Technical Report

JUNE 1992

APPLIED RESEARCH: GROUND-WATER CONTAMINANT TRANSPORT AT ROCKY MOUNTAIN ARSENAL

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LIST OF ACRONYMS

Bicyclo[2,2,1]hepta-2,5-diene
Dicyclopentadiene
cubic centimeter per gram
Critical Micelle Concentration
Concentration of Organic Carbon
Colloids
p-Chlorophenylmethyl sulfide
Comprehensive Monitoring Program
Diisopropylmethylphosphonate
Dimethylmethylphosphonate
Dissolved Oxygen
Dissolved Organic Carbon
Electron Impact
Environmental Protection Agency (U.S.)
feet below ground surface
gram per cubic centimeter
Granular Activated Carbon
Gas Chromatograph/Mass Spectrometer
Gas Chromatograph
Gallons Per Minute
Liter
Bithium Bromide
Methylene Blue Active Substance
milliliters
North Boundary Containment System
Internal Standards
Polycyclic Aromatic Hydrocarbons
Tetrachloroethene
Photo Ionization Detector
Program Manager Office
Program Manager, Rocky Mountain Arsenal
2,2-Bis(para-chlorophenyl)-1,1-dichloroethene
2,2-Bis(para-chlorophenyl)-1,1,1-trichloroethane
Poly Vinyl Chloride
quart
R.L. Stollar and Associates Inc.
Rocky Mountain Arsenal
Selected Ion Monitoring
Solid Phase Extraction
Trichloroethane
Trichloroethene
United States Geological Survey
University of Waterloo
Volatile Organic Carbon
microliters per kilogram
microgram per liter
microliters

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EXECUTIVE SUMMARY

An applied research project was conducted in a high-concentration, mixed-contaminant plume in Section 23 of the Rocky Mountain Arsenal. The purpose of the project was to assess the contaminant transport characteristics in the aquifer to provide information to support remedial system design. The transport behavior of the contaminants was assessed in the field by monitoring the rate at which contaminants were flushed from the aquifer by organic-free water. Organic-free water was injected into the aquifer at one well and ground water was extracted at the same rate from another well 30 feet away. The flushing rates of the organic contaminants were monitored with a cluster of short-screened monitoring wells located between the injection and extraction wells. Monitoring was conducted during a 10-day period of injection/extraction (the "forced-gradient" portion), and for approximately 2 weeks after injection/extraction ceased (the "natural-gradient" portion).

An assessment of background conditions conducted prior to extraction/injection indicated that existing contamination was stratified vertically and contained a wide range of contaminants including halogenated volatile organics, aromatic hydrocarbons, organochlorine pesticides and other, more polar, organic compounds. The total dissolved organic carbon concentration (DOC) in the ground water was approximately 100 mg/L; approximately 70 mg/L are uncharacterized by historical ground water chemistry data.

Monitoring conducted during the forced-gradient portion of the experiment revealed that, contrary to expectations, the concentrations of the measured organic contaminants were reduced at the monitoring wells to near or below detection levels, essentially at the same time that the clean water flush arrived at the monitoring points (i.e. no retardation). This occurred despite the fact that the properties and expected mobilities of the contaminants varied widely.

Laboratory retardation tests indicated that aquifer sediments obtained by coring exhibited a significant capacity for sorption. Laboratory analysis of extracts from the sediments, however, indicated that in the aquifer the sorbed component of contamination was near or below detection levels. These apparently contrasting results indicate that the increased mobility measured during the forced gradient portion of the experiment results from constituents in the ground water rather than from properties of the aquifer sediments.

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Several potential ground water constituents which could cause or contribute to the observed enhanced mobility were considered:

- Colloidal material was found to be present and mobile in the aquifer. Analysis of the colloidal material indicated a significant load of sorbed contaminants per unit mass of colloid. The concentration of colloids in the ground water, however, is low enough that the net effect on contaminant migration in this aquifer is not significant.
- Surfactants were investigated in the aquifer because of their ability to increase the solubility of hydrophobic compounds. Results indicate that anionic surfactants are not present in concentrations sufficient to explain the level of mobility enhancement. Non-ionic surfactants could not be investigated during the experiment, but could be a factor in the enhanced mobility if they constitute the majority of the uncharacterized DOC. This is unlikely because non-ionic and anionic surfactants are commonly associated, so the lack of anionic surfactants suggests that the non-ionic surfactants are also unlikely to be present.
- Co-solvent effects are judged to not be a significant factor because the concentrations required to enhance contaminant mobility greatly exceed the concentrations of total DOC in the ground water.

Monitoring conducted during the natural gradient portion of the experiment revealed that contaminant concentrations in the monitoring wells rose slowly over a period of two weeks. Tracer data indicate that the concentration increases likely result from diffusion and possible advection from reservoirs of contamination which were not flushed during the forced gradient portion of the test, rather than advection of the displaced plume back to the monitoring wells. Possible reservoirs may include the underlying Denver Formation bedrock, the overlying capillary fringe, and/or thin, lower-permeability strata distributed randomly within the aquifer.

This project demonstrates that small-scale pilot tests may be conducted within existing plumes of contamination to yield contaminant transport information which can strongly facilitate the design of remediation systems. In particular, the technique of extracting, treating, and reinjecting ground water to flush contaminants from the aquifer was shown to be effective at this site. Further application of this

technique could involve a specialized treatment system which takes advantage of the mobility-enhancing constituents determined to be present in the site ground water. This specialized system, ideally, would remove targeted contaminants from the ground water while allowing the mobility-enhancing agents to be reinjected with the flush. A first step toward researching such a system - lab verification of the extracting capabilities of the site ground water - will be conducted by D. Mackay and students under limited funding available at the University of Waterloo. In addition to this work, it is recommended that the unidentified fraction of the total DOC be fully characterized.

Other information gained that is applicable to remedial design includes the finding that dissolved oxygen moves conservatively through the aquifer. This suggests that oxygen could be delivered relatively easily to large volumes of the subsurface if it proved desirable to encourage in-situ aerobic bioremediation for enhancement of aquifer remediation. Additionally, the identification of potential contaminant reservoirs which remained in the aquifer after the initial flush suggests that a pulsed pumping approach might be more efficient than standard continuous pump-and-treat practices. Also, innovative and cost effective analytical systems developed and utilized during this experiment could be applied to remediation system performance monitoring, as well as general ground-water monitoring efforts, for potentially significant cost savings. Finally, the pilot study provided information on the performance of a treatment technology application (granular activated carbon) under actual site conditions.

1.0 INTRODUCTION

R.L. Stollar and Associates (RLSA), in collaboration with the University of Waterloo (UW) Centre for Groundwater Research, conducted an applied research project involving contaminant transport in ground water at Rocky Mountain Arsenal(RMA), Commerce City, Colorado. The project was conducted for, and primarily funded by, the Program Manager for Contamination Cleanup, Rocky Mountain Arsenal. Additional funding was provided by RLSA, and the UW. Other organizations that contributed to the project include the United States Geological Survey (USGS); Texas A&M University; Analytic and Remedial Technology, Inc.; and the Grundfos Pump Corporation.

1.1 <u>THE PROBLEM</u>

A significant challenge associated with many environmental restoration programs involves the remediation of contaminated ground water. The Environmental Protection Agency (EPA) has estimated that 45 percent of large public water systems served by ground water are contaminated with human-introduced chemicals (Miller, 1990). Consequently, the ability to remove contaminants from ground-water aquifers is important.

The ability to remove contaminants from an aquifer depends on the transport characteristics of the contaminants in the particular aquifer. Historically, contaminant transport characteristics have been estimated primarily through laboratory studies on saturated sediments. Unfortunately, laboratory results can vary considerably from the transport characteristics observed in actual aquifers. This has been observed at the RMA where the travel distances of several contaminants often exceed the travel distances expected based on laboratory tests. Variations between laboratory results and field results have also been observed in a significant number of pump and treat programs across the country which report that projected clean-up time frames will be greatly exceeded because the contaminants are removed from the aquifer at much slower rates than predicted from laboratory studies.

Inconsistencies between laboratory data and actual field observations can be at least partially attributed to the large number of complex chemical and physical processes which contribute to contaminant transport behavior in aquifers. These processes are difficult to accurately duplicate in a laboratory. They include sorption, volatilization, degradation, transformation, filtration, advection, and diffusion. The effects of these processes on contaminant transport characteristics are related to physical and chemical properties of the contaminants, the aquifer materials, and the ground water. Hence, transport characteristics can vary between contaminants in an aquifer, between aquifers for a contaminant, and, given the heterogeneous nature of many aquifer systems, between locations within an aquifer. Failure to identify the transport characteristics of contaminants in an actual aquifer can result in misapplied or ineffective remedial actions.

1.2 <u>PURPOSE/GOALS</u>

The purpose of this research is to assess the transport characteristics of contaminants in a complex, highlevel, mixed-contaminant plume at RMA. Specific goals of the research include estimating the retardation factors of target contaminants in the aquifer, and evaluating potential in-situ transport-enhancing mechanisms. The results are intended to support remediation system design.

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2.0 APPROACH/DESIGN

The transport characteristics of the contaminants in the RMA aquifer were assessed by measuring their elution rates in the presence of a constant flush of organic-free water. Similar tests have been conducted in the western tier of RMA (Mackay, 1990) and in Gloucester, Ontario (Bahr, 1989). The organic-free water was pumped into the aquifer at an injection well; contaminated water was pumped from the aquifer at an extraction well 30 ft away. The contaminated water pumped from the extraction well was treated and re-injected as the flush of organic-free water. The elution rates of the contaminants in the aquifer were measured through repeated sampling and analysis of ground water from three partially penetrating monitoring wells. The monitoring wells were located between the injection and extraction wells. Each monitoring well screened a unique interval in the aquifer. Together, the monitoring wells screened the entire thickness of the aquifer. Samples from the monitoring wells were collected and chemically analyzed during the period of extraction and injection, and for roughly two weeks after pumping was stopped. Sample frequencies at each monitoring well ranged from 1 every two hours to 1 per day, depending on the timeframe of the experiment. The pumping portion of the experiment is referred to as "forced gradient"; the non-pumping portion is referred to as "natural gradient".

The implementation of the experiment involved several elements: the selection of an appropriate site; a baseline characterization of the site (including the simultaneous development of appropriate analytical protocols), the modeling, design, and installation of a well network for injection, extraction, and monitoring; and the engineering and construction of a process system to treat and re-inject ground water.

2.1 <u>SITE SELECTION</u>

The RMA is a 27-square-mile industrial and military facility with a complex and extensive ground-water contamination scenario resulting from:

- many contaminant sources, some areally separated, some overlapping;
- a 40-year span of contaminant releases involving changes in waste stream make-up, discharge/spill periods, and ground-water flow patterns; and

many contaminants.

In choosing the site for the experiment, a high level, multi-component plume was desired. Three locations were initially examined: Basin A Neck, South Plants, and Section 23 (Figure 2.1-1). The site selection process was based on the following desired characteristics:

1. Shallow depth to ground water (approximately 20 ft maximum).

This would allow peristaltic pumps to be used to obtain the ground-water samples, and minimize drilling efforts.

2. Relatively simple hydrogeology.

The presence of fractures, heterogeneous lithologies, nearby pumping influences, and other complexities could make the interpretation of measured elution rates difficult.

3. Contaminants of interest in the aquifer at concentrations at least an order of magnitude higher than the laboratory detection limit.

This ensures that reliable concentration measurements can be made across a range which clearly defines an elution curve.

4. Relatively flat hydraulic gradient.

One purpose of the extraction/injection pumping is to dominate and control ground-water flow. An initially flat gradient can generally be more easily influenced than an initially steep gradient.

5. Stable water table.

Water table fluctuations can alter the saturated zones in the aquifer, ground-water flow rates and directions, contaminant concentrations, and pumping rates - all increase the complexity of interpreting elution measurements.

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6. Relatively permeable aquifer.

Allows the test to be conducted in a reasonable time frame.

7. Limited saturated thickness.

Reduces the depth of well screens, injection/extraction rates, and treatment process capacity, minimizing experimental costs and efforts.

8. Located away from plume boundaries.

The experiment site should include a plume volume that is laterally uniform at least over the volume expected to be impacted by the experiment.

The Basin A location was not selected because of fluctuations in water levels and flow that could be caused by the nearby ground-water intercept system. The South Plants location was not selected because it overlies the plume boundaries of several target contaminants.

The Section 23 location was chosen as the experimental site because its characteristics most closely match the desired site characteristics. It is located in the southeastern quadrant of section 23, approximately one-quarter mile downgradient of historical Basin F (Figure 2.1-1).

2.1.1 HYDROGEOLOGY

An unconfined alluvial aquifer exhibiting a high-level, multi-component plume exists beneath the Section 23 site. The geology of the aquifer was investigated through visual logging during drilling of wells and soil borings. Figure 2.1-2 presents a schematic cross section of the site geology based on these drilling

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Horizontal & Vertical Scales : 1 inch equals 3 ft. Vertical Exageration : None Well Bore Diameters not to Scale

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Plan Orientation of Cross-Section (See Section 2.3 for Discussion of Well Network)



logs. A description of the well network is presented in Section 2.3. The geology consists of gravelly fine to medium sand with occasional thin (1 to 2 in) clay layers. The clay layers do not appear contiguous from boring to boring. Gravelly zones are more predominant between 44 and 48 feet. There appears to be a general grain size coarsening with depth between approximately 38 and 48 ft in borings I, B2, and E. The unconfined aquifer is bounded below by the Denver Formation bedrock, which consists of claystone and siltstone. The bottom foot of the unconfined aquifer exists within a zone of weathered Denver Formation claystone. The depth to water is approximately 38 ft below ground surface (bgs). The saturated thickness of the unconfined aquifer is approximately 14 ft.

One drawback to the section 23 site was the depth to water (38 ft) which precluded the use of a simplified peristaltic sampling system, and increased drilling expenses. However, this drawback was determined to be less detrimental than the drawbacks associated with the other potential sites.

Historical information obtained from the Comprehensive Monitoring Program (CMP) (Stollar 1989, 1990, and 1991) indicates that water levels in the area are relatively stable (Figure 2.1-3). The hydraulic gradient across the site appeared, from Fall 1990 data, to be approximately 0.004 ft/ft toward the north-northeast.

Additional hydraulic information on the aquifer was provided prior to the experiment by Dr. Jim Warner of Colorado State University (Warner, 1991). Dr. Warner is modeling a ground-water containment system located approximately 3,000 ft north of the experiment site. The experimental data were gathered in 1990 and the hydraulic information supplied are based on the model calibration results:

- hydraulic conductivity: 120 to 400 ft/day
- flow velocity: approximately 0.04 ft/day
- transmissivity: approximately 3000 ft²/day
- storage coefficient: 0.05

2.1.2 CHEMISTRY

The chemistry of the aquifer is strongly influenced by its location downgradient from a former disposal basin, Basin F (Figure 2.1-1). Basin F was used as a containment basin for RMA waste streams from 1956 to 1980.



Ground water from the aquifer exhibits a large number of contaminants with widely varying physical/chemical properties. The known contaminants include organochlorine pesticides, organophosphorous compounds, organosulfur compounds, volatile aromatics, volatile organohalogens, volatile hydrocarbons, phosphonates, metals, cations, and anions.

Historical ground-water monitoring efforts indicated the presence of high-level contaminants, including chloroform (1,000 to 10,000 μ g/L), dicyclopentadiene (500+ μ g/L), diisomethylphosphonate (1,000+ mg/L), organosulfur compounds (100 to 1,000 μ g/L, summed), chloride (1,000+ μ g/L), and fluoride (5,000+ μ g/L) (RLSA, 1991). Specific historical chemical ground-water data are presented in Appendix A, which contains the most recent three years of analytical data collected from an existing, fully penetrating well (23095) in the vicinity of the experiment site.

2.2 BASELINE CHARACTERIZATION/ANALYTICAL PROTOCOLS

A baseline physical and chemical characterization of the aquifer was conducted to support the design parameters of the experiment and to facilitate the interpretation of the final results. This characterization included information on:

- · Chemical composition of the ground water,
- · Chemical stratification in the aquifer,
- · The quantitation of contaminants sorbed on the aquifer sediments,
- The sorption capacity of the aquifer sediments,
- · The presence of colloids and dissolved organic carbon, and
- The presence of surfactants in the aquifer.

Each of these aspects of the baseline characterization is discussed below. Analytical protocols were developed during the baseline characterization, and are included in the following discussions.

2.2.1 CHEMICAL COMPOSITION OF THE GROUND WATER

The measurement of contaminant elution rates during the experiment involved the periodic sampling and analysis of ground water from three monitoring wells at frequencies which ranged from one sample per hour at the beginning of the experiment to one sample per day near the end of the experiment. The development of appropriate analytical protocols for this monitoring required preliminary testing to characterize baseline ground-water chemistry in order to support methods development. The analytical protocols involved the determination of field parameters, concentrations of inorganic constituents, and concentrations of organic compounds.

2.2.1.1 Field Parameters

Field parameter data were collected from the three monitoring wells, the extraction well, and the injection well during the experiment. The field parameters consisted of pH, electric conductivity, alkalinity, dissolved oxygen, and temperature. Measurements were taken at each well approximately once every six hours during the first 72 hours of the experiment. Thereafter, field parameter measurements were conducted once per day. The measurements were obtained using a YSI model electric conductivity meter, a YSI dissolved-oxygen meter, a Beckman pH meter and thermometer, and a Hach alkalinity titration kit. The field parameter measurements from the monitoring wells provided information on the effect of the organic-free-water flush on the aquifer. The field parameter measurements from the extraction and injection wells were primarily used to monitor the effect of the treatment system on the extraction water before re-injection. Baseline field parameters were primarily measured to identify possible interferences with the analytical methods, as well as possible complications (i.e., corrosivity, salinity) with treatment and re-injection of the extraction water.

2.2.1.2 Inorganics

The inorganic analyses were conducted by the USGS National Water Quality Inorganic Laboratory. The inorganic methods and associated analytes were:

Ion Chromatography

- 1. chloride
- 2. fluoride
- 3. sulfate

Colorimetric

- 4. Bromide
- 5. Ammonia
- 6. Nitrite
- 7. Nitrite + Nitrate
- 8. Ortho Phosphate

Atomic Absorption Spectroscopy

- 9. sodium
- 10. potassium
- 11. magnesium
- 12. calcium

Descriptions of these methods can be found in Methods for the Determination of Inorganic Substances in Water and Fluvial Sediments (USGS, 1989a).

Because bromide was used as the tracer, samples for the bromide analysis were collected and analyzed separately from the other inorganic analytes.

2.2.1.3 <u>Organics</u>

The organic analyses were conducted using two innovative techniques designed to allow the large number of samples generated during the experiment to be processed accurately and cost effectively. The first technique applies to the volatile organic compound (VOC) fractions which were analyzed in an on-site laboratory. Ground-water samples were pumped via a semi-automatic manifold into one of seven sample loops, each with a different volume. The appropriate sample loop was chosen based on the anticipated concentration range of volatile contaminants in the sample and the volume of the sample loop. Once the sample was pumped into the appropriate loop, it was sent to a purge and trap concentrator, then a small gas chromatograph (GC) (SRI Inc.) with a Hall/Photo Ionization Detector (PID) combination. This technique provided nearly real-time analysis for selected samples, allowing essentially continuous monitoring of the elution in the aquifer. It also allowed real-time monitoring of the performance of ground-water treatment (discussed in Section 2.4).

The second innovative technique was used to analyze samples for the semi-volatile fractions. Solid phase extraction (SPE) of the sample was conducted in the on-site lab. The SPE was conducted using a small cartridge (approximately 3 in. long, 0.5 in .diameter) filled with an 18-Carbon backbone resin. Approximately 180 ml of the sample was pumped through the SPE cartridge after preparing the cartridge with methanol and adding a surrogate mixture to the sample. The cartridges were then delivered to the laboratory for analysis on a GC/Mass Spectrometer (MS) in selected ion mode (SIM). This technique exhibited several advantages. First, a large suite of analytes was analyzed with a single run through the GC/MS. Second, once prepared, samples could be stored with an essentially unlimited holding time. This allowed the set of samples analyzed to be chosen from a much larger set of samples collected (and prepared). The samples analyzed were chosen based on the results of previous analyses. This resulted in an overall reduction in the number of samples analyzed. Although a large number of samples were collected to ensure that the elution curves could be delineated, the results from a limited initial set of screening samples were used to determine when additional samples should be analyzed to delineate the elution curves.

The organic analytes and methods are summarized below:

Volatile Fractions - Field lab: purge and trap GC

- 1. Chloroform
- 2. Bicyclo[2,2,1]hepta-2,5-diene (BCHD)
- 3. Dicyclopentadiene (DCPD)
- 4. Trichloroethane (TCA)
- 5. Trichloroethene (TCE)
- 6. Tetrachloroethene (PCE)
- 7. Benzene
- 8. Toluene
- 9. Xylene

Semi-Volatile Fractions - USGS Organic Lab: SPE GC/MS SIM

- 10. Atrazine
- 11. Chlordane (alpha, cis)
- 12. Chlordane (gamma, trans)
- 13. 2,2-Bis(para-chlorophenyl)-1,1-dichloroethene (PPDDE)
- 14. 2,2-Bis(para-chlorophenyl)-1,1,1-trichloroethane (PPDDT)
- 15. Dieldrin
- 16. Diisomethylphosphonate (DIMP)
- 17. Dimethylmethylphosphonate (DMMP)
- 18. Endrin
- 19. Isodrin
- 20. p-chlorophenylmethyl sulfide (CPM sulfide)
- 21. p-chlorophenylmethyl sulfone (CPM sulfone)
- 22. p-chlorophenylmethyl sulfoxide (CPM sulfoxide)
- 23. CL₅-1,3-Butadiene
- 24. Tetrachlorobenzene
- 25. Unknown Multihalogenated Compound Molecular Weight 364
- 26. Unknown Multihalogenated Compound Molecular Weight 378

2.2.2 CHEMICAL STRATIFICATION IN THE AQUIFER

Water samples were collected from a fully penetrating 2 in. well using a double packer system to isolate 2-ft. sections of the aquifer for sampling. The well was installed without a gravel pack to minimize short circuiting along the outside of the well during this sampling. The water samples were analyzed by the on-site VOC system and the USGS methods described in Section 2.2.1.

Figure 2.2-1 presents depth profiles of background concentrations of chloroform, DCPD and DIMP. Note the strong stratification in concentrations of chloroform and DCPD, contrasted to the rather uniform DIMP concentration with depth. Significant vertical stratification is shown in Figure 2.2-2 for CPM sulfide, CPM sulfone and CPM sulfoxide, and in Figure 2.2-3 for PCE and TCA.



Figure 2.2-1 Depth Profiles of Background Concentrations of Chloroform, DCPD and DIMP. Note the strong stratification in concentrations of chloroform and DCPD, contrasted to the rather uniform DIMP concentration with depth.

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Figure 2.2-2 Depth Profiles of Background Concentrations of CPM Sulfide, CPM Sulfoxide and CPM Sulfone. Note strong vertical stratification.





2.2.3 QUANTITATION OF CONTAMINANTS SORBED ON AQUIFER SEDIMENTS

The sorbed concentrations of contaminants were investigated by analyzing extracts from six sediment cores obtained from two borings (three cores per boring). The cores were collected from depth intervals which correspond with the depth intervals of ground-water monitoring during the experiment (discussed in Section 2.3.3). The sediment samples were collected in a stainless steel split-spoon, hammer driven along two-ft runs. The runs were designed to allow preparation of composites of the sediments which would correlate with the depth intervals planned for ground-water monitoring. The composites were prepared by cutting the core at the correct depth intervals, peeling to remove exposed surfaces, and then peeling freshly exposed sediments into 1-quart (qt) wide-mouth glass jars. The samples were then prepared for analysis by conducting a 12-hour soxlet extraction at the RMA laboratory.

The extracts were delivered to the USGS National Water Quality Lab, Organics Section, for analysis of semi-volatile compounds. The sample extracts were concentrated on a steam bath at 60°C using a Kudema-Danish/Snyder Column concentration apparatus, to a volume of about 5 milliliters (ml). The volumes were then slowly reduced to 0.9 ml by evaporation under a gentle stream of nitrogen. Next, 0.1 ml of toluene containing 6 deuterated-Polycyclic Aromatic Hydrocarbon (PAH) internal standards (ISTDs) was added to the extract prior to analysis. Two microliters (ml) of the extract was then injected for GC/MS analysis in the electron impact (EI) SIM mode. At least 3 characteristic ions were monitored for each contaminant. Calibration curves were constructed for all of the compounds, equivalent to a range of 15-500 micrograms per kilogram (µg/kg) of sediment. No special attempts were made to determine recoveries, but USGS experience suggests that the organochlorine compounds (e.g. dieldrin) would have been recovered at over 70 percent and the rest at 30-60 percent.

The initial intent was to collect and analyze one set of sediment core prior to the aquifer flush, and a second set of core immediately following the flush. The second set of cores would be collected from as close as possible to the locations of the first set of cores. The locations of the borings utilized to obtain the cores are discussed in Section 2.3.4. It was hoped that the analytical results from both sets of cores would provide insight about the sorbed component that existed prior to the experiment, and the fraction of that sorbed component that was removed during the aquifer flush. The sediment cores were to be collected from the same depth intervals screened by the monitoring wells (discussed in Section 2.3.3), allowing the results of the sediment analysis to support interpretation of the behavior of dissolved contaminants in the depth intervals monitored during the flushing process.

Table 2.2-1 lists the results for analyses of the first set of sediment cores collected prior to the aquifer flush, expressed as µg of contaminant per dry kg of sediment. These results are corrected for residual pore water present in the sediment samples. No corrections were made for potentially low analytical recovery of the sorbed mass. It is evident that the only contaminants detected were DIMP and CPM sulfone. The concentrations of the other contaminants which could be analyzed by this technique were at or below the detection limit. The below-detection level results for dieldrin are generally consistent with low or below-detection level concentrations of dieldrin measured in Section 23 sediment cores by Shell Development Company (Shell, 1991).

Unfortunately, at the time the RLSA Contaminant Transport Research sediment core analyses were done, the method was not able to quantify the concentrations of tetrachlorobenzene and the two unknown multihalogenated compounds that were later detected and quantified in water samples. Thus, the background concentrations of these compounds in the sediments was not determined. In addition, more volatile compounds (e.g. DCPD, TCE, PCE, etc.) could not be analyzed by this technique, so information is not available on the background concentrations of these compounds in the sediment.

Depth interval (ft bgs)	39-43	39-43	43-47	43-47	47-51	47-51	
Sample number	B1A	B2A	B1B	B2B	B1C	B2C	
DIMP CPM sulfide CPM sulfoxide CPM sulfone Dieldrin	83.9 <1.5 <1.5 7.1 <1.5	116.1 <1.5 <1.5 8.1 <1.5	46.0 <1.5 <1.5 6.6 <1.5	37.3 <1.5 <1.5 4.1 <1.5	44.2 <1.5 <1.5 21.0 <1.5	64.5 <1.5 <1.5 27.8 <1.5	

 Table 2.2-1 Background Contaminant Concentrations in Sediments (µg/kg) (Corrected for Residual Pore-Water Analyte Concentrations)

Because the majority of the contaminants (including the more hydrophobic contaminants) were below the method detection limits, it was determined that collection and analysis of the second set of cores was not warranted.

These data are discussed in Section 3.0 of the report in light of the contaminant elution rates measured during the field experiment.

2.2.4 LABORATORY RETARDATION TESTS

Because the retardation of solutes during ground-water transport is caused by sorption of the solutes by the aquifer media, it is common to measure sorption using batch analyses in the laboratory as a means of estimating retardation in the field. This approach was used in the work described below, and relies on the assumptions that sorption is linear, reversible and rapid. When these assumptions are met, the ratio of the average velocity of the water to the average velocity of the solute of interest is a constant termed retardation factor (R), where R can be estimated from batch sorption analyses using the following equation:

$$R = 1 + \frac{(1 - \theta)}{\theta} \rho_s K_d$$

where θ is the porosity (-), ρ_s is the solid density of the sediments grams per cubic centimeter (g/cm³), and K_d is the sorption distribution coefficient in cubic centimeters per gram (cm³/g). The first two variables may be estimated or cores may be analyzed by standard methods. The batch sorption methods discussed below are used to determine the k_d.

A series of cores spanning the saturated alluvium at the field site were taken during drilling of the injection well borehole. The cores were divided into depth intervals, which were then shipped to Duke University for analysis under the supervision of Prof. W. Ball (Civil and Environmental Engineering Department). Each interval was air dried and homogenized, then split for analysis. The subsamples were analyzed by one or both of two batch methods to determine the sorption distribution coefficient for PCE.

The first method used to estimate the sorption capacity of the sediments is hereafter referred to as headspace analysis and is a modification of the approach outlined by Garbarini and Lion (1985). This method was used to estimate the degree of sorption and retardation in the field experiment. The method was used in subsequent analyses to generate more reliable estimates of the K_d for PCE. PCE was used as the model solute because 1) it is present in ground water at the site, 2) it is volatile enough for the headspace method, and 3) it is relatively easy to analyze.

In this work, the headspace analysis method was conducted as follows. For each experiment, a known mass of sediment was put in a 120-ml bottle along with a known volume of synthetic ground water (0.033 N CaCl₂), leaving a known volume of headspace. A known mass of PCE was then injected into the water and the bottle immediately capped with a teflon-lined septum. Similarly, an equal mass of PCE was added to a second bottle containing only water and headspace. The bottles were gently shaken for 24 hours to allow partitioning among the air, water and sediments to reach or at least closely approach equilibrium (see additional discussion below). After equilibration, 0.5 ml of headspace gas from each bottle was extracted and injected onto a Varian 3300 GC to determine gas-phase PCE concentration. Based on mass balances in each bottle, the K_d of the sample is calculated as a function of the mass of sediment, the gas and liquid volumes in the bottles. The approach and equations for calculating K_d are described in detail by Garbarini and Lion (1985).

The second batch method is most commonly used for sorption, which will be referred to herein as water analysis. In this method, a known mass of sediments was exposed to a known volume of synthetic ground water (0.033 N CaCl₂) into which a known mass of PCE had been injected (creating a known initial water concentration of PCE). After 72 hours of gentle mixing to allow the PCE sorption to reach or at least closely approach equilibrium, the bottles were centrifuged to separate the solids and the water. Then 6 ml of the water was transferred to another bottle and extracted by shaking with 2.3 ml of hexane. Three µlsof the hexane extract were then injected onto a Varian 3300 GC. Comparison of the results with those of calibration standards allowed the concentration of PCE in the hexane to be determined. Given the extraction ratio, the water concentration was calculated from the hexane concentration. The sorbed mass of PCE was calculated as the difference in mass of PCE in the water before and after equilibration. The K_d (cm³/g), was calculated as the ratio of the sorbed concentration (g PCE/g sediment) to the water concentration (g PCE/cm³ of water).

The results of the sorption analyses are presented in Table 2.2-2, which indicates that the two methods yielded different average values on the sample interval (S1) to which both were applied. It is possible that part of the difference may be attributed to the different equilibration times; in liquid analysis, the equilibration time was 72 hours compared to 24 hours for the headspace method. From prior work with PCE sorption by sandy media from a different site (in Borden, Ontario), Ball and Roberts (1991) found that PCE equilibration required about 10 days. It may be that PCE sorption onto the sediments from the

Sample	Depth Interval	Sorption distribution coefficie Headspace analyses			ficient (K _d) <u>Water</u>	nt (K _d) Water analyses		
code	(feet bgs)	Ave.	Range	n	Ave.	Range	n	
S1	37-39	0.22	0.07-0.36	6	0.37	0.21-0.59	2	
S2	39-41				0.36	0.00-0.61	3	
S 3	41-43	0.58	0.29-1.03	3				
S4	43-45				0.28	0.08-0.40	3	
S5	46-46	0.97	0.61-1.32	2				
S6	46-48				0.02	0.00-0.06	3	
S7	48-50	0.56	0.50-0.62	2				
S8A	50-50.8	0.52	0.14-0.86	4				
S8B	50.8-51.5				0.02		1	

Table 2.2-2 Laboratory Estimates of PCE Sorption Distribution Coefficient for Various Depth Intervals in the Aquifer (n=number of analyses, units of K_d are cm³/gram)

RMA site requires similar or longer periods to reach equilibration. A longer period of equilibration should yield an apparent K_d closer to the true equilibrium value.

In both methods, the K_d estimates from replicate analyses ranged fairly widely, but the ranges for analysis of sample interval S1 from the two methods overlapped. Such variation may arise from analytical difficulties, but may also be due to real differences in the solids analyzed if the sample is not perfectly homogenized. Perfect homogenization of samples was certainly the goal, but cannot be confirmed.

It should be noted that both sets of analyses used high concentrations of PCE, on the order of several hundred $\mu g/L$ iter (L) after equilibration. At lower concentrations more typical of the site (i.e. 1-50 $\mu g/L$), the sorption distribution coefficient may be higher if the sorption isotherm is nonlinear. For example, Ball and Roberts (1991) found that the K_d for PCE for a sand from another site (Borden, Ontario) was approximately 0.9 cm³/g for low concentrations (<50 $\mu g/L$), whereas the K_d was estimated as

approximately 0.6 cm³/g for concentrations ranging up to several hundred μ g/L. Without considerably more work, however, it is not possible to determine or estimate the magnitude of such an increase for sorption of PCE by the RMA sediments at the typically low solution concentrations of PCE in the aquifer.

Despite the uncertainties, it is apparent from this work that the sediments do have the ability to sorb PCE from synthetic ground water, although the measured sorption capacity varies somewhat in the aquifer. From analyses of the core strata, the porosity of the aquifer appears to range from 0.19 to 0.35, with the low values resulting from the presence of large cobbles in the analyzed cores. For cobble-free sands, the porosity probably averages on the order of 0.30. If we assume the porosity is relatively uniform and that the solid density of the sands is 2.7 g/cm³ (approximately the density of quartz), we can use the equation above to estimate the range of retardation factors expected for PCE in the aquifer. This estimate assumes that the site conditions were well represented by the conditions in the sorption analyses (other than the PCE concentration). The estimated retardation factors for PCE range from slightly greater than 1 in the strata with low K_d to approximately 7 in the stratum with the highest average K_d (S5). As mentioned above, the retardation may conceivably be higher in the aquifer since PCE concentrations in the aquifer are lower than those used in the laboratory analyses.

The data in Table 2.2-2 may also be used to estimate the depth-averaged properties of the aquifer. Using the average of the two estimates for S1, and weighing the estimates by the depth interval of the stratum, the depth-averaged K_d for PCE in this aquifer is approximately 0.4 cm³/g. This value is in the range of PCE K_d values estimated for other sandy aquifers; for example Ball and Roberts (1991) reported values of approximately 0.6 to 0.9 cm³/g for various PCE concentrations and various subsamples of a large bulk sample of the sandy aquifer in Borden, Ontario.

In summary, the laboratory analyses of strata from the site indicate that the sandy sediments have the capacity to sorb PCE from synthetic ground water. The depth-averaged K_d for PCE is estimated as 0.4 cm³/g (based on batch analyses conducted at equilibrium concentrations on the order of several hundred µg/L), but may be higher for the low PCE concentrations of interest at the site. Given the assumed solid density and measured porosity, the aquifer has the capability to retard PCE transport by a factor of approximately 3 to 4 (or more) if site conditions aside from PCE concentration are similar to those in the laboratory analyses. Furthermore, based on these analyses, it would be expected that the aquifer could more strongly retard the transport of more hydrophobic compounds such as tetrachlorobenzene, dieldrin and others present in ground water at the site. However, these expectations are based on the assumption

that the complex ground-water chemistry at the site has no effect on the partitioning of contaminant between the ground water and the solids (compared to that observed with the simple synthetic ground water used in the lab tests). As is evident from results presented and discussed later, this assumption appears to be incorrect.

2.2.5 COLLOIDS/DISSOLVED ORGANIC CARBON

Traditional methods of separating a fluid phase into "particulate" and "dissolved" factors rely on physical means such as filtration or centrifugation. Filtration methods usually imply 0.4 μ m filters to operationally define the dissolved fraction of the bulk fluid. Unfortunately, significant amounts of non-dissolved material are able to pass standard 0.4 μ m filters; this material has been implicated in the enhanced mobilization of insoluble chemical components.

The possible role of sub-micron, non-aqueous phase material (i.e., colloids) in the transport of contaminants was investigated as part of the applied research project. This investigation involved the use of a 10,000-molecular weight cut-off ultrafilter (\approx 2 nm effective diameter) in series with 0.5 and 0.4 µm pre-filters, and the analysis of organic carbon and analyte concentrations in each size fraction.

2.2.5.1 <u>Prefiltration</u>

i

The colloid separation system is shown schematically in Figure 2.2-4. Groundwater was pumped from the aquifer to the surface using submersible, 2 in. Grundfos pumps. The sample was pre-filtered, using polypropylene filtration cartridges (0.5 μ m median pore diameter; 25 cm long, 6 cm in diameter) in series with 0.4 μ m Nuclepore cartridges (12 cm long, 6.5 cm diameter, 1m² surface area), at a pumping flow rate of approximately 5 L/min. All pre-filtration cartridges were pre-conditioned to minimize the leaching of organic carbon from filter constituents during the pre-filtration step. The effluent was temporarily stored in a low-density, acid-washed polyethylene drum liner supported by a high-density polypropylene drum (Drum #1). Polyethylene or Tygon tubing was used for the entire pumping and filtration system.

2.2.5.2 <u>Ultrafiltration</u>

The 0.4 µm pre-filtered ground water, held in Drum #1, was subsequently ultra-filtered using an Amicon cross-flow, hollow-fiber ultrafiltration cartridge (Model H10P10-20) with a molecular weight cutoff of

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10,000 Dalton. The cartridge has a surface area of 0.88 m^2 , is 0.64 m in length, a diameter of 4.5 cm, an internal filter diameter 0.5 mm and consists of approximately 1,000 hollow fibers. Fluid from Drum #1 was circulated through the Amicon ultrafilter using a Masterflex peristaltic pump at a flow rate of about 15 L/min. The major portion of the pumped fluid passed unhindered through the filter, traveling through the fiber interiors and returning to Drum #1. The returned fluid is called the retentate. A small volume of the circulating fluid, however, passed through the fiber walls carrying with it constituents of 10,000 molecular weight or less. This material, the ultrafiltrate, was collected in Drum #2. With time, the volume of fluid in Drum #1 decreased and the volume in Drum #2 increased. When the volume of the filtration process continued. When the volume of the retentate reached about 1L, 10 M-ohm Nanopure water was added to the carboy to the 5L level, and the ultrafiltration processes continued until the retentate was approximately 0.5-1.0L in volume. This final retentate was transferred to sample bottles, which had been cleansed of residual organic carbon, for later processing and analysis.

2.2.5.3 Dissolved Organic Carbon (DOC)

Fluid from each stage of the colloid extraction system was subsampled for DOC analysis. Subsamples were: 1) post pre-filtration, i.e., Drum #1 prior to ultrafiltration; 2) the retentate from the ultrafiltration step (i.e., colloid organic carbon, COC, concentration); 3) the ultrafiltered dosing with phosphoric acid; bottles were capped using teflon liners and filled to zero headspace. DOC and Total Organic Carbon (TOC) concentrations were determined using a Shimadzu TOC 5000 analyzer using oxidation at 680°C in the presence of a platinum catalyst.

The approximately 1L final retentate solutions were freeze-dried and the mass of colloidal material determined. Recovered colloid masses ranged from 0.56 to 6.6 g; this corresponds to colloid concentrations in the ground water ranging from 15.9 to 161 mg/L (Table 2.2-3). Background (i.e., pre-experiment) DOC concentrations were 93 mg/L (ppm), as determined from a composite sample (Wells A, B, and C). Mass balances on carbon are presented in the column headed by %PFW.

2.2.6 PRESENCE OF SURFACTANTS

The second potential contaminant-transport-enhancing mechanism investigated was surfactants which can increase the solubility of hydrophobic contaminants. Samples were collected and analyzed for anionic

Date	Sample	Volume (L)	DOC (mg/L) (in sample)	DOC (mg/L) in situ	SD (mg/L)	TOC(mg) (mg/L x L)	% PFW [-]	Coll(mg) (sum)	COC(mg) (sum)	Coll/COC [-]	t sinœ inj. (h)
10/11/91	B-1PFW1	107.5	93.53	93.53	2.09	10054					
	B-1PFW2	107.5	93.23	93.23	3.72	10022					
	B-1COL	0.89	44.89	0.37	0.47	40	0.40	560	40	14.0	0
	B-1UFW	106.5	84.98	84.98	. 1	9050	90.30				

Table 2.2-3 Summary of Background (pre-experiment) Colloid Information

PFW1: Prefilter Water (original sample)

PFW2: Prefilter Water (duplicate of PFW1)

COL: Colloids

UFW: Ultra Filtrate Water

Volume: Volume of aquifer fluid in each fraction; PFW is the total volume pumped; COL is the final colloid retentate volume.

DOC: "dissolved" organic carbon concentration in each fraction.

DOC(in situ): the DOC concentration estimated to be in the aquifer fluids; e.g., for COL: DOC (in situ) = $44.89 \times 0.89 = 0.3$

SD: standard deviation on replicate samples for DOC measurement.

TOC: total carbon in fraction volume, e.g., 93.53 mg/L x 107L = 10054 mg.

% PFW: percent of fraction relative to total; e.g., COL = 40/10054 = 0.4%.

Coll(sum): colloid mass in total sample.

COC(sum): colloid organic carbon in total sample.

surfactants by the USGS National Water Quality Inorganics Laboratory. Analytical protocols for the analysis of non-ionic surfactants could not be implemented due to funding and contractual limitations. The anionic surfactants were analyzed by using a modification of the Methylene Blue Active Substances (MBAS) method as described in "Standard Methods for the Examination of Water and Wastewater," 17th edition, (USGS, 1989). This modification reduces interference from high chloride concentrations, such as is exhibited in the site ground water.

Table 2.2.4 summarizes the results of the surfactant analyses. Two replicate samples were analyzed and contained 0.021 and 0.013 mg/L MBAS. About 80 percent of an MBAS spike added to the ground-water matrix was recovered. Additionally, four fortified reagent-water samples were prepared with known amounts of chloride and MBAS to determine the level of interference of chloride with the modified method. Chloride content from 0 to 2,500 mg/L did not appear to significantly affect the analyte recovery.

From discussions with USGS personnel and information in the scientific literature (Kile, 1989), it was determined that the types of surfactants most likely to have been used/disposed at RMA and to have migrated in ground water were anionic and nonionic. Anionic surfactants are commonly used in detergents and various industrial formulations, including pesticides. Nonionic surfactants are less common in domestic products, but are often used in pesticide and other industrial formulations, often in combination with anionic surfactants.

Sample ID	Chloride Concentration (mg/L)	Amount MBAS added (mg/L)	Amount MBAS Analyzed (mg/L)	Recovery of Added MBAS (percent)
Site Ground water		0	.021	
Site Ground water		0	.013	
Site Ground water Spil	ke	0.5	0.417	80
Fortified Reagent-wate	ſ			
#1	2500	.025	0.013	53
Fortified Reagent-wate	r			
#2	2500	0.50	0.443	89
Fortified Reagent-wate	er (
#3		.025	.011	45
Fortified Reagent-wate		-	105	
#4		.5	.427	85

 Table 2.2-4
 Summary of Surfactant Analysis Results

These results suggest that there are insignificant concentrations of anionic surfactants in the ground water at the site. To have a significant effect on partitioning and mobility of contaminants ranging from chloroform to dieldrin, the scientific literature suggests that the anionic surfactants would have to be present at much higher concentrations, i.e. above their critical micelle concentrations (CMCs) which generally are on the order of 1,000 mg/L or higher. Although no data were obtained on nonionic surfactants, the fact that they are often used in combination with anionic surfactants suggests that they too are unlikely to be present in site ground water at significant concentrations. On the other hand, if all of the detected DOC at the site (approx. 100 mg/L) were nonionic surfactants, it is possible that they could exert a significant effect on contaminant partitioning and mobility since the CMC for some nonionic surfactants is at or below 100 mg/L. For reasons discussed later, further investigation of the importance of nonionic surfactants may be warranted.

2.3 WELL NETWORK

The well network for the experiment was designed to perform several functions. An extraction well was used to remove contaminated water from the aquifer. An injection well was used to deliver the clean water flush into the aquifer. A series of monitoring wells were used to collect the ground-water samples. A series of soil borings were used to obtain the sediment core for the laboratory analyses, and a series of piezometers were used to monitor the hydraulics during the experiment.

The flush of clean water through the aquifer was accomplished utilizing an injection well/extraction well combination. This method provided several advantages over the use of a single injection well:

Increased ground-water flow control.

The combination of an injection and extraction well reduces the potential mounding that might occur at a single injection well. The negative aspect of mounding at the injection well is that sediments above the water table, which may not have been exposed to the contaminated ground water, might be exposed to the clean-water flush. This situation could distort the measured elution rates. The addition of the extraction well provides a steeper gradient away from one side of the injection well, theoretically increasing flow away from that side of the injection well and reducing the magnitude of mounding. Further, because the water for the injection flush is obtained from the nearby extraction well, there is no net volume gain of water in the aquifer.

The combination of injection and extraction wells also increases ground-water flow control by developing a preferred flow path in the aquifer. A single injection well would deliver flow radially to the aquifer, assuming homogeneous hydraulic conductivity in the aquifer. Because hydraulic conductivities could not be assumed homogeneous based on the lithologic descriptions and core study results discussed above, a monitoring well adjacent to a single injection well may or may not receive the anticipated flux of clean water. The use of the extraction well increased the confidence that a flux of clean water would flow through the aquifer at the monitoring points.

Eliminated external source for injection water.

The use of an injection well/extraction well combination precluded the necessity to obtain an external source of injection water because water pumped out of the extraction well was treated (discussed in Section 2.4) and then re-injected as the clean-water flush.

Minimized waste management efforts.

By treating and re-injecting the extraction water, large volumes of contaminated water did not have to be containerized, piped, or transported to disposal facilities.

The wells in the network were located designed with the aid of computer models. The extraction, injection, and monitoring wells were oriented in line with the approximated natural ground-water flow direction to minimize the potential for flow deviations from the desired flow path during the experiment (Figure 2.3-1).

Construction details for the wells in the network are in Appendix B. A brief summary is presented below.



2.3.1 INJECTION WELL

The injection well was drilled first and screened through the entire aquifer. Its purpose was to deliver the clean water flush into the aquifer. Preliminary information on aquifer properties was gained from lithologic logging (Section 2.1.1) and laboratory testing (Section 2.2.4) of the drill core. The information gained during the installation of the injection well contributed to the design parameters of the extraction and monitoring wells.

2.3.2 EXTRACTION WELL

The extraction well was used to pump contaminated ground water out of the aquifer. It was located 30 ft from the injection well, downgradient along the estimated natural ground-water flow path (Figure 2.3-1). The distance between the injection and extraction well was determined by integrating hydraulic conductivity estimates with pump rate limitations, the desired velocity field in the forced gradient portion of the test, and the proposed time frame of the experiment. Hydraulic conductivity was estimated based on the lithology encountered at the injection well, and modelling efforts conducted at the North Boundary Containment System (NBCS, located roughly 3000 ft to the north of the site) by Colorado State University (Warner, 1991) (Section 2.1.1). The lithology of the aquifer was logged during drilling of the extraction well. This information was used to design for the extraction well and monitoring wells. The construction of the extraction well was similar to the construction of the injection well.

2.3.3 CHEMICAL MONITORING WELLS

Chemical monitoring of the ground water was conducted at a 3-well cluster, located approximately 10 ft downgradient of the injection well (Figure 2.3-1). The distance between the monitoring cluster and the injection well was based on the flow velocities anticipated under the forced gradient, combined with the range of expected mobilities of the contaminants. Because the physical/chemical characteristics of the contaminants varied widely, it was expected that the mobility of at least some of the contaminants would be significantly retarded. The intent was to locate the chemical monitoring wells to allow monitoring of the broadest range of contaminant mobilities within the time frame of the experiment.

Each of the three wells in the cluster screened a separate, 4-ft interval in the aquifer. The screened intervals were initially planned to investigate separate lithologic zones, if present. Because the logging

conducted while drilling the other wells revealed no distinct, separate lithologic zones, the aquifer was instead screened along three, roughly equal depth intervals. Well A screened the top portion of the aquifer from 38.2 to 42.2 ft bgs; well B screened the middle portion, from 42.6 to 46.6 ft bgs; and well C screened the bottom portion, from 47.0 to 51.0 ft bgs.

The monitoring points were constructed by augering one 8-in. hole to just above the top of the water table. Then three stainless steel drive points and screens (2-in. diameter) were pushed with an electric jackhammer from the bottom of the augered hole into the aquifer at the desired depth intervals. The initial plan was to hammer in the drive points with the drill rig, but the hammer impact destroyed the screens, and its force could not be reduced. Thus, a jackhammer and a set of scaffolding was substituted for the drill rig. The impact force was conveyed to the drive point tip through a drill rod installed in the well casing. This reduced the tendency to distort the screen and casing during installation. Two-in. stainless steel casing was used from the top of each drive-point screen up to 3 ft above the water table. Schedule 40 PVC casing (2-in. diameter) extended from the stainless casing to the surface. The construction details are presented in Appendix B.

2.3.4 SEDIMENT BORINGS

Two sediment borings were drilled to obtain the core needed for the quantification of the contaminants sorbed to the aquifer sediments (discussed in Section 2.2.3). The first boring was located 4 ft from the injection well; the second boring was located 10 ft from the injection well (Figure 2.3-1). The borings were located at these distances from the injection well because it was theorized that the desorption of the more highly retarded contaminants would need to be measured close to the injection well where the degree of flushing would be higher, whereas the desorption of the more mobile compounds would need to be measured farther away from the injection well where the degree of flushing would be lower. Both of the borings were located slightly offset from the predominant ground-water flow path anticipated during the flushing portion of the experiment to minimize their effect on ground-water flow between the injection and monitoring wells. A second set of cores were planned to be collected from two additional borings, each located as close to the initial borings as possible. The second set of cores were not collected because the results from the first set of cores indicated that the majority of the contaminants were not present above detection levels (discussed in Section 2.2.3).

The initial set of cores were collected prior to the aquifer flush from soil borings B1 and B2 (Figure 2.3-1).

2.3.5 WATER LEVEL PIEZOMETERS

Five piezometers were used to monitor the hydraulic gradient and water level fluctuations during the experiment. One of the piezometers, (Well 23095) (Figure 2.3-1) existed prior to the experiment. Two of the piezometers were constructed by completing the two initial soil borings, B1 and B2, as wells. The last two piezometers were installed specifically to monitor water levels between the monitoring cluster and the extraction well. Water levels were not monitored in the chemical monitoring wells because the transducer cables did not fit through packers installed above the sample pumps.

The construction details of the piezometers are presented in Appendix B. All the piezometers screened the entire saturated thickness of the aquifer except Well 23095, for which construction details are inconsistent. The two piezometers which were completed in the boreholes of the sediment borings were installed without an artificial filter pack. This technique was utilized because a naturally caved aquifer would minimize the increased vertical hydraulic conductivity that an artificial filter pack can create immediately outside the screen. Because the depth discrete sampling conducted during the baseline characterization (discussed in Section 2.2.2) utilized inflatable packers on each end of a submersible pump, a reduction in vertical communication immediately outside the well screen was desirable to provide increased definition of the sample intervals. Additionally, the introduction of filter pack, bentonite, and/or grout into these boreholes would increase the risk that the sorption capacity and/or hydraulic conductivity of the sediments could be unfavorably altered. The other piezometers, P1 and P2, were installed following standard United States Army Toxic and Hazardous Materials Agency (USATHAMA) protocol.

Water level data collected from the piezometers are reviewed and compared to sorption and other chemical data in section 3.0 to describe the effect of the flow field on the experimental results. A dedicated automated system was chosen because it provided the most sensitive measurements and did not require continuous monitoring by site personnel.

The water levels at the site were monitored via a computer-driven datalogger with six pressure transducers. Five-pounds-per-square-inch, gauge, (psig) Druck PDCR 950 pressure transducers were placed approximately 4 feet below the water table in wells I, B1, B2, P1, P2, E, 095. The accuracy of the



transducers was rated at ± 0.002 ft. A Geokon Micro 10 Datalogger was used to store the measurements, and was controlled through a portable computer located in the on-site lab. The computer software provided by Geokon was used to program the timing of the measurements. Data from the datalogger was periodically dumped to a floppy disk for backup and subsequent flow system analysis.

2.4 PROCESS SYSTEM

The process system consists of the pumps, filters, treatment canisters, piping, and related equipment that allows ground water to be pumped from the aquifer, treated, sampled and/or injected. Figure 2.4-1 schematically diagrams the components of the process system: extraction, treatment, injection, and monitoring.

2.4.1 EXTRACTION

Ground water was pumped from the extraction well using a 4-in. stainless steel electric submersible pump. The extraction rate was 4 gpm. The extraction rate was monitored with a flow meter and adjustments were made with a gate valve on the pump discharge line. Fluctuations in the flow rate were minor, estimated at ± 0.2 gpm. Polyvinyl Chloride (PVC) tubing was utilized to transport the water to the surface where a sand trap and a series of filters reduced particulate matter prior to treatment.

2.4.2 TREATMENT

Downstream from the initial filters, the extracted water was routed through an activated carbon system to remove dissolved organic compounds before it was reinjected. Two Tigg C25 modular disposable activated carbon units were used. These units contain 330 pounds of 12 X 40 U.S. sieve carbon and can withstand a maximum flow of 25 gpm. The estimate of needed activated carbon capacity was based on the chemical results from historical monitoring programs (RLSA, 1989, 1990, 1991).

The number and size of the carbon units was determined by the manufacturer using the following assumptions:

1. Flow rate = 10 gpm2. Water pH = 7

CMP-TRACER.FNL 06/30/92 3. Water temp. $= 60^{\circ}F (15.6^{\circ} C)$

4.

Total concentrations of organics = 12 mg/L (approximately 9 mg/L CCl₃ and 3 mg/L for other organics.

The canisters were connected in series to provide extra capacity. The net treatment system water pressure resulting from the combination of Granular Activated Carbon (GAC) canisters and several pre- and post-treatment filters necessitated a centrifugal booster pump that was located between the GAC canisters.

Because the manufacturer of the GAC canisters indicated that chloroform was the contaminant in the aquifer that would break through a canister first, the performance of the treatment system was monitored by frequently analyzing chloroform concentrations in water collected from sample ports located immediately upstream and downstream of each canister (Figure 2.4-1). These analyses were conducted in the field on the automated GC system discussed in Section 2.2-1. The potential for breakthrough of other unknown compounds was also monitored by collecting samples from these ports for semi-volatile analysis at the USGS laboratory. These results were not real-time, as was the on-site GC analysis for chloroform.

Note that the total organic carbon concentration of the site ground water (approximately 100 mg/L) was not known at the time the activated carbon supplier was estimating the operating parameters of the units. Initial information on DOC was later available from USGS. Since the actual DOC was about an order of magnitude higher than the value initially assumed, the estimate of activated carbon use per day was probably about an order of magnitude too low. Thus, while the two activated carbon tanks were initially expected to last about 60 days, hindsight suggests that they should have been expected to last only 6 days (about 150 hours). As determined later, the capacity of the activated carbon treatment system was indeed exceeded after about 6 days during the experiment.

The carbon treatment system did not remove salts from the extraction water. This was favorable because the natural chemistry of the treated water should ideally be the same as the aquifer water in order to control the number of variables that could affect the contaminant elution rates. Similarly, temperature variations in the process water were also minimized by enclosing and insulating the entire process system. A thermostatically controlled heat tape was used to provide a constant temperature of the process water. By maintaining consistent natural chemistry and temperature between the injected water and the aquifer water, the number of variables affecting contaminant elution rates was controlled.

2.4.3 INJECTION

The third element of the process system involved the injection of the organic-free water flush. After treatment, organic-free water was delivered to the injection well at the same rate that contaminated water was extracted (4 gpm). A lithium bromide (LiBr) tracer was added to the injection water for a period of time using a peristaltic pump drawing from a 55-gallon drum containing a concentrated LiBr solution. The drum was continuously stirred to ensure uniformity of the LiBr concentrations. The addition of the tracer solution to the flow of injection water was kept constant using a flow meter and adjusting a needle valve and/or the peristaltic pump speed. To ensure that the tracer was uniformly mixed into the injection water, an in-line motionless mixer followed by a two-liter mixing canister were utilized. Downstream of this mixing apparatus, a sample port was used to monitor the concentrations of the tracer being injected into the aquifer. The concentration of the tracer in the injected water was approximately 250 mg/L. This sample port was also used to collect samples for organic analyses to ensure that the injection water did not contain organic contaminants.

The LiBr tracer was added to the injection water at the beginning of the experiment when the injection/extraction pumping was started. In aqueous solution, the LiBr salt dissociates into lithium and bromide ions (Li+ and Br-), which can then migrate independently. Bromide (Br-) is widely used in ground-water studies as a conservative tracer since it interacts with aquifer media only to a negligible extent, and travels at the same mean velocity as the ground water. By monitoring for the Br- tracer at the monitoring wells, the arrival time of the leading edge of the organic-free flush was determined. This arrival time allows the ground-water flow velocity under the forced gradient to be estimated for each monitoring interval in the aquifer. Comparing the arrival time of the organic-free flush with the elution rates of the contaminants allows quantificaton of the contaminant transport characteristics. Flow velocities and hydraulic information about the aquifer can also be determined from the tracer information.

2.4.4 MONITORING

The fourth element of the process system involves the collection of water samples from the partially penetrating monitoring wells. A stainless steel electric submersible pump (Grundfos Rediflo 2) was installed in each of the three wells of the monitoring cluster. The pump intake was located at or slightly above the top of the screen section. Packers were used to isolate the pump and screen section from the rest of the well casing. Samples were taken after a small purge volume was pumped to remove any

stagnant water in the pump, tubing, or well screen. The packers minimized the volume of purge water, the impact on the aquifer flow field, and the time required to sample. A gauge to monitor the inflation pressure in each packer was installed at the surface to ensure that the packers remained properly inflated. Stainless steel tubing was used to transport the water from the pumps up to a surface sample port. Waste water generated by purging and sampling was directed into the process system for subsequent treatment.

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3.0 FIELD EXPERIMENT RESULTS

The field experiment results are presented for the forced gradient portion of the experiment in Section 3.1, and for the natural gradient portion of the experiment (after injection/extraction is ceased) in Section 3.2. Hydraulic and chemical monitoring results are discussed for each of these portions of the experiment.

3.1 FORCED-GRADIENT PORTION OF THE EXPERIMENT

The forced gradient portion of the field experiment, including ground-water extraction, treatment, and reinjection, began at 10:46 p.m. on November 3, 1991.

3.1.1 HYDRAULICS

Hydraulic effect of the forced gradient on the aquifer was monitored by 7 transducers installed in wells I, B2, B1, 095, P1, P2, and E (Figure 2.3-1). The transducers were wired to a data logger in the site trailer.

The information desired from the hydraulic monitoring included:

- nearly continuous tracking of the ground-water gradient (flow direction and slope),
- the time required for the aquifer to reach steady-state conditions following the start and/or stop of extraction/injection, and
- hydraulic conductivity estimates.

Hydraulic monitoring was initiated approximately 25 days before the start of injection/extraction in order to obtain background information on the natural ground-water gradient and the stability of the water table. The background water-level information from each well exhibits a fair amount of time-dependent variability. This variability typically involves fluctuations in calculated water levels of 0.03 ft over as little as one minute. The accuracy of the transducers is rated at ± 0.002 ft over the pressure range which was encountered in each well application. Several reasons for this variability were postulated:

- 1. The water table is dynamic at this scale.
- 2. Baseline characterization activities (collecting method development samples, depthdescrete samples, testing pumps, etc.), disturbed the water table.
- 3. Other electrical equipment caused power supply fluctuations affecting transducer voltages.
- 4. The site location was influenced by external factors such as pumping at the Basin F and/or boundary containment systems.
- 5. Random pressure disturbances occurred at ground level.

The actual cause of the water table elevation variability has not been determined.

Figure 3.1-1 presents the water level elevations measured approximately 48 hours prior to starting the extraction/injection. Water level elevations ranged from 5,139.70 ft mean sea level (msl) at Well I to 5,139.63 ft msl at Well E. It is difficult to interpret the background hydraulic gradient at the site from these data because of the extremely small elevation change exhibited across the horizontal extent of the site (30 ft), combined with the time-dependent variability. The flatness of the background water table at the site confirms the more regional estimate of 0.004 ft/ft obtained from the CMP data (RLSA, 1991; Section 2.1.1).

During the forced gradient portion of the test water level responses were minimal. Figure 3.1-2 presents the water table elevations measured 48 hours after starting the extraction/injection. The maximum elevation change occurs at the injection well (I) where water levels increased only 0.07 feet. Earlier hydraulic data indicate that this degree of water level response had occurred within three minutes from the start of extraction/injection, suggesting that a steady state was reached very quickly. This information is consistent with the hydraulic conductivity estimates (120 to 400 ft/day) discussed in section 2.1.1. However, this high hydraulic conductivity, combined with the time-dependent variability in the data, precludes accurate gradient calculations.





3.1.2 CHEMISTRY

3.1.2.1 Treatment System Performance

The extracted water was reinjected after passage through a treatment train consisting of two consecutive GAC tanks with particulate filters before and after the tanks. Samples of the extracted and injected water were collected at a regular frequency.

The various frames of Figures 3.1-3 and 3.1-4 show the concentrations of several organic analytes versus time for the extraction and injection wells, respectively. Samples of water which had passed though the first GAC tank were also taken at a regular frequency in order to help identify the incipient failure of the carbon treatment system. However, since the failure occurred much earlier than anticipated (as discussed below), so the samples collected immediately downstream of the first GAC tank were of little use, and most were not analyzed.

Comparison of the frames in Figures 3.1-3 and 3.1-4 indicates that the treatment system was reducing the concentrations of the contaminants to below their detection limits for the first 125 hours of operation. This was also true for other target analytes not illustrated in Figures 3.1-3 and 3.1-4. In addition, yellowness in the contaminated ground water was absent in the injection water during this period. At about 125-150 hours, as shown in Figure 3.1-4, the carbon treatment system began to fail and the concentrations of chloroform, methylene chloride, TCA, and DIMP began to rise rapidly. These were the only detected organic analytes that broke through the carbon treatment system at significant concentrations. The failure of the treatment system was noted by on-site VOC analysis (detecting the chloroform initially) and by visual inspection of the injection water samples, which regained the yellow tint typical of the contaminated ground water at the site. The breakthrough of the organics occurred much earlier than was anticipated. The carbon supplier had estimated that the system would provide organic-contaminant-free injection water for up to 2 months; their estimate, however, was based on historical chemistry data which does not quantify the high levels of uncharacterized dissolved organic carbon, identified later in this experiment. The uncharacterized dissolved organic carbon apparently affected the efficiency of the activated carbon treatment.

When the breakthrough of the VOCs was noticed, it was not known whether any of the semivolatiles were also breaking through; such information was not obtained for several weeks after the regular turn around









Figure 3.1-4 Concentration Histories of DIMP, Chloroform, Methylene Chloride and 1,1,1-Trichloroethane in the Injection Well. Note that the carbon treatment system reached capacity for these compounds, which were then recycled into the injection well, after about 100 hours.

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time for analysis at the USGS laboratory. However, after consultation with the Program Manager's Office (PMO) project officer and others, the decision was made to stop the extraction and injection. The primary reason for this decision was to avoid reinjecting the contaminants, which would confound the interpretation of the results. A significant concern, however, was that by stopping the experiment so early, the opportunity to observe the flushing of some of the more hydrophobic and presumably less mobile contaminants would be lost. As discussed later, this concern turned out to be unwarranted.

3.1.2.2 <u>Tracer injection</u>

The injected water was spiked with LiBr from 0-62 hours. The bromide ion (Br) was included as a conservative tracer and thus is of primary interest for this report. The behavior of the lithium ion (Li⁺) was primarily of academic interest; for this reason and also because the lithium analyses have yet to be completed (they are being conducted under separate funding), the lithium results are not discussed in this report.

The top frame of Figure 3.1-5 shows that the spiking system achieved the desired result, i.e., a bromide concentration in the injected water that was reasonably constant at about 275 mg/L during the spiking interval. As evident in the lower frame of Figure 3.1-5, the injected bromide began to appear in the extraction well sometime between 50 and 75 hours. Since the bromide was not removed by the activated carbon treatment system, the extracted bromide was recycled into the injection well. This recycling is the cause of the elevated bromide concentration in the injection well which exists after the initial bromide spike until about 278 hours. From 278.12 to 281.62 hours, a second spike of bromide was added to the injection line, resulting in the concentration rise to about 500 mg/L in the top frame of Figure 3.1-5. This final spike was added to aid in interpreting data collected from the monitoring points after the injection-extraction system was shut off at 281.62 hours; the period after 281.62 hours is hereafter referred to as the natural-gradient portion of the experiment.

Figures 3.1-6 and 3.1-7 present the alkalinity, dissolved oxygen (DO) and pH measured in the extraction and injection wells. Both alkalinity and pH are relatively constant in the water pumped from the extraction well and the water pumped into the injection well during the period they were measured. The DO, however, changed slightly with time in both wells. The DO in the water pumped from the extraction well started at a relatively high value, considering the background monitoring (Section 2.2.1)



Figure 3.1-5 Concentration Histories of the Bromide Tracer in the Injection Well and the Extraction Well During the Forced-Gradient Portion of the Experiment.

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Figure 3.1-6 Concentration Histories of Alkalinity, Dissolved Oxygen (DO) and pH in the Injection Well During the Forced-Gradient Portion of the Experiment.



Figure 3.1-7 Concentration Histories of Alkalinity, Dissolved Oxygen (DO) and pH in the Extraction Well During the Forced-Gradient Portion of the Experiment.

which indicated that background DO was quite low (2 mg/L) throughout the aquifer. This elevated DO in the extracted water which was almost certainly due to aeration of the water within the well bore resulting from a portion of the extracted water which was wasted back into the extraction well in order to achieve an overall extraction rate of 4 gpm with a pump capable of a considerably higher flow rate. As the water cascaded back down the well bore from the surface, it would certainly have been aerated.

For the first 20 hours or so, the DO in the injected water was significantly lower than measured in the extracted water. This suggests that oxygen was consumed in the activated carbon tanks during this period. Nevertheless, the DO in the injected water was significantly higher than that within the aquifer. Thus the injected DO serves as an additional signal and, indeed, a DO increase is noted in limited samples from the extraction well after about 50 hours, at roughly the same time the bromide appears (compare DO in Figure 3.1-6 with Bromide in Figure 3.1-5). This suggests that DO may behave relatively conservatively in the aquifer, at least under the conditions and over the short time frame of this experiment. This observation is consistent with observations noted at other (although relatively uncontaminated) sites (Roberts, 1990). This observation suggests that oxygen-consuming reactions, if they occurred at all within the aquifer, were relatively insignificant during the forced-gradient portion of the experiment. Examples of potential oxygen-consuming reactions of interest would be biochemical oxidation of some of the organic constituents in the water or chemical oxidation of reduced mineral species.

3.1.2.3 <u>Aquifer monitoring</u>

3.1.2.3.1 Tracer advection

As mentioned previously, the aquifer chemistry was monitored via 1) a cluster of three partiallypenetrating wells (A, B, and C) spanning the full aquifer thickness at one location approximately 10 ft. from the injection well, and 2) a fully-penetrating well approximately 13 ft. from the injection well.

Figure 3.1-8 presents the breakthrough curves for the bromide tracer at Wells A, B, and C. It is evident that breakthrough is fastest in Well B and slowest in Well A, indicating a slight vertical variation in hydraulic conductivity in the aquifer. In Wells B and C, the bromide concentration peaks at or near the initial injected concentration of approximately 250-300 mg/L. In Well A, the breakthrough curve appears to be made up of a series of pulses, perhaps representing different rate travel through various strata within



Figure 3.1-8 Bromide Breakthrough Curves for Wells A, B and C. The different positions of the breakthrough curves indicate that there is some variability in hydraulic conductivity with depth in the aquifer. Water (and bromide) travelled fastest in strata sampled by Well B; overall, travel was slowest in strata sampled by Well A.

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the depth interval sampled by the well. Overall, the breakthrough curve in Well A is more dispersed than in Wells B and C, and the peak concentration does not reach the initially injected value. The breakthrough curves for all three of the wells have long tail which result from the recirculation of the bromide tracer from the extraction well into the injection well. The tails all reach the same plateau of about 60-80 mg/L as observed in the injection well. Overall, the tracer behavior in Wells A, B, and C indicates that the injected water swept through the entire vertical interval sampled by these wells, i.e. through the entire saturated thickness of the aquifer.

Figure 3.1-9 presents the breakthrough curve for bromide in Well P1. Monitoring for the Bromide Tracer at Well P1 was not started until the collection of colloidal samples, so only the falling side of the bromide pulse was detected. As was the case for Wells A, B, and C, the concentration of bromide in Well P1 drops to a plateau which represents the recirculated bromide. These limited data, indicate that the injected also swept through the entire interval sampled by Well P1, which screens the entire saturated thickness of the permeable aquifer. This also supports the conclusion that the injected water swept through the entire saturated thickness of the aquifer.

3.1.2.3.2 Changes in geochemistry

The three frames of Figure 3.1-10 present bromide, alkalinity, dissolved oxygen (DO) and pH versus time in Well A. Figures 3.1-11 and 3.1-12 present the same plots for Wells B and C, respectively. The behavior of the alkalinity is consistent with the fact that the aquifer was initially stratified with respect to alkalinity, with alkalinity increasing with depth (Section 2.2.2). The alkalinity in the extracted and injected water is in a sense an average of that in the aquifer, since the extraction well draws from the entire aquifer thickness. Thus, the alkalinity of the injected water (approximately 1500 μ g/L) is higher than initially present in the upper portion of the aquifer sampled by Well A (approximately 600 μ g/L). Figure 3.1-10 shows that the alkalinity in Well A gradually increases to that of the injected water, with the change occurring over the same time interval that the bromide pulse is detected.

In Well B, the initial alkalinity (approximately 900 μ g/L) is lower than that of the injected water. Figure 3.1-11 shows that the alkalinity rises rapidly to that of the injected water, essentially over the same time interval that bromide breaks through at the well. In Well C (Figure 3.1-12), the initial alkalinity is higher than that of the injected well, so the alkalinity decreases to the injection value at the same time as the



Figure 3.1-9 Bromide Breakthrough Curve for Well P1.



Figure 3.1-10 Concentration Histories of Bromide, Alkalinity, Dissolved Oxygen (DO) and pH at Well A During the Forced-Gradient Portion of the Experiment.



Figure 3.1-11 Concentration Histories of Bromide, Alkalinity, Dissolved Oxygen (DO) and pH at Well B During the Forced-Gradient Portion of the Experiment.



Figure 3.1-12 Concentration Histories of Bromide, Alkalinity, Dissolved Oxygen (DO) and pH at Well C During the Forced-Gradient Portion of the Experiment.

bromide breaks through. Because alkalinity is a conservative quantity, the convergence of alkalinity values reflects the depth interval mixing of extracted ground water prior to reinjection.

The behavior of DO in all three wells is similar. The initial values are very low, as found in the analyses of the depth discrete samples prior to the experiment. The DO rises in each of the wells at roughly the same time that the bromide arrives, reaching a plateau approximately equal to the DO in the injected water. This is another indication that there are no oxygen-consuming reactions occurring at a significant rate in any depth interval of the aquifer during the forced-gradient portion of the experiment.

3.1.2.3.3 Flushing of the organic contaminants

• Well A. The contaminants were flushed from the aquifer by the injected, initially contaminant-free water. Figure 3.1-13 illustrates the flushing (elution) of DIMP, CPM sulfone and dieldrin from the strata monitored by Well A. Despite the marked difference in properties (solubility, octanol-water partition coefficient, etc.) of the first two compared to those of dieldrin, the three contaminants were apparently flushed from the aquifer at essentially the same rate. Concentrations of all three contaminants began to decrease significantly at about 25 hours, roughly the same time as the bromide tracer (i.e. the injected water) began to arrive at the well (Figure 3.1-13). The concentrations of all three contaminants were reduced to very low levels by 150-200 hours, which corresponds relatively well to the concentration plateau in the bromide breakthrough curve during the same time interval (Figure 3.1-13). The low concentration tail on the DIMP elution curve beyond 150 hours resulted largely from the reinjection of DIMP due to failure of the treatment system described earlier (see Figure 3.1-4).

Figure 3.1-14 illustrates the concentration histories of chloroform, DPCD, TCE and PCE in the strata sampled by Well A. Since there was little chloroform in these strata initially, the first frame shows primarily the arrival at approximately 200 hours of the chloroform that was injected after about 150 hours due to failure of the treatment system (see Figure 3.1-4). For the other contaminants in Figure 3.1-14, which were initially present in the strata in significant concentrations, it is again observed that flushing is apparently complete after about 150 hours.

Figure 3.1-15 illustrates the flushing of tetrachlorobenzene and two multihalogenated unknowns. These, like dieldrin, would normally be expected to be retarded in their migration through the aquifer.



Figure 3.1-13 Elution Curves at Well A for DIMP, CPM Sulfone and Dieldrin During the Forced-Gradient Portion of the Experiment.

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Figure 3.1-14 Elution Curves at Well A for Chloroform, Dicyclopentadiene, Trichloroethylene and Tetrachloroethylene During the Forced-Gradient Portion of the Experiment.



Figure 3.1-15 Elution Curves for Tetrachlorobenzene and Two Unknown Multihalogenated Compounds the During Forced-Gradient Portion of the Experiment.

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However, they too are apparently flushed from the aquifer at the same rate as the clean water migrates through. Ground-water concentrations of all three decrease to the detection limit by about 150 hours.

• Well B. Figure 3.1-16 shows the rapid elution of DIMP, CPM sulfone and dieldrin from the strata sampled by Well B. The concentrations of all three were reduced essentially to the detection limit by 50 hours, which is roughly the same time as the maximum of the bromide peak (see Figure 3.1-11). Figure 3.1-17 shows equally rapid elution of chloroform, dicyclopentadiene, trichloroethylene and tetrachloroethylene. The rise in chloroform concentrations after 150 hours is certainly a result of the reinjection of chloroform after 150 hours. The final data point for the TCE curve (approximately 270 hours) is likely to be an analytical artifact since there was no evidence that TCE was reinjected or ever broke through the treatment system. Figure 3.1-17 shows rapid elution of tetrachlorobenzene and the two unknown multihalogenated compounds; the ground-water concentrations of all three were reduced to the detection limit by about 45 hours. Although no monitoring data were available for these three compounds for the period 50-230 hours, the confirmation of non-detection thereafter (Figure 3.1-18) suggests that the elution behavior of these compounds was probably similar to that of the others.

• Well C. Figure 3.1-19 shows that the ground-water concentrations of DIMP, CPM sulfone and dieldrin are reduced in strata sampled by Well C to very low values by about 100 hours. This is roughly equal to the mean arrival time of the injected water, judging from the shape of the bromide peak and the plateaus of the alkalinity and dissolved oxygen changes in Figure 3.1-12. Figure 3.1-20 shows that elution of chloroform, dicyclopentadiene, trichloroethylene and tetrachloroethylene also occurred by about 100 hours. Figure 3.1-21 shows that the elution of tetrachlorobenzene and the two unknown multihalogenated compounds may have occurred at roughly the same rate, with the groundwater concentrations at or below the detected in significant concentrations only in the strata sampled by Well C: benzene, o-xylene, bicycloheptadiene, pentachloro-1,3-butadiene, CPM sulfide and CPM sulfoxide. Ground-water concentrations of these contaminants were reduced to low or non-detect values by 70-100 hours.

• Well P1. As described earlier, monitoring Well P1 was begun after 85 hours in conjunction with the collection of colloid samples. Apparently, contrary to preliminary estimated elution rates, all significant concentration decreases had occurred at Well P1 by this time. Numerous analyses of the semivolatiles confirm that the concentrations were at or below the detection limit for many of the analytes (e.g. tetrachlorobenzene, CPM sulfoxide and the unknown multihalogenated compounds). Plots of the data are therefore relatively uninteresting and are not included herein.



Figure 3.1-16 Elution Curves at Well B for DIMP, CPM Sulfone and Dieldrin During the Forced-Gradient Portion of the Experiment.

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Figure 3.1-17 Elution Curves at Well B for Chloroform, Dicyclopentadiene, Trichloroethene and Tetrachloroethene During the Forced-Gradient Portion of the Experiment.



Figure 3.1-18 Elution Curves at Well B for Tetrachlorobenzene and Two Unknown Multihalogenated Compounds During Forced-Gradient Portion of Experiment.



Figure 3.1-19 Elution Curves at Well C for DIMP, CPM Sulfone and Dieldrin During the Forced-Gradient Portion of the Experiment.

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Figure 3.1-20 Elution Curves at Well C for Chloroform, Dicyclopentadiene, Trichloroethene and Tetrachloroethene During the Forced-Gradient Portion of the Experiment.



Figure 3.1-21 Elution Curves at Well C for Tetrachlorobenzene and Two Unknown Multihalogenated Compounds During the Forced-Gradient Portion of the Experiment.



Figure 3.1-22 Elution Curves at Well C for Benzene, o-Xylene, Bicylcoheptadiene (BCHD) and an Unknown Isomer of Pentachloro-1,3-butadiene During the Forced-Gradient Portion of the Experiment.

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Figure 3.1-23 Elution Curves at Well C for CPM Sulfide and CPM Sulfoxide During the Forced-Gradient Portion of the Experiment.

3.1.2.3.4 <u>Colloids</u>

The analytical procedure for separating colloidal (10,000 Dalton <0.4 μ m) material from bulk solution (discussed in Section 2.2.5) was implemented four times; 1) on the pre-experiment ground-water fluids (reported in Table 2.2-3); 2) on the injection (i.e., post-treatment) fluids; 3) at Well P1, 64 hours into the injection experiment; 4) again at P1, 114 hours into the experiment. These samples correspond to sample numbers B-1 through B-4 in Table 3.1-1.

The background DOC concentration, that is, the in situ, pre-experiment concentration, of the aquifer fluid was about 93 mg/L; the injection water had a DOC of 8.6 mg/L. In comparison, post pre-filter DOC values for the aquifer fluids sampled at 64 and 114 hours after injection show DOC levels roughly intermediate between the two end members: this suggests a mixing of injection and in situ fluid masses. Colloid organic carbon concentration values are, for both of the samples taken during the injection period, higher than are the background levels, suggesting some mobilization of colloidal material.

The ratio of colloid mass (Coll) to the organic carbon content of the colloids (COC) in the background (pre-injection) fluids is 14 (Table 3.1-1; 2.2-3). Natural organic material, e.g., humic and fulvic substances, have mass-to-carbon ratios of about 2.5; thus, the colloidal material of the study site is significantly carbon-poor compared to substances of natural origin. An even greater departure from natural material is seen with the colloids remaining in the post-treatment system: not only was the mass of colloidal material reduced via the treatment, but the colloids were significantly depleted in carbon content (a Coll/COC ratio of 82 compared to 14). Fluid samples taken during the injection experiment show ratios intermediate between the two end members, and exhibit a progression from the background value to the injection value, with time. Coll/COC ratio of the colloidal material, particularly the highly carbon-depleted injection fluid colloids, can be used to trace colloid movement through the aquifer system.

The data presented in Figure 3.1-25 are compared to those for bromide (Br), a presumably conservative tracer. Bromide breakthrough occurred more rapidly at P1 than anticipated and the data reflect the "back" side of the bromide pulse. Although the leading edge of the bromide tracer pulse is missing, the data suggest that the colloidal material is substantially retarded relative to bromide.

Date	Sample	Volume (1)	DOC (mg/L) (in sample)	DOC (mg/L) in situ	SD (mg/L)	TOC(mg) (mg/L x l)	% PFW [-]	Coll(mg) (sum)	COC(mg) (sum)	Coll/COC [-]	t since inj. (h)
10/11/91	B-1PFW1	107.5	93.53	93.53	2.09	10054					
	B-1PFW2	107.5	93.23	93.23	3.72	10022					
	B-1COL	0.89	44.89	0.37	0.47	40	0.40	560	40	14.0	0
	B-1UFW	106.5	84.98	84.98	1	9050	90.30				
										·	
11/03/91	B-2-PFW	202.9	8.65	8.65	0.32	1755					
	B-2-COL	0.68	15.87	0.1	0.26	11	0.61	885.4	11	82.0	0
	B-2UFW	201.9	14.92	14.92	0.17	3012	171.64				
	· .										
11/05/91	B-3FPW	110.2	53.18	53.18	1.64	5860					
	B-3-COL	1.18	161.2	0.94	2.09	190	3.25	4880	190	25.7	64.17
	B-3-UFW	109	43.07	43.07	0.13	4695	80.11				
11/07/91	B-4-PFW	135.9	49.05	49.05	0.66	6666					
	B-4-COL	0.85	122.4	0.94	1.31	104	1.56	6642	104	63.8	113.67
	B-4-UFW	134.8	50.46	50.46	0.52	6807	102.12				

Table 3.1-1 Summary of Baseline Colloid Information

PFW1: Prefilter Water (original sample)

PFW2: Prefilter Water (duplicate of PFW1)

COL: Colloids

UFW: Ultra Filtrate Water

Volume: Volume of aquifer fluid in each fraction; PFW is the total volume pumped; COL is the final colloid retentate volume.

DOC: "Dissolved" organic carbon concentration in each fraction.

DOC(in situ): The DOC concentration estimated to be in the aquifer fluids; e.g., for COL: DOC (in situ) = $44.89 \times 0.89 = 0.37$

SD: Standard deviation on replicate samples for DOC measurement.

TOC: Total carbon in fraction volume, e.g., 93.53 mg/L x 107L = 10054 mg.

% PFW: Percent of fraction relative to total; e.g., COL = 40/10054 = 0.4%.

Coll(sum): Colloid mass in total sample.

COC(sum): Colloid organic carbon in total sample.





The four colloid samples were analyzed to characterize the sorbed component of contamination. The sorbed component of contamination was analyzed by the USGS National Water Quality Organics Laboratory. First, each colloid sample was pulverized, added to 10 mL of methylene chloride, and then sonicated for 10 minutes. The solvent was then decanted, while the sample was again added to a fresh 10 mL of methylene chloride. This procedure was repeated four times for each sample, and then the extracts from each sample were combined and filtered (0.2 microns glass fiber), producing roughly 40 mL of extract per sample. The extract was then concentrated, on a micro Kuderna-Danish condenser, to 1 to 2 mL, and then finally further concentrated to 0.1 mL using a gentle stream of nitrogen. An internal standard was then added to the extract which was then injected into the GC.

The organic analyte composition of the colloidal material is given in Table 3.1-2. All values in the table are µg compound/kg colloid mass. The colloid material contains substantial amounts of a range of compounds, including some which are highly insoluble (e.g., DDT). Table 3.1-3 presents calculated concentrations (µg/L) of target analytes associated with colloidal material. While the mass associations of the organic analytes with the colloids are, on a mass basis, significant, the extremely low colloid concentration in the ground-water results in a small contribution of the colloid phase to the distribution of the target organics in the system (i.e., dissolved, colloid and aquifer sediment). Never the less, the combination of colloid movement through the aquifer, as evidenced by the near breakthrough of the injected (high Coll/COC) colloids and the associated organic contaminants, provides documentation of the ability of colloidal material to transport insoluble materials through an aquifer system.

3.2 NATURAL-GRADIENT PORTION OF THE EXPERIMENT

The natural-gradient portion of the experiment began on November 14, 1991 at 5:47 pm, roughly 281 hours after the injection/extraction was started. At this time the injection system and extraction well pump were turned off. A second spike of LiBr tracer was injected for approximately 3.5 hours immediately preceding shut-off. This spike was intended as a signal of the shut-off point which could be monitored in the aquifer. Monitoring of both hydraulic and chemical data were continued for approximately 240 more hours (10 days) in order to investigate the post-flushing response of the aquifer.

Table 3.1-2	Organic A	Analysis o)f	Colloidal	Material	Sam	pled	from	Well	Pl
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Target Compound	Pre-Injection	Injection Water	Time = 63 h	Time = 114 h
Atrazine	100	<3	03 h	
Cl ₅ -1,3-butadiene	40	<3	30	100
Cl ₅ -1,3-butadiene	50	<3	10	200
alpha-chlordane	400	10	<2	<2
gamma-chlordane	600	10	2	<2
CPM-sulfide	<6	<3	<2	2
CPM-sulfone	400	50	600	400
CPM-sulfoxide	<6	<3	<2	<2
C ₇ -H ₃ Cl ₇	<6	<3	<2	<2
DCPD	400	<6	300	50
DDE	3200	80	<2	<2
DDT	400	<3	<2	<2
Dieldrin	<6	<3	<2	<2
DIMP	1600	4	4	20
Endrin	<6	<3	<2	<2
Isodrin	200	70	4	<2
Multihalogen (MW364)	20	100	80	20
Multihalogen (MW378)	100	50	30	<2
Tetrachlorobenzene	<6	4	<2	20

All values are µg compound/kg colloid mass. Pre-injection values represent composite (i.e., Wells A, B, and C), pre-experiment colloid organic loadings; injection water values correspond to post-treatment-system colloids. Values in time columns are for colloids sampled at 63 and 114 hours after the initiation of injection.

Table 3.1-3	Calculated Concentration (µg/L) of Organic Compounds Associated with Colloidal Material.
	Based upon a background colloid concentration of $5.2 \times 10^{-6} \text{ kg/L}^*$

Target Compound	Pre-Injection concentration on colloids (µg/L) (compound colloid)	In situ colloid-contaminant concentration (µg/L)
Atrazine	100	5.2 x 10 ⁻⁴
Cl ₅ -1,3-butadiene	40	2.1 x 10 ⁻⁴
Cl ₅ -1,3-butadiene	50	2.6 x 10 ⁻⁴
alpha-chlordane	400	2.1 x 10 ⁻³
gamma-chlordane	600	3.1 x 10 ⁻³
CPM-sulfide	<6	
CPM-sulfone	400	2.1 x 10 ⁻³
CPM-sulfoxide	<6	
C ₇ -H ₃ Cl ₇	<6	
DCPD	400	2.1 x 10 ⁻³
DDE	3200	1.7 x 10 ⁻²
DDT	400	2.1 x 10 ⁻³
Dieldrin	<6	
DIMP	1600	8.3 x 10 ⁻³
Endrin	<6	
Isodrin	200	1.0 x 10 ⁻³
Multihalogen (MW364)	20	1.0 x 10 ⁻⁴
Multihalogen (MW378)	100	5.2 x 10 ⁻⁴
Tetrachlorobenzene	<6	

* For example:

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$$\left(\frac{100 \ \mu g \ atrazine}{kg \ colloids}\right) \times \left(\frac{5.2 \times 10^{-6} \ kg \ colloids}{L}\right) = 5.2 \ \times \ 10^{-4} \ \mu g/L$$

3.2.1 HYDRAULICS

After the extraction/injection was ceased, the response of the aquifer as it returned to natural gradient conditions was monitored. Water-level data from this period indicate that water table elevations returned to background elevations within a few minutes of stopping the extraction/injection, except at Well E. At Well E, water levels increased immediately after stopping injection/extraction and then took 24 hours to reach the background water elevations. This behavior probably reflects a leaky valve in the extraction pump which may have allowed water in the discharge line to recharge back into the well. Aside from this observation, a situation similar to that described in Section 3.1.1 (extremely small elevation changes combined with background data variability) precluded accurate gradient calculations.

3.2.2 CHEMISTRY

3.2.2.1 <u>Treatment System Performance</u>

After stopping the injection/extraction, the treatment system was not used, so no monitoring was conducted.

3.2.2.2 Aquifer monitoring

During the natural-gradient portion of the test, samples were periodically collected from Wells A, B and C. The wells were pumped at approximately 500 ml/min for approximately 5 minutes in order to flush the screen section and tubing. The samples were collected as usual and the pump was turned off. The following interpretations of the monitoring data are preliminary and will be corroborated by future modeling efforts and/or laboratory studies.

3.2.2.2.1 Tracer behavior

Figure 3.2-1 presents the bromide concentrations in Wells A, B and C for the natural-gradient portion of the experiment. In Well A, the bromide concentration remains for about 50 hours at the value measured at the end of the forced-gradient portion of the test, then begins to drop off slowly. In Well B, the bromide concentration remains relatively stable for 50-100 hours, with a slight decrease in concentration noticeable thereafter. The behavior of bromide in Well C is similar to that observed in Well B.



Figure 3.2-1 Bromide Concentrations in Monitoring Wells A, B and C During the Natural-Gradient Portion of the Experiment.

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No bromide spike was detected at any of the wells during the monitored period. This suggests that the bromide spike added at the end of the forced-gradient portion of the experiment either was too narrow or dispersed to be noted or had not yet migrated to the sampling point. If the second explanation is accepted, it would imply that the average linear ground-water velocity was less than 1 foot per day, i.e. less than 10 feet in approximately 250 hours (the travel distance to Well A divided by the total time of monitoring during the natural-gradient portion of the test). This is consistent with expectations from modeling discussed in Section 2.1.1, which estimate average linear ground-water velocity at less than 0.1 ft/day.

In any case, the fact that the bromide concentrations remain relatively constant in Wells B and C for approximately 250 hours suggests that the wells are not sampling contaminated water resulting from advection of the pre-existing plume. The reason for the gradual decrease in bromide concentration in Well A beginning about 50 hours into the natural-gradient portion of the test is not so clear. This issue will be addressed more in the next section.

3.2.2.2.2 Organic contaminant behavior

• Well A. Figure 3.2-2 presents the concentration histories for DIMP, CPM sulfone and dieldrin in the strata sampled by Well A. During the natural-gradient portion of the experiment, the concentrations of all three contaminants are observed to rise to roughly the values measured before beginning the forced-gradient portion of the experiment. The same is true for tetrachlorobenzene and the two unknown multihalogenated compounds (Figure 3.2-3). The DIMP increase is likely to be due in part to the DIMP that was reinjected toward the end of the forced-gradient portion of the test. However, this cannot be the only source, since the maximum reinjected DIMP concentration was approximately 35 μ g/L, whereas the concentration rises to over 800 μ g/L in Well A during the natural-gradient portion of the test. Presumably the source of the DIMP is the same source that causes the observed concentration increases for the other contaminants.

Because the bromide concentration during the natural-gradient portion of the experiment is not completely flushed out (Figure 3.2-1), this increase in organic concentrations is unlikely the result of the plume migrating back into the experimental zone. Thus, there are possibly sources for the contaminants within the portion of the aquifer which had been swept by the bromide-enriched injection water. Two possible



Figure 3.2-2 Concentration Histories for DIMP, CPM Sulfone and Dieldrin at Well A During the Natural-Gradient Portion of the Experiment.



Figure 3.2-3 Concentration Histories of Tetrachlorobenzene and Two Unknown Multihalogenated Organics at Well A During the Natural-Gradient Portion of the Experiment.

sources of organic contaminant mass that might have remained after flushing of the zone with the bromide-enriched water are 1) desorption of organics sorbed to the geologic media, and 2) diffusion of the organics out of relatively low permeability layers within the medium.

As described in Section 2.2.4, the analysis of core samples taken from the same depth horizon as sampled by Well A indicated that there were generally very low or nondetectable concentrations of contaminants associated with the geologic media. For example, CPM sulfone was detected at 3.5 and 3.8 μ g/kg in two different cores. This sorbed mass is insufficient to explain the observed rise in CPM sulfone concentrations in the ground water during the natural-gradient portion of the test. The sorbed concentration is assumed to be 3.8 μ g/kg (the higher of the above two values), the porosity of the aquifer is assumed to be 0.3, and the solid density of the grains is 2.7 g/cm³, both reasonable values for a sandy aquifer. Furthermore the porosity value was estimated in this work to be on the order of 0.3. Finally, <u>all</u> of the sorbed mass is assumed to desorb instantaneously during the natural-gradient portion of the test. The pore water, which is contaminant-free due to the flushing in the forced-gradient portion of the test. The pore water concentration resulting from complete, instantaneous desorption of CPM sulfone can then be calculated as approximately 24 μ g/L. This value is lower than the observed plateau value of approximately 40 μ g/L. A similar calculation for DIMP yields an estimated pore water concentration from instantaneous desorption of 200 μ g/L, which is also considerably lower than the observed value of 800 μ g/L.

Furthermore, the simple method of estimating the solution concentrations outlined above almost certainly overstates the concentration increase expected from desorption. In reality, it is extremely unlikely that all of the sorbed mass would desorb; it is more likely that a portion of the sorbed mass would desorb to raise the solution concentration to a point at which the equilibrium ratio between solution and sorbed concentrations is similar to that before the flushing occurred (this would be the expected effect if the sorption isotherm were "linear"). Thus, the expected concentrations from desorption would be significantly lower than measured, which suggests that other sources must contribute some or all of the contaminant mass found in the pore water sampled by Well A toward the end of the natural-gradient portion of the experiment.

The other possible source is contaminant mass retained in relatively lower permeability layers within the strata sampled by Well A. These may not have been flushed by the injected water during the forced-gradient portion of the experiment, since the advective rate through these layers may have been very slow

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compared to the rate through adjacent, more permeable layers. Thus the contaminant mass in these layers was left behind, except for that small fraction that may have been able to diffuse from the outer portions of the lower permeability layers into the rapidly moving water in the adjacent higher permeability layers. During the natural-gradient portion of the experiment, when the ground-water flow rate through the adjacent, more permeable layers was much lower, diffusion may have been rapid enough to allow the concentrations in the more permeable layers to rise. Since the monitoring wells would draw water preferentially from these more permeable layers, this concentration rise would be detected in the monitoring results.

The hypothesis that lower-permeability layers are in contact with the strata sampled by Well A is also supported by the decrease in bromide concentrations noted in Well A during the natural-gradient portion of the experiment. Since the injected water would not flush through these lower-permeability layers, they would, at the beginning of the natural-gradient portion of the experiment, contain water with very low background concentrations of bromide. The water in the more permeable layers, on the other hand, would have the higher concentrations of bromide resulting from injection. Thus the lower-permeability layers may act as sinks for some of the bromide mass, as bromide diffuses into them. This phenomenon would gradually lower the bromide concentration in the water in the more permeable layers, explaining the results noted by monitoring Well A.

Well B. Figure 3.2-4 shows the concentration histories of DIMP, chlorophenlymethyl sulfone and dieldrin at Well B during the natural-gradient portion of the experiment. Figure 3.2-5 presents similar information for tetrachlorobenzene and the two unknown multihalogenated compounds. If we take the last data point for the multihalogenated compound (MW 378) to be a possible artifact, the only contaminants for which concentrations rise significantly in Well B are DIMP and CPM sulfone. The rise in the DIMP concentrations to approximately 300 μ g/L cannot be due only to the DIMP that was reinjected during the final 100 hours of the forced-gradient portion of the experiment, since the maximum reinjection concentration was approximately 35 μ g/L. Thus there must be some other source for the DIMP that appeared in Well B. The same source presumably contributes all of the CPM sulfone, which rises to about 15 μ g/L, since it was not reinjected. Since the bromide concentrations do not decrease significantly, as discussed earlier, it would appear that a source for the contaminant mass must be that sorbed to the geologic media. Calculations such as described above indicate that instantaneous desorption of all the maximum detected sorbed mass would yield about 200 μ g/L of DIMP and about 35 μ g/L of CPM sulfone.

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Figure 3.2-4 Concentration Histories of DIMP, CPM Sulfone and Dieldrin at Well B During the Natural-Gradient Portion of the Experiment.



Figure 3.2-5 Concentration Histories of Tetrachlorobenzene and Two Unknown Multihalogenated Compounds at Well B During the Natural-Gradient Portion of the Experiment.

As before, however, not all of the sorbed mass would be likely to desorb instantaneously, leading to lower expected concentrations in the ground water over time.

• Well C. Figure 3.2-6 presents the concentration histories for DIMP, CPM sulfone and dieldrin at Well C during the natural-gradient portion of the experiment. Figure 3.2-7 presents the same information for tetrachlorobenzene and the two unknown multihalogenated compounds. Figure 3.2-8 presents the results for DCPD, benzene, o-xylene, TCE and PCE. For all of the compounds except dieldrin, the concentrations rise significantly during the natural-gradient portion of the experiment. However, the bromide concentrations do not decrease significantly, as discussed earlier.

Calculations again suggest that desorption of the detected sorbed mass is insufficient to lead to the observed rises in ground-water concentrations. However, a possible reservoir of contaminants for stratasampled by Well C is the upper portion of the Denver Formation which forms the base of the aquifer. The Denver Formation is known to be weathered and somewhat permeable and potentially has significant concentrations of the contaminants in its pore water and sorbed to the solid media. Especially if an upward vertical gradient exists at the site (as have been sporadically detected elsewhere at RMA), the Denver Formation may be the source for contaminants reappearing in strata sampled by Well C during the natural-gradient portion of the experiment.



Figure 3.2-6 Concentration Histories for DIMP, CPM Sulfone and Dieldrin at Well C During the Natural-Gradient Portion of the Experiment.



Figure 3.2-7 Concentration Histories of Tetrachlorobenzene and Two Unknown Multihalogenated Organics at Well C During the Natural-Gradient Portion of the Experiment.



Figure 3.2-8 Concentration Histories for DCPD, Benzene, o-Xylene, TCE and PCE at Well C During the Natural-Gradient Portion of the Experiment.

4.0 SUMMARY AND IMPLICATIONS

4.1 RESULTS OF THE FIELD TEST ON FLUSHING OF ORGANIC CONTAMINANTS

An applied research project was conducted in a high-concentration, mixed-contaminant plume in Section 23 upgradient of the NBCS system at the RMA. The transport behavior of the contaminants was assessed in the field by monitoring the rate at which they were flushed from the aquifer by organic-free water. The organic-free water, obtained by treatment of ground water withdrawn from an extraction well, was returned to the aquifer through an injection well approximately 30 feet upgradient. The injection water was spiked for various periods with a conservative, nontoxic tracer. The migration rate of the tracer and the flushing rates of the organic contaminants were monitored in a cluster of short-screened monitoring wells located between the injection and extraction wells. The monitoring wells A, B, and C monitored the top third, middle third, and bottom third of the aquifer, respectively. Samples from the monitoring wells were collected and analyzed during the 10-day period of injection/extraction (the "forced-gradient" portion of the test), and for approximately 2 weeks after injection/extraction was ceased (the "natural-gradient" portion of the test).

The existing contamination was stratified vertically and contained a wide range of contaminants including halogenated VOCs, aromatic hydrocarbons, organochlorine pesticides and other, more polar, organic compounds. It is noteworthy that the total DOC concentration of the ground water (approximately 100 mg/L) was significantly higher than the sum of the concentrations of the analytes detected during historical monitoring (approximately 30 mg/L). In this work, despite some effort and identification of several previously unrecognized organic contaminants, it was not possible to determine the composition of the vast majority of the previously unidentified fraction of the DOC.

The properties and expected mobilities of the identified organic compounds varied widely. Some of the compounds would ordinarily be expected to travel at or near the velocity of the ground water (e.g. DIMP and chloroform), whereas others would ordinarily be expected to be retarded significantly in their migration due to sorption by the aquifer media (e.g. dieldrin and tetrachlorobenzene).

Contrary to expectations, during the forced-gradient portion of the test, the ground-water concentrations of all contaminants were reduced rapidly, essentially at the same time as the bromide tracer (i.e. the clean water flush) arrived at the monitoring points. There was no significant retardation of any of the

contaminants based on the ground-water concentration histories. The ground-water concentrations of all contaminants were rapidly reduced to near or below the detection limit. Nevertheless, after cessation of injection/extraction, the concentrations of many of the contaminants in the monitoring wells rose slowly over a period of two weeks, in some cases to values very similar to that observed before the forced-gradient portion of the test began. These latter results suggested that after the flushing there remained contaminant mass in one or more reservoirs which could impact the ground water in the aquifer. Insight into the reasons for the observed behavior was gained from a variety of other information collected during the study, as discussed below.

Laboratory studies showed that the aquifer media was able to sorb PCE to a significant degree from synthetic ground water (with inorganic chemistry similar to site ground water, but no additional organic chemicals). Thus, the aquifer media should have been able to sorb substantial quantities of dieldrin and the other more hydrophobic organic chemicals. However, chemical analysis of samples of the aquifer sediments taken before the flushing experiment indicated that the sorbed concentrations of organic chemicals was very low, in most cases below the detection limit. These facts suggested that there were agents in the ground water which were capable of preventing significant sorption of the contaminants by the sediments.

Recently there has been considerable interest in the possibility that colloidal material in ground water may facilitate the transport (i.e. reduce the retardation) of organic chemicals. This effect would occur only if the colloidal material were mobile, strongly sorbed the contaminants, and present in relatively high concentrations. Considerable effort was made during the forced-gradient portion of the field test to monitor the movement and composition of colloidal matter in the aquifer. The results indicate clearly that colloids were present and migrate through the aquifer. Furthermore, chemical analysis of the colloids indicated that they carry a significant load of sorbed contaminants per unit mass of colloid. However, the concentration of colloids in the ground water was relatively low. Overall, the evidence from this work suggests that the colloids were not a significant factor in controlling contaminant mobility at this site. However, it is conceivable, but cannot be proven from this work, that colloidal enhancement of contaminant mobility could be significant at other sites at RMA or elsewhere if the colloid concentrations are higher.

Another potentially important mechanism for enhancement of contaminant mobility is the increase of contaminant solubility which can be caused by the presence of co-solvents such as alcohols, ketones, etc.

(Nkdei-Kizza, 1985, 1987). This effect, however, is likely to be significant only if the co-solvents are present as a significant fraction of the "ground water" (on the order of 5 percent or more, which, for methanol as an example, is on the order of 40,000 mg/L). Since the total DOC in the site ground water was on the order of 100 mg/L, it appears unlikely that the effect of co-solvent on contaminant mobility at the site is significant (unless extremely high concentrations of co-solvents are lost during processing of the samples prior to the DOC analysis).

A third potentially important mechanism for enhancement of contaminant mobility is the interaction of contaminants with surfactants in solution in ground water. The concentrations of anionic surfactants, however, were much too low to have resulted in a detectable reduction of sorption and increase in mobility. The concentrations of nonionic surfactants was not measured. If all of the DOC measured in the site ground water were nonionic surfactants, then it is conceivable that a significant effect on mobility of other contaminants could occur. It seems unlikely, however, that all of the measured DOC is nonionic surfactants. Even if it were, as long as the nonionic surfactants were assumed to be the only mobility-enhancing mechanism, theory would not predict that all of the contaminants would be unretarded, as observed. Instead, it would be expected that the retardation of each would be reduced somewhat (leading to a range of enhanced mobilities, with dieldrin still much slower than chloroform, for example).

In summary, the extensive efforts of this work were not able to pinpoint the reasons that the contaminants were so rapidly flushed from the aquifer. Nevertheless, it is clear that unknown properties of the ground water, rather than the properties of the aquifer sediments, are responsible for the insignificant concentrations of contaminants sorbed to the sediments, and the corresponding unretarded migration rate of the contaminants within the aquifer. This issue will be pursued to the extent possible by D. Mackay and students through limited funding currently available through the University of Waterloo.

The rebounding of contaminant concentrations observed during the natural-gradient portion of the test appears to be a consequence of contamination residing in portions of the subsurface around or adjacent to the flushed portion of the aquifer, which for some reason, were not flushed during the forced-gradient portion of the test. This conclusion is based on the expectation that the very low natural-gradient ground-water velocity would have been much too slow to result in the plume around the zone impacted by injection being advected back to the monitoring location. This conclusion is strengthened by the continued presence of the injected tracer in the water sampled by the monitoring points, which otherwise would have been flushed away. Thus, the reservoirs for contaminant mass which could conceivably have impacted

the monitoring wells must exist within the areal extent impacted by the tracer injection. It is clear that the Denver Formation, which lies beneath the aquifer, could be a contributor of contaminants to the lower portion of the aquifer. It appears also, that the capillary fringe, or the portion of the aquifer just above it, may be a contaminant reservoir contributing contaminants to the upper portion of the aquifer. Finally, it appears possible that lower permeability strata distributed within the aquifer may also have remained unflushed after the forced-gradient portion of the experiment and then, during the natural-gradient portion of the experiment, contributed contaminants by diffusion or slight advection to the more permeable strata surrounding them. However, examination of the cores suggests that most of these strata are likely to be sandy with only slightly less permeability than the bulk of the aquifer, with occasional exceptions of thin clayey and silty strata.

4.2 IMPLICATIONS OF THE FINDINGS FROM THIS WORK

This project demonstrates that small-scale pilot tests may be conducted within existing plumes of contamination to yield insight into the processes which most significantly impact the efficiency of pump and treat remediation. In particular, the results suggest that although the contaminants were removed essentially unretarded from the permeable and highly contaminated portion of the aquifer in this work, there would appear to be value in a pulsed pumping approach to allow more efficient removal of contaminants residing in lower-permeability portions of the subsurface. If the zone at or above the water table proves to be a significant reservoir of contaminants, as suggested by this work, it may be advantageous to raise and lower the water table during remediation. However, a reservoir which is likely to contribute contaminants to the permeable aquifer over a longer term is the contaminated portion of the Denver Formation, as suggested by the results of this work.

Two innovative techniques for organic chemical analysis were successfully used in this project and may warrant consideration in other efforts at RMA. SPE followed by GC/MS analysis proved to be a relatively simple and economical way to monitor for semivolatiles. An on-site, semi-automated GC system proved useful for real-time analyses of VOCs, allowing, among other things, rapid detection of failure of the activated carbon system for treatment of the extracted water.

Probably related to the rapid failure of the activated carbon system is the determination during the experiment that much of the organic carbon load has not been characterized. This is a common observation at highly contaminated sites (Bramlett, et al, 1987). Estimates derived from historical RMA

ground-water monitoring programs suggest that the dissolved organic carbon load of the ground water is roughly 30 mg/L. Organic carbon measurements obtained during the experiment indicate that the actual dissolved organic carbon load is closer to 100 μ g/L. Treatment system breakthrough estimates using the actual DOC values (100 mg/L) correlate closely with the treatment system breakthrough times observed during the experiment. The uncharacterized DOC possibly contains the answer to the enhanced contaminant mobility observed during the experiment. Thus, full chemical characterization of the ground water at the site is needed prior to further remedial design efforts.

Although not of direct interest to this work, it was noted that the elevated concentrations of DO in the injected water were propagated rapidly through the aquifer. This suggests that there are not significant, rapid sinks for oxygen in the aquifer, such as reduced mineral species. Thus it appears that oxygen could be delivered relatively easily to large volumes of the subsurface to encourage in-situ aerobic biodegradation during aquifer remediation.

Lastly, there appear to be at least two types of agents in the ground water which enhance the mobility and removal rate of the identified organic contaminants. The most clearly identified agent is the colloidal matter; although the effect of the colloidal matter is quite insignificant at this site, it may not be at others. The other agent or agents are unidentified, but appear to be very significant in enhancing the mobility of the identified contaminants. In fact, the results suggest that the contaminated ground water is an extremely efficient fluid for extracting contaminants from the subsurface. This fact suggests that the ideal pump and treat remediation scheme might involve a specialized treatment system capable of removing identified and targeted contaminants without altering the extracting capabilities of the ground water. If this were possible, the treated ground water might then be reinjected to extract more contaminant mass. Implementation of this strategy would require more research to determine what components give the contaminated ground water its extracting capabilities.

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6.0 ACKNOWLEDGEMENTS

The successful completion of this project would not have been possible without the support and funding from the Project Manager, Rocky Mountain Arsenal (PMRMA) and the analytical services provided by the United States Geological Survey National Water Quality Laboratory. In particular, the authors would like to acknowledge the support from Greg Ward and Brian Anderson of the PMRMA, and Bob Middelberg of the USGS. Additionally, extraordinary contributions from the following people were integral to the outcome of the project:

• For 24 hour shift duty involving collection, preparation, analysis, and shipment of groundwater samples, and the continuous maintenance of the experiment's many electric and mechanical systems...

Dietrich Whitesides - RLSA Todd Sullivan - RLSA Dan Kenney - RLSA Larry Hudnall -RLSA

• For long hours of preliminary design and preparation, rig oversight, and well development...

Ed Young - RLSA

For set-up and trouble shooting of the semi-automatic field GC...

Gary Hopkins - Analytic and Remedial Technology

· For analytical insight, support, and responsiveness...

Steve Zaugg - USGS National Water Quality Lab

• For coordination of the analytical requirements and last minute procurement of the LiBr tracer...

Jim Seeley - USGS National Water Quality Lab

· For Analytical Methods development support...

Mark Sandstrom - USGS National Water Quality Lab

· For donating the organic carbon analyses and equipment for the colloid studies...

Peter Santschi - Texas A&M University M. Baskaran - Texas A&M University

· For Moral Support...

Jennifer Myller Lynda Smith Joyce Schroeder Christina Branch

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APPENDIX A

HISTORICAL GROUND WATER CHEMICAL DATA WELL 23095

· FALL/WINTER 1989 TO WINTER 1991

· FALL 1991

FALL/WINTER 1989 TO WINTER 1991

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Data Definition: RKPMCGW

Site	Site_ID	Test	Sample	Field	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Corrected	UOM	Acc	Flag Anal
Туре		Name	Date	Number	Depth		Number	Date	Date	Num		Value			Code Type
LL	23095	111TCE	89307	M4484,85	48.0	UB	HQN008	89315	89315	N8	1 T	0 760000		0 070	64
WELL	23095	111TCE	89307	M4492,93	48.0	UB	HQW002	89319	89319	UN21	1.T	100 000000	UCL	1 020	U (
LL	23095	111TCE	90057	M9005,06	48.5	UB	I ZB009	90067	90067	N8	11	0 760000	HCL	0 070	AI
LL	23095	111TCE	90141	N1747,48	48.0	UB	JRH005	90147	90147	NB	17	0.760000	UGL	0.930	C1
WELL	23095	111TCE	90233	N3012	48.4	ED	GYR016	90242	90242	TT8	LT	1.090000		0.730	C1
սելլ	23095	111TCE	91039	N5191,92	48.0	UB	NEQ013	91043	91043	N8	1 T	0 760000	UCI	0.771	C1
; LL	23095	112TCE	89307	M4484,85	48.0	UB	HQN008	89315	89315	NB	11	0 780000	HCL	0.750	C1
wall	23095	112TCE	89307	M4492,93	48.0	UB	HQW002	89319	89319	10021	11	100 000000		1 070	LI 1A
WELL	23095	112TCE	90057	M9005,06	48.5	UB	1ZB009	90067	90067	NR	11	0 780000	UGL	0.940	1A 01
LL	23095	112TCE	90141	N1747,48	48.0	UB	JRH005	90147	90147	NR	iT	0.780000		0.000	UI
LL	23095	112TCE	90233	N3012	48.4	ED	GYR016	90242	90242	TTR		1 630000		0.000	U1
WELL	23095	112TCE	91039	N5191,92	48.0	UB	NEQ013	91043	010/3	NR	17	0.790000	UGL	0.958	U1
""LL	23095	11DCE	89307	M4484,85	48.0	UB	HONOOS	80315	80315	NO		1.700000	UGL	0.860	C1
LL	23095	11DCE	89307	M4492.93	48.0	UR	HOUOO2	80310	80710	11021	17	1.700000	UGL	0.890	C1
WELL	23095	11DCE	90057	M9005.06	48.5	UR	178000	00067	00067	NO	LI 1 T	100.000000	UGL	0.896	14
WELL	23095	11DCE	90141	N1747.48	48.0	UB	.IRH005	001/7	001/7	NO	L1 1 T	1.700000	UGL	0.890	C1
LL	23095	11DCE	90233	N3012	48.4	FD	CYP016	002/2	90147	NO		1.700000	UGL	0.890	C1
LL	23095	11DCE	91039	N5191.92	48.0	UB	NEO013	010/3	90242	110		1.850000	UGL	0.948	C1
WELL	23095	11DCLE	89307	M4484.85	48.0	LIB	HONOOR	20715	91043	NO		1.700000	UGL	0.890	C1
11	23095	11DCLE	89307	M4402 03	48.0	110	401003	90710	07313	NÖ		0.730000	UGL	0.890	C1
LL	23095	11DCLE	90057	M9005 06	40.0		170000	00047	09319	UM21	LT	100.000000	UGL	0.918	18
WELL	23095	11DCLE	90141	N1747 48	48.0	110	120009	90007	90067	NÖ	LT	0.730000	UGL	0.890	C1
WELL	23095	11DCLE	90233	N3012	40.0	50	JKRUUJ	90147	90147	NS	LT	0.730000	UGL	0.890	C1
LL	23095	11DCLE	91039	N5101 02	-0.4 /8 0		UTRUID	90242	90242	118	LT	1.930000	UGL	0.919	C1
LL	23095	120CF	89307	M//8/ 85	40.0	08	NEQUIS	9104.5	91043	N8		1.440000	UGL	0.890	C1
WELL	23095	120CE	89307	M//02 07	40.0	UB	HUNUUS	89315	89315	N8	LT	0.760000	UGL	0.930	C1
LL.	23095	120CE	90057	MO005 04	40.0	UB	HUWUUZ	89319	89319	UM21	LT	500.000000	UGL	0.982	1A
LL	23095	120CE	90141	N1767 /9	40.7	UB	128009	90067	90067	N8	LT	0.760000	UGL	0.930	C1
WELL	23095	12005	00277	NZ012	48.0	UR UR	JRH005	90147	90147	N8	LT	0.760000	UGL	0.930	C1
HELL	23095	12005	01030	N5101 02	48.4	ED	GYR016	90242	90242	TT8	LT	1.750000	UGL	0.924	C1
u	23095	12001 5	80307	W//0/ 05	48.0	OB	NEQU13	91043	91043	N8	LT	0.760000	UGL	0.930	C1
WELL	23095	12001 5	80307	M4404,00	48.0	OB	HQN008	89315	89315	N8	LT	1.100000	UGL	0.820	C1
VELL	23095	12001 5	00057	HOODE 04	48.0	UB	HQWOOZ	89319	89319	UM21	LT	100.000000	UGL	0.996	1A
LL	23095	12001 5	90161	N17/7 /0	48.0	UB	128009	90067	90067	N8	LT	1.100000	UGL	0.820	C1
ι.	23095	120000	00277	N1/4/,40	48.0	UB	JRH005	90147	90147	N8	LT	1.100000	UGL	0.820	C1
WELL	23095	120015	90233 01030	N5101 00	48.4	ED	GYR016	90242	90242	TT8		12.300000	UGL	0.960	C1
1-LL	23095	120010	80307	N//02.07	48.0	UB	NEQ013	91043	91043	N8	LT	1.100000	UGL	0.820	C1
11	23005	130010	07307 80707	M4492,93	48.0	UB	HQWOOZ	89319	89319	UM21	LT	100.000000	UGL	0.933	1A
in-11	23005	13000	07307 80707	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	100.000000	UGL	1.050	14
UFII	23005	17040	09307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	480.000000	UGL	1.050	1A
11	22005	17040	89307	M4482,83	48.0	UB	HQM008	89315	89315	AV8	LT	1.320000	UGL	0.876	C1
·	23093	120MB	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	100.000000	UGL	0.991	1A
5 66. UCTI	23093	TSUMB	90057	M9003,04	48.5	UB	I ZA009	90067	90067	AV8		2.270000	UGL	0.876	C1
WEEL	23093	TSDMB	90141	N1745,46	48.0	UB	JRG005	90147	90147	AV8	LT	1.320000	UGL	0.876	c1
	23095	1.50MB	90233	N3010	48.4	ED	GZR016	90234	90234	SS8	LT	1.040000	UGL	0.934	C1
i LL.	22005	1.50MB	91039	N5189,90	48.0	UB	NEP013	91043	91043	AV8	LT	1.320000	UGL	0.876	C1
WELL	23095	236TCP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	1.700000	UGL	0.984	1A
WELL	23095	245TCP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	2.800000	UGL	1.120	14
	23095	246TCP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	3.600000	UGL	1.080	14
W LL	25095	24DCLP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	8.400000	UGL	1.030	14

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Data Definition: RKPMCGW

Site	Site_ID	Test	Sample	Field	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Corrected	UOM	Acc	Flag	Anal
Ţvpe		Name	Date	Number	Depth		Number	Date	Date	Num		Value			Code	Type
																.,
heel	23095	24DMPN	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	4.400000	UGL	1.220		18
WELL	23095	24DNP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	176.000000	UGL	0.572		14
1 .L	23095	2CLEVE	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	350.000000	UGL	1.290		14
1 _L	23095	2CLP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	2.800000	UGL	0.955		14
WELL	23095	2MP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	3.600000	UGL	0.932		14
Del L	23095	2NP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	8.200000	UGL	1.080		14
1_L	23095	4CL3C	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	8.500000	UGL	1.100		14
WELL	23095	4MP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	2.800000	UGL	0.878		14
WELL	23095	4NP	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	96.000000	UGL	0.664		14
L	23095	ACET	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	800.00000	UGL	0.405		14
' _L	23095	ACRYLO	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	840.000000	UGL	0.844		18
WELL	23095	ALDRN	89307	M4488	48.0	UB	HQQ008	89314	89316	KK8		0.328000	UGL	0.861		C1
1 TL	23095	ALDRN	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	13,000000	UGL	0.539		14
1 .L	23095	ALDRN	90057	M9009	48.5	UB	1XQ006	90064	90065	KK8		0.880000	UGL	0.861		C1
WELL	23095	ALDRN	90141	N1751	48.0	UB	JR8005	90145	90151	KK8		0.293000	UGL	0.861	C	C1
WELL	23095	ALDRN	90233	N3016	48.4	ED	QL0016	90236	90239	MM8A	LT	0.083000	LIGI	0.851	U	C1
1 _L	23095	ALDRN	91039	N5195	48.0	UB	NET013	91042	91047	KK8	LT	0.050000	LIGI	0.861		C1
1. L	23095	AS	89307	M4498	48.0	UB	HPT020	89333	89335	AX8	LT	2.350000	UGI	0.001	F	C1
WELL	23095	AS	90057	M9016	48.5	UB	JB0029	90072	90085	AX8	LT	2,350000	UGI	0.74	-	C1
1 .L	23095	AS	90141	N1758	48.0	UB	JTC005	90162	90164	AX8		12,800000	UGI	0.774 0.974		C1
(.L.	23095	AS	90233	N3023	48.4	ED	QI F016	90247	90248	VV8		23.200000	IIGI	0.001	г	C1
WELL	23095	AS	91039	N5202	48.0	UB	NEN023	91045	91046	AX8		15.400000	LICI	0.774	E	C1
VELL	23095	ATZ	89307	M4497	48.0	UB	HQR008	89314	89318	UH11		120.000000	UGL	0.774	F	C1
_L	23095	ATZ	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	5.900000		1 200		14
hL	23095	ATZ	90057	M9015	48.5	UB	IXR006	90064	90066	UH11	LT	4 030000		0 084		
WELL	23095	ATZ	91039	N5201	48.0	UB	NEU013	91042	91051	UH11		74 000000	HCL	0.700		C1
ι.L	23095	BCHPD	89307	M4490	48.0	UB	HQT008	89314	89321	P8		25 000000	HIGH	0.700		1
Į "L	23095	BCHPD	90141	N1753	48.0	UB	JRE005	90145	90156	P8		22 200000	HCL	0.500		C1
WELL	23095	BCHPD	91039	N5197	48.0	UB	NEW013	91042	91046	PR		10 000000		0.500		
¥=1.L	23095	BRDCLM	89307	M4492,93	48.0	UB	HQW002	89319	89319	11121	IT	100.000000		1 100		C1
۱.۱	23095	BTZ	89307	M4487	48.0	UB	HQP008	89313	89314	AAAR	17	5 000000		0.050		AI
WELL	23095	BTZ	90057	M9008	48.5	UB	IXS006	90064	90067	AAAB	11	5.000000	UCL	0.750		C1
WELL	23095	BTZ	90141	N1750	48.0	UB	JRA005	90145	90150	AAAR	11	5 000000		0.730		C1
4 .L	23095	BTZ	90233	N3015	48.4	ED	QB0016	90236	90239	PPRA	17	1 1/0000		0.750		C1
∖ .L	23095	BTZ	91039	N5194	48.0	UB	NES013	91042	91047	AAAR	1.1	5 000000		0.924		
WELL	23095	BTZ	91105	03611	48.0	UB	010006	91109	91118	AAAS	17	5.000000		0.930		U
V-".L	23095	C2H3CL	89307	M4492,93	48.0	UB	HQW002	89319	89319	1.1121	11	1200 000000	UGL	0.930		U
V L	23095	C2H5CL	89307	M4492,93	48.0	UB	HQW002	89319	89319	LIM21	11	800.000000	UGL	0.033		10
WELL	23095	C6H6	89307	M4482,83	48.0	UB	HQM008	89315	89315	AV8	1.T	1 050000	UCL	0.972		10
VELL	23095	C6H6	89307	M4492,93	48.0	UB	HQWOOZ	89319	89319	UM21	LT	100 000000	UCI	1 200		44
¥.,L	23095	C6H6	90057	M9003,04	48.5	UB	1ZA009	90067	90067	AV8		16 900000	HCL	1.270		1A 01
₩.L	23095	C6H6	90141	N1745,46	48.0	UB	JRG005	90147	90147	AVR	17	1.050000		0.009		C1
WELL	23095	C6H6	90233	N3010	48.4	ED	67R016	90234	00236	892	CT	1.00000	UGL	0.009		U1
₿.L	23095	C6H6	91039	N5189,90	48.0	UB	NEP013	91043	010/3	AVR		10.00000	UGL	0.042		UI
₹.L	23095	CA	89307	M4500	48.0	UB	HSX008	90022	21073 200023	Ce12		310000 00000	UGL	0.859		C1
WELL	23095	CA	90057	M9018	48.5	UR	JBP020	9007/	00079	5012		30000.000000	UGL	0.990	• F	C1
VELL	23095	CA	90141	N1760	48.0	UB	JRUNIO	901AT	00168	0312		370000.000000	UGL	0.990	• F	C1
V.L	23095	CA	91039	N5204	48.0	UB	NEO023	91044	0106	5312		330000.000000	UGL	0.990	• F	C1
Wert	23095	CCL3F	89307	M4492.93	48.0	UB	HOUDOS	80710	80710	1112	1 T	100.00000	UGL	0.990	F	C1
								07319	07319	UNZ I	L I	100.000000	UGL	1.040	I	18

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Data Definition: RKPMCGW

Site	Site_ID	Test	Sample	Field	Sample	Lab	Lot	Prep	Anal	Neth	Bool	Corrected		100	Floo	An-1
⊺}/ ≈e		Name	Date	Number	Depth		Number	Date	Date	Num	0000	Value	COM	ALL	Code	Anal
					•							Tatuc			code	rype
4ELL	23095	CCL4	89307	M4484,85	48.0	UB	HQN008	89315	89315	N8	LT	0,990000	UGL	0.910		C1
WELL	23095	CCL4	89307	M4492,93	48.0	UB	HQW002	89319	89319	UH21	LT	100.000000	UGL	1.090		14
~ L	23095	CCL4	90057	M9005,06	48.5	UB	I ZB009	90067	90067	N8	LT	0,990000	UGL	0 010		-10 -11
k L	23095	CCL4	90141	N1747,48	48.0	UB	JRH005	90147	90147	N8	LT	0.990000	UGI	0.710		C1
WELL	23095	CCL4	90233	N3012	48.4	ED	GYR016	90242	90242	TT8	LT	1.690000	UGL	0.710		C1
wт: Г	23095	CCL4	91039	N5191,92	48.0	UB	NEQ013	91043	91043	N8	LT	0.990000	HGL	0.740		C1
W L	23095	CD	89307	M4500	48.0	UB	HSX008	90022	90023	SS12	11	6 780000	HCI	0.710	F	C1
WELL	23095	CD	90057	M9018	48.5	UB	JBP029	90074	90078	5512	11	6 780000		0.752	r F	C1
WELL	23095	CD	90141	N1760	48.0	UB	JRW019	90163	90168	SS12	IT	6 780000	UGL	0.752	r E	
W.L	23095	CD	90233	N3025	48.4	ED	QSK016	90247	90248	800	11	5 000000	UGL	0.752	r	C1
WL	23095	CD	91039	N5204	48.0	UB	NE0023	91046	91048	\$\$12	11	6 780000	HGL	0.773	-	C1
WELL	23095	CH2CL2	89307	M4484,85	48.0	UB	HQN008	89315	89315	NR	11	7 400000		0.732	r	C1
W L	23095	CH2CL2	89307	N4492.93	48.0	UB	HOUOO2	80310	80310	11421	17	100,000000		0.090		
W L	23095	CH2CL2	90057	M9005.06	48.5	UB	178009	90067	00067	NR	LI	28 500000	UGL	0.990		18
WELL	23095	CH2CL2	90141	N1747.48	48.0	LIR	128005	90167	001/7	NO		20.00000	UGL	0.090		C1
WFIL	23095	CH2CL2	90233	N3012	48.4	FD	CYP016	002/2	002/2	RO TTO	LI	7.400000	UGL	0.890		C1
ΨL.	23095	CH2CL2	91039	N5191 92	48.0	118	NEODIZ	9024 <u>2</u> 010/3	90242	110		16.300000	UGL	0.979		C1
WLLL .	23095	CH3BR	89307	M4492 93	48.0	110		90710	90740	NO		22.500000	UGL	0.890		C1
WELL	23095	CH3CL	89307	N6402 03	48.0	110	100002	90710	07319	Um21	L(1400.000000	UGL	0.882		18
W.L	23095	CHBR3	89307	M4402 03	40.0	110		90710	07317	UM21		120.000000	UGL	0.879		18
K L	23095	CHCL3	89307	M4472,75	40.0		NONUOS	07319	09319	UMZ1	LI	1100.000000	UGL	1.090		18
WELL	23095	CHCL3	89307	M6602 03	40.0		HOLIDOS	07313	07313	NO		12000.000000	UGL	0.880		C1
W ^{er} L	23095	CHCL3	90057	M9005 04	40.0		170000	07317	09319	UM21		10000.000000	UGL	1.000		18
K L	23095	CHCL3	90141	N1747 48	40.5	110	IDUOOF	90007	90067	Nð		10000.000000	UGL	0.880		C1
WELL	23095	CHCL3	00233	N3012	40.0	05	JKRUUJ	90147	90147	NS	GT	200.000000	UGL	0.880		C1
WELL	23095	CHCI 3	01030	N5101 02	40.4	50	GTKU10	90242	90242	118		11000.000000	UGL	0.928		C1
Ψ L	23095	CL	80307	N//01	40.0	05	NEWUIS	91045	91043	NB		9700.000000	UGL	0.880		C1
ν. Γ	23095	CI	90141	N1754	40.0	05	120005	09313	89313	1109		430000.000000	UGL	0.961		C1
WELL	23095	0	00733	NT010	40.0	08	JIGUUS	90164	90164	TT09		4700000.000000	UGL	0.961		C1
WT'L	23095	CI	910233	N5109	40.4	ED		90248	90249	NN8		760000.000000	UGL	0.993		C1
V L	23095		91037	M// 90	48.0	UB	NFBU15	91065	91065	TT09		560000.000000	UGL	0.961		C1
WELL	23095		80307	M4400	40.0	UB	HQQUUS	89514	89316	KK8		0.636000	UGL	0.802		C1
WELL	23095	CI &CD	07007	H0000	40.0	UB	HQXUUZ	89314	89319	UM25	LT	54.000000	UGL	1.100	l	18
L L	23095		90057	M7009	40.7	UB	120005	90064	90065	KK8		0.950000	UGL	0.802		C1
77	23095		00277	N1/21	48.0	UR	JRBUUS	90145	90151	KK8		0.350000	UGL	0.802	C	C1
WETT	23005		90203	N3016	48.4	ED	QL0016	90236	90239	mm8a	LT	0.170000	UGL	0.590	I	C1
	23095		073U7	M4484,85	48.0	UB	HQN008	89315	89315	N8	LT	8.200000	UGL	0.880	l	C1
	23005		09307	M4492,93	48.0	UB	HQWOOZ	89319	89319	UM21	LT	100.000000	UGL	1.040	ł	18
	23075		90057	M9005,06	48.5	UB	I ZB009	90067	90067	N8	LT	0.820000	UGL	0.880	:	C1
UCIL	23095		90141	N1747,48	48.0	UB	JRH005	90147	90147	N8	LT	0.820000	UGL	0.880)	C1
WELL U I	23095	CLCOHS	90233	N3012	48.4	ED	GYR016	90242	90242	TT8	LT	1.360000	UGL	1.010	1	C1
щ. ь. 17. г.	23095	CLUGHS	91039	N5191,92	48.0	UB	NEQ013	91043	91043	N8	LT	0.820000	UGL	0.880)	C1
Wielele 10711	23095	CLUAN	89307	M4488	48.0	UB	HQQ008	89314	89316	KK8		5.200000	UGL	0.828	3	C1
WELL	23095	CLDAN	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	37.000000	UGL	0.506		18
# L	23093	CLDAN	90057	M9009	48.5	UB	I XQ006	90064	90065	KK8	LT	0.095000	UGL	0.828	6	C1
₩ L.	23095	CLDAN	90141	N1751	48.0	UB	JRB005	90145	90151	кк8		21.000000	UGL	0.828	c c	C1
WELL	25095	CLDAN	90233	N3016	48.4	ED	QL0016	90236	90239	MM8A	LT	0.300000	UGL	0.977	,	C1
WFL	23095	CLDAN	91039	N5195	48.0	UB	NET013	91042	91047	KK8	LT	0.095000	UGL	0.828		C1
W L	23095	CPMS	89307	M4487	48.0	UB	HQP008	89313	89314	8888		48.500000	UGL	0.942		C1
WELL	23095	CPMS	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		109.000000	UGL	0.795		14

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Site	Site_ID	Test	Sample	Field	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Corrected	UOM	Acc	Flag Anat
Ivne		Name	Date	Number	Depth		Number	Date	Date	Num		Value			Code Type
i Weel	23095	CPMS	90057	M9008	48.5	UB	1XS006	90064	90067	8848		40.40000	HG	0 0/2	c1
WELL	23095	CPMS	90141	N1750	48.0	UB	JRA005	90145	90150	AAA8		32 000000	UCI	0.742	U1
¥ .L	23095	CPMS	90233	N3015	48.4	ED	QB0016	90236	90239	PP8A		210 000000		0.942	C1
€ .L	23095	CPNS	91039	N5194	48.0	UB	NES013	91042	91047	AAAB		38,400000	HCL	0.900	U1
WELL	23095	CPMS	91105	03611	48.0	UB	01D006	91109	91118	8444		38 100000	HCL	0.742	
Ver.L	23095	CPMSO	89307	M4487	48.0	UB	HQP008	89313	89314	88484		150,000000	HCL	0.742	U1
V.L	23095	CPNSO	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	15 000000	UCL	0.703	U1
Well	23095	CPMSO	90057	M9008	48.5	UB	IXS006	90064	90067	AAA8		150 000000	UGL	0.004	IA Of
WELL	23095	CPMSO	90141	N1750	48.0	UB	JRA005	90145	90150	AAAB		180,000000	UGL	0.903	U1
¥ .L	23095	CPMSO	90233	N3015	48.4	ED	QB0016	90236	90239	PPRA	1 T	1 020000	UGL	0.903	C1
ι.L	23095	CPHSO	91039	N5194	48.0	UB	NES013	91042	91047	AAAR		22 700000		0.932	C1
WELL	23095	CPNSO	91105	03611	48.0	UB	010006	91109	91118	AAAR	ŧΤ	120,000000	UGL	0.963	C1
V.L	23095	CPMSO2	89307	M4487	48.0	UB	HQP008	89313	89314	AAAR		360.000000	UGL	0.963	C1
J.L	23095	CPMSO2	89307	M4494	48.0	UB	HQX002	89314	80310	11125		352 000000	UGL	1.160	C1
WELL	23095	CPMSO2	90057	M9008	48.5	UB	IXS006	90064	90067	AAAR		370,000000	UGL	0.795	14
WELL	23095	CPMSO2	90141	N1750	48.0	UB	JRA005	90145	90150	AAAR		270.00000	UGL	1.160	C1
4 .L	23095	CPMSO2	90233	N3015	48.4	ED	QR0016	90236	00230	DDSA		520.000000	UGL	1.160	C1
4 .L	23095	CPMSO2	91039	N5194	48.0	UB	NES013	91042	910239	AAAR		320.000000	UGL	0.823	C1
WELL	23095	CPMSO2	91105	03611	48.0	UB	010006	01100	01118	AAAQ		220.000000	UGL	1.160	C1
₩.L	23095	CR	89307	M4500	48.0	LIR	RUUASH	00022	00022	0000		250.000000	UGL	1.160	C1
۱.L	23095	CR	90057	M9018	48.5	UR	.IRP020	00022	00023	0012	17	16.800000	UGL	1.010	F C1
WELL	23095	CR	90141	N1760	48.0	LIR		00147	90076	3312		16.800000	UGL	1.010	F C1
VELL	23095	CR	90233	N3025	48.4	FD	054016	00267	003/9	3312		16.800000	UGL	1.010	F C1
J.L	23095	CR	91039	N5204	48.0	LIR	WEO023	90247	90240	KYU 6010	LI	22.000000	UGL	0.960	C1
WLLL	23095	CU	89307	M4500	40.0	112	HEVOOR	71040	91046	3512		26.400000	UGL	1.010	F C1
WELL	23095	cu	90057	M9018	48.5	112	100020	90022	90023	5512		50.400000	UGL	0.958	F C1
√.L	23095	cu	90141	N1760	48.0	110	10010	90074	90078	2212		47.700000	UGL	0.958	F C1
ι.L	23095	CU	90233	N3025	40.0	ED	00014	90105	90100	5512		41.800000	UGL	0.958	F C1
WELL	23095	cu	91039	N5204	48.0	110	43K010	90247	90248	KYU CO17		31.800000	UGL	0.978	C1
V.L	23095	CYN	90233	N3021	40.0	50	NEUUZJ	71040	91040 000/7	5512		86.200000	UGL	0.958	F C1
1.1	23095	DBCP	89307	MAARA	40.4	ED	WARD 14	90242	90243	CNI	LT	8.900000	UGL	0.954	C1
WELL	23095	DBCP	89307	MAAQA	48.0		HOVOOD	07313	09314	AYS	LT	0.195000	UGL	0.991	C1
WELL	23095	DBCP	90057	NO007	40.0	110		09314	89319	UM25	LT	12.000000	UGL	0.841	1A
H.L	23095	DBCP	90141	N1749	48.0		107005	90001	90061	ATO	LT	0.195000	UGL	0.991	C1
ιL	23095	DBCP	90233	N3014	40.0	50	J42003	90144	90144	ATO	LT	0.195000	UGL	0.991	C1
WELL	23095	DBRCIN	80307	M//02 07	40.4	20	VIIIO	90235	90235	89	LT	0.130000	UGL	0.904	C1
W L	23095	DCIR	80307	M//02 07	40.0	08	HWWUUZ	89319	89319	UM21	LT	100.000000	UGL	1.190	1A
	23095	DCDD	80307	M4472,73	40.0	UB	HQWUUZ	89319	89319	UM21	LT	200.000000	UGL	1.070	1A
WELL	23095	DCPD	80307	M4470	40.0	UB	ROLLOR	89314	89321	P8		1200.000000	UGL	0.550	C1
VELL	23095	DCDD	00161	114474	40.0	UB	HQXUUZ	89314	89319	UM25	LT	5.500000	UGL	0.794	1A
 L 1	23095	DCPD	00277	N 1723	48.0	UB	JRE005	90145	90156	P8		1100.000000	UGL	0.550	C1
u -	23095	DCPD DCPD	90233 01030	NJU10	48.4	ED	GXP016	90240	90242	R8		650.000000	UGL	0.930	C1
VELL	23095	DOND	91039	NJ 197	48.0	UB	NEW013	91042	91046	P8		1100.000000	UGL	0.550	C1
	23005		07307 90707	M449/	48.0	UB	HQR008	89314	89318	UH11	LT	0.384000	UGL	0.891	C1
	23095	DOVP	07307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	8.500000	UGL	1.150	1A
יי, יי עובוו	23095	0046	90037	19015	48.5	UB	IXR006	90064	90066	UH11	LT	0.384000	UGL	0.891	C1
UCII	23005		70141 01070	NE204	48.0	UB	JRC005	90145	90161	UH11	LT	0.384000	UGL	0.891	C1
J	23093	DUVP	90707	ND2U1	48.0	UB	NEU013	91042	91051	UH11	LT	0.384000	UGL	0.891	C1
	22093		07207	M4489	48.0	UB	HQS008	89314	89318	AT8		580.000000	UGL	0.908	C1
466.	23473	UIMP	04201	M4474	48.0	UB	HQX002	89314	89319	UM25	GT	200.00000	UGL	1.060	1A

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Site	Site_ID	Test	Sample	Field	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Connected		_		
⁻ ype		Name	Date	Number	Depth		Number	Date	Date	Num	5001	Value	UUM	Acc	Flag	Anal
												vatue			Code	Туре
J.	23095	DIMP	90057	M9010	48.5	UB	IXT006	90064	90068	AT8		750 000000		• • • •		
ÆLL	23095	DIMP	90141	N1752	48.0	UB	JRD005	90145	90150	AT8		310,000000	UGL	0.908		C1
.L	23095	DIMP	90233	N3017	48.4	ED	QA0016	90239	90262	008		1100.000000	UGL	0.908		C1
У., L	23095	DITH	89307	M4487	48.0	UB	HQP008	89313	89314	AAAR		76.000000	UGL	0.994		C1
JELL	23095	DITH	89307	M4494	48.0	UB	HQX002	89314	89319	11425		50.000000	UGL	0.853		C1
JEI L	23095	DITH	90057	M9008	48.5	UB	IXS006	90064	90067	AAAR		50.000000	UGL	1.100		1A
- L	23095	DITH	90141	N1750	48.0	UB	JRA005	90145	90150	AAAR		57.000000	UGL	0.853		C1
wert	23095	DITH	90233	N3015	48.4	ED	9B0016	90236	001230	DDRA		43.000000	UGL	0.853		C1
WELL	23095	DITH	91039	N5194	48.0	UB	NESO13	91042	91047	AAAR		52.000000	UGL	1.100		C1
r L	23095	DITH	91105	03611	48.0	UB	010006	91109	01118	4448		49.000000	UGL	0.853		C1
κ L	23095	DLDRN	89307	M4488	48.0	UB	HQQ008	89314	80316	~~~		42.000000	UGL	0.853		C1
WELL	23095	DLDRN	89307	M4494	48.0	UB	HQX002	89314	80310		17	0.050000	UGL	1.120		C1
Werl	23095	DLDRN	90057	M9009	48.5	UB	TX0006	00064	00045	VVO	LI	26.000000	UGL	1.250		1A
W L	23095	DLDRN	90233	N3016	48.4	ED	01.0016	00234	90000	KKO		5.600000	UGL	1.120		C1
WELL	23095	DMDS	89307	M4487	48.0	UR	HOPOOR	80717	90239	MMOA		1.600000	UGL	1.020	C	C1
WELL	23095	DMDS	90057	M9008	48.5	UB	1125006	07313	07314	AAAO		0.550000	UGL	0.955		C1
W L	23095	DMDS	90141	N1750	48.0	HR	104005	70004 001/F	90067	8AA8	LT	0.550000	UGL	0.955		C1
WL	23095	DMDS	91105	03611	48.0	HR	010004	90142	90150	BAAA	LT	0.550000	UGL	0.955		C1
WELL	23095	DMMP	89307	M4489	48.0	LIB	NUC008	91109	91118	8AAA	LT	0.550000	UGL	0.955		C1
s,≣° L	23095	DMMP	89307	M4494	48.0	UR UR	HOVODO	07314 9071/	89318	AT8	LT	0.188000	UGL	0.925		C1
H L	23095	DMMP	90057	M9010	48.5	110	IVTOOL	09314	89319	UM25	LT	130.000000	UGL	0.179		1A
WELL	23095	DMMP	90141	N1752	48.0	112	100005	90004	90068	AT8	LT	0.188000	UGL	0.925		C1
WELL	23095	DMMP	90233	N3017	48.4	ED	040016	90145	90150	AT8	LT	0.188000	UGL	0.925		C1
W.L	23095	ENDRN	89307	M4488	48.0		440010	90239	90262	QQ 8	LT	16.300000	UGL	1.050		C1
WL	23095	ENDRN	89307	M4494	48.0	08		89314	89316	KK8		0.689000	UGL	1.220		C1
WELL	23095	ENDRN	90057	MOUUD	40.0 /8 C	00	HQXUUZ	89314	89319	UM25	LT	18.000000	UGL	0.979		1A
W L	23095	ENDRN	90233	N3016	40.5	08	120005	90064	90065	KK8		17.000000	UGL	1.220		C1
W.L	23095	ENDRN	91030	N5105	40.4	EU	QL0016	90236	90239	MM8A	LT	0.120000	UGL	1.060		C1
WELL	23095	ETC685	80307	W//80 97	40.0	UR	NET013	91042	91047	KK8	LT	0.050000	UGL	1.220		C1
WELL	23095	ETC685	80307	M//02 07	40.0	UR	HQM008	89315	89315	AV8	LT	1.370000	UGL	0.946		C1
W.L	23095	FTCANS	00057	HOOOT 0/	48.0	08	HQWOOZ	89319	89319	UM21	LT	100.000000	UGL	1.000		14
WL	23095	FTCANS	90037	M7003,04	48.5	UB	1ZA009	90067	90067	AV8	LT	1.370000	UGL	0.946		C1
WELL	23095	FTCANS	00277	N1743,40	48.0	UB	JRG005	90147	90147	AV8	LT	1.370000	UGL	0.946		C1
W. L	23095	FTC645	70233 01070	N5010	48.4	ED	GZR016	90234	90234	SS8		2.030000	UGL	0.950		C1
W L	23095	F	91039	N2 109,90	48.0	UB	NEP013	91043	91043	AV8	LT	1.370000	UGL	0.946		C1
WELL	23095	Е	001/1	114491 NA 75 (48.0	UB	HQV008	89313	89313	TT09		22000.000000	UGL	0.971		C1
MELT	23095	r E	90141	N1/04	48.0	UB	JTG005	90164	90164	TT09		24000.000000	UGL	0.971		C1
W.L	23095	r E	90233	N5019	48.4	ED	QQD016	90248	90249	NN8		9000.000000	UGL	1.110		C1
	23095	r No	91059	N5198	48.0	UB	NFB013	91065	91065	TT09		10000.000000	UGL	0.971		C1
VELL	23005	лц 80	09307	M4499	48.0	UB	HRU008	89334	89334	CC8		0.175000	UGL	1.000	F	C1
u. :	23005	60 110	90057	M9017	48.5	UB	1ZP030	90066	90067	CC8		0.251000	UGI	1 000	E	C1
u .	23005	nu vo	90141	N1759	48.0	UB	JTD005	90164	90164	CC8		0,758000	ligi	1 000	- E	C1
- 15 11	22005	16	90233	N3024	48.4	ED	GWS016	90243	90243	WW8	LT	0.500000	UCL	1.000	r	C1
HCLL (JEL)	22072	HG	91039	N5203	48.0	UB	NEM023	91043	91043	CC8		0.102000		1 000	-	
W?	23095	ISODR	89307	M4488	48.0	UB	HQQ008	89314	89316	KK8		8 570000		0.000	F	C1
WL _	23095	I SODR	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	ŧτ	7 200000		0.819		C1
WELL	23095	ISODR	90057	M9009	48.5	UB	1X9006	90064	90065	KKR		7 40000	UGL	0.582		1A
WELL	23095	ISODR	90141	N1751	48.0	UB	JRB005	90145	90151	KK8	LT	7.000000	UGL	0.819		C1
WIL _	23095	ISODR	90233	N3016	48.4	ED	QL0016	90236	90239	MM8A	 IТ	0.051000	UGL	0.819		C1
WL	23095	ĸ	89307	M4500	48.0	UB	HSX008	90022	90023	SS12		47500 00000	UGL	0.772		C1
													UGL	0.945	F	C1

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S.Le Site_ID Test Sample Field Sample Lah Lot Prep TYDE Anal Meth Bool Name Corrected UOM Date Number Acc Flag Anal Depth Number Date Date Num Value Code Type W L 23095 κ 90057 M0018 48.5 UB **JBP029** 90074 90078 ÆLL 23095 SS12 90141 52500.000000 K N1760 UGL 0.945 48.0 UB F C1 JRW019 90163 ₩. 90168 23095 SS12 ĸ 57000.000000 90233 60600*12 UGL 0.945 48.4 F C1 ED **QJH016** 90290 90291 L 23095 XX8 46300.000000 K 91039 N5204 UGL 0.828 48.0 UB C1 NE0023 91046 91048 WELL 23095 SS12 MEC6H5 49500.000000 89307 M4482,83 UGL 48.0 0.945 F LIR C1 HQM008 89315 YELL 23095 89315 **AV8** LT MEC6H5 89307 1.470000 M4492,93 UGL 0.906 48.0 UB C1 HQV002 89319 4 89319 23095 MEC6H5 UM21 LT 90057 100.000000 M9003,04 UGL 1.000 48.5 UB 1A 1ZA009 90067 90067 4 . 23095 AV8 MEC6H5 9.580000 90141 N1745,46 UGL 0.906 48.0 UB **JRG005** Cĺ 90147 WELL 90147 23095 AV8 LT MEC6H5 90233 N3010 1.470000 UGE 0.906 48.4 ED C1 **GZR016** 90234 90234 WC11 23095 **SS8** LT MEC6H5 91039 N5189,90 2.100000 UGL 0.932 48.0 UR CI NEP013 91043 WE 23095 91043 AV8 . LT MEK 89307 1.470000 M4492,93 UGL 0.906 48.0 UB C1 HQW002 89319 WELL 89319 23095 UM21 LT MG 89307 1000.000000 M4500 UGL 48.0 UB 0.435 14 HSX008 90022 90023 WELL 23095 SS12 MG 90057 211000.000000 M9018 UGL 0.942 48.5 F UB C1 **JBP029** 90074 WE. 90078 23095 SS12 MG 223000.000000 90141 N1760 UGL 48.0 0.942 F C1 UB **JRW019** 90163 90168 ١Æ 23095 SS12 236000.000000 MG 91039 N5204 UGI 0.942 48.0 F UB C1 NE0023 91046 WELL 23095 91048 SS12 262000.000000 MIBK 89307 M4490 UGL 0.942 48.0 F UB C1 HQT008 89314 89321 VE 23095 P8 LT MIBK 89307 4.900000 UGL M4492,93 48.0 0.650 UB HQW002 C1 89319 WE 23095 89319 UM21 LT MIBK 90141 140.000000 N1753 UGL 48.0 0.779 UR 1A **JRE005** 90145 90156 WELL 23095 P8 LT MIBK 90233 N3018 4.900000 48.4 UGL 0.650 ED C1. **GXP016** 90240 ÆU 90242 23095 R8 ιT MIBK 91039 12.900000 N5197 UGL 48.0 0.890 UB C1 **NEW013** 91042 91046 Æ 23095 P8 LT MLTHN 89307 4.900000 M4407 ligi 48.0 0.650 UR C1 HQR008 89314 89318 Æ. 23095 UH11 MLTHN 89307 2.990000 M4494 UGL 0.891 48.0 UB HQX002 C1 89314 **FELL** 23095 89319 UM25 MI THN LT 90057 M9015 21.000000 UGI 0.893 48.5 UB **IXR006** 90064 14 Æ 90066 23095 UH11 MLTHN LT 91039 0.373000 N5201 UGL 48.0 0.891 UB NEU013 C1 91042 Æ 91051 23095 **UH11** NA 2.820000 89307 M4500 UGL 48.0 0.891 UB C1 HSX008 90022 90023 ÆLL 23095 SS12 NA 3200000.000000 90057 M9018 48.5 UGL 0.942 UB F C1 **JBP029** /EU 23095 90074 90078 SS12 NA 3900000.000000 90141 N1760 UGL 48.0 UB 0.942 F C1 **JRW019** Έ: 90163 90168 23095 SS12 NA 3500000.000000 91039 N5204 UGL 48.0 0.942 UB F C1 NE0023 91046 91048 ÆL. 23095 SS12 NIT 3500000.000000 89307 M4495 UGI 0.942 48.0 UB F C1 HRS008 89333 89334 ÆLL 23095 NIT LL8 90057 M9013 5200.000000 UGL 48.5 0.989 UR C1 1ZK030 90072 90073 Æ 23095 LL8 NIT 90141 2300.000000 N1755 UGL 48.0 0.989 UR C1 **JRR038** NEL 90156 90155 23095 LL8 NIT 7100.000000 90233 N3020 UGL 48.4 0.989 FS SMN016 C1 90257 WELL 23095 90257 **TF22** NIT 91039 10000.000000 N5199 UGI 48.0 0.999 UB Cİ **NEL023** 91057 WEL? 23095 91058 LL8 OXAT 89307 30000.000000 M4487 UGI 0.989 48.0 UB HQP008 C1 89313 WEL 89314 23095 **AAA**8 OXAT 89307 17.600000 M4494 UGL 48.0 0.932 UB C1 HQX002 ELL 89314 89319 23095 UM25 OXAT LT 90057 M9008 27.000000 LIGI 48.5 0.916 1 IR IXS006 1A 90064 **FELL** 90067 23095 OXAT AAA8 90141 15.000000 N1750 UGL 48.0 UB 0.932 C1 **JRA005** 90150 /ELÍ 90145 23095 AAA8 OXAT 90233 N3015 16.300000 UGL 0.932 48.4 FD C1 **QBOO16** 90236 EL 23095 90239 PP8A OXAT 91105 20.800000 03611 48.0 UGL 0.965 UB C1 01D006 ELL 91109 91118 23095 8448 PB 89307 M4500 4.540000 UGI 48.0 0.932 UB C1 HSX008 90022 ΞĽ. 23095 90023 SS12 PB 90057 M9018 LT 43.400000 UGL 48.5 0.980 F UB C1 **JBP029** 90074 EL 23095 90078 SS12 PB LT 90141 N1760 43.400000 UGL 48.0 0.980 F UB C1 JRW019 90163 ELĹ 90168 23095 SS12 PB 90233 LT N3025 43.400000 UGL 48.4 0.980 ED F **QSK016** C1 90247 ELL 90248 23095 R90 PB 91039 LT 52.000000 N5204 48.0 UGL 0.936 UB C1 NE0023 91046 91048 ΞĽĹ 23095 SS12 PCP LT 89307 M4494 43.400000 UGL 0.980 48.0 F UR C1 HQX002 89314 ELL 23095 89319 UM25 PHENOL LT 89307 9.100000 M4494 UGL 48.0 1.040 UB 1A HQX002 89314 89319 -1 E 23095 UM25 PPDDE LT 89307 M4488 2.200000 UGI 48.0 0.401 UB HQQ008 89314 18 89316 ШÉ 23095 KK8 PPDDE 89307 16.000000 M4494 UGI 0.917 48.0 UB HQX002 C1 89314 ΞLĽ 89319 23095 UM25 PPODE 90057 LT M9009 14.000000 48.5 UGL 1.040 UR 1XQ006 18 90064 90065 KK8 6.100000 UGL 0.917

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Data Definition: RKPMCGW

Sita	Site ID	Test														
Time	SILE_ID	lest	Sample	Field	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Corrected	UOM	Acc	Flag	Anal
i Æ		Nane	Vate	Number	Depth		Number	Date	Date	Num		Value			Code	Туре
Q=11	23095	PPNDE	001/1	114764				<u>.</u>								-
VELL	23095	PPODE	00277	N1/21	48.0	UB	JRB005	90145	90151	KK8		1.700000	UGL	0.917	С	C1
I I	23095	DDDDT	90233	OIUCN	48.4	ED	QL0016	90236	90239	MM8A	LT	0.092000	UGL	0.977		C1
1.1	23095	PPODT	80307	M4400	48.0	UB	HQQUUB	89314	89316	KK8		0.263000	UGL	0.906		C1
MELL	23095	PPONT	07507	M4474	48.0	UB	HQX002	89314	89319	UM25	LT	18.00000	UGL	0.951		1A
V71	23095	PPDDT	90037	M9009	48.5	UB	IXQ006	90064	90065	KK8		6.700000	UGL	0.906		C1
5 F	23005	DDDDT	00277	N1731	48.0	UB	JRB005	90145	90151	KK8		3.000000	UGL	0.906	C	C1
VELL	23095	DDTWN	90207	N3010	48.4	ED	QL0016	90236	90239	MM8A	LT	0.120000	UGL	0.915		C1
WELL	23095	DDTUN	80707	M4497	48.0	UB	HQR008	89314	89318	UH11	LT	0.647000	UGL	0.790		C1
k t	23095	DDTUN	07307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	37.000000	UGL	0.684		1A
ч.с. к. г.	23095	DOTUM	90037	MYU15	48.5	UB	1XR006	90064	90066	UH11	LT	0.647000	UGL	0.790		C1
UFII	23005	DDTUN	90233	NSUZZ	48.4	ES	PG0016	90240	90242	UN07	LT	0.250000	UGL	0.983		C1
	23095	PRI NN	91039	N5201	48.0	UB	NEU013	91042	91051	UH11		26.100000	UGL	0.790		C1
ч.с. с. т	23005	504	09307	M4491	48.0	UB	HQV008	89313	89313	TT09		1800000.000000	UGL	0.856		C1
UEIT	23095	504	90141	N1/54	48.0	UB	JTG005	90164	90164	TT09		1700000.000000	UGL	0.856		C1
UCII	23095	504	90255	N3019	48.4	ED	QQD016	90248	90249	NN8		1500000.000000	UGL	1.000		C1
e e	23095	504	91039	N5198	48.0	UB	NFB013	91065	91065	TT09		200000.000000	UGL	0.856		C1
ų .u	23095	SUPONA	07507 90707	M4497	48.0	UB	HQR008	89314	89318	UH11		0.930000	UGL	0.889		C1
UFII	23095	SUPUNA	89307	M4494	48.0	UB	HQX002	89314	89319	UM25	LT	19.000000	UGL	0.630		1A
WELL	23095	SUPUNA	90057	M9015	48.5	UB	I XR006	90064	90066	UH11	LT	0.787000	UGL	0.889		C1
	23095	SUPUNA	90141	N1757	48.0	UB	JRC005	90145	90161	UH11	LT	0.787000	UGL	0.889		C1
ueri	23005	JOPUNA	91059	N52U1	48.0	UB	NEU013	91042	91051	UH11	LT	0.787000	UGL	0.889		C1
UPLI	23095	TOLEA	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	150.000000	UGL	0.992		18
т. К. 1	23093	TOLEE	89307	M4484,85	48.0	UB	HQN008	89315	89315	N8		31.000000	UGL	0.910		C1
tion a t	23095	TOLEE	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	100.000000	UGL	1.080		14
UELL	23095	TOLEE	90057	M9005,06	48.5	UB	I Z8009	90067	90067	N8		18.500000	UGL	0.910		C1
	23095	TOLEE	90141	N1/47,48	48.0	UB	JRH005	90147	90147	N8		2.430000	UGL	0.910		C1
	23095	TOLEE	90233	N3012	48.4	ED	GYR016	90242	90242	TT8		24.600000	UGL	1.070		C1
UCII	23095	TOOLEE	91039	N5191,92	48.0	UB	NEQ013	91043	91043	N8		22.000000	UGL	0.910		C1
WEEL	23095	TROLE	89307	M4484,85	48.0	UB	HQN008	89315	89315	N8		21.000000	UGL	0.910		C1
• .L	23073	TROLE	89307	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	100.000000	UGL	1.000		14
UEFT.	23093	TRULE	90057	M9005,06	48.5	UB	I ZB009	90067	90067	N8	LT	0.560000	UGL	0.910		C1
	23093	TACLE	90141	N1747,48	48.0	UB	JRH005	90147	90147	N8		0.776000	UGL	0.910		C1
	23005	TRULE	90233	N3012	48.4	ED	GYR016	90242	90242	T T 8		17.200000	UGL	0.984		C1
1.5	23095	INUCITA	91039	N5191,92	48.0	UB	NEQ013	91043	91043	N8	LT	0.560000	UGL	0.910		C1
UFIL	23005	1111/677	80707	M4492,93	48.0	UB	HQW002	89319	89319	UM21		4900.000000	UGL	0.000	s	14
V I	23005	UNK555	07307 90707	M4494	48.0	UB	HQX002	89314	89319	UM25		6.000000	UGL	0.000	S	1A
	23095	UNK347	69307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	S	14
T - L	23095	UNKSST	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		200.000000	UGL	0.000	S	18
UCII	23095	UNKODO	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		5.000000	UGL	0.000	S	14
1977 C.	23095	UNK557	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0.000	s	14
i ata i a	23095	UNK559	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	Ð	14
NGLL .	23095	UNKSSY	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0.000	s	14
WCLL Ú ·	22005	UNK561	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		5.000000	UGL	0.000	ŝ	14
	23095	UNK562	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0,000	s	14
η	22005	UNK564	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0,000	ت د	14
WELL	22005	UNK565	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		4.000000	UGL	0,000	s	14
	23005	UNK569	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0,000	S	14
₩	22075	UNK570	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0,000	S	14
WELL	23093	UNK571	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0.000	s	14
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Data Definition: RKPMCGW

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ite	Site ID	Test	Commilia	et	· ·											
,	Site_ID	Nomo	Sample	Fleid Number	Sample	Lab	Lot	Ргер	Anal	Meth	Bool	Corrected	UOM	Acc	Flag	Anal
•••		IN CALIFY	Date	Number	vepth		Number	Date	Date	Num		Value			Code	Туре
ELL	23095	114/572	80707	W/ / O/	(0.0											
ELL	23095	1101/572	80307	M4474	48.0	UB	HQX002	89314	89319	UM25		30.000000	UGL	0.000	D	1A
1 -	23095	UNK572	80307	M4474	40.0	UR	HQXUU2	89314	89319	UM25		7.000000	UGL	0.000	D	14
A _	23095	UNK574	80307	M4474	40.0	UB	HQXUUZ	89314	89319	UM25		20.000000	UGL	0.000	S	18
IELL	23095	1111577	80307	M4474	48.0	UB	HQXUO2	89314	89319	UM25		20.000000	UGL	0.000	S	1A
π.	23095		80307	M4474	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0.000	S	18
n _	23095	1111570	80307	M4474 W//O/	48.0	UB	HQX002	89314	89319	UM25		5.000000	UGL	0.000	Ð	1A
JELL.	23095	LINK581	80707	M4474	48.0	UB	HQX002	89314	89319	UM25		50.000000	UGL	0.000	S	1A
WELL	23095	LINK282	80307	M4474	48.0	UB	HQX002	89314	89319	UM25		60.000000	UGL	0.000	S	1A
¥.	23095	INK583	80307	M4474	48.0	UB	HQX002	89314	89319	UM25	·	400.000000	UGL	0.000	S	1A
WL _	23095	INK28/	80307	M4494	48.0	UB	HQX002	89314	89319	UM25		30.000000	UGL	0.000	S	1A
WELL	23095	INK285	80307	M4494	48.0	UB	HQX002	89314	89319	UM25		7.000000	UGL	0.000	S	1A
u .	23095	UNK285	80307	M4494	48.0	UB	HQX002	89314	89319	UM25		80.000000	UGL	0.000	S	18
4.	23095	LINKSRR	80307	M4474	40.0	UB	HQX002	89314	89319	UM25		30.000000	UGL	0.000	D	18
WELL	23095	LINK 500	80307	M4494	48.0	UB	HQXOOZ	89314	89319	UH25		80.000000	UGL	0.000	S	1A
WELL	23095		80307	M4474	48.0	UB	HQX002	89314	89319	UM25		5.000000	UGL	0.000	D	18
W L	23095	UNK501	80307	M4474	48.0	UB	HQX002	89314	89319	UM25		7.000000	UGL	0.000	S	1A
w	23095	INK501	80707	M4474	48.0	UB	HQX002	89314	89319	UH25		4.000000	UGL	0.000	D	1A
WELL	23095	LINKSOS	80707	M4494	48.0	UB	HQX002	89314	89319	UM25		8.000000	UGL	0.000	S	18
á.	23095	110/505	80707	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	S	1A
	23095	INK595	90707	M4494	48.0	UB	HQX002	89314	89319	UM25		30.000000	UGL	0.000	D	1A
WELL	23095	LINKADO	07307 80707	M4494	48.0	UB	HQX002	89314	89319	UH25		10.000000	UGL	0.000	s	1A
UPIN	23095	UNKOUZ	07307 90707	M4494	48.0	UB	HQX002	89314	89319	UM25		30.000000	UGL	0.000	S	1A
	23095		07307 90707	M4494	48.0	UB	HQX002	89314	89319	UH25		4.000000	UGL	0.000	S	18
14 - L	23095	UNKOU7	07307 80707	M4494	48.0	UB	HQX002	89314	89319	UM25		7.000000	UGL	0.000	S	18
JET 1	23095		073U/ 07307	M4494	48.0	UB	HQX002	89314	89319	UM25		6.000000	UGL	0.000	D	18
UE :	23095	LINKATO	07307 90707	M4494	48.0	UB	HQX002	89314	89319	UM25		6.000000	UGL	0.000	S	18
U C	23095	UNKOIZ	89307 80707	M4494	48.0	UB	HQX002	89314	89319	UM25		10.000000	UGL	0.000	S	14
JET 1	23005		09307	M4494	48.0	UB	HQX002	89314	89319	UM25		7.000000	UGL	0.000	S	14
	23095	UNKOIY	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	S	14
	23095	UNKOZZ	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		4.000000	UGL	0.000	S	14
un i	23095	UNKO24	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	S	14
MEL I	23005	UNKOZO	89307	M4494	48.0	UB	HQX002	89314	89319	UM25		9.000000	UGL	0.000	S	14
J.	23005	UNKOZO	09307	M4494	48.0	UB	HQX002	89314	89319	UM25		4.000000	UGL	0.000	s	14
	23005	UNKOZY	69307 80707	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	S	14
UFIL	23005	UNKOJZ	09307	M4494	48.0	UB	HQX002	89314	89319	UM25		4.000000	UGL	0.000	S	14
	23005		09307	M4494	48.0	UB	HQX002	89314	89319	UM25		20.000000	UGL	0.000	s	14
u ·	23005	VYLEN	07207 00707	M4482,83	48.0	UB	HQM008	89315	89315	AV8	LT	1.360000	UGL	0.889		C1
STELL.	23005	VYLCH	00057	M4492,93	48.0	UB	HQW002	89319	89319	UM21	LT	200.000000	UGL	1.010		14
	23095	ATLEN	90057	M9003,04	48.5	UB	1ZA009	90067	90067	AV8	LT	1.360000	UGL	0.889		C1
	23095	XTLEN	90141	N1745,46	48.0	UB	JRG005	90147	90147	AV8	LT	1.360000	UGL	0.889		C1
на не	23095	XTLEN	90233	N3010	48.4	ED	GZR016	90234	90234	SS8		2.150000	UGL	0.932		C1
	22005	XTLEN	91039	N5189,90	48.0	UB	NEP013	91043	91043	AV8		1.700000	UGL	0.889		C1
WELL UE	22005	ZN	89307	M4500	48.0	UB	HSX008	90022	90023	SS12	LT	18.000000	UGL	0,969	F	C1
ик, . 1 к	23095	ZN	90057	M9018	48.5	UB	JBP029	90074	90078	SS12		20.200000	UGL	0,960	F	C1
WК	22005	ZN	90141	N1760	48.0	UB	JRW019	90163	90168	SS12	LT	18.000000	UGL	0,960	F	C1
WELL	23095	ZN	90233	N3025	48.4	ED	QSK016	90247	90248	R9D	LT	20.000000	UGI	0.962	r	C1
WIC T.	23093	ZN	91039	N5204	48.0	UB	NE0023	91046	91048	SS12	LT	18.000000	UGL	0,960	F	C1

Number of chemical records printed: 390

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RMA Chemical Reject Report

Data Definition: RKPMREJ

Site Tyne	Site_ID	Test Name	Sample Date	Meth Num	Lab	Lot Number	Bool	Unc Mant	Unc Exp	UOM	Dil Mant	Dil Exp	Perc Moist	Acc	Corr Mant	Corr Exp	Corrected Value	Anal Type
Weel	23095	ATZ	90141	UH11	UB	JRC005		9.26	1	UGL				-986	0 70	1	07 00000	
WELL	23095	CL6CP	91039	кк8	UB	NET013		2.82	-1	UGL				802	3 52	- 1	93.90000	C1
W L	23095	CYN	89307	TF20	UB	HQU008		7.25	0	UGL				1 01	7 18	-1	0.55200	C1
W L	23095	CYN	90057	TF20	UB	1XY006	LT	5.00	0	UGL				1 01	5 00	0	7.18000	C1
WELL	23095	CYN	90141	TF20	UB	JRF005		6.65	0	LIGI				1 01	5.00	0	5.00000	C1
WT'L	23095	CYN	91039	TF34	UB	NEX013	LT	5.00	0	1161				1.01	5.00	U	6.58000	C1
W L	23095	DBCP	91039	AY8	UB	NER013	LT	1.95	- 1	HGI				001	5.00	0	5.00000	C1
WELL	23095	DIMP	91039	AT8	UB	NEV013		5.71	0	LICI	1 0	2		. 771	1.95	•1	0.19500	C1
WELL	23095	DLDRN	90141	KK8	UB	JRB005		2.26	- 1	LICI	1.0	4		.908	0.3	2	630.00000	C1
W L	23095	DLDRN	91039	KK8	UB	NET013		3 20	- 1	UCL.	1.0			1.12	2.0	0	2.00000	C1
W L	23095	DMDS	90233	PP8A	ED	QB0016	iΤ	1 16	- 1		1.0	• •		1.12	2.9	0	2.90000	C1
WELL	23095	DMDS	91039	8AAA	UB	NES013	17	5 50	- 1	UGL				1.02	1.16	0	1.16000	C1
₩ L	23095	DMMP	91039	AT8	LIR	NEV013	11	1 00	- 1	UGL				.955	5.50	•1	0.55000	C1
W L	23095	ENDRN	90141	KKR	1 IR	100005	-	0.74	• 1	UGL				.925	1.88	-1	0.18800	C1
WELL	23095	ISODR	91039	KK8	110	NET017		0.31	-1	UGL				1.22	6.81	•1	0.68100	C1
WFLL	23095	MLTHN	90141	11411	110	IPCOOS		4.49	•1	UGL				.819	5.48	-1	0.54800	C1
W L	23095	OXAT	91030	AAA9	100	JKC003		2.81	U	UGL				.891	3.15	0	3.15000	C1
WLLL	23095	PPDDE	91039	rve		NESUIS		1.22	1	UGL				.932	1.31	1	13.10000	C1
WELL	23095	PPDDT	91039	NNO VVR		NEIUIS		3.85	-1	UGL	1.0	1		.917	4.2	0	4.20000	C1
W L	23095	PRTHN	90141	NNO	100	NE1015		8.34	-1	UGL	1.0	1		.906	9.2	0	9.20000	C1
			20141	UNII	UB	JKCUUS		2.82	1	UGL				.790	3.57	1	35.70000	C1

Number of chemical records printed: 20

*** End of Report ***

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LEVEL 3

Site	Sample	Sample	Parameter	Value		Flag	Units	Dilution	Method
ID	Date	Depth							
23095	91105	48.0	111TCE	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	111TCE	LT	.760000		UGL		N8
23095	91105	48-0	112TCE	LT	100.000000		UGL	100.000000	UM21
23005	01105	48.0	112TCE	1 T	.780000		UGL		N8
23075	01105	48.0	11005	1.7	100.000000		LIGI	100,000000	UM21
23073	01105	40.0	11000	17	1 700000		UCI		N8
23095	91105	40.0		17	100.00000		HCI	100 00000	11121
23095	91105	48.0	I IDULE		770000			100.00000	NR
23095	91105	48.0	11DCLE	L1	.730000		UGL		11425
23095	91105	48.0	125108		5.800000		UGL		UNDE
23095	91105	48.0	1241CB	LT	2.400000		UGL	100 00000	0822
23095	91105	48.0	12DCE	LT	500.000000		UGL	100.00000	UMZI
23095	91105	48.0	12DCE	LT	.760000		UGL		NO
23095	91105	48.0	12DCLB	LT	1.200000		UGL		UNZO
23095	91105	48.0	12DCLE	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	12DCLE		7.840000		UGL		N8
23095	91105	48.0	12DCLP	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	12DPH	LT	13.00000		UGL		UM25
23095	91105	48.0	13DCLB	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	13DCLB	LT	3,400000		UGL		UM25
23095	91105	48.0	130CP	LT	480,000000		UGL	100.000000	UM21
23005	01105	48.0	130MB	LT	100,000000		UGL	100.000000	UM21
23005	01105	48.0	130MR	1.1	1.320000		UGL	n.	AV8
23073	01105	40.0	140CLR	11	1 500000		UGL		UM25
23093	01105	40.0	236100	1 T	1 700000		LIGI		UM25
23093	91105	40.0	230107		2 800000		1101		LIM25
23095	91105	40.0	243168		7 400000		UGL		LIN25
23095	91105	48.0	24010		3.00000		UGL		11425
23095	91105	48.0	24UCLP		8.40000		UGL		11425
23095	91105	48.0	240MPN		4.40000		UGL		UN25
23095	91105	48.0	J Z4DNP	LT	1/6.000000		UGL		UNZJ
23095	91105	48.0	24DNT	LT	5.800000		UGL		UNZS
23095	91105	48.0) 260NA	LT	8.800000		UGL		0825
23095	91105	48.0) 26DNT	LT	6.700000		UGL		UM25
23095	91105	48.0	2CLEVE	LT	350.00000		UGL	100.000000	UM21
23095	91105	48.0) 2CLP	LT	2.80000		UGL		UM25
23095	91105	48.0	D 2CNAP	LT	2.600000	ł.	UGL		UN25
23095	91105	48.0	D 2MNAP	LT	1.300000	l .	UGL		UH25
23095	91105	48.0	0 2MP	LT	3.60000)	UGL		UM25
23095	91105	48.0	D 2NP	LT	8.200000	1	UGL		UM25
23095	91105	48.0	0 330C80	LT	5.00000)	UGL		UM25
23095	91105	48.	0 350NA	LT	21.000000)	UGL		UM25
23005	01105	48	O SNANTI	IT	15,000000)	UGL		UN25
23093	01105	48.1	0 3NT	1 T	2,90000)	UGL		UM25
22005	01105	40.	A ARPDOF	1 T	22.000000)	UGL		UM25
23073	01105	40.		1 T	8 500000	, 1	UGL		UM25
23093	91103	40.		1 T	23 000000	, 1	UGL		UN25
23093	91103	40.	O ACLEFE	17	2 80000	, ,	LICI		UM25
23095	91105	40.	0 489		2.00000	, ,	UCL		11125
23095	91105	48.	U ANP	L1	90.00000				11125
23095	91105	48.	O ABHC	LT	5.30000		UGL	400.00000	UM21
23095	91105	48.	0 ACET	LT	800,00000	0	UGL	100.000000	UMZI
23095	91105	48.	O ACRYLO	LT	840.00000	0	UGL	100.00000	UM21
23095	91105	48.	0 AENSLF	LT	23.00000	0	UGL		UH25
23095	91105	48.	0 ALDRN	LT	13.00000	0	UGL		UM25
23095	91105	48.	0 ANAPNE	LT	5.80000	0	UGL		UM25
23095	91105	48.	O ANAPYL	LT	5.10000	0	UGL		UH25
23095	91105	48.	0 ANTRC	LT	5.20000	0	UGL		UN25
23095	91105	48.	.0 AS		16.10000	0 F	UGL		8XA
23005	91105	48.	O ATZ	LT	5.90000	0	UGL		UH25
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LEVEL 3

Site	Sample	Sample	Parameter	Value		Flag	Units	Dilution	Method
ID	Date	Depth							
23095	91105	48.0	ATZ		88.900000		UGL		UH11
23095	91105	48.0	B2CEXM	LT	6.800000		UGL		UH25
23095	91105	48.0	B2CIPE	LT	5.000000		UGL		UM25
23095	91 105	48.0	B2CLEE	LT	.680000		UGL		UH25
23095	91105	48.0	B2EHP	LT	7.700000		UGL		UM25
23095	91105	48.0	BAANTR	LT	9.800000		UGL		UM25
23095	91105	48.0	BAPYR	LT	14.000000		UGL		UH25
23095	91105	48.0	BBFANT	LT	10.000000		UGL		UN25
23095	91105	48.0	BBHC	LT	17.000000		UGL		UM25
23095	91105	48.0	BBZP	LT	28.000000		UGL		UH25
23095	91105	48.0	BCHPD		12.700000		UGL		P8
23095	91105	48.0	BENSLF	LT	42.000000		UGL		UM25
23095	91105	48.0	BGHIPY	LT	15.000000		UGL		UM25
23095	91105	48.0	BKFANT	LT	10.000000		UGL		UM25
23095	91105	48.0	BRDCLM	LT	100.000000		UGL	100.000000	UN21
23095	91105	48.0	BRMCIL	LT	2.900000		UGL		UM25
23095	91105	48.0	BZALC	LT	4.000000		UGL		UH25
23095	91105	48.0	C13DCP	ND	500.000000	R	UGL	100.000000	UN21
23095	91105	48.0	C2AVE	ND	1000.000000	R	UGL	100.000000	UN21
23095	91105	48.0	C2H3CL	LT	1200.000000		UGL	100.000000	UN21
23095	91105	48.0	C2H5CL	LT	800.00000		UGL	100.000000	UN21
23095	91105	48.0	C6H6	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	C6H6		11.800000		UGL		AV8
23095	91105	48.0	CA		300000.000000	F	UGL	10.000000	SS12
23095	91105	48.0	CCL3F	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	CCL4	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	CCL4	LT	.990000		UGL		N8
23095	91105	48.0	CD	LT	6.780000	F	UGL		SS12
23095	91105	48.0	CH2CL2	LT	100.000000		UGL	100.000000	UH21
23095	91105	48.0	CH2CL2	LT	7.400000		UGL		N8
23095	91105	48.0	CH3BR	LT	1400.000000		UGL	100.000000	UM21
23095	91105	48.0	CH3CL	LT	120.000000		UGL	100,000000	UM21
23095	91105	48.0	CHBR3	LT	1100.000000		UGL	100.000000	UM21
23095	91105	48.0	CHCL3		8200.000000		UGL	100.000000	UM21
23095	91105	48.0	CHCL3		9400.000000		UGL	100.000000	N8
23095	91105	48.0	CHRY	LT	7.400000		UGL		UN25
23095	91105	48.0	CL	GT	500000.000000		UGL	500,000000	TTOP
23095	91105	48.0	CL6BZ	LT	12.000000		UGL		UM25
23095	91105	48.0	CL6CP	LT	54.000000		UGL		11425
23095	91105	48.0	CL6ET	LT	8,300000		UGL		11125
23095	91105	48.0	CLC6H5	LT	100.000000		UGL	100.000000	UN21
23095	91105	48.0	CLC6H5	LT	.820000		UGI		NR.
23095	91105	48.0	CLDAN	LT	37,000000		UGI		HM25
23095	91105	48.0	CPMS		38,100000		LIGI		AAAR
23095	91105	48.0	CPMS		239,000000		UGI		11425
23095	91105	48.0	CPMSO	IТ	15,000000		LICI		LM25
23095	91105	48.0	CPMS02		230,000000		UGI	10 00000	AAAR
23095	91105	48.0	CPMS02		264.000000		UGI	10.00000	11425
23095	91105	48.0	CR	LT	16.800000	F	LIGI		5512
23095	91105	48.0	CS2	ND	500.000000	R	UGI	100 00000	11121
23095	91105	48.0	CU		93.000000	F	UGI		SS12
23095	91105	48.0	CYN	LT	5.000000	•	HGI		TEZ/
23095	91105	48.0	DBAHA	17	12 000000				18/25
23095	91105	40.0 48 fi	DRCP	11	12 000000				
27005	01105	-0.0 /g n			200000		UGL		
23005	01105	40.U /g /		ND	.200000	в			ATO
23073	91103 01105	40.0			3.000000	ĸ	UGL	400 000000	UM2D
23083	ALIOD	40.0	UBKULM	LÍ	100.000000		UGL	100.00000	UM21

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LEVEL 3

Site	Sample	Sample	Parameter	Value		Flag	Units	Dilution	Method
ID	Date	Depth							
23095	91105	48.0	DBZFUR	LT	5.100000		UGL		UH25
23095	91105	48.0	DCLB	LT	200.000000		UGL	100.000000	UM21
23095	91105	48.0	DCPD	GT	300.000000		UGL		UM25
23095	91105	48.0	DCPD		910.000000		UGL	10.000000	P8
23095	91105	48.0	DOVP	LT	8.500000		UGL		UH25
23095	91105	48.0	DOVP	LT	.384000		UGL		UH11
23095	91105	48.0	DEP	LT	5,900000		UGL		UH25
23095	91105	48.0	DIMP	GT	200.000000		UGL		UM25
23095	91105	48.0	DIMP		570.000000		UGL	100.000000	AT8
23095	91105	48.0	DITH		42.000000		UGL	10.000000	8844
23095	91105	48.0	DITH		42.700000		UGL		UN25
23095	91105	48.0	DLDRN	LT	26.000000		UGL		UH25
23095	91105	48.0	DMMP	LT	130.000000		UGL		UH25
23095	91105	48.0	DHIMP	LT	.188000		UGL		AT8
23095	91105	48.0	DMP	LT	2.200000		UGL		UM25
23095	91105	48.0	DNBP	LT	33.000000		UGL		UM25
23095	91105	48.0	DNOP	LT	1.500000		UGL		UM25
23095	91105	48.0	ENDRN	LT	18.000000		UGL		UM25
23095	91105	48.0	ENDRNA	LT	5.000000		UGL		UM25
23095	91105	48.0	ESFSO4	LT	50.000000		UGL		UM25
23095	91105	48.0	ETC6H5	LT	100.000000		UGL	100:000000	UM21
23095	91105	48.0	ETC6H5	LT	1.370000		UGL		AV8
23095	91105	48.0	F		18000.000000		UGL	10.000000	1109
23095	91105	48.0	FANT	LT	24.000000		UGL		UM25
23095	91105	48.0	FLRENE	LT	9.200000		UGL		UM25
23095	91105	48.0	HCBD	LT	8.700000		UGL		UN25
23095	91105	48.0	KG		.235000	F	UGL		CC8
23095	91105	48.0	NPCL	LT	38.000000		UGL		UM25
23095	91105	48.0	HPCLE	LT	28.000000		UGL		UM25
23095	91105	48.0	ICDPYR	LT	21.000000		UGL		UM25
23095	91105	48.0	ISODR	LT	7.800000		UGL		UM25
23095	91105	48.0	I SOPHR	LT	2.400000		UGL		UM25
23095	91105	48.0	ĸ		44200.000000	F	UGL		SS12
23095	91105	48.0	LIN	LT	7.200000		UGL		UN25
23095	91105	48.0	MEC6H5	LT	100.000000		UGL	100.000000	UM21
23095	91105	48.0	MEC6H5	LT	1.470000		UGL		AV8
23095	91105	48.0	MEK	LT	1000.000000		UGL	100,000000	LM21
23095	91105	48.0	MEXCLR	LT	11.000000		UGL		11125
23095	91105	48.0	MG		239000.000000	F	UGL		SS12
23095	91105	48.0	MIBK	LT	140.000000		UGL	100.000000	LIM21
23095	91105	48.0	MIBK	LT	4.900000		UGL		P8
23095	91105	48.0	MIREX	LT	24.000000		ÜGL		UM25
23095	91105	48.0	MLTHN	LT	21.000000		UGL		11125
23095	91105	48.0	MLTHN		4.260000		UGL		1111
23095	91105	48.0	MNBK	ND	1000.000000	R	UGI	100 000000	18421
23095	91105	48.0	NA		3600000.000000	F		100.000000	SC12
23095	91105	48.0	NAP	1.1	500000	•	HCL	100.000000	11425
23095	91105	48.0	NB	LT	3 700000				11425
23095	91105	48.0	NIT		31000-000000		lici	500 00000	118
23095	91105	48.0	NNDMEA	LT	9.700000		UCI	200.00000	11425
23095	91105	48.0	NNONPA	LT	6_800000		UGI		LM25
23095	91105	48.0	NNDPA	 LT	3 700000		UCI		111/25
23095	91105	48.0	OXAT		4 540000		UGI		
23095	91105	48.0	OXAT	LT	27 00000		UGL		LM25
23095	91105	40.0 48 0	PB	17	LT.000000	F	lici		0527
23095	91105	48.0	PCP	11	Q 100000				3312
23095	91105	48.0	PHANTR	с. 1 Т	0 00000				
		7010		-	7.700000		UUL		UMCJ

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LEVEL 3

Site	Sample	Sample	Parameter	Value		Flag	Units	Dilution	Method
ID	Date	Depth							
23095	91105	48.0	PHENOL	LT	2.200000		UGL		UN25
23095	91105	48.0	PPDDD	LT	18.00000		UGL		UN25
23095	91105	48.0	PPDDE	LT	14.000000		UGL		UH25
23095	91105	48.0	PPDDT	LT	18.00000		UGL		UH25
23095	91105	48.0	PRTHN	LT	37.000000		UGL		UM25
23095	91105	48.0	PRTHN	LT	.647000		UGL		UH11
23095	91105	48.0	PYR	LT	17.000000		UGL		UH25
23095	91105	48.0	SO4		1700000.000000		UGL	250.000000	TT09
23095	91105	48.0	STYR	ND	500.000000	R	UGL	100.000000	UN21
23095	91105	48.0	SUPONA	LT	19.00000		UGL		UH25
23095	91105	48.0	SUPONA	LT	.787000		UGL		UH11
23095	91105	48.0	T13DCP	ND	500.000000	R	UGL	100.000000	UH21
23095	91105	48.0	TCLEA	LT	150.000000		UGL	100.000000	UN21
23095	91105	48.0	TCLEE	LT	100.000000		UGL	100.000000	UH21
23095	91105	48.0	TCLEE		20.300000		UGL		N8
23095	91105	48.0	TRCLE	LT	100.000000		UGL	100.000000	UH21
23095	91105	48.0	TRCLE		10.100000		UGL		N8
23095	91105	48.0	UNK559		20.000000	S	UGL		UM25
23095	91105	48.0	UNK571		30.000000	S	UGL		UN25
23095	91105	48.0	UNK572		20.000000	S	UGL		UN25
23095	91105	48.0	UNK574		30.000000	S	UGL		UM25
23095	91105	48.0	UNK579		40.000000	S	UGL		UN25
23095	91105	48.0	UNK581		40.000000	S	UGL		UN25
23095	91105	48.0	UNK581		400.000000	D	UGL		UH25
23095	91105	48.0	UNK583		30.000000	S	UGL		UH25
23095	91105	48.0	UNK585		90.000000	S	UGL		UH25
23095	91105	48.0	UNK587		80.00000	S	UGL		UH25
23095	91105	48.0	UNK595		50.000000	S	UGL		UH25
23095	91105	48.0	UNK605		30.00000	S	UGL		UM25
23095	91105	48.0	XYLEN	LT	200.000000		UGL	100.000000	UM21
23095	91105	48.0	XYLEN		1.820000		UGL		AV8
23095	91105	48.0	ZN	LT	18.000000	F	UGL		SS12

APPENDIX B

WELL CONSTRUCTION INFORMATION

- · DRILLING LOGS/WELL CONSTRUCTION LOGS
- · SURVEY DATA

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DRILLING LOGS/WELL CONSTRUCTION LOGS

F	R. L. STC	LLA	387	ASSC	OCIATES,	INC.		S	ITE T	(PE		S	te ic)						
F	FIELD L	.0G	OF	BO	RING			1	BORIN	IG	Ī,	· . ec)	tion .	wel						~
P	ROJECT NA		ID LO	CATION	1		IPROJ		Tere	VATA	~						SHE	ЕТ_/	0	<u>ج_ع</u>
	RMA -	- 77	4100	-				170		VAU		AND D	ATUM							
	RILLING CO	MPAN	Y	,			DRILL	ER	DAT			ME ST	ARTE	<u> </u>		IDAT	E AND	TH 45 4		
Ь	<u>Layne</u> RILLING FC	- 20	NT:	ere METUC			OnH	as Weiner	9	15/9	/	11	30	0		2		11ME (O O	2 ETED
	CMI	E - 7	25	<i>4</i> ,		1			coi	MPLE	TION	OEP	TH C			TOT	AL NO.	OF SA	MPL	ES
s	ZE AND TY	PEOF	BIT	1144	K MOUNT	200			-	05	כ דר			<u> </u>		000	8	<u>}</u>	·	
	<u>634</u>	ID		9 - <u> 4 - 4</u>	5			·	SA	MPLE	s:		ľ	ັ8			E			BORATORY 9
ľ		510	N	one	-				W/	TER	1	FIRST	2	e '	<u> </u>		AFTER	24	HOU	JRS
S	AMPLER HA	MMER	75	" 10		Kroliau).	c has					<u> </u>		5 				_3:	8'	
<u>n</u>	(PE 5/1	+ 510	200		DRIVING WT	440	150	DROP 30	E	1	Key.	.0015 Ae	/ 5	- 5/5/	31	CHEO	jked e	SY/DAT	Έ	
	·	SAMPLI	ES T >							ES	TIM		Ť.	18	<u>.</u>					
E		1 AL	VER.	₹₹					ي الا	PE	RCE	NT OF	۲ ۲	Ň	б					
	NUMBER	٩Ë	0 2 2	52		DESCR	IPTION		USC SYN				OST	Sis	ğ			COMN	ENT	s
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	Americ		╂──							 			1							
1	Cuttings		╞──		Tight big	ma fine	SANO	, little silt	·		86	20	Dry	 	10y	/4/	6			
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	TYPE AND NUMBER	INTER	RECOV		DESCRIPTION	USC: SYME	GR	SA	FI	MOIST	CONSIS-	COLO	COMMENTS
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	Auger				light bearing free sandy set with			40	60	Dry	10	4	HAUNCH at 25'
<u> </u>	Cuttings				clay								
						_							
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	Arger				fight brown silly fin SAND			50	50	Dry	10	~	5/4
	cothings											_	
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<u>A</u> 2	inger ittings				light brown sulty course to five sand, trace grovel		5	<u>5</u> 5	30	an	10	yr	4/4
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,	9-7=1				ight brown medium to five SAND		5	90	5	0 a a		100	15/4
<u>a</u>	Hings				Little Ceave sand, trace grave	4-1	_					-/	

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S DEDTUKE	TYPE AN NUMBER		INIEHVAL	RECOVERY	BLOW COUNT	DESCRIPTION	USCS	ES PE GR	RCEI		MOISTURE	CONSISTENCY	COLOR	COMMENTS
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20	·		1				+			+-			┨	
37	·	127		-	100							<u> </u>	┼──	(unter at ser)
8	51	Ţ.	1.	69	35	light brown coarse to fine SAND little proval sites			0-		+			
			-	_	30	grains			05	5	wet		roy.	514 shipped
9		39.	0	-	- 30 5	i. It he are to the								
0-	52		1	55	10	SAND, Jubrounded, coarser			100		het			
		41.1		+	<u>15</u>	towards better of speen							<i>loy</i>	3/4 Shipped
1		41		ļ	4	light because to fine	$\left - \right $				$\left - \right $			· .
2-	53		14	8	25	SAND less fin Sond than			100		wet		loyr	5/4 1
,		43.0	,		23	above, subscurded	$\left - \right $							
	<u> </u>	43.0	2	4		light brown coarse to fine					┼─┤			
Ҡ	34			33	<u> </u> 20	SAND, trace gravel becoming		5	95		wet		1020	5/4 Shined
ļ		45.0		1	30	caaiser with depth					┠─┤			
$\left \right $	55	45.0	<u> </u>	<u> </u>	z_↓	light breme coarse to fine sand								Flering C. 1
Ĭ		46.0	(0)	6	50	fittle gravel, colle in tip	/	0	10		wet	10	7.5	your picture shin
╀	SE_		1.3	5/1	2	1. ght brown coarse to Fine sand		5	95		ind			
ŀ		48.0		┼	24	Frace grand								
┞		48.0		6		trace graved at 47.0 ft		د '	75		uet		10	(/4/4
╀	57		14	73	o lu	ight blows medium to fine sand to			100		wet		100	4/4 Share
L		500			40	will been coarse to fine sand	,	0					1	
┞	50.0	\$0.0		10		. He gravel, subranded					wer	-	¹ × 1	5/4
	58B		<u>[,0:</u>		4 G 3	$\frac{1}{2} \frac{1}{2} \frac{1}$				-1			1	Shipped
		52.0			40 5	Ily Redium to Fin Sand with 12'		-+'	-014		wet_	-+-		Shingel
				-	_ <u> </u> s	came of coarse to fine sond			1	1				
					<u></u>	19-51.6 graver - brown C						_		
					_ <u>_</u>	andy S.It. 1. He clay						//	er/	5/2
						TD = 52.0 ft		÷	-F	\neg			1	
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R. L. STOLLAR &	ASSOCIATES, INC.	SITE T	YPE			
WELL CONSTR	RUCTION SUMMARY	WEL	L IN	ject:	n we	
1	PROJECT RMA - Tracer					
L · · ·	PERSONNEL Ed Young					
	LOCATION OR COORDS.	ELE				
		····			cL	
				P OF CASI	NG	
	DRILLING SUMMARY	C	ONSTRUCTIO		OG	
5 5	TOTAL DEPTH 5 2.8-AL	TASK	SI	ART	FIN	ISH
tk k	BOREHOLE DIAMETER	DBILLING	DATE	TIME	DATE	TIME
	DRILLER Dallas Werner		9/5/4	1130	9/6/91	0745
	RIG CME-75 truck mounted					
K S	BIT(S) HSA					
L (5 .	DRILLING FLUID	GEOPHYSICAL LOGGING:				
	SURFACE CASING					
_ 5 5						
1 2 31.1	WELL DESIGN					
	BASIS: GEOLOGIC LOG					·
		SCREEN PLACEMENT:	9/6/91	0915	9/6/91	0950
36.0		FILTER:	9/6/91	0.950	7/4 /0	1073
		SEAL:	3/1.191	1073	6/1/2	1025
37.0		GROUT:	9/6/91	1125	9/1/51	10.55
		DEVELOPMENT	<i>4°_/ ''</i>	1125	~~~~~~	1150
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	S2		COMM	ENTS		
	\$3					······
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过三日 经	MATERIAL: 10/20 Silica Sand					
Auge Bestenk	SEAL: "Hole Plus" Bestante dise				<u> </u>	
1/52.1 1/52.1	GROUT: <u>Cement/Bentonite</u>					
	OTHER:			- · · · · · · · · · · · ·		
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R.	L. STO	LLA	7 & <i>I</i>	ASSC	C	IATES	s, inc	Э.			:	SITE T	YPE	<u> </u>		SIT	E 10)							
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					Little clay								
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R. L. STOLLAR & ASSOCIATES, INC. SITE TYPE FIELD LOG OF BORING SITE ID CONTINUATION SHEET: PROJECT NUMBER_ BORING 1170 Extraction wear SHEET 3 OF 3 SAMPLES DEPTHVFEET ESTIMATED CONSISTENCY RECOVERY INTERVAL PERCENT OF MOISTURE BLOW USCS SYMBOL COLOR TYPE AND DESCRIPTION NUMBER COMMENTS 35 GR SA FI 36 37 water at 37.2' 37.C ۱٩ light biene coarse to fine 5 85 10 mit ily 5/4 S 1.7 23 SAND, Little silt 1. He gravel 38 19 subscieded grains more 39 39 39.0 Laft brown coarse to f. 5 195 uet 104 5/4 52 8 70 trace gravel, for 12 towards hatten of speer Subconded grains 41.0 ZÛ 41 41.0 light brown coarse to f. in VISTY Sand running into 100 iet 53 0.7 SAND Subrended guins 3 42 augers 6 43,0 43 Ŷ3 ٤ light buing course to fine 10 90 wet 104 5/4 54 0.9 3 SAND, 1. He gravel. ceauser 44 20 howards settem 1 COSSIE in Arse speen. Subrended grains 45.0 25 of 45 45.0 1/961 course to fine SAND brown 20 80 wet 16x1 5/4 55 1.1 5 1. He gravel one cobble 76 subranded grains 35 47.0 ΫZ 47 47.0 to 47.8 Same S6A 42.0 20 80 ve 10×r 5/4 47.8 tothin 47.9 gray 56**b** 1.2 10 48 100 1 4/4 1. He silt 17 SEC 49.0 429 - 48.2 light bisma coarse to 20 49 20 80 ivef ioy 5/4 49.0 SAND, 1ttle gravel 1. ttle sil flowing Sand <u>574</u> 2.9 + Sampled inside Augers -50 Rasan , لر w 0' 5 1.+ Spoor Same as above to 51.5 20 80 vet bravely zone From 49,7 to 51.2 51 30 70 ret 1 5/4 Topof perver formation 20 80 80 S7B at 51.5 gray-bring fine 52.0 52 5/2 mois 10 sandy SILT , 1: He clay 53 T.D. 52.0 54 55

R. L. STOLLAR &	ASSOCIATES, INC.	Ś				•	
WELL CONSTR	UCTION SUMMARY		WELL	= 1			
	PROJECT RMA - Tracer	L.		CXT/	action	Well]
$\mathbf{F} \cdot \cdot \cdot$	PERSONNEL Ed Young						
	LOCATION OR COORDS.		ELEVAT	ION: GRO		·····	
- in is				TOF	POF CASI		
THE ST	DBILLING SUMMARY						
		ТА	CONS			ж Г	······
		14		ST	ART	FIN	
		DRILLING	:			DATE	TIME
	BIG CONE - 75 tout to	<u> </u>		9 <i>/11/</i> 91_	0910	<u>4////91</u>	1930
							
	DRILLING FLUID Nene	GEOPHY	SICAL		<u> </u>		
	SUBFACE CASING STYRE ST CARLING						
5 5 31.0		<u></u>					
0 4	WELL DESIGN						
- · · · · · ·	BASIS: <u>GEOLOGIC LOG</u> GEOPHYSICAL LOG	SCREEN					
	CASING STRING (S): C = CASING S = SCREEN	PLACEME	ENT:	9/11/91	1445	a/"/aı	1500
- <u>A</u> <u>A</u> <u>36,0</u>		FILTER:		9 /11 / 91	1500	9/1/91	1540
7.		SEAL:		9/1/91	1540	9/11/91	1600
- i - i - 510		GROUT:		9/11/91	1640	9/11/21	1700
		DEVELOF	PMENT:				ļ
		OTHER:					
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Allurium			W		LOPMENT		
	CASING C1 9 PZ C			·····		<u></u>	
	C2					<u></u>	
	. C3			·	<u> </u>		
	C4	<u></u>		<u> </u>			
	SCREEN S1, 010 5101 FVC (9-)			COM/4	INTO		·····
	\$2						<u>-</u>
	\$3				·		
	FILTER						
31.5	SEAL: " If the Pl " As to the						<u> </u>
	GROUT: <u>Coment/bentonite</u>						
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EXX:							
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R. L. STOLLAR & ASSOCIATES, INC.	SITE TYPE SITE ID
FIELD LOG OF BORING	BORING P1
PROJECT NAME AND LOCATION	SHEET 1 OF 3
RMA - Tracks	ER ELEVATION AND DATUM
DRILLING COMPANY	
Layne - western Dallas We	DATE AND TIME STARTED DATE AND TIME COMPLETED
DRILLING EQUIPMENT: METHOD	COMPLETION DEPTH 17/2/9/ 14/5
CME 75 truck meanted	53.0 SAMPLES
3 ³ /4" NSA	NO. OF BULK SS DRIVE LABORATORY
IRILLING FLUID	SAMPLES: 8
None	WATER FIRST
AMPLER HAMMER	
YPE Andrauli DRIVING WT. 150 DROP 3	G FI Same Shala
SAMPLES	
	J PERCENT OF H Z c
TYPE AND E S SS DESCRIPTION	
	36 3 3 COMMENTS 36 5 5 5
0.0	
	-++++++++++
Auger	
Cuttings (114)	
City Sada	2 70 30 114 1041 474
Fine SHIVO	
50	
5.0	
Avari	
Cuttures Carlo Carlo I	1~ 70 30 Dig 10 y 5/6
SAND, trace clay.	
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/4.4	
August 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1 1	
with a fit brown sitty redium to fin	- 60 40 Dry 1041 5/6
withings SAND, trace clay	
15.5	

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DEPTHVF	TYPE AND NUMBER	INTERVAL	RECOVERY	BLOW COUNT	DESCRIPTION	USCS	PE	RCEN		OISTURE	NSISTENC	COLOR	COMMENTS
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17	Auger				light brown first sandy STLT			40	10	A.,			
_	Ce Hings				trace clay	1				7		104	<u> </u>
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2	Auger			_	light brown fine sondy SILT,			40	60	R;		104	s]y
3	y				trace clay	┼╌┨							
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R. L. STOLLAR & ASSOCIATES, INC. SITE TYPE SITE ID FIELD LOG OF BORING CONTINUATION SHEET: PROJECT NUMBER BORING 1170 P1 SHEET 3 OF 3 SAMPLES DEPTHVFEET ESTIMATED CONSISTENCY INTERVAL RECOVERY PERCENT OF MOISTURE USCS SYMBOL BLOW COLOR TYPE AND DESCRIPTION COMMENTS NUMBER GR SA FI 35 36 360 hf light brown coarse to fine Ory h 10 90 104 5/4 1.74 SAND, I'ttle gravel, subject of 51 28 37. water at 36.9' 35 grains_ 380 28 38 35.0 light brown coarse to fine SAND 90 10 Wet SZ 8 7 slight color change <u>l:ttle</u> silt 39 at 39.0 1045/4 to 1045 5/6 D 40.0 M 1/2 40 sandy SILT layer at 39.6" wet 40 60 40.0 light brow. SAND COURCE te Gin int 100 Ŝ3 Ь 8 color change 41 at 410 to a dracks 12 0 nac - brown 10 90 Coarce 1 a wef 420 20 41.0 with 1. He gravel subscirated 42 VZD light brown coarse to fine 90 wŧ 10 SY 1,2' 8 SAND trace gravel relatively finer 4 3 42.6 to 42.8 gravely frem 18 42.8 43.2 Subround SO 44.0 Y 4 4.0 29 24 55 Iralt brews coarse to fine SAND 80 20 wrf 16 1. Hele gravel schounded 45.0 27 4 5 21 z 46 46.0 2135 light brown coarse to fine SAND 20 80 wet 56 3 23.41 Little gravel, subrearded 47 38 48.0 41 48 48.0 17 light brown coarse to fine SAND 10 90 wet 57 z7 C.9 4 9 trace gravel Subcound 30 50.0 35 50 500 27 Same to 50,3 ST 50.5 30 to 50.5 weathered zne 51 50.3 10 70204 iron stainty. Silly coarse to SAND L'HLE gravel 52 Bedrock at 51.5 TD = 53.0'53 54 5 5

R.L.	STO		& ASSOCIATES, INC.	5	SITE TYPI	Ξ			
r			The Transform		WELL	P1			
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њ ·	•		LOCATION OF COORDS.		ELEVAT	TION: GR	OUND LE	VEL	
l						то	P OF CAS	NG	
. ·	•		DRILLING SUMMARY		CON	STRUCT	ON TIME L	.OG	
			TOTAL DEPTH 5 3. 0	TA	SK	S	rart .	FIN	 //SH
- •	•	•	BOREHOLE DIAMETER			DATE	TIME	DATE	TIME
			DRILLER Dallas Werner			9/12/m,	1040	9/12/11	1415
- ·	•	•	RIG_CME 75 truck nousted					<u> </u>	1
			BIT(S) 3 ³ 4 NSA				 	1	<u> </u>
. ·	•	•	DRILLING FLUID	GEOPHYS LOGGING:				†	<u> </u>
			SURFACE CASING 8×8 × 5 Stainless steel	·			<u> </u>	<u> </u>	<u> </u>
- •	•	•	WELL DESIGN				<u> </u>	<u> </u>	<u> </u>
			BASIS: GEOLOGIC LOG				<u> </u>		<u> </u>
_ •	•	•	GEOPHYSICAL LOG CASING STRING (S): C = CASING S = SCREEN	SCREEN PLACEMEI	NT:	9 Jizhi	1425	1/12/51	1430
- ·	•	•		FILTER:		9/12/11	1430	9/12/91	1510
			·	SEAL:		9/12/91	1510	9/12/41	15 30
- •	•	•	·	GROUT:		9/2/21	1610	9/12/41	1630
				DEVELOP	VENT:				
. •	•	•		OTHER:					
				<u> </u>					
	•	•							
					WE	ELL DEVE	LOPMENT		
. •	•	•	CASING C1_2" Black 40						
			C2						
. •	•	•	С3			······································			
			C4						
. •	•	•	SCREEN S1 2" ,010 slot 15'						
			S2			СОММЕ	INTS		
• •	•	•	\$3						
_			SI TEO			·			
	•		MATERIAL: 10/20 silica Sand					<u></u>	
_			SEAL: Bentonite Pellets				# <u>************************************</u>		•
• •	•		GROUT: <u>Cement/Bentenite</u>						
			OTHER:						
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RMA - Tracer 11 PRILLING COMPANY DRILLI PRILLING COMPANY DRILLI PRILLING EQUIPMENT: METHOD CME 75, truck man tel IZE AND TYPE OF BIT 3 ³ 14 HSA	ECT NUMBER 70 ER -3 Weiner	EL	EVAT	ION .	AND (DATUN	A		
AILLING COMPANY Layne - Western AILLING EQUIPMENT: METHOD CME 75, truck man tel IZE AND TYPE OF BIT 3 ³ /4 HSA	ER S Weiger	DA							
RILLING EQUIPMENT: METHOD CME 75, truck man tol IZE AND TYPE OF BIT 3 ³¹ 4 HSA	s Weiger					TADT			
CME 75, truck mounted IZE AND TYPE OF BIT 3 314 45A		4	7/13	19	/	089	20 70		DATE AND TIME COMPLETED
3 ³ /4 45A		co	MPLE	TION	N DEP	тн			TOTAL NO. OF SAMPLES
				1	BULK		22		8
AILLING FLUID		SA	MPLE	s:			8		LABORATO
		U U	ATER	F	FIRST	2	> <))	AFTER HOURS
PE Hydraulis DRIVING WIT		нүт	DROG	EOL	OGIS	T/DAT	E.		CHECKED BY/DATE
SAMPLES	DROP 30		<u>e</u>	4	200		<u>1]13</u>	171	
		S S	ES PE	TIMA RCE	NT OF	: ₩	TENCY	Б	
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light brown silts a	edim					-+			· · · · · · · · · · · · · · · · · · ·
to Fine sand ten	u clay			-		7		<u>eył</u> -	5/6
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R. L. STOLLAR & ASSOCIATES, INC. WELL LOG

WELI	LLOG					Project No Well/Boring	. <u> </u>		-19-3	,——-			
Project N	ame and Location		Flowation			Combo/Ke	y No.		- 2	•			
Drilling Co	ompany		Cievation			Coordinate	s		Tov	wnshi	ρ	Range	ectio
Orilling F	Olioment	·	Driller			Date and T	ime St	arted	<u> </u>				
0						Date and T Total Driller	ime Co d Depri	omple h	ted		molec		
Urilling M	ethod		Boring Diam	eter		Samel					******	un Dep	pth
Size and 1	Type of Casing		1			Sampler	No	of Sa	amples	s Ca	1	Τ	SS
Type of Pe	erforation	Erom				Water Eleva	ation	Firs	1	0	mpletic	on	24 Hrs
Type of Se			To		(Ft-BGS)	Pack Size a	nd Typ	ve ve		I Fre			(FI-RC)
		From	То		(Ft-BGS)	Hydrogeolod				1	ected	Burb.	
0				1-1						<u> </u>			le
(feet)	Dee	viotion		3 B		Piezomolor	ž		suma % of	16	5		Remarks
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Page 2 of 3

I. L. STOLLAR &	ASSOCIATES, INC.	SITE TYPE								
WELL CONSTI	HUCTION SUMMARY Π		WELL	P7						
	PROJECT RMA - Trais			1]			
├ · · ·	PERSONNEL Ed Young									
	LOCATION OR COORDS.		ELEVAT	ion: Gr	OUND LEV	'EL				
				TOF	OF CASI	NG				
<u> </u>	DRILLING SUMMARY	CONSTRUCTION TIME LOG								
	TOTAL DEPTH53.0	TASK START FINISH								
			~.	DATE	TIME	DATE TIME				
	DRILLER Pallas werner		J. ·	alista,	0840	9/13/91	1130			
· <u>s</u> · <u>z</u> ·	RIG CME-75 truck mounted									
2	BIT(S) <u>3 1/4 HSA</u>		(0)041							
	DRILLING FLUID	LOGGIN	G:							
	SURFACE CASING 8 x8 x 5 5 to but she	<u> </u>								
2 31.0	WELL DESIGN									
	BASIS: GEOLOGIC LOG	<u></u>								
		SCREEN PLACEM	ENT:	9/13/11	1170	9/13/71	1145			
		FILTER:		9/13/41	1145	9/13/21	12.30			
b 1 <u>37.0</u>		SEAL:		9/13/91	1230	9/13/91	1250			
364	·	GROUT:		9/13/11	1330	9/13/41	1350			
		DEVELO	PMENT:							
		OTHER:								
		<u></u>			· · · · · · · · · · · · · · · · · · ·					
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		WELL DEVELOPMENT								
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1.1.1	C3					<u> </u>	· · · · · · · · · · · · · · · · · · ·			
2:	64	·			·····		<u></u>			
	SCREEN S1_ 2.0" PVC,010 slot 15'									
	S2	<u></u>		COMME	ENTS					
	\$3		<u></u>		·······					
	S4									
F SI.0	MATERIAL: 10/20 S.l.ca Sand	<u> </u>								
520	SEAL: <u>Bentonite Pellets</u>						·			
Slout	GROUT: <u>Cenent / Bentonite</u>									
T0=53.0	OTHER:		<u> </u>							
/ Belsock										

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R. L. STOLLAR & ASSOCIATES, INC. SITE TYPE SITE ID																			
FIELD LOG OF BORING							В	ORIN	G	<	R	4							
PROJECT NAME AND LOCATION PROJECT									Truc	VATIO			<u></u>]	SHEET _1_OF_3			
RMA - Tracer 117																			
DI	DRILLING COMPANY DRILLER									DATE AND TIME STARTED							DATE AND THE COURSE		
DRULING FOLIDATION DATION DATION WERE								9	9/18/91 11/0						9/18/9/ 1415				
CME 75 truck man Lad								COMPLETION DEPTH						TOTAL NO. OF SAMPLES					
SIZE AND TYPE OF BIT								52.0							9				
3 3/4 45A								SAM	SAMPLES: 4						UNIVE	LABORATORY			
								WA	TER	FI	RST	-1-	<u> </u>		AFTER HOURS				
SAMPLER HAMMER																			
TYPE Hydraulis DRIVING WE LCO DOGO 20								F Y Click							CHECKED BY/DATE				
SAMPLES										- <u>-</u> -	FST	10 <u>(</u>	MATED 7/18/51						
1		₹	ERY								D PERCENT OF								
P	TYPE AND	LE R	§	β		DESCRIP	TION			N ME					SIST	ğ	COMMENTS		
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	<u>Auger</u> (cHines				light brown silly fine			60	40	Diy		10,	(4/4		
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7	Auger				light brougs sity coarse to		5	55	40			[1]	n chu		
8	Cettings.				fine SAND trace gravel										
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3	Huger				light brew coase to fine SAND 1: He silt, fittle acave		20	60	20	Dry		10	vr s/3		
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R. L. STOLLAR & ASSOCIATES, INC.									SIT	ETY	1 5E		SIT	E ID								
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PRO	DJECT NAM	E AND	LOC	ATION				PROJ	CENUME	H:R	ELEN	VATIC	N AN	D DA	[UM							
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DR	LLING EQUI	PMEN	IT: M	ETHO				Var			COM	IPLET	NI ION I	<u>/</u> DEPTH	<u>63</u> .	>		TOTAL NO. OF SAMPLES				
c	mE 75		tru	ck r	neur te	J					52.0'								SAMPLES			
SIZ	E AND TYPE	OF B	IT C A								NO.OF BULK SS							DRIVE LABORATORY			RATORY	
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Augel				light brown silts meduce		┼──	10	14.			╂──	16 -
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Huger	┨╴┨			light brown silly fing			50	50	neis		10	V1 5/3
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LU LUG	D LOG OF BORING IUATION SHEET: PROJECT NUMBER 多日 マ)170					NG	Γ.							
- s	SAMPLES							B	2.					
	TERVAL	COVERY	SOUNT	DESCRIPTION	USCS SYMBOL	EST PER	RCEN		DISTURE	ASISTENCY	COLOR	COMMENTS		
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Cutting	1			CALLA VILLETALE		3	<u>45</u>		and the					
	1			Stills, tille grave					ŕ					
8	1													
	39.0													
			9	meding & Fine SAND 1-Hu			100							
			23	Concer Sand to 397			140		wet					
SBZA		1.5'	32	Cooks to find SAND to YAL			101)							
			33	Course to Fine SANA Have			100	_		\neg		P. A		
·			25	acovel to 41.5		10	90		no L			Redlove 2 spoon		
2			10	Coarse to fine SAND train	_	- v						Part 11 TO 43		
			10	ornoel								Reavered, 1		
,	43.0		19									······································		
	430		12	Coarse to Fine SAND to			100		wef		\neg			
SBZA		1.4	14	44.0 Coarse to Fine SAND										
			16	true gravel to 44.4		10	70		wef					
;	45.0		28											
	Y5.0		13	Consete Fin SAND	_		100		we					
SB2B		1.3'	17	to 45.6 , gravely										
			28	Coase to Fine SAND to		40	60		wef			· · · · · · · · · · · · · · · · · · ·		
+	47.0		_ 37	46.3										
10000	47.0		9											
15820		0.8	19	Coarse for from SAND		10	90		wet					
			27	true gravel near tip					-+					
+	47.0		39											
SAZE	77.0			LOARSE TO FILL SAND		10	90	<u></u> ł	wet					
1000			-C7	TRUE GRAVES			-+							
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	SITE TYPE											
WELL CONSTRUCTION SUMMARY	WELL	SB	52									
PROJECT <u>RMA-Tracer</u>	<u> </u>		·									
PERSONNEL <u>Rá Yourg</u>												
LOCATION OR COORDS.	ELEVAT	'ION: GRO	OUND LEV	EL								
F ·] ·		TOP	OF CASI	łG								
DRILLING SUMMARY	CONSTRUCTION TIME LOG											
► TOTAL DEPTH52.0	TASK	ST	ART	FINISH								
BOREHOLE DIAMETER 7.0"+		DATE	TIME	DATE	TIME							
DRILLER Dallas Werner	ING:	9/18/21	1635	9/18/41	1920							
0 . 0 . RIG _ CME - 75												
6 8 8 BIT(S) 3 ³ /4 MSA												
GEOP	HYSICAL ING:											
SURFACE CASING <u>y"xy"x5</u> metal	<u> </u>	<u> </u>										
		k										
			· · · · · · · · · · · · · · · · · · ·									
BASIS: <u>GEOLOGIC LOG</u> GEOPHYSICAL LOG PLACE	EN EMENT:	9/18/4	1930	9/18/4.	1930							
CASING STRING (S): C = CASING S = SCREEN	 R:		100	510								
		9/18	alala	Gludor	gliala.							
		9/19/4	414 .	aliala	11.11.11							
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52		СОММ	ENTS									
S3	Backhal	0.110										
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+ FILTER None (see coments)	(10/20) placed for 35 to 36 N La											
ability STS_ SEAL: Bentinite Pellete	- <u>(10/20) placed from so to so ft. Las</u> <u>Screen and Brank steam cleane</u> <u>be for in shukation</u>											
GROUT: Cement Bentonite bel												
OTHER:	Bentonit. sellet hydrated with											
5	gallens	<u>of _w</u>	ater.									

B. L. STOLLAR & ASSOCIATES INC								SITE TYPE SITE ID (Monitony Cluster)								
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R. L. STOLLAF	& ASSOCIATES, INC.	SITE TYPE										
VELL CONS	TRUCTION SUMMARY	WELL	mw el	'uster								
	PROJECT <u>RMA - Tracci</u>			<u></u>								
- · · · ·	PERSONNEL											
	LOCATION OR COORDS.	ELEVATION: GROUND LEVEL										
		······································	TOP OF CA	SING	<u> </u>							
	DRILLING SUMMARY	CON	ISTRUCTION TIME	LOG								
a	TOTAL DEPTH 37.5 '	TASK	START	FINIS	 эн							
0	BOREHOLE DIAMETER 12" ±		DATE TIM	E DATE	TIME							
4	DRILLER Dallas Werker		9/16	9/16								
6	RIG_CME-75											
	BIT(S) X 1/4 " USA	4			······							
-	DRILLING FLUID	GEOPHYSICAL LOGGING:										
A CLER	SURFACE CASING 12 x 12 stainless steel											
	WELL DESIGN				-							
- 7	BASIS: GEOLOGIC LOG GEOPHYSICAL LOG	SCREEN PLACEMENT:	9/20	9/20	M							
>	CASING STRING (S): $C = CASING S = SCREEN$		1									
- 6		CEAL.										
4	·	SEAL:	9/24	9/24								
			9/24	9/24								
25		DEVELOPMENT:										
3.		OTHEH:										
25 4		· · · · · · · · · · · · · · · · · · ·	WELL DEVELOPM	 ENT								
	CASING C1											
	C2											
- 35.2	<u>7.5</u> c3											
	C4											
- · [] ·	SCREEN ST 40 2" ALA ST A LAS											
427	Yerb S2 4.0 2", 010 stat standers steel		COMMENTS									
- · [-] ·	53 4.6 2° , 010 slat stainless steel	Drive De	ints dr.	were in el								
-	S4	with a in	skhemper.	Blank								
- =	FILTER MATERIAL: Natural soil &	carina is	slainless	steel	intil							
11.1 三	47.0 SEAL: Bentonto Polkk	abose us	ter task	then Pu	(
- · · [=]	GROUT: <u>Cenert</u> / Berlon te											
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SURVEY DATA

GEONEX North American Operations, Inc.

1215 S. Huron, Denver, CO 80223

R.L. STOLLAR & ASSOCIATES WELL LOCATION SECTION 23 - ROCKY MOUNTAIN ARSENAL SEPTEMBER 27, 1991 STATE PLANE COORDINATES - COLORADO NORTH ZONE - US FEET - NAD27 **ELEVATIONS IN US FEET NGVD 1929** PT# NORTHING EASTING ELEVATION DESCRIPTION 201 192109.8689 2181023.8373 5177.951 CONCRETE # 1 - INJECTION WELL (I) 5180.086 TOP OF CASING 5180.111 TOP OF PVC 192113.7852 2181022.5956 5177.685 GROUND 202 # 2 - PIEZOMETER (P1) 5179.836 TOP OF CASING 5179.913 TOP OF PVC 203 192119.4567 2181024.9543 5177.601 GROUND # 3 - PIEZOMETER (P2) 5179.758 TOP OF CASING 5179.694 TOP OF PVC 204 2181027.2427 5177.551 GROUND 192118.0926 # 4 - MONITORING CLUSTER -(8) 5179.415 TOP OF CASING 5179.602 TOP OF PVC 205 192118.3925 2181027.4176 5177.551 GROUND # 5 - MONITORING CLUSTER -(A) 5179.415 TOP OF CASING 5179.603 TOP OF PVC 206 192118.0926 2181027.5926 5177.551 GROUND # 6 - MONITORING CLUSTER - (C) 5179.415 TOP OF CASING 5179.573 TOP OF PVC 207 192113.1974 2181034.7442 5177.603 CONCRETE # 7 - PIEZOMETER (23095) 5179.808 TOP OF PVC 208 192121.2064 2181028.8382 5177.571 GROUND #8 - PIEZOMETER (B2) 5180.175 TOP OF CASING 5180.000 TOP OF PVC 192129.7152 2181032.5419 5177.690 CONCRETE 209 # 9 - PIEZOMETER (B1) 5180.057 TOP OF CASING 5180.999 TOP OF PVC 210 192136.8979 2181035.5271 5177.599 CONCRETE #10 - EXTRACTION WELL (E)5179.824 TOP OF CASING 5179.856 TOP OF PVC