194 +or - 10	-								- 232	- 5.4	- 1700	- 1.175	5.5
MWS-3 9/20/1978					-										1					
			41.36 +or - 3.6													-	-	-		
MWS-3         11/16/1978         -           MWS-3         12/21/1978         < 0	< 0.01 1200 <	0.05 < 0.05	41.50 1.01 - 5.0	7.7 +or - 0.5	17 +or - 5	< 0.02	503 +or - 60	< 0.01								- 36	- 8.22	500	3.95	4.5



# TABLE 2 Water-Quality Results Compilation (page 3 of 3)

	Sample	Arsenic	Selenium	TDS	Molybdenum	Vanadium	Radium 226	Radium 228	Radium	Thorium 230	Lead 210	Lead	Gross Alpha	Zinc	Uranium	Uranium-234	Uranium-235	Uranium-238	Calcium	Magnesium	Potassium	Sodium	Chloride	рН	Sulfate	Alkalinity L		Nitrate
Location ID	Date	mg/l	mg/l	mg/l	ma/l	mg/l	pCi/L	pCi/L	(226 & 228) pCi/L	pCi/L	pCi/L	mg/l	pCi/L	mg/l	mg/l	pCi/L	pCi/L	pCi/L	mg/l	mg/l	mg/l	mg/l	mg/l	log[H]	ma/l		mg/l	NO <sub>3</sub> as N) mg/l
North Vent Pipe	6/19/1985	0.011	< 0.005	495	0.3	< 0.1	pci/L	μείγε	pci/t	μείγε	pci/L	< 0.1	perr	< 0.1	iiig/i	perre	pci/L	ρειγε	6.3 or 48		ilig/i	iiig/i	11.9	log[H]	205		ing/i	iiig/i
Tap Water	Unknown			480															35	11.4				7.72	100			2.77
Tap Water	27687 7/5/1977	< 0.01	< 0.01	452	0.009	0.03	2.34 +or - 0.23 2.7 +or - 0.03			_	-		0 +or - 2 19 +or - 5		0.008													
UG-4 UG-4	9/30/1977	< 0.01	< 0.01	432	< 0.1	< 0.1	2.7 +or - 0.29				-		19 +0r - 3 11 +or - 7						-									
UG-4	10/14/1977	< 0.01	< 0.01	469	< 0.1	< 0.1	3.29 +or - 0.35			-	-		5 +or - 3															
UG-4	10/29/1977 11/15/1977	< 0.01 < 0.01	< 0.01 < 0.01	- 468	< 0.1	< 0.1 < 0.1	2.09 +or - 0.28 3.16 +or - 0.41			-	-		9 +or - 6 10 +or - 5						_									
UG-4 UG-4	12/7/1977	< 0.01	< 0.01	400	< 0.1	< 0.1	0.49 +or - 0.29			-	-		10 +or - 5 0 +or - 5						-									
UG-4	1/6/1978	< 0.01	< 0.01	430	< 0.1	< 0.1	0 +or - 0.05			-	-		0 +or - 5															
UG-4 UG-4	2/8/1978 3/9/1978	< 0.01 < 0.01	< 0.01	- 454	< 0.1	< 0.1	2.92 +or - 0.47 1.49 +or - 0.4			-	-		18 +or - 3 16 +or - 5						_									
UG-4 UG-4	4/3/1978	< 0.01	< 0.01	434	< 0.1	< 0.1	-			-	-		-						-									
UG-4	5/8/1978	< 0.01	< 0.01	-	< 0.1	< 0.1	-			-	-		-															
UG-4 UG-5	1/25/1978	< 0.01	< 0.01	- 567	< 0.1	< 0.1	5.75 +or - 0.06 5.9 +or - 0.6			-	-		18 +or - 9 11 +or - 5						_									
UG-5	9/30/1977	< 0.01	< 0.01	-	< 0.1	< 0.1	1.09 +or - 0.22			-	-		11 +or - 7						-									_
UG-5	10/14/1977	< 0.01	< 0.01	609	< 0.1	< 0.1	0.68 +or - 0.17			-	-		9 +or - 5															
UG-5 UG-5	10/29/1977 11/15/1977	< 0.01 < 0.01	< 0.01	- 640	< 0.1	< 0.1	1.79 +or - 0.35 4.94 +or - 0.49			-	- 9.7 +or - 7.	7	9 +or - 6 11 +or - 4						-									
UG-5 UG-5	11/15/19// 12/7/1977	< 0.01	< 0.01	-	< 0.1	< 0.1	4.94 +or - 0.49 1.01 +or - 0.22			-	9.7 +or - 7. 0 +or - 2		11 +or - 4 0 +or - 5															
UG-5	1/6/1978	< 0.01	< 0.01	570	< 0.1	< 0.1	0.8 +or - 0.18			-			0 +or - 5															
UG-5	1/25/1978 2/8/1978	< 0.01 < 0.01	< 0.01	-	< 0.1	< 0.1	1.86 +or - 0.28 2.05 +or - 0.4			- 204 0.9	-		0 +or - 5 103 +or - 11						_									
UG-5 UG-5	3/9/1978	< 0.01	< 0.01	- 680	< 0.1	< 0.1	0.79 +or - 0.3			- 204 0.9	6 +or - 2		20 +or - 7						_									
UG-5	4/3/1978	< 0.01	< 0.01	-	< 0.1	< 0.1	-			-	-		-															
UG-5	5/8/1978	< 0.01	< 0.01	-	< 0.1	< 0.1	-		24		-		-	0.3					_				8	7.30	75		0.044	
UG-5 UG-5	7/18/1978 8/13/1978	< 0.1	< 0.01	908	< 0.05	< 0.02			2.1 +or - 0.6	< 0.5 +or - 1	< 1 +or - 0.	5 < 0.02	26 +or - 3	-									-	7.38	- 75		0.011	0.2
UG-5	8/14/1978	-	-	-	-	-			-	-		-		-									-	-	-		-	-
UG-5	9/13/1978 9/14/1978	< 0.05	0.1	632	0.1	< 0.02			1.07 +or - 0.09	1.66 +or - 0.4	< 1 +or - 0.	5 < 0.02	44 +or - 5	0.22					_				10.4	6.48	-		0.04	0.5
UG-5 UG-5	9/14/1978 9/20/1978	-	-	-					-	-	-		-	-					_					-	- 175		-	
UG-5	10/19/1978	-	-	-	-	-			-	-	-	-	-	-									-	-	-		-	-
UG-5	11/16/1978 12/21/1978	- < 0.05	- < 0.01	- 728	- < 0.02	- < 0.01		-	- 1.56 +or - 5.7	- 1.08 +or - 0.5	-	- 5 < 0.02	- 70 +or - 10	- 0.01			-						-	- 8.62	- 325		- 0.002	- 0.2
UG-5 UG-6	7/5/1977	< 0.03	< 0.01	407	0.054	0.03	5.8 +or - 0.6		1.56 +01 - 5.7	- 108 +01- 0.5	- +0r - 0.	5 0.02	47 +or - 8	0.01									6	0.02	323		0.002	0.2
UG-6	9/30/1977	< 0.01	< 0.01	-	< 0.1	< 0.1	6.17 +or - 0.48			-	9.9 +or - 6.	7	40 +or - 11															
UG-6 UG-6	10/14/1977 10/29/1977	< 0.01 < 0.01	< 0.01	459	< 0.1	< 0.1	6.26 +or - 0.47 7.39 +or - 0.5			-	-		13 +or - 6 23 +or - 3						-									
UG-6	11/15/1977	< 0.01	< 0.01	440	< 0.1	< 0.1	10.1 +or - 0.9			-	12 +or - 6		26 +or - 6															
UG-6	12/7/1977	< 0.01	< 0.01	-	< 0.1	< 0.1	0 +or - 0.1			-	12 +or - 6		0 +or - 5															
UG-6 UG-6	1/6/1978 1/25/1978	< 0.01 < 0.01	< 0.01	414	< 0.1	< 0.1	4.69 +or - 0.61 97.8 +or - 4			- 4.81 +or - 0.39	-		14 +or - 9 178 +or - 31						_									
UG-6	2/8/1978	< 0.01	< 0.01	-	< 0.1	< 0.1	4.17 +or - 0.57			+.01 +01 - 0.39	- 5 +or - 2	+	178 +or - 31 37 +or - 5	+														
UG-6	3/9/1978	< 0.01	< 0.01	516	< 0.1	< 0.1	5.36 +or - 0.76			-	-		28 +or - 5															
UG-6	4/3/1978 5/8/1978	< 0.01 < 0.01	< 0.01 < 0.01	-	< 0.1	< 0.1	-			-	-		-	+					_									
UG-6	7/18/1978	0.2	< 0.01	832	< 0.05	< 0.02			1.75 +or - 0.6	< 0.5 +or - 0.1	< 1 +or - 0.	5 < 0.02	42 +or - 4	0.2									5	7.26	130		0.03	0.2
UG-6	8/13/1978	-	-	-	-	-			-	-	-	-	-	-			_			l .	1		-		-		-	-
UG-6 UG-6	8/14/1978 9/13/1978	- < 0.05	- 0.05	- 496	- 0.05	- < 0.02			- 0.2 +or - 0.07	- 1.02 +or - 0.32	- < 1 +or - 0	- 5 < 0.02	- 27 +or - 3	- 0.02					_				- 9.2	- 6.36	-		- 0.02	- 0.6
UG-6	9/13/1978	-	-		-	- 0.02		1	-	-	-		-	-							1		-	-	160		-	-
UG-6	9/20/1978	-	-	-	-	-			-	-	-	-	-	-			_			l .	1		-		-		-	-
UG-6 UG-6	10/19/1978 11/16/1978	-		-	-	-			-	-	-	-	-	-									-	-	-		-	-
UG-6	12/21/1978	< 0.05	< 0.01	560	< 0.02	< 0.01			1.28 +or - 5.5	2.02 +or - 0.8	< 1 +or - 0.	5 0.1	20 +or - 1	< 0.01									4	8.54	235	<	0.002	0.1
Water Supply (WW)	Unknown														0.024				40	8					140			7.26
Water Supply (WW) Water Supply (WW)	Unknown Unknown			265 480										< 0.1					38 34	7.3	5.6	11.8		7.11	100			2.99
Water Supply (WW)	Unknown			480										< 0.1					34	11.4	7	55		7.65	230			1.05
Water Supply (WW)	Unknown			290										< 0.1					4	6.9	8.5	60		7.1				
Water Supply (WW) Water Supply (WW)	26846 27089																										0.3	
Water Supply (WW)	Unknown			423															22	11.1	10.4	63		7.6			0.1	
Water Supply (WW)	Unknown																		40	53					270		0.34	0.17
Weir Water	27687						2.92 +or - 0.26						13 +or - 9		0.04													



Appendix B

Applicable or Relevant and Appropriate Requirements (ARARs)

	Description	А	RAR Type		Po	otential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
Federal								
40 CFR 122	The NPDES program requires permits for the discharge of pollutants from any point source into waters of the United States.	Х	Х	X			Х	The remedial alternative will not result in point source dischargers. Moreover, no federal permit is required for work on the site per 42 U.S.C. §9621(e)(1) and 40 CFR 300.400.
40 CFR 141	These regulations protect the health-based quality of public drinking water supplies through regulation of maximum contaminant levels (MCLs), some of which correspond to COPCs.	Х				Х		The MCLs for COPCs in mine-related material may be relevant and appropriate depending on the land use and institutional controls at the site. No ground or surface water is currently being impacted by the site COPCs.
40 CFR 50	Implementing regulations to support the Clean Air Act providing primary and secondary ambient air quality standards	Х					Х	No stationary sources exist or are being constructed within the project area; therefore, this regulation is not applicable or relevant and appropriate.

	Description	А	RAR Type		Pc	otential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
40 CFR 61	Regulations containing the National Emission Standards for Hazardous Air Pollutants (NESHAPs).	х				Х		The NESHAP standard for radon-222 emission from uranium mill tailings piles contained in 40 CFR 40.61 of a 20 pCi/m <sup>-2</sup> s <sup>-1</sup> cover design criteria is not applicable but is relevant and appropriate to the protectiveness of the selected removal action.
40 CFR 192	Implementing regulations of the Uranium Mill Tailing Radiation Control Act. Establishes enforceable standards for cleanup levels of radionuclides for sites affected by uranium mill tailings. Governs stabilization, disposal, and control of uranium and thorium mill tailings by setting health and environmental protection standards.	Х				Х		The UMTRCA standard of 5 pCi/g above background for radium-226 in soil is not applicable but is relevant and appropriate since the one of the COPCs for the project area is radium-226

	Description	А	RAR Type		Ро	tential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
16 U.S.C § 470f; 36 CFR Part 800; and 36 CFR Parts 63 and 80	National Historic Preservation Act (NHPA) and Implementing regulations of NHPA		Х				X	Requires the federal agencies to take into account the consequences of their "undertakings," on a district, site, building, structure, or object that is included in, or is eligible for, inclusion in the National Register of Historic Places. Regulates the inventory, assessment, and consultation on removal project efforts and protection measures for cultural properties on Federal land. The site is not eligible for inclusion on the National Register, nor does the site contain any such buildings structures or objects. Moreover, no federal permit is required for work on the site per 42 U.S.C. §9621(e)(1) and 40 CFR 300.400. Therefore, these requirements are not applicable or relevant and appropriate.

	Description	А	RAR Type		Po	tential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
16 U.S.C. § 469	Archaeological and Historic Preservation Act of 1974		Х			Х		Provides for preservation of significant scientific, prehistoric, historic, and archaeological data which may be affected by removal efforts. These requirements are likely not applicable because no archaeological items have been identified within the project area. However, if archaeological items are found during project work, this may be relevant and appropriate.
43 CFR Part 7	Implementing regulations of the Archaeological Resources Protection Act of 1979		Х			Х		Requires permits for excavation of archaeological resources on public or tribal lands. These requirements are not applicable because the site is not on public or tribal lands and, moreover, no archaeological items have been identified within the project area. However, if archaeological items are found during project work, these requirements may be relevant and appropriate.

	Description ARAR Type Potential ARA				otential ARAR Cat	egory	Comment	
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
43 CFR Part 10	Implementing regulations of the Native American Graves Protection and Repatriation Act		X			X		Establishes regulations pertaining to the identification, protection, and appropriate disposition of human remains, funerary objects, sacred objects, or objects of cultural patrimony. Because the Act does not apply to private land, these requirements are not applicable. However, if cultural objects are found during project work, these requirements may be relevant and appropriate.
40 CFR Part 257	Resource Conservation and Recovery Act Subtitle D Regulations, Criteria for Classification of Solid Waste Disposal Facilities and Practices.		Х			Х		These regulations establish federal criteria for use in siting the disposal of solid waste in certain locations. Although these requirements are not applicable as an exemption applies here, these regulations are relevant and appropriate to the design of the onsite waste repository.

	Description	ARAR Type			Potential ARAR Category			Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
33 CFR 330	Implementing regulations of Section 404, Clean Water Act.		Х			X		Regulates discharge of dredge or fill materials into waters of the U.S. Permit is not required per 42 U.S.C. § 9621(e)(1) and 40 CFR 300.400. However, substantive requirements of the regulations may be relevant and appropriate to any dredge or fill activities, to the extent waters of the United States are impacted in project area.
16 U.S.C. § 662	Fish and Wildlife Coordination Act (FWCA)		Х				X	Requires consultation with Federal and State agencies to provide adequate protection of fish and wildlife resources, specifically when modification of any stream or other water body is proposed. No fish or wildlife resources will be impacted by removal action. Not applicable or relevant and appropriate.

	Description	А	RAR Type		Po	otential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
16 U.S.C. § 1531 et seq.; 50 CFR, Part 402	Endangered Species Act of 1973 (ESA), and ESA implementing regulations		Х				Х	Regulates the protection of threatened or endangered species and critical habitat. Requires action to conserve endangered plant and animal species within critical habitats upon which they depend. A proposed action may not jeopardize continued existence of endangered species or destroy or adversely modify a critical habitat. No evidence of endangered species or critical habitat within project area. Not applicable or relevant and appropriate.
29 CFR	Implementing regulations of Occupational Safety and Health Act (OSHA)			Х	Х			Regulates worker health and safety. Applicable.
49 CFR 107, 171-177	Implementing regulations of Hazardous Materials Transportation Act – Standards Applicable to Transport of Hazardous Materials			х	Х			Applicable if mine-related material is transported on pubic roadways.
40 CFR Part 192	Health and Environmental Protection Standards for Uranium and Thorium Mill Tailings			X		Х		Standards for the control of residual radioactive materials from inactive uranium processing sites. Relevant and appropriate

	Description	А	RAR Type		Po	tential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
10 CFR Part 20	NRC Standards for Protection Against Radiation			Х		Х		Sets permissible dose levels, radioactivity concentration limits for effluents, precautionary procedures, waste disposal requirements for NRC licensees, and establishes protocols for protection of workers, protection of the public, discharges to air and water, and waste treatment and disposal. No NRC license is needed so not applicable but relevant and appropriate.
40 CFR Part 440	Ore Mining and Dressing Point Source Category address in NPDES			Х		Х		Radionuclide concentration limits for surface water discharges of radioactive waste. Potentially relevant and appropriate to the extent waters of the united States are impacted in project area.

	Description	А	ARAR Type		Potential ARAR Category			Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
40 CFR 122	Stormwater discharges from construction sites.			X		Х		Requires a permit (general or individual) for point source discharges of stormwater from construction activities. Permit is not required per 42 U.S.C. § 9621(e)(1) and 40 CFR 300.400. However, substantive requirements of the regulations may be relevant and appropriate to activities during response action implementation, to the extent waters of the United States are impacted in project area.
State 20.2.3 NMAC	New Mexico Ambient Air Quality Standards establish standards for air pollutants in order to prevent or improve air quality deterioration.	Х				Х		This regulation provides the maximum allowable concentrations of total suspended particulate in the ambient air, and is applicable. As no stationary sources exist, these requirements are not applicable. However, they may be relevant and appropriate to dust control efforts during response action implementation.

	Description	А	RAR Type		Po	tential ARAR Cat	egory	Comment
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
20.6.2 and 20.6.4 NMAC	New Mexico Ground and Surface Water Protection Standards.	Х				Х		Regulations governing water quality protection in New Mexico. Establishes water quality standards for the state, including description of the designated use(s), the criteria necessary to protect the use(s), and anti- degradation policies. No ground or surface water is currently being impacted by the site COPCs. Moreover, no discharge permit will be required for the selected removal action. However, the New Mexico groundwater protection standards for COPC in mine-related material are relevant and appropriate as they relate to the protectiveness of the selected remedial alternative.
20.3.13.1317 NMAC	New Mexico Protection of the General Population from Release of Radioactivity			X		Х		Same as 40 CFR 192. Not applicable but relevant and appropriate.
20.3.4 NMAC	Standards for Protection Against Radiation			Х		Х		Same as 10 CFR 20. Not applicable but relevant and appropriate.

	Description	А	ARAR Type			otential ARAR Cat	Comment	
Regulation		Chemical	Location	Action	Applicable	Relevant and Appropriate	Not Applicable, Relevant and Appropriate	
19.10.5 NMAC	Performance and Reclamation Standards and Requirements for Non-Coal Mining, Existing Mining Operations.			X		Х		This regulation provides re- vegetation requirements for existing, non-coal mining operations as well as other reclamation requirements. Relevant and appropriate.

POT	ENTIAL ADVISORIES, CRITERIA, POLICY OR	GUIDANCE TO BE CONSIDERED
Item	Citation	Discussion
Federal		
Establishment of Cleanup Levels for CERCLA Sites with Radioactive Contamination	EPA OSWER 9200.4-18	Clarifies that cleanups of radionuclides are governed by the risk range for carcinogens established in the $10^{-6}$ to $10^{-4}$ range when ARARs are not achievable or not sufficiently protective. To be
Regional Soil Screening Levels, Region 6.	http://www.epa.gov/earth1r6/6pd/rcra_c/pd- n/screen.htm	consideredThe screening levels (SLs) presented for this site were developedusing risk assessment guidance from the EPA Superfund program.They are risk-based concentrations derived from standardizedequations combining exposure information assumptions with EPAtoxicity data. SLs are considered by the Agency to be protectivefor humans (including sensitive groups) over a lifetime; however,SLs are not always applicable to a particular site. The SLscontained in the SL table are generic; they are calculated withoutsite-specific information. However, they will be considered indeveloping site-specific screening levels.
US EPA Technical Report on Technologically Enhanced Naturally Occurring Radioactive Materials ("TENORM") from uranium mining	TENORM Vol. 2: Investigation of Potential Health, Geographic, and Environmental Issues of Abandoned Uranium Mines. EPA 402-R-05-007 August 2007	Provides non-regulatory soil screening levels for external exposure, incidental ingestion, and inhalation of fugitive dust to certain radionuclides common to uranium mine sites. Also provides guidance on assessing human health and ecological risk at uranium mine sites.
State		
New Mexico Soil and Water Conservation District Act	73-20-25 NMSA	Establishes state authority to control and prevent soil erosion, prevent floodwater and sediment damage to soil, and conserve natural resources. To be considered.

Appendix C

**EE/CA Removal Alternatives Cost Estimate** 

### JOHNNY M MINE EECA REMOVAL ALTERNATIVES COST ESTIMATE

#### Removal to 3.5 pCi/g Residential Use Criterion

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### **OFF-SITE ALTERNATE ENERGY SOLUTIONS, CLIVE UTAH**

11/10/2014

Item #	Description	Material(s)	Units	\$/Unit	Quantity	Cost, \$	Cost Reference	Quantity Reference
	Direct Reclamation Costs							
1	Mobilization/ Demobilization - Construction Contractor	labor, equipment	1	lump sum	1	\$ 91,774	NM MMD Financial Assurance Guidebook	(3% of other total on-site direct)
2	Worker Health and Safety				-			
2.1	RSO, full-time	labor	hrs	\$ 130		\$ 338,000		full time for 12 months, 50 hrs/wk, 2600 hrs
2.2	Technician	labor	hrs	\$ 80	1300	\$ 104,000	ERG	half time for 12 months, 50 hrs/wk, 1300 hrs
3	Radiological Surveying, Sampling and Testing		1	I .	1 .			
3.1	Mobilization/ Demobilization	labor,equipment	lump sum	\$ 2,000			erg	
3.2	Remediation support surveys	labor,equipment	hrs	\$ 80		\$ 208,000		item 4.3
3.3 3.4	Field gamma surveying Soil sample testing	labor,equipment	lump sum each	\$ 40,000 \$ 130		\$ 40,000 \$ 179,400		Estimate based on 2000 $\text{m}^2$ SU segments for residential use (assumed), 10000 $\text{m}^2$ for industrial use. Table 4-26 SIR for areas and volumes.
4	Construction Management			I	1			
4.1	Construction superintendent	labor	wk	\$ 2,050			RSM 2014 01 31 13.20 0240	full time for 7 months, 50 hrs/wk, 1515 hours
4.2	Field engineer	labor	wk	\$ 1,350		\$ 70,200		full time for 7 months, 50 hrs/wk, 1515 hours
4.3 4.4	Ground control and volumetrics surveying Field lab operations for soil testing	labor,equipment	day wk	\$ 802 \$ 1,025		\$ 73,183 \$ -	RSM 2014 01 71 23.13 1100	quarter time
4.4	Field QC technicians for soil liner/cover	labor,equipment	wk wk	\$ 1,025			RSM 2014 01 31 13.20 0100 RSM 2014 01 31 13.20 0100	one-third time one-third time
4.6	QC records and reports clerk	labor	wk	\$ 1,023		\$ 22,620		
4.7	Field Office - 32' x 8' unit and 20' x 8' unit, furnished; 2 portable toilets		month	\$ 1,171	18	\$ 21,078		Office trailer remains on site for 18 months
4.8	Truck and equipment pressure washer	labor,equipment	month	\$ 540	12	\$ 6,480	RSM 2014 01 54 33 40 5460	
4.9	Dust control	labor,equipment	day	\$ 1,625	365	\$ 593,125	5 RSM 2014 31 23.23.20 2510	
5	Clearing of Vegetation	labor,equipment	acre	\$ 252	111	\$ 27,972	RSM 2014 31 13 13.10 0550	
6	Removal of Contaminated Mine-Related Materials							
6.1	Excavation of contaminated soil and mine waste rock		1	I	I			
	by scraper, 5000 ft haul	labor,equipment	BCY	\$ 5.26	28139	\$ 148,01	RSM 2014 31 23 16.50 2440; Caterpillar Performance Handbook	CAT 621;
	by scraper, 3000 ft haul	labor,equipment	BCY	\$ 4.31	59276	\$ 255,479	RSM 2014 31 23 16.50 2430; Caterpillar Performance Handbook	CAT 621
	by scraper, 1500 ft haul	labor,equipment	BCY	\$ 3.32	129835	\$ 431,05	RSM 2014 31 23 16.50 2420; Caterpillar Performance Handbook	CAT 621
	by excavator and truck, 0.5 m.i RT	labor,equipment	BCY	\$ 4.13	23054	\$ 95,21	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4014; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
	by excavator and truck, 1.0 mi. RT	labor,equipment	BCY	\$ 4.35	173428	\$ 754,41	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4016; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
	by excavator and truck, 2.0 mi. RT	labor,equipment	BCY	\$ 4.75	2870	\$ 13,63	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4018; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
6.2	Load for off-site haul	labor,equipment	LCY	\$ 0.50	499921	\$ 249,960	RSM 2014 31 23 16.43 0450; Caterpillar Performance Handboo	k CAT 990F with 10 cv bucket: total BCY *1.2
6.3	Transport by rail from Milan, NM siding	· · · ·		1.	1			
	Haul by truck to Milan siding	labor,equipment	LCY	\$ 5.86	499921	\$ 2,929,536	RSM 2014 31 23 23.20 4110-4112	20 CY truck, 18 miles site to siding
	Loading into gondola cars	labor,equipment	LCY	\$ 0.95	499921	\$ 474,925	5 RSM 2014 31 23 16.42 1601	load from truck to rail car with CAT 980
	Gondola mobilization fee	equipment	each	\$ 5,500	100	\$ 550,000	MHF Services, email 3/13/14	per 110 T capacity gondola car
	Gondola lease rate	equipment	month	\$ 625			MHF Services, email 3/13/14	per 110 T capacity gondola car
	Rail rate to EnergySolutions	equipment	car trip	\$ 10,000	5454	\$ 54,536,808	MHF Services, email 3/13/14	per 110 T capacity gondola car, 1.2 T/CY
6.4	Transport by truck from mine site		4.5		00040	<b>A A A A A A A A A A</b>		
7	Haul by truck and dump - EnergySolutions, Clive, UT	labor,equipment	trip	\$ 6,240	26312	\$ 164,184,49	MHF Services, email 3/13/14	23 tons or 19 cy/trip
7	Disposal at Licensed Facility EnergySolutions	fee	LCY	\$ 45	499921	\$ 22,496,433	EnergySolutions email 4/1/14	\$30/T, est. 1.2 T/cy
8	Site Restoration	100	101	μ 40	700021	ψ 22,490,43		φου, τ, σαι. τ.2. τ/σχ
Ť	Final grading	labor,equipment	acre	\$ 774.40	111	\$ 85,958	RSM 2014 31 22 16.10 3300	Grade fo drainage and revegetation
	Revegetation - seeding with mulch and fertilizer	labor,equipment	acre	\$ 938.28			P RSM 2014 32 92 19.14 5300	Drll seeding with mulch and fertilizer
9	Post-Removal Site Controls							
	Annual Inspections and Reports	labor	Yr	\$ 1,714.00	12	\$ 20,568	RSM 2014 01 13 13.20	one trip per year, scientist and technician for two days
TOTAI	DIRECT COST			By Rail		\$ 85,593,063	k	
				By Truck	C C	\$ 190,723,790	·	
	Estimated dynation of construction –		monthe	50 (	woole		5 days two 6 month work noriads concreted by 6 mor	

Estimated duration of construction =

12 months 52.0 weeks 365 days, two 6-month work periods separated by 6 months

Waste volume for removal. Based on clean-up to 3.5 pCi/g Ra-226, is 318,502 m<sup>3</sup> or 416,601 CY bank volume; or ~500,000 LCY Time-based work assumes half the waste volume is removed in each of two consecutive years, six months each year

#### JOHNNY M MINE EECA REMOVAL ALTERNATIVES COST ESTIMATE

Removal to 3.5 pCi/g Residential Use Criterion

#### OFF-SITE ALTERNATE WASTE CONTROL SPECIALISTS, ANDREWS, TEXAS

11/10/2014

Item #	Description	Material(s)	Units	\$/Unit	Quantity	Cost, \$	Cost Reference	Quantity Reference
	Direct Reclamation Costs		I			L		
1	Mobilization/ Demobilization - Construction Contractor	labor,equipment	1	lump sum	1	\$ 91,774	NM MMD Financial Assurance Guidebook	(3% of other total on-site direct)
	Worker Health and Safety		r			r I		
2.1	RSO, full-time	labor	hrs	\$ 130	2600	\$ 338,000		full time for 12 months, 50 hrs/wk, 2600 hrs
2.2	Technician Radiological Surveying, Sampling and Testing	labor	hrs	\$ 80	1300	\$ 104,000	ERG	half time for 12 months, 50 hrs/wk, 1300 hrs
3.1	Mobilization/ Demobilization	labor,equipment	lump sum	\$ 2.000	1	\$ 2.000	FRG	
3.2	Remediation support surveys	labor,equipment	hrs	\$ 80	2600	\$ 208,000		item 4.3
3.3	Field gamma surveying	labor,equipment	lump sum	\$ 40,000	1	\$ 40,000		
3.4	Soil sample testing	contract lab	each	\$ 130	1380	\$ 179,400		Estimate based on 2000 m <sup>2</sup> SU segments for residential use (assumed), 10000 m <sup>2</sup> for industrial use. Table 4-26 SIR for areas and volumes.
4	Construction Management				I	· · · · · · · · · · · · · · · · · · ·		
4.1	Construction superintendent	labor	wk	\$ 2,050	52.0		RSM 2014 01 31 13.20 0240	full time for 7 months, 50 hrs/wk, 1515 hours
4.2	Field engineer	labor	wk	\$ 1,350	52.0		RSM 2014 01 31 13.20 0120	full time for 7 months, 50 hrs/wk, 1515 hours
4.3	Ground control and volumetrics surveying	labor,equipment	day	\$ 802	91	\$ 73,183	RSM 2014 01 71 23.13 1100	quarter time
4.4	Field lab operations for soil testing	labor,equipment	wk	\$ 1,025	0	\$-	RSM 2014 01 31 13.20 0100	one-third time
4.5	Field QC technicians for soil liner/cover	labor	wk	\$ 1,025	0	\$-	RSM 2014 01 31 13.20 0100	one-third time
4.6	QC records and reports clerk	labor	wk	\$ 435	52	\$ 22,620	RSM 2014 01 31 13.20 0020	
4.7	Field Office - 32' x 8' unit and 20' x 8' unit, furnished; 2 portable toilets	equipment	month	\$ 1,171	18	\$ 21,078	RSM 2014 01 52 32.20; 01 52 13 40; 01 54 33 40 6410	Office trailer remains on site for 18 months
4.8	Truck and equipment pressure washer	labor,equipment	month	\$ 540	12	\$ 6,480	RSM 2014 01 54 33 40 5460	
4.9	Dust control	labor,equipment	day	\$ 1,625	365	\$ 593,125	RSM 2014 31 23.23.20 2510	
5	Clearing of Vegetation	labor,equipment	acre	\$ 252	111	\$ 27,972	RSM 2014 31 13 13.10 0550	
6	Removal of Contaminated Mine-Related Materials					•		
6.1	Excavation of contaminated soil and mine waste rock							
	by scraper, 5000 ft haul	labor,equipment	BCY	\$ 5.26	28139		RSM 2014 31 23 16.50 2440; Caterpillar Performance Handbook	CAT 621;
	by scraper, 3000 ft haul	labor,equipment	BCY	\$ 4.31	59276	\$ 255,479	RSM 2014 31 23 16.50 2430; Caterpillar Performance Handbook	CAT 621
	by scraper, 1500 ft haul	labor,equipment	BCY	\$ 3.32	129835	\$ 431,051	RSM 2014 31 23 16.50 2420; Caterpillar Performance Handbook	CAT 621
	by excavator and truck, 0.5 m.i RT	labor,equipment	BCY	\$ 4.13	23054	\$ 95,211	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4014; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
	by excavator and truck, 1.0 mi. RT	labor,equipment	BCY	\$ 4.35	173428	\$ 754,411	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4016; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
	by excavator and truck, 2.0 mi. RT	labor,equipment	BCY	\$ 4.75	2870	\$ 13,631	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4018; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
6.2	Load for off-site haul	labor,equipment	LCY	\$ 0.50	499921	\$ 249,960	RSM 2014 31 23 16.43 0450; Caterpillar Performance Handbook	CAT 990F with 10 cy bucket; total BCY *1.2
6.3	Transport by rail from Milan, NM siding							
	Haul by truck to Milan siding	labor,equipment	LCY	\$ 5.86	499921		RSM 2014 31 23 23.20 4110-4112	20 CY truck, 18 miles site to siding
	Loading into gondola cars Gondola mobilization fee	labor,equipment	LCY each	\$ 0.95 \$ 5,500	499921 100		RSM 2014 31 23 16.42 1601 MHF Services, email 3/13/14	load from truck to rail car with CAT 980 per 110 T capacity gondola car
	Gondola mobilization ree Gondola lease rate	equipment equipment	month	\$ 5,500 \$ 625	900		MHF Services, email 3/13/14 MHF Services, email 3/13/14	per 110 T capacity gondola car per 110 T capacity gondola car
	Rail rate to Waste Control Specialists	equipment	car trip	\$ 7,825	5454		MHF Services, email 3/13/14	per 110 T capacity gondola car, 1.2 T/CY
6.4	Transport by truck from mine site			· ·				
	Haul by truck and dump - WCS, Andrews, TX	labor,equipment	trip	\$ 4,440	26312	\$ 116,823,583	MHF Services, email 3/13/14	23 tons or 19 cy/trip
7	Disposal at Licensed Facility	6	1.01/	¢ 000	400001	¢ 00.004.110		\$200/m ant 4.0 T/m
8	Waste Control Specialists	fee	LCY	\$ 200	499921	\$ 99,984,148	Jeff Havlichak, WCS, 4/25/14	\$200/cy, est 1.2 T/cy
0	Site Restoration Final grading	labor equipment	acre	\$ 774.40	111	\$ 85.958	RSM 2014 31 22 16.10 3300	Grade fo drainage and revegetation
	Revegetation - seeding with mulch and fertilizer	labor,equipment labor,equipment	acre	\$ 938.28			RSM 2014 32 92 19.14 5300	Drll seeding with mulch and fertilizer
9	Post-Removal Site Controls	assi,oqupriont						
-	Annual Inspections and Reports	labor	Yr	\$ 1,714.00	12	\$ 20,568	RSM 2014 01 13 13.20	one trip per year, scientist and technician for two days
τοται	DIRECT COST			By Rail	12	\$ 151,219,022		ene any per year, colonies and technician for two days
				By Truck		\$ 220,850,593		
	Estimated duration of construction =	40	months		weeks		days, two 6-month work periods separated by 6 mor	

Estimated duration of construction =

12 months 52.0 weeks

365 days, two 6-month work periods separated by 6 months

Waste volume for removal. Based on clean-up to 3.5 pCi/g Ra-226, is 318,502 m<sup>3</sup> or 416,601 CY bank volume; or ~500,000 LCY

Time-based work assumes half the waste volume is removed in each of two consecutive years

#### JOHNNY M MINE EECA REMOVAL ALTERNATIVES COST ESTIMATE

Removal to 3.5 pCi/g Residential Use Criterion

Item # Description	Material(s)	Units	\$/Unit	Quantity	Cost, \$	Cost Reference	Quantity Reference
Direct Reclamation Costs							
1 Mobilization/ Demobilization - Construction Contractor	labor, equipment	1	3% of other direc	t 1	\$ 147,206	NM MMD Financial Assurance Guidebook	(3% of other total on-site direct)
2 Worker Health and Safety				1			
2.1 RSO, full-time	labor	hrs	\$ 130	1213	\$ 157,733	ERG	full time for 7 months, 50 hrs/wk, 1213 hrs
2.2 Technician	labor	hrs	\$ 80	607	\$ 48,533	ERG	half time for 7 months, 50 hrs/wk, 607 hrs
3 Radiological Surveying, Sampling and Testing							
3.1 Mobilization/ Demobilization 3.2 Remediation support surveys	labor, equipment labor, equipment	lump sum hrs	\$ 2,000 \$ 80		÷ _,	ERG	item 4.3
3.3 Field gamma surveying	labor, equipment	lump sum	\$ 40,000			ERG	item 4.5
3.4 Soil sample testing	contract lab	each	\$ 130		\$ 179,400		Estimate based on 2000 m <sup>2</sup> SU segments for residential use (assumed), 10000 m <sup>2</sup> for industrial use. Table 4-26 SIR for areas and volumes.
4 Construction Management							
4.1 Construction superintendent	labor	wk	\$ 2,050	30.3	\$ 62,183	RSM 2014 01 31 13.20 0240	full time for 7 months, 50 hrs/wk, 1515 hours
4.2 Field engineer	labor	wk	\$ 1,350		\$ 40,950	RSM 2014 01 31 13.20 0120	full time for 7 months, 50 hrs/wk, 1515 hours
4.3 Ground control and volumetrics surveying	labor, equipment	day	\$ 802		\$ 85,380	RSM 2014 01 71 23.13 1100	half itme
4.4 Field lab operations for soil testing 4.5 Field QC technicians for soil liner/cover	labor, equipment labor	wk	\$ 1,025 \$ 1,025			RSM 2014 01 31 13.20 0100 RSM 2014 01 31 13.20 0100	one-third time
4.6 QC records and reports clerk	labor	wk	\$ 435			RSM 2014 01 31 13.20 0020	full time for 7 months, 50 hrs/wk, 1515 hours
4.7 Field Office - 32' x 8' unit and 20' x 8' unit, furnished; 2 portable toilets	equipment	month	\$ 1,171			RSM 2014 01 52 32.20; 01 52 13 40; 01 54 33 40 6410	
4.8 Truck and equipment pressure washer	labor, equipment	month	\$ 540			RSM 2014 01 54 33 40 5460	
4.9 Dust control	labor, equipment	day	\$ 1,625	213		RSM 2014 31 23.23.20 2510	
5 Clearing of Vegetation	labor, equipment	acre	\$ 252	125	\$ 31,438	RSM 2014 31 13 13.10 0550	SIR Table 4-26 and Figure 4-33
6 Removal of Contaminated Mine-Related Materials		1		1	1		
6.1 Excavation of contaminated soil and mine waste rock							
by scraper, 5000 ft haul	labor,equipment	BCY	\$ 5.26	28139	\$ 148.011	RSM 2014 31 23 16.50 2440; Caterpillar Performance Handbook	CAT 621;
by scraper, 3000 ft haul	labor,equipment	BCY	\$ 4.31				
by scraper, 1500 ft haul		BCY	\$ 3.32	00210	\$ 431.051	RSM 2014 31 23 16.50 2430; Caterpillar Performance Handbook	CAT 621
	labor,equipment		+		+	RSM 2014 31 23 16.50 2420; Caterpillar Performance Handbook	CAT 621
by excavator and truck, 0.5 m.i RT	labor,equipment	BCY	\$ 4.13	23054	\$ 95,211	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4014; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
by excavator and truck, 1.0 mi. RT	labor,equipment	BCY	\$ 4.35	173428	\$ 754,411	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4016; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
by excavator and truck, 2.0 mi. RT	labor,equipment	BCY	\$ 4.75	2870	\$ 13,631	RSM 2014 31 23 16.13 1030; RSM 21 23 23.20 4018; Caterpillar Performance Handbook	CAT 350 with 3 cy bucket, CAT D300E with 20 cy capacity
6.2 Spread and compact in repository location	labor, equipment	LCY	\$ 1.48	499921	\$ 739.883	RSM 2014 31 23 16.46 6006; Caterpillar Performance Handbook	CAT D10, < 100 ft push; SIR Table 4-26
7 Construction of Repository			1	1	1.		Cover area is 11 acres
7.1 Excavation of soil for liner and cover							
Rip shale	labor, equipment	BCY	\$ 1.63	58817	\$ 95,872	RSM 2014 31 23 16.32 2200	CAT D8 with ripper;
Disc and windrow shale	labor, equipment	BCY	\$ 2.05	58817	\$ 120,576	RSM 2014 31 23 16.32 3200	CAT D8 short push
Load, haul, and place reworked shale	labor, equipment	LCY	\$ 3.58	70581	\$ 252,680	RSM 2014 31 23 16.50 2300; Caterpillar Performance Handbook	CAT 621, 1500 ft ave.
7.2 Moisture conditioning of shale clay	labor, equipment	LCY	\$ 1.50	70581		RSM 2014 31 23 23.23 9030	6000 gal water truck, 1/2 mile haul
7.3 Spread and compact - clay liner	labor, equipment	LCY	\$ 1.01			RSM 2014 31 23 23.23 5620	CAT 825C sheepsfoot w/ blade, 3 passes
7.4 Spread and compact - low permeability layer of cover	labor, equipment	LCY	\$ 1.01			RSM 2014 31 23 23.23 5621	CAT 825C sheepsfoot w/ blade, 3 passes
7.5 Excavation of sandy soil for evapotranspiration layer of cover	labor, equipment	BCY	\$ 3.32			RSM 2014 31 23 16.50 2300; Caterpillar Performance Handbook	Excavate, load, haul and place with CAT 621 scrapers; 1500 ft ave.
7.6 Spread and compact - evapotranspiration layer	labor, equipment	LCY	\$ 1.48	35171	\$ 52,053	RSM 2014 31 23 16.46 6006; Caterpillar Performance Handbook	CAT D10 short push;
8 Erosion Protection							Cover area is 11 acres
8.1 Rock fragmentation Ripping sandstone	labor, equipment	BCY	\$ 4.06	11799	\$ 47,902	RSM 2014 31 23 16.32 1600	CAT D8 with ripper; assume 30% loss from intact volume
Drilling and blasting	labor, equipment	BCY	\$ 11.81		\$ 41,502	RSM 2014 31 23 16.30 0100	only for resistant rock, not rippable
		BCY	\$ 1.67	-	+		MB America BF90 crusher bucket on CAT 320 excavator, minus 4 inche
8.2 Crushing	labor, equipment					MB America Robbett Eyler, pers comm Nov 2013; RSM 2014 31 23 16.42 0300	46 CY/hr
8.3 Screening	labor, equipment	Day	\$ 1,424			RSM 2014 01 54 33 3710; crew A-3C	150-200 CY/day
8.4         Load, haul and dump           8.5         Riprap placement by machine	labor, equipment	LCY	\$ 3.19	15338	\$ 48,928	RSM 2014 31 23 16.42 1601; RSM 2014 31 23 23.20 0026	CAT 980, 8 CY trucks, 0.5 mi RT
Channel riprap	labor, equipment	CY	\$ 149	444	\$ 66,222	RSM 31 37 13.10 0200	Machine placed, up to 1.5 ft thick
Slope riprap/ rock mulch	labor, equipment	CY	\$ 27.52			RSM 31 37 13.10 0300	Placed on slope for rock mulch or spreading in finish grading
9 Site Restoration		<u> </u>	1		1		,
9.1 Final grading	labor, equipment	acre	\$ 774.40	125	\$ 96,800	RSM 2014 31 22 16.10 3300	Grade all disturbed ground for drainage and revegetation
9.2 Revegetation	, (						······································
Mulch	labor, equipment	acre	\$ 2,427			RSM 2014 32 91 13.16 0350	hay straw mulch, power mulcher
Seeding	labor, equipment	acre	\$ 938.28			RSM 2014 32 92 19.14 5300	tractor spreader
	labor, equipment	LF	\$ 0.27	3000	\$ 5,865	RSM 2014 32 31 26.20 0015; 32 31 13.40 2360	three strand barbed wire w/ one gate, 3400 ft.
9.3 Fencing							
10 Post-Removal Site Controls		1					
10         Post-Removal Site Controls           10.1         Annual Inspections and Reports	labor	Yr	\$ 1,714			RSM 2014 01 13 13.20	one trip per year, scientist and technician for two days
10 Post-Removal Site Controls	labor labor, equipment	Yr Day	\$ 1,714 \$ 1,447.32			RSM 2014 01 13 13.20 RSM 2014 Crew B-11C	one trip per year, scientist and technician for two days two person crew with backhoe loader for light repairs, two days annually
10         Post-Removal Site Controls           10.1         Annual Inspections and Reports							

Waste volume for removal based on clean-up to 3.5 pCi/g Ra-226, is 317,183 m<sup>3</sup> or 414875 BCY. All of this volume is removed and placed in the C North repository. Repostory surface area is: 10.9 acres 474,804 sq ft LCY = BCY x 1.2