



UNDERSTOOD
MINERAL RESOURCES LTD.

TECHNICAL REPORT ON THE ANGILAK PROPERTY, NUNAVUT, CANADA

Prepared for:
ATHA ENERGY CORP.

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ATHA
ENERGY CORP.

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1 SUMMARY

1.1 Executive Summary

Understood Mineral Resources Ltd. (UMR) was retained by ATHA Energy Corp. (ATHA or the Company) to prepare an independent technical report on the Angilak Property (the Property), located in southern Nunavut, Canada. This Technical Report has been prepared in accordance with the Canadian Securities Administration's (CSA's) National Instrument 43-101 Standards of Disclosure for Mineral Projects (NI 43-101) and guidelines for technical reporting from Canadian Institute of Mining, Metallurgy and Petroleum (CIM) "Best Practices and Reporting Guidelines".

The report includes a summary of exploration activities conducted on the Property to date and discloses potential uranium quantity and grade for the Lac 50 Uranium Deposit (Lac 50 Deposit, or the Deposit), expressed as ranges, as a target for further exploration. The stated potential quantity and grade are conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource. The effective date of this report is October 14, 2025.

ATHA is a Canadian exploration company engaged in the acquisition, evaluation, and development of uranium assets. ATHA is listed on the Toronto Stock Exchange Venture (symbol SASK), on the OTCQB Best Markets (symbol SASKF), and on the Frankfurt Stock Exchange (symbol X5U.F).

1.2 Technical Summary

1.2.1 Property Description and Location

The Property, which hosts Lac 50 Deposit, is located 350 kilometres west of Kangiqliniq (Rankin Inlet) and 225 kilometres southwest of Baker Lake in the Kivalliq Region of Nunavut. The Property is bound between Latitudes 62° 14' and 62° 48' North and Longitudes 98° 21' and 99° 44' West, (North American Datum 1983 (NAD83), Universal Transverse Mercator (UTM) Zone 14 coordinates: 6901449m N and 6960000m N and 463830m E to 533463m E) and is within the 1:50:000 National Topographic (NTS) map sheets 065 J/05, J/06, J/07, J/09, J/10, J/11, J/12, J/14 and J/15. The Property comprises 135 Crown issued mineral claims and one mining lease, as well as an Inuit Owned Land (IOL) parcel (RI30-001) for a total area of 192,913 hectares. Latitude Uranium Inc. (LUR), and subsequently ATHA, has acquired the right to conduct exploration work on the IOL parcel under a Mineral Exploration Agreement (MEA) with Nunavut Tunngavik Inc. (NTI). Land use permits enabling exploration work to be conducted on the Property have been issued, amended and renewed by the Kivalliq Inuit Association (KIA) for parts of the Property covering the IOL and by Crown-Indigenous Relations and Northern Affairs Canada (CIRNAC) for the Crown Lands.

1.2.2 Accessibility, Climate, Local Resources, Infrastructure, and Physiography

Access to the Property is reliant on helicopters and fixed wing aircrafts. There is a gravel airstrip located at the Nutaaq drilling camp and a secondary airstrip located 1.5 kilometres west of the Nutaaq drilling camp. Exploration at the Property is typically conducted between the months of February and October. Local access to and around the Project site is by either helicopter, float plane or wheeled fixed wing aircraft such as a Single Otter. Due to the commercial-grade airport and the relatively close distance, Baker Lake and Rankin Inlet are the logical mobilization points for all supplies and people. All required infrastructure and supplies for exploration can be brought

in each field season by air or via an over-land haulage during the winter season. The Property is situated in the "barren lands," a large region of almost flat, treeless tundra characterized by poor bedrock exposure and extensive swampy areas with abundant small, shallow lakes. Elevation at the Property ranges from 150 to 250 metres above sea level (asl). Locally maximum relief ranges from 30 to 75 metres but is more commonly less than 20 metres. Glacial deposits in the area are extensive thus limiting rock exposure to less than a few percent of the total Property area.

The climate is best described as continental-arctic with short cool summers and long cold winters with minimal precipitation. Average summer high temperatures can reach up to 20°C, while average winter temperatures are in the order of -30°C to -35°C. Snow is generally on the ground until the first week of June and ice does not leave the mid-sized lakes until the third week of June. Nearby Yathkyed Lake has ice cover usually until early or mid-July. Smaller lakes freeze over around the end of September, therefore, most of the year the Angilak Property is covered with snow, except between June and the end of September. Permafrost is present from one metre to unknown depths in mid-summer. The thawed active layer is thick enough by mid to end June to allow till sampling and induced polarization surveys. Diamond drilling to 200 metre depths can be accomplished without salt or propane based upon experience.

1.2.3 History

Airborne radiometric surveys, geologic mapping and lake sediment sampling for uranium in the Project area began in 1970 but halted shortly after. Between 1976 and 1982, various operators completed regional and targeted exploration work including soil sampling, geophysical surveys, and drilling. As historical property boundaries are not the same as the current property boundaries, some of the historical work completed during this timeframe may fall outside of the current margins of the Property. The Lac 50 Deposit was discovered and partially delineated by Pan Ocean (later named Aberford Resources) during this period, but there is very little documentation or data that exists for the work completed. The long section of the Lac 50 Deposit provided by Miller et al. (1986) shows at least 58 drillholes over a strike length of 1 kilometre down to a depth of close to 250 metres below surface. Exploration for uranium ceased abruptly at Lac 50 and the surrounding area when Pan Ocean divested its uranium projects in 1982.

In 1993, NTI was formed to manage land and implement the Nunavut Land Claims Agreement (NLCA), which itself was established in 1993. Along with the formation of the territory of Nunavut in 1999, came the establishment of 37,000 km² of subsurface land parcels of Inuit Owned Land, including IOL Parcel RI30-001, which is situated over the historic Lac 50 Uranium Deposit. In 2007, NTI announced its new pro-uranium policy and expressed interest in forming a partnership with exploration companies to conduct uranium exploration on IOL parcels in Nunavut. That same year, NTI and Kaminak Gold Corporation (Kaminak) signed a landmark uranium partnership to explore IOL parcel RI30-001 and Kaminak's surrounding federal mineral claims. This led to the creation of Kivalliq Energy Corporation (Kivalliq Energy) as a spin out company of Kaminak in 2008, formed with the express purpose to explore and advance the Angilak Project. Kivalliq Energy was renamed ValOre Metals Corporation (ValOre) in 2018.

In 2007, Kaminak commissioned GeoVector Management Inc. (GeoVector) to conduct a detailed compilation followed by a reconnaissance field program that included geological mapping, prospecting, and field verification of historical work. Between 2008 and 2012, exploration work on the Property included ground geophysical surveys, airborne geophysical surveys, diamond drilling, reverse circulation (RC) drilling, soil sampling, rock sampling, geological mapping, and prospecting. The diamond and RC drilling further delineated the Lac 50 Deposit, as well as tested regional exploration targets.

ValOre continued exploration on the Property from 2013 to 2016, with rock sampling, soil sampling, geophysical surveys, trenching and channel sampling, limited RC and diamond drilling, and heavy mineral sampling. No exploration work was completed on the Property from 2017 to 2021, but in 2022, ValOre conducted geophysical surveys, a soil sampling program, RC drilling and diamond drilling.

On June 20, 2023, Labrador Uranium Inc. announced the acquisition of the Property from ValOre and subsequently changed their company name to Latitude Uranium Inc. (which is previously defined as LUR in this report). Exploration work completed on the Property by LUR included a high resolution radiometric and aeromagnetic airborne survey during the spring of 2023 and diamond drilling completed in the summer of 2023. The diamond drilling consisted of 18 diamond drillholes and successfully increased the extent of known mineralization at Lac 50 and identified new mineralization horizons.

ATHA acquired LUR in April 2024 including 100% of all assets owned by LUR and its wholly owned subsidiary 5833 Nunavut Ltd. Claims staked by ATHA are currently registered under 5833 Nunavut Ltd. and ATHA Energy (NU) Corp, another wholly owned subsidiary of ATHA. In addition, the Mineral Exploration Agreement with NTI for RI30-001 has been assigned from 5833 Nunavut Ltd. to ATHA Energy (NU) Corp.

1.2.4 Geology and Mineralization

The Angilak Property is located within the Western Churchill Province, a large Archean craton that experienced significant crustal shortening and uplift during the Proterozoic, where the subsequent gravitational collapse led to the deposition of several rift basins, including the Baker Lake Basin. Two major structural corridors surround the Property: The Snowbird Tectonic Zone to the northwest, and the Tyrrell Shear Zone to the southeast. The structural corridors formed because of the assembly of the Churchill Province and were later reactivated by tectonic activity in the Proterozoic. The Archean basement rocks underlying the Property consist of tonalite-granodiorite gneisses and granitoids, as well as the metasedimentary and metavolcanic greenstones of the Henik Group. These are unconformably overlain by the Angikuni and Yathkyed sub-basins (Baker Lake Group). The Lac 50 Uranium Deposit is located adjacent to the northeastern margin of the Angikuni sub-basin and is hosted in Archean metasedimentary and metavolcanic rocks of the Henik Group. Mineralization at the Lac 50 Deposit is structurally controlled and bears similarities to Beaverlodge-type vein deposits.

1.2.5 Exploration

In March 2024, ATHA completed the acquisition of 100% of the issued and outstanding common shares of LUR, which became a wholly owned subsidiary of ATHA. The transaction included the Angilak Property, which hosts the Lac 50 Uranium Deposit.

A 25-hole diamond drill program, a Mobile MagnetoTellurics (MobileMT) airborne geophysical survey, bedrock mapping and soil sampling were conducted during the 2024 summer field program between the end of May and mid September.

In 2025, ATHA completed a 22-hole diamond drill program, ground geophysical surveys including gravity and fixed-loop electromagnetic (EM) over the KU and RIB target areas, and a regional airborne gravity survey. All field programs were completed between the end of April to late September.

1.2.6 Drilling

Drilling has been conducted on the Property by several previous exploration companies prior to ATHA acquiring the claims in 2024. Essex Minerals, Urangesellschaft, Noranda Exploration and Pan Ocean conducted diamond drilling during the early exploration period from the 1970s to 1982, and Western Mining Corporation in 1994-1995. Kivalliq Energy conducted consistent exploration between 2008 and 2018 with diamond drilling in 2009, 2010, 2011, 2013 and 2015 (total 78,835 metres), and RC drilling in 2011 and 2012 (total 11,684 metres), further delineating the Lac 50 Deposit in 2013 and discovery of new mineralized zones or extensions of existing zones. ValOre completed a twenty-six hole, 3,590 metre diamond drillhole program and a twenty-seven hole, 3,165 metre RC program in 2022 focusing on Dipole, J4 West and Yat zones. LUR completed a diamond drill program in the Lac 50 Deposit area specifically targeting the Main Zone. The 2023 drill program successfully increased the extent of known mineralization and identified new mineralization horizons. A total of 18 diamond drillholes totalling 5,662 metres were completed.

ATHA conducted a 10,052-metre helicopter-supported diamond drill program between June 4 and August 22, 2024. A total of twenty-five drillholes were completed, not including one lost drillhole. This drilling program had several key objectives: to expand the footprint of known mineralized zones within the Lac 50 Trend (Western Extension, Eastern Extension, Main Zone, J4, and Ray zones) by testing along strike, down-dip, and down-plunge. Additionally, it aimed to investigate historical data by testing previously identified VLF anomalies and mineralized showings in under-drilled areas identified as the Lac 48, Lac 52 and Lac 54 Trends (which include the Blaze, Hot, Pulse, and Mushroom Lake zones).

In 2025, ATHA conducted a 10,774-meter diamond drill program between May 31 and September 6, 2025. A total of twenty-two drillholes were completed, excluding one lost drillhole. The drilling program had several key objectives; to expand the mineralization footprint within the Lac 50 trend, specifically targeting the down-dip extension of the J4 and Ray Zones, and down-dip extension of surficial mineralization identified at the Mushroom Lake zone. Additionally, the 2025 campaign was designed to test high priorities targets at the KU and RIB regional areas located within the 31-km RIB-Nine Iron structural corridor.

1.2.7 Data Verification

LUR was provided the geologic data and information for the Property after acquisition (2023) as Microsoft Excel spreadsheets, Microsoft Access databases, and ESRI shapefiles. These data were imported into MXDeposit™, a database management system made by Seequent, by LUR, and ArcGIS software was used to check for geospatial errors. For the 2023 drilling program, LUR logged the drilling data into Excel files and then transferred the data into MXDeposit™. Drilling data collected by ATHA was entered directly into MXDeposit™.

ATHA personnel completed an internal audit of the Angilak Project drillhole database by comparing approximately 10% of the drilling data to the original drill logs, assay certificates, and collar coordinates. Original assay certificates and geological logs were used to check the MXDeposit™ database after import from previous drilling. Minor typos and column mismatches were found and rectified, but overall, the drillhole database was found to be satisfactory. However, the RC drilling assay results were deemed to be imprecise relative to the validated core drilling results, and thus the RC drilling was not considered in the exploration target model. The QP reviewed the audit work completed by ATHA and considers the results, methods, and conclusions to be accurate and appropriate. The QP further validated the diamond drilling database via the following digital queries:

- Header table: searched for incorrect or duplicate collar coordinates and duplicate hole IDs.

- Survey table: searched for duplicate entries, survey points past the specified maximum depth in the collar table, and abnormal dips and azimuths.
- Lithology, alteration, and structure tables: searched for duplicate entries, intervals past the specified maximum depth in the collar table, overlapping intervals, negative lengths, missing collar data, missing intervals, and incorrect logging codes.
- Geochemical, density, and assay tables: searched for duplicate entries, sample intervals past the specified maximum depth, negative lengths, overlapping intervals, sampling lengths exceeding tolerance levels, missing collar data, missing intervals, and duplicated sample IDs.

No significant issues were identified.

During the QP's two-day site visit, he reviewed ongoing, recent, and historic core from the Deposit, confirmed the location of three collar locations with a handheld GPS, verified the geological setting, and reviewed drilling, logging, sampling, analytical and QA/QC procedures. A comparison of the drill logs and assay results with the laid-out drill core showed that the information recorded in the drill database matched well with the drill core. As part of the core review, the QP verified the occurrences of mineralization visually and by way of a hand-held scintillometer.

In Mr. Batty's opinion, the Angilak Project exploration data are free of any material or systematic errors and are considered well validated and of sufficient quality for use in this Technical Report. ATHA and its predecessors had a robust QA/QC process in place for the data collected on the Angilak Property.

1.2.8 Mineral Processing and Metallurgical Testing

In June 2012, the Saskatchewan Research Council (SRC) commenced a metallurgical testing program that built on first pass work completed in 2010. The initial 2010 results indicated alkaline leaching as the most effective extraction process for the Lac 50 Deposit uranium mineralization. The objective of the 2012 program was to investigate uranium alkaline leaching optimization and perform a preliminary evaluation of the purity levels of a final yellowcake product. The SRC aggregated a master composite sample weighing approximately 60 kilograms by blending and homogenizing 166 quarter-split and half-split pulp reject samples from 51 core holes. The sampled 2010 and 2011 core holes represent 3.2 km of strike length of uranium mineralization along the Lac 50 Main Zone, Western Extension and Eastern Extension.

A head grade sample from the 2012 composite assayed 0.737 % U, 0.217% Mo, 0.667% Cu, 0.221% Zn, 0.231% Pb and 26.7 g/t Ag. Optimized results from alkaline leaching indicate that 94.1% of uranium can be extracted in 48 hours and 95.9% of the uranium extracted in 72 hours with a final yellowcake product that contained 71.9% uranium. It is encouraging at this early stage that the assayed impurities in the yellowcake product are below the maximum allowable concentration limits without penalty for uranium ore concentrate specifications. Additional metallurgical work is warranted.

1.2.9 Mineral Resource Estimations

There is no current resource estimate on the Property.

1.2.10 Adjacent Properties

Other operators in the area with a focus on uranium include Orano Canada (previously Areva Canada), which completed an initial feasibility study of the Kiggavik Uranium Deposit and

submitted a Draft Environmental Assessment Study to the Nunavut Impact Review Board in 2007 (Areva, 2008). The Kiggavik Deposit is located approximately 90 kilometres west of Baker Lake and 210 kilometers north of the Angilak Project. Following public hearings in March 2015, the Nunavut Impact Review Board (NIRB) recommended Kiggavik not be approved at that time. NIRB stated it does not intend for the project not to proceed at any time, but that it should be resubmitted when a project start date and development schedule can be provided. The federal government supported NIRB's decision (NIRB website). Orano stills retains ownership of the mining lease covering the Kiggavik deposit. The Project is owned jointly by Orano (66.2%), Denison Mines (16.9%) and Uranium Energy Corp (16.9%). The information above is sourced from Orano's 2022 Activities Report.

In 2022, Forum Energy Metals Corp. (Forum) expanded their land position around the Orano leases to encompass 95,518 ha of prospective land (Forum's website). Forum's Nunavut Uranium Project (located approximately 195 kilometres north of the Angilak Project) covers two high-grade unconformity style uranium deposits – Tatiggaq and Qavvik – and the Ayra uranium showing (Forum's website).

The Author of this report has not verified the information pertaining to adjacent properties in the area and the information is not necessarily indicative of the mineralization on the Angilak Property.

1.2.11 Interpretation and Conclusions

The Angilak Project is host to the Lac 50 Uranium Deposit. ATHA's maiden 2024 Angilak Exploration Program built upon the work completed by its predecessors and included the completion of twenty-five diamond drillholes between early June and late August for a total of 10,052 metres. The drill program was focused on the expansion of the historic footprint of mineralization along the Lac 50 Trend and the identification of uranium mineralization within new parallel mineralized trends called the Lac 48, Lac 52, and Lac 54 Trends. The Lac 48, Lac 50 (host to the Lac 50 Deposit), Lac 52, and Lac 54 Trends, as well as the untested areas between the trends, remain prospective for future drill programs.

A total of twelve holes were completed in the Lac 50 Trend, targeting expansion of uranium mineralization, for a total of 4,884 metres. All holes achieved the objective of intersecting uranium mineralization outside of the historic mineralized extents and expanding the footprint of mineralization of the known zones, along with identification of new tuff horizons.

Within the Jay4/Ray Zones, a total of four holes were completed during 2024, all successfully intersecting uranium mineralization. Notably, mineralization was intersected approximately 400 metres along strike to the west of the J4/Ray Zones.

A total of eight holes were drilled within the Western-Extension, Main Zone, and Eastern-Extension Zones, all intersecting uranium mineralization, expanding on known zones of mineralization and identifying new tuff horizons.

Thirteen holes were completed at prospective regional targets proximal to the Lac 50 Trend for a total of 5,167 metres. All holes intersected uranium mineralization, expanded on previously discovered showings, or identified prospective structures. Three prospective trends, all parallel to the Lac 50 Trend, were tested, inclusive to the Lac 48 Trend, Lac 52 Trend, and Lac 54 Trend.

Mr. Batty, the Qualified Person (QP) for this technical report provided ranges for potential uranium quantity and grade as a target for further exploration on the Lac 50 Deposit using the available verified diamond drillhole data, including the 2024 drill program results (Table 1-1). The ranges were derived from vein wireframes, drill core assays, grade interpolation, and an applied uncertainty range. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.

Table 1-1: Lac 50 Tabulated Exploration Target Model Ranges
Lac 50 Exploration Target

Cutoff (% U₃O₈)	Tonnes (Mt)	Grade (% U₃O₈)	Metal Content (Mlbs U₃O₈)
0.1	7.4 - 9.3	0.37 - 0.48	60.8 - 98.2

Notes:

1. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.
2. The ranges were derived from interpreted vein wireframes, drill core assays, grade interpolation/extrapolation, and applied uncertainty ranges.
3. An assumed cut-off of 0.1% U₃O₈ was used for the tabulation of the exploration target model.

Surficial mapping and sampling programs were also conducted as part of the 2024 exploration program, which discovered uranium mineralization on surface beyond the extents of the Lac 50 Deposit footprint. Uranium mineralization was identified between Lac 48, Lac 50, Lac 52, and Lac 54 trends. A zone of extensive bedrock outcrop with radioactivity up to >60,000 counts per second (cps), measured with a hand-held RS-120 scintillometer, was identified over a 3-kilometre strike length located between the Mushroom Lake zone on the Lac 52 trend and the Hot zone on the Lac 54 trend.

An airborne MobileMT geophysical survey was also completed within the Lac 50 Deposit area in 2024.. This survey identified an approximate 31-km prospective trend identified as the RIB-Nine-Iron corridor.

In 2025, ATHA completed its second exploration campaign on the Angilak Project which was focused on continued expansion of the Lac 50 deposit mineralization footprint at the J4/Ray Zone, testing the down-dip continuity of mineralization identified on surface at the Mushroom Lake zone, and testing of high-priority regional targets within the KU and RIB areas located within the prospective RIB-Nine Iron corridor. As part of the 2025 campaign, a total of twenty-two diamond drill holes for a total 10,774-metres of drilling was completed. To date, final drill core sample assays have not yet been received and all results are based on preliminary down-hole radiometric probe results obtained from a Mount Sopris 40TGU-1000 Triple Gamma Geiger probe.

In the Lac 50 deposit area a total of three drill holes and 1,608 metres were completed at the J4/Ray and Mushroom Lake zones. At the J4/Ray zone, drilling successfully extended mineralization down-dip by approximately 100 metres and remains open both along strike and at depth. At Mushroom Lake, two drill holes were completed to test the depth extent of uranium mineralization identified in outcrop. Both drill holes successfully intersected uranium mineralization as identified by preliminary down-hole radiometric probe results and remains open at depth and along strike.

At the KU regional target area, a total of six drill holes and 3,427 metres of drilling were designed to test stacked gravity and EM geophysical anomalies coincident with interpreted NW-SE and E-

W trending structures within the 31-km RIB-Nine Iron corridor. All drill holes successfully intersected mineralized structures with elevated radioactivity, highlighted by drill hole KU-DD-001

Within the RIB regional target area, a total of thirteen drill holes and 5,739 metres were completed to test stacked gravity and EM anomalies coincident with interpreted NE-SW, E-W and N-S trending structures along the regional RIB-Nine-Iron corridor. All drill holes successfully intersected mineralized structures associated with graphitic shears zones over an interpreted 12-kilometre conductive EM trend within the RIB area, with the most significant intersection coming from the RIB North area in drill hole RIBN-DD-001.

1.2.12 Recommendations

Based on the historical exploration work discussed in this Technical Report and the exploration programs completed by ATHA, it is the opinion of the Author of this Technical Report that the Angilak Property warrants further exploration work.

Based upon the results of exploration conducted to date, the Author recommends that the following work be completed at the Angilak Property:

- 1) Additional drilling along the Lac 48, 50, 52 and 54 Trends to further expand mineralization immediately along strike, and at depth, and along parallel and cross-cutting mineralized structural corridors identified by previous drilling,
- 2) Additional drilling within the RIB target area to continue to test the prospective 12-kilometre structural corridor following on drill results from the 2025 exploration campaign,
- 3) Exploration drilling including:
 - testing of geophysical conductors proximal to the Lac 50 Deposit, including conductors along strike that could represent extensions and parallel trends prospective to host uranium mineralization.
 - further drill testing at the KU, Nine Iron, and Dipole showings, and
 - reconnaissance drilling of additional exploration targets within the Angikuni Basin.
- 4) Further airborne geophysical surveys to help characterize, de-risk and prioritize regional targets across the Property,
- 5) Baseline environmental monitoring in support of future project evaluation studies, and
- 6) Ongoing community consultation.

A preliminary cost estimate for the recommended work to be carried out in 2026 is provided in Table 1-2.

Table 1-2: 2026 Cost Estimate for Recommended Work

Item	Cost Estimate (CDN\$M)
Geophysical Surveys	\$2.0
Drilling (10,000m) and Logistical Support	\$9.0
Baseline Environmental Monitoring	\$0.5
Community Consultation	\$0.1
TOTAL	\$11.6

2 INTRODUCTION

Understood Mineral Resources Ltd. (UMR) was retained by Atha Energy Corp. (ATHA or the Company) to prepare an independent Technical Report on the Angilak Property (the Property), located in southern Nunavut, Canada. This Technical Report has been prepared in accordance with the Canadian Securities Administration's (CSA's) National Instrument 43-101 Standards of Disclosure for Mineral Projects (NI 43-101) and guidelines for technical reporting from Canadian Institute of Mining, Metallurgy and Petroleum (CIM) "Best Practices and Reporting Guidelines".

The purpose of this report is to disclose exploration activities conducted on the Property to date as well as the exploration target model for the Lac 50 Uranium Deposit (Lac 50, Lac 50 Deposit, or the Deposit). The exploration target model represents potential uranium quantity and grade, expressed as ranges, as a target for further exploration on the Deposit. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource. The effective date of this report is November 25, 2024, coincident with the final signoff of the exploration target model.

ATHA is a Canadian exploration company, primarily engaged in the acquisition, evaluation, and development of uranium properties with a view to commercial production. ATHA is listed on the Toronto Stock Exchange Venture (symbol SASK), on the OTCQB Best Markets (symbol SASKF), and on the Frankfurt Stock Exchange (symbol X5U.F).

2.1 Qualified Person Site Visit

This Technical Report was prepared by, and in parts under the supervision of, Matt Batty, MSc, P. Geo, of UMR, who visited the Property from August 14 to 15, 2024. During Mr. Batty's site visit, he examined historic and recent drill core, confirmed collar locations, reviewed drilling, logging, sampling, analytical and QA/QC procedures, and reviewed site facilities.

By virtue of his education, membership to a recognized professional association (Association of Professional Engineers and Geoscientists of Saskatchewan), independence from ATHA, and relevant work experience, Mr. Batty is eligible to be the Qualified Person (QP) for the Project as this term is defined by National Instrument 43-101.

2.2 Sources of Information

This Technical Report is based on the following sources of information:

- Discussions with ATHA personnel;
- Inspection of the Angilak Property as detailed above and in Section 12.1;
- Drill data, geologic models, cross sections, and other geologic data that were transferred to the QP as a Leapfrog Project
- Documentation and other sources of information listed in Section 27 including from the public domain.

2.3 Effective Date

The effective date of this technical report is October 14, 2025.

2.4 List of Abbreviations

Units of measurement used in this report conform to the metric system. All currency in this report is Canadian dollars (C\$) unless otherwise noted.

Abbreviation	Description	Abbreviation	Description
a	annum	kWh	kilowatt-hour
A	ampere	L	litre
bbl	barrels	lb	pound
btu	British thermal units	L/s	litres per second
°C	degree Celcius	m	metre
C\$	Canadian dollars	M	mega (million)
cal	calorie	m ²	square metre
cfm	cubic feet per minute	m ³	cubic metre
cm	centimetre	μ	micron
cm ²	square centimetre	MASL	metres above sea level
d	day	μg	microgram
dia	diameter	m ³ /h	cubic metres per hour
DGM	discrete gaussian model	mi	mile
dmt	dry metric tonne	min	minute
dwt	dead-weight ton	μm	micrometre
°F	degree Fahrenheit	mm	millimetre
ft	foot	mph	miles per hour
ft ²	square foot	MVA	megavolt-amperes
ft ³	cubic foot	MW	megawatt
ft/s	foot per second	MWh	megawatt-hour
g	gram	OK	Ordinary Kriging
G	giga (billion)	oz	Troy ounce (31.1035g)
Gal	Imperial gallon	oz/st,opt	ounce per short ton
g/L	gram per litre	ppb	part per billion
Gpm	Imperial gallons per minute	ppm	part per million
g/t	gram per tonne	psia	pound per square inch absolute
gr/ft ³	grain per cubic foot	psig	pound per square inch gauge.
gr/m ³	grain per cubic metre	RL	relative elevation
ha	hectare	s	second
hp	horsepower	st	short ton
hr	hour	stpa	short ton per year
Hz	hertz	stpd	short ton per day
in.	inch	t	metric tonne
in ²	square inch	tpa	metric tonne per year

Abbreviation	Description	Abbreviation	Description
J	joule	tpd	metric tonne per day
k	kilo (thousand)	US\$	United States dollar
kcal	kilocalorie	USg	United States gallon
kg	kilogram	USgpm	US gallon per minute
km	kilometre	V	volt
km ²	square kilometre	W	watt
km/h	kilometre per hour	wmt	wet metric tonne
kPa	kilopascal	wt%	weight percent
kVA	kilovolt-amperes	yd ³	cubic yard
kW	kilowatt	yr	year

3 RELIANCE ON OTHER EXPERTS

For the purpose of this report, the QP relied on ATHA to provide all pertinent information concerning the legal status of ATHA and the current legal title information for the mineral claims and material environmental information that relate to the Property. The QP reviewed an internal memorandum authored by Mr. Cliff Revering, Vice President of Exploration for ATHA, titled “Angilak Project Land Status” (dated October 10, 2025) that summarizes the legal status of the Company, the land tenure status and material environmental information that relate to the Property as of the effective date of this report. The reliance is limited to these matters and applies to the relevant information summarized in Section 4 of this Technical Report.

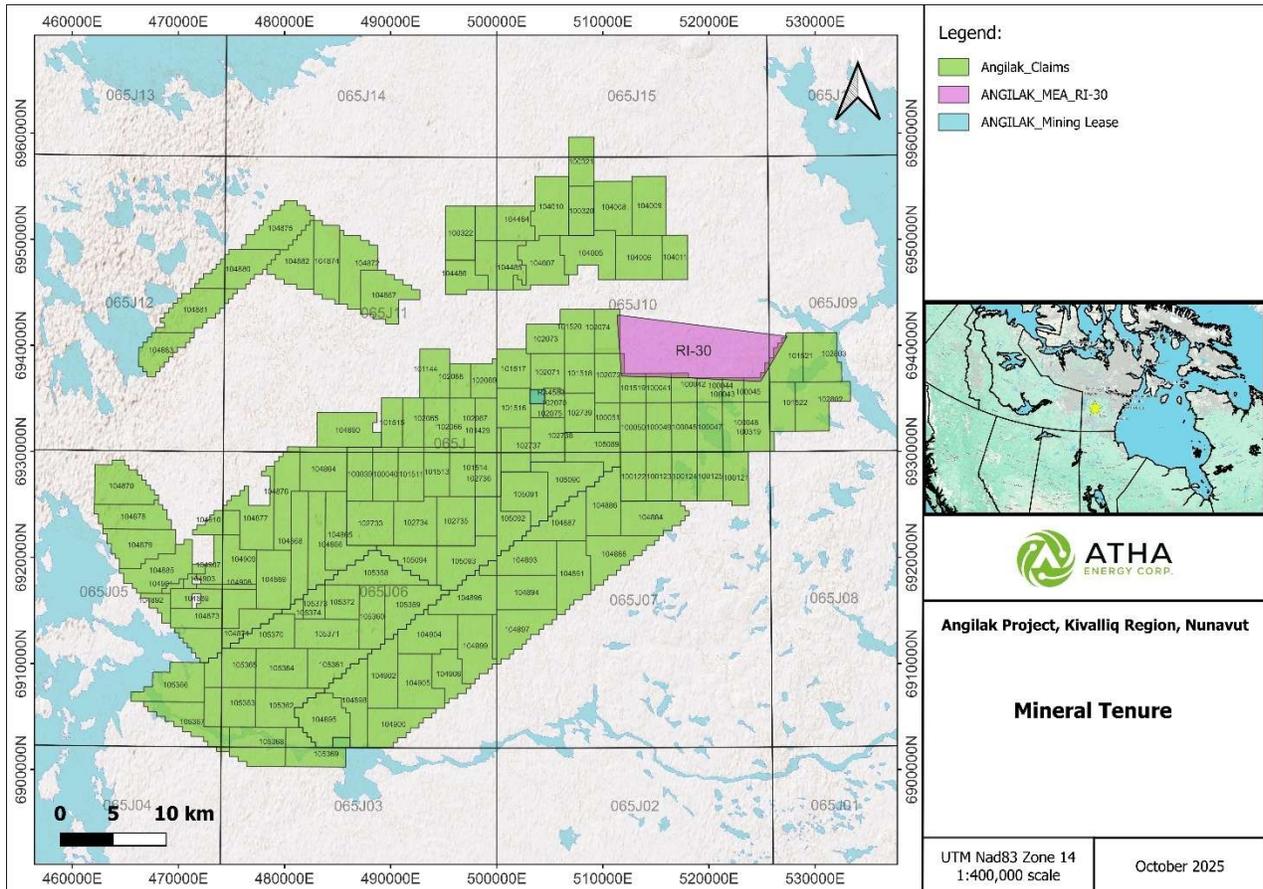
All mineral claim and mining lease information and status is available online and was confirmed on the CIRNAC website under the Nunavut Map Viewer as of the signed date of this Report by the Author.

4 PROPERTY DESCRIPTION AND LOCATION

4.1 Description and Location

The Angilak Property, which hosts the Lac 50 Uranium Deposit, is located 350 kilometres west of Kangiqiniq (Rankin Inlet) and 225 kilometres southwest of Baker Lake in the Kivalliq Region of Nunavut. The Property is bound between Latitudes 62° 14' and 62° 48' North and Longitudes 98° 21' and 99° 44' West, (North American Datum 1983 (NAD83), Universal Transverse Mercator (UTM) Zone 14 coordinates: 6901449 mN and 6960000 mN and 463830 mE to 533463 mE) and is within the 1:50:000 National Topographic (NTS) map sheets 065 J/05, J/06, J/07, J/09, J/10, J/11, J/12, J/14 and J/15. The Property comprises 135 Crown issued mineral claims and one mining lease, as well as an Inuit Owned Land (IOL) parcel (RI30-001) for a total area of 192,913 hectares. ATHA has acquired the right to conduct exploration work on the IOL parcel under a Mineral Exploration Agreement (MEA) with Nunavut Tunngavik Inc. (NTI). Land use permits enabling exploration work to be conducted on the Property have been issued, amended and renewed by the Kivalliq Inuit Association (KIA) for parts of the Property covering the IOL and by CIRNAC for the Crown Lands. Figure 4-1 depicts the general location of the Angilak Property.

Figure 4-1: Angilak Property Land Tenure



4.2 Land Tenure

The Property comprises 135 Crown issued mineral claims (185,329 ha) and one mining lease (198 ha), as well as IOL Parcel RI30-001 (7,386 ha), which is administered by NTI (Table 4-1). A portion of the Crown issued claims were registered in the name of a holding company named 5833 Nunavut Ltd. when LUR (formerly Labrador) acquired the claims in 2023. ATHA acquired LUR in April 2024 including 100% of all assets owned by LUR and its wholly owned subsidiary 5833 Nunavut Ltd. Claims staked in 2024 by ATHA are currently registered under 5833 Nunavut Ltd. whereas claims staked by ATHA in 2025 are register under ATHA Energy (NU) Corp, another wholly owned subsidiary of ATHA. In addition, the MEA with NTI for RI30-001 has been assigned from 5833 Nunavut Ltd. to ATHA Energy (NU) Corp.

Under the Nunavut Mining Regulations (NMR), the duration of a recorded mineral claim is 30 years, beginning on its recording date, plus any extensions, unless the recorded claim is taken to lease or cancelled. In order to keep a mineral claim in good standing, a holder of a recorded claim must do work that incurs a cost annually beginning on the day on which the claim is recorded for each unit (approximately 18 to 19 ha) included in the recorded claim as follows:

- \$45 in respect of the first year;
- \$90 in respect of the second to fourth years;
- \$135 in respect of the fifth to seventh years;
- \$180 in respect of the eighth to tenth years;
- \$225 in respect of each of the eleventh to twentieth years; and
- \$270 in respect of each of the twenty-first to thirtieth years.

To maintain the mineral claim in good standing a report of work (assessment report) is required to be filed within 120 days after the second anniversary of the recording of the claims or any subsequent anniversary date. Work reported in one report must have been performed within a period of not more than 12 consecutive months during the four years immediately preceding the day on which the report was submitted and after the day on which the claim was recorded. Expenditure costs are required to be filed with the assessment report along with a table setting out the cost of work (expenditure) that is allocated to each claim. The Mining Recorder will evaluate the assessment report to assess their compliance with NMR and determine the cost of work to be set out in a certificate of work. Once the expenditures are approved an allocation of work can be completed on NMS and will result in the updating of expiry dates of the claims.

An assessment report for the 2023 field program covering all the claims (including those with expiry dates in 2023) comprising the Angilak Property was filed in December 2023. The report and expenditures are currently under review. Once the report and expenditures are approved the costs will be allocated to the claims and expiry dates will be updated. The expenditures reported in the 2023 Assessment report totalled \$10,023,455.29 and are sufficient to extend the expiry dates of the claims by 6 to 10 years.

Table 4-1: Land Tenure Status for the Angilak Property

Tenure Type	Claim Number	Claim Name	OWNERS	Issue Date	Anniversary Date	AREA_HA
Claim	100039	DIP 01	5833 Nunavut Ltd. (100%)	2021-11-08	2052-11-08	1234.914
Claim	100040	DIP 02	5833 Nunavut Ltd. (100%)	2021-11-08	2052-11-08	1234.914
Claim	100041	KU 1	5833 Nunavut Ltd. (100%)	2021-11-08	2031-11-08	671.97
Claim	100042	KU 2	5833 Nunavut Ltd. (100%)	2021-11-08	2032-11-08	634.652
Claim	100043	KU 3	5833 Nunavut Ltd. (100%)	2021-11-08	2037-11-08	560.016
Claim	100044	KU 4	5833 Nunavut Ltd. (100%)	2021-11-08	2036-11-08	466.68
Claim	100045	KU 5	5833 Nunavut Ltd. (100%)	2021-11-08	2035-11-08	634.645
Claim	100046	KU 6	5833 Nunavut Ltd. (100%)	2021-11-08	2035-11-08	934.335
Claim	100047	KU 7	5833 Nunavut Ltd. (100%)	2021-11-08	2034-11-08	1121.202
Claim	100048	KU 8	5833 Nunavut Ltd. (100%)	2021-11-08	2032-11-08	1121.202
Claim	100049	KU 9	5833 Nunavut Ltd. (100%)	2021-11-08	2032-11-08	1121.202
Claim	100050	KU 10	5833 Nunavut Ltd. (100%)	2021-11-08	2031-11-08	1121.196
Claim	100051	KU 11	5833 Nunavut Ltd. (100%)	2021-11-08	2032-11-08	672.528
Claim	100121	KU 21	5833 Nunavut Ltd. (100%)	2021-11-08	2034-11-08	1197.646
Claim	100122	KU 17	5833 Nunavut Ltd. (100%)	2021-11-08	2031-11-08	1122.732
Claim	100123	KU 18	5833 Nunavut Ltd. (100%)	2021-11-08	2035-11-08	1122.738
Claim	100124	KU 19	5833 Nunavut Ltd. (100%)	2021-11-08	2035-11-08	1122.738
Claim	100125	KU 20	5833 Nunavut Ltd. (100%)	2021-11-08	2034-11-08	1122.738
Claim	100319	VK 1	5833 Nunavut Ltd. (100%)	2021-09-13	2047-09-13	1195.981
Claim	100320	TAL 2	5833 Nunavut Ltd. (100%)	2021-11-01	2030-11-01	1114.326
Claim	100321	TAL 7	5833 Nunavut Ltd. (100%)	2021-11-01	2030-11-01	1112.766
Claim	100322	VGR-5	5833 Nunavut Ltd. (100%)	2021-05-18	2053-05-18	1430.968
Claim	101144	KV 16	5833 Nunavut Ltd. (100%)	2021-09-03	2053-09-03	1306.053
Claim	101429	KV 27	5833 Nunavut Ltd. (100%)	2021-09-03	2053-09-03	1121.15
Claim	101511	ANG1	5833 Nunavut Ltd. (100%)	2021-10-26	2038-10-26	1234.914
Claim	101513	ANG2	5833 Nunavut Ltd. (100%)	2021-10-26	2049-10-26	1122.57
Claim	101514	ANG3	5833 Nunavut Ltd. (100%)	2021-10-26	2049-10-26	1122.57
Claim	101515	ANG4	5833 Nunavut Ltd. (100%)	2021-10-26	2051-10-26	934.195
Claim	101516	ANG10	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	1195.272
Claim	101517	ANG12	5833 Nunavut Ltd. (100%)	2021-10-26	2032-10-26	1175.29
Claim	101518	ANG14	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	1044.841
Claim	101519	ANG16	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	671.967
Claim	101520	ANG18	5833 Nunavut Ltd. (100%)	2021-10-26	2032-10-26	1229.923
Claim	101521	ANG22	5833 Nunavut Ltd. (100%)	2021-10-26	2029-10-26	1286.962
Claim	101522	ANG23	5833 Nunavut Ltd. (100%)	2021-10-26	2036-10-26	1120.59
Claim	102065	ANG5	5833 Nunavut Ltd. (100%)	2021-10-26	2052-10-26	934.195
Claim	102066	ANG6	5833 Nunavut Ltd. (100%)	2021-10-26	2050-10-26	1121.034
Claim	102067	ANG7	5833 Nunavut Ltd. (100%)	2021-10-26	2049-10-26	1121.034
Claim	102068	ANG8	5833 Nunavut Ltd. (100%)	2021-10-26	2033-10-26	653.165
Claim	102069	ANG9	5833 Nunavut Ltd. (100%)	2021-10-26	2035-10-26	802.449

Tenure Type	Claim Number	Claim Name	OWNERS	Issue Date	Anniversary Date	AREA_HA
Claim	102070	ANG11	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	560.205
Claim	102071	ANG13	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	1119.436
Claim	102072	ANG15	5833 Nunavut Ltd. (100%)	2021-10-26	2030-10-26	1306.117
Claim	102073	ANG17	5833 Nunavut Ltd. (100%)	2021-10-26	2032-10-26	1006.533
Claim	102074	ANG19	5833 Nunavut Ltd. (100%)	2021-10-26	2031-10-26	1006.32
Claim	102075	ANG20	5833 Nunavut Ltd. (100%)	2021-10-26	2034-10-26	168.096
Claim	102733	ANG31	5833 Nunavut Ltd. (100%)	2021-11-19	2051-11-19	1854.947
Claim	102734	ANG32	5833 Nunavut Ltd. (100%)	2021-11-19	2037-11-19	1742.482
Claim	102735	ANG33	5833 Nunavut Ltd. (100%)	2021-11-19	2048-11-19	1686.195
Claim	102736	ANG34	5833 Nunavut Ltd. (100%)	2021-11-20	2049-11-20	1010.382
Claim	102737	ANG35	5833 Nunavut Ltd. (100%)	2021-11-20	2030-11-20	1177.974
Claim	102738	ANG36	5833 Nunavut Ltd. (100%)	2021-11-20	2029-11-20	1345.888
Claim	102739	ANG37	5833 Nunavut Ltd. (100%)	2021-11-20	2030-11-20	1046.01
Claim	102802	ANG38	5833 Nunavut Ltd. (100%)	2022-02-14	2029-02-14	1867.495
Claim	102803	ANG39	5833 Nunavut Ltd. (100%)	2022-02-14	2029-02-14	1566.656
Claim	104005		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1840.848
Claim	104006		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1841.095
Claim	104007		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1841.363
Claim	104008		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1745.591
Claim	104009		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1782.712
Claim	104010		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1856.987
Claim	104011		5833 Nunavut Ltd. (100%)	2023-12-14	2026-10-10	1004.238
Claim	104484	ANG1	5833 Nunavut Ltd. (100%)	2024-04-01	2026-10-10	1764.986
Claim	104485	ANG2	5833 Nunavut Ltd. (100%)	2024-04-01	2026-10-10	1822.757
Claim	104486	ANG3	5833 Nunavut Ltd. (100%)	2024-04-01	2026-10-10	1042
Claim	104864	DA1	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1852.123
Claim	104865	ANG_CB_1	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1836.819
Claim	104866	ANG_CB_2	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1555.967
Claim	104867	KT1	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1861.642
Claim	104868	ANG_CB_3	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1649.887
Claim	104869	ANG_CB_4	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1801.624
Claim	104870	NN_01	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1871.817
Claim	104871	ANG_CB_5	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1879.555
Claim	104872	KT2	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1785.792
Claim	104873	ANG_CB_6	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1803.288
Claim	104874	KT3	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1692.484
Claim	104875		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1783.851
Claim	104876	ANG_CB_7	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1797.365
Claim	104877	ANG_CB_8	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1854.726
Claim	104878	NN_02	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1554.948

Tenure Type	Claim Number	Claim Name	OWNERS	Issue Date	Anniversary Date	AREA_HA
Claim	104879		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1874.929
Claim	104880	DA3	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1785.9
Claim	104881	DA4	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1862.483
Claim	104882	KT4	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1673.9
Claim	104883	KT5	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1230.663
Claim	104884	KT6	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1798.447
Claim	104885		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1876.255
Claim	104886	KT7	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1816.545
Claim	104887	KT8	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1648.81
Claim	104888	KT9	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1875.3
Claim	104889		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	807.417
Claim	104890	MC1	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1850.126
Claim	104891	KT10	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1876.326
Claim	104892		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	450.734
Claim	104893	KT11	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1875.608
Claim	104894	KT12	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1839.768
Claim	104895	MC2	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1884.028
Claim	104896	KT13	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1783.737
Claim	104897	KT14	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1879.275
Claim	104898	MC3	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1826.504
Claim	104899	KT15	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1880.128
Claim	104900	MC4	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1658.133
Claim	104901		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	131.398
Claim	104902	MC5	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1712.338
Claim	104903		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	131.381
Claim	104904	KT16	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1879.555
Claim	104905		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1882.073
Claim	104906	KT17	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1185.396
Claim	104907		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	112.569
Claim	104908		5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	262.794
Claim	104909	KT18	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	1857.096
Claim	104910	KT19	5833 Nunavut Ltd. (100%)	2024-09-02	2026-09-02	786.988
Claim	105089		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1570.58
Claim	105090		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1833.943
Claim	105091		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1834.687
Claim	105092		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1798.684
Claim	105093		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1875.668
Claim	105094		5833 Nunavut Ltd. (100%)	2025-03-14	2027-03-14	1256.657
Claim	105358		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1744.955
Claim	105359		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1784.088

Tenure Type	Claim Number	Claim Name	OWNERS	Issue Date	Anniversary Date	AREA_HA
Claim	105360		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1446.527
Claim	105361		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1486.087
Claim	105362		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1713.812
Claim	105363		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1808.076
Claim	105364		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1806.072
Claim	105365		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1655.626
Claim	105366		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1882.252
Claim	105367		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1658.087
Claim	105368		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1847.729
Claim	105369		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1621.688
Claim	105370		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1296.921
Claim	105371		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1691.595
Claim	105372		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	1051.6
Claim	105373		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	582.164
Claim	105374		ATHA Energy (NU) Corp. (100%)	2025-09-08	2027-09-08	225.393
Lease	RA4583	L-6247		8/29/2018	8/29/2021	198
IOL	RI30-001			4/1/2007	Annual Renewal	7386
Total Hectares						192,912.96

At any time during the life of the mineral claim, the holder may apply to convert all or a portion of the mineral claim to a mining lease, as long as a certificate of work has been issued in respect of the claim that allocates to the claim a total cost of work of at least \$1,260 per unit. No exploration work is required once the application to convert the mineral claim to a lease is filed with the mining recorder. The application to convert a mineral claim to a mining lease must be accompanied by a legal survey. No exploration is required for granted mining leases. A mining lease is normally granted for a term of 21 years and is renewable for further terms. Mining of any mineral product may only be conducted on a mining lease.

The holder of the mining lease that was issued before November 1, 2020, is required to pay an annual rental fee of \$2.50 per hectare during the first term and \$5.00 per hectare during each renewed term before that date. The annual rent for a lease that is issued on or after November 1, 2020, and for any lease that is renewed on or after that date is \$10 per hectare.

Work and fees for IOL Parcel RI30-001 are described in a Mineral Exploration Agreement (MEA RI30-001) between the Company and NTI, and are as follows:

Annual fees:

- \$1.00 per hectare in respect of the first year;
- \$2.00 per hectare in respect of the second to fifth years;
- \$2.50 per hectare in respect of the sixth to tenth years; and
- \$4.00 per hectare in respect of the eleventh to twentieth years.

Exploration Work:

- \$4.00 per hectare in respect of the first and second years;
- \$10.00 per hectare in respect of the third to fifth years;
- \$18.00 per hectare in respect of the sixth to tenth years;
- \$30.00 per hectare in respect of the eleventh to fifteenth years; and
- \$40.00 per hectare in respect of the sixteenth to twentieth years.

4.3 Mineral Rights

Obtaining surface rights is required to obtain a mining lease.

The surface rights for the 135 mineral claims and the single mining lease are owned by the Crown and administered by CIRNAC. Under the Territorial Land Use Regulations (TLUR), a Land Use Permit must be obtained from CIRNAC to conduct any work, including ground disturbing work such as drilling, mining or establishment of a camp.

The surface rights for the IOL parcel are owned by the Inuit and administered in the Kivalliq Region by the Kivalliq Inuit Association (KIA). Under the 1993 Nunavut Land Claims Agreement (the NLCA), the Regional Inuit Associations (RIAs) administer access through the issuance of Land Use Licences and Surface Leases, as well as other forms of authorization. A Land Use Licence must be obtained from the regional RIA prior to any access to an IOL.

The Nunavut Planning Commission (NPC), Nunavut Impact Review Board (NIRB) and the Nunavut Water Board (NWB) are institutions of the Nunavut government also established under the Agreement, which provide a regime for land use planning and project assessment.

Under the NLCA and the Nunavut Planning and Project Assessment Act (NUPPA) all activities that require a land or water use authorization from CIRNAC, NWB or an RIA must be submitted as a Project Proposal to the NPC to ensure conformity to the Regional Land Use Plan, if one exists, and to determine whether the activities require screening from NIRB to assess the potential environmental and socioeconomic impacts prior to approval of the required project authorizations. The NWB primary function is to license uses of water and deposits of waste within the Nunavut Settlement Area.

Any future mining on a mineral claim will require conversion to a mining lease, in addition to obtaining surface leases from CIRNAC. On the subsurface IOL Parcel, a production lease must be obtained from the KIA prior to mining.

4.4 Royalties and Other Encumbrances

The NMR employ a sliding royalty scheme that ranges from 0 to 14% of the “value” of the output of the mine, with allowable deductions including mining and processing, storage, handling and transportation, reclamation, depreciation, exploration, etc., essentially representing a “Net Profits Interest” (NPI) Royalty. This royalty will be applicable to mining on any of the Crown mineral claims or mining leases.

The IOLs are subject to an underlying 12% NPI Royalty payable on all minerals to NTI. The MEA (as defined below) requires annual exploration work to be done or payments made in lieu of work, advance royalty payments of C\$50,000/year (to be credited against the 12% NPI Royalty), and a bonus payment of C\$1,000,000 within 60 days of receipt a NI43-101 report that demonstrates a measured mineral resource of at least 12 million pounds of uranium oxide. Upon a production decision at the Angilak Property, NTI can elect to have a 25% participating interest in the Project or collect a 7.5% NPI royalty (in addition to the 12% NPI Royalty).

In 2017, ValOre granted a 1% Net Smelter Returns (NSR) Royalty to Sandstorm Gold Ltd. (ValOre News Release dated January 16, 2017) payable on all mineral products produced from the Angilak Property.

4.5 Environmental Liabilities, Permitting and Significant Factors

Physical work within the mineral claims, other than indirect (airborne) surveys, requires several permits and approvals. The mineral claims are subject to land use rules administered by CIRNAC on behalf of the Federal Government. The 1993 NLCA gave Inuit title to 356,000 km² of land. IOLs comprise several parcels for which Inuit hold surface and/or subsurface title. Work within IOLs requires notification of the applicable RIA. In the case of the Angilak Property and IOL Parcel RI30-001, ATHA must obtain and hold land use licenses issued by the KIA. To conduct any surface disturbances including trenching, drilling and mining or to construct a camp, appropriate land use permits are required. The KIA administers the surface rights on behalf of the Inuit people. NTI administers the subsurface rights for IOL Parcel RI30-001 and has a MEA in place with ATHA.

Below is a list the active permits and licences issued for exploration activities on the Angilak Property. A Nunavut Water Board (NWB) licence authorizes ATHA's water use on the Property.

Issuer/Agency, Date Issued, File Number

- KIA, July 21, 2023, KVL308C09
- NIRB, July 31, 2008, 08EN052
- CIRNAC, July 23, 2023, N2019C0013
- NWB, April 12, 2022, 2BE-ANG2227

Currently, there are a number of 45-gallon drums (370) that contain drill cuttings from the prior drilling campaigns and are stored in a containment storage area west of the main Angilak (Nutaag) camp. Most of these drums contain non-radioactive cuttings or background radioactivity and will need to be disposed of in a local sump. There are several drums (estimated at fifteen) that contain radioactive drill cuttings. These drums will need to be eventually removed and disposed of in a government approved facility.

The Author is not aware of any environmental liabilities to which the Property may be subject. The Author understands that ATHA has yet to perform any ground disturbance work other than drilling and to the Author's knowledge, there is no significant historical work which would result in any environmental liabilities on the Property.

The Author is not aware of any other significant factors or risks that would affect access, title, or the ability to perform work on the Property.

5 ACCESSIBILITY, CLIMATE, LOCAL RESOURCES, INFRASTRUCTURE AND PHYSIOGRAPHY

5.1 Accessibility

The Angilak Project is located 350 kilometres west of Kangiqliniq (Rankin Inlet) and 225 kilometres southwest of Baker Lake in the Kivalliq Region of Nunavut. Access to the Property is reliant on helicopters and fixed wing aircrafts. There is a gravel airstrip located at the Nutaaq camp and a secondary gravel airstrip 1.5 kilometres west of the Nutaaq drill camp. Exploration at the Property is typically conducted between the months of February and October. Local access to and around the Project site is by either helicopter, float plane or wheeled fixed wing aircraft such as a Single Otter. Commercial-grade airports in Baker Lake and Rankin Inlet are used as mobilization points for supplies and people. Required infrastructure for exploration can be brought in each field season with a Single Otter typically available in Baker Lake.

5.2 Climate

The climate is best described as continental-arctic with short cool summers and long cold winters with minimal precipitation. Average summer high temperatures can reach up to 20°C, while average winter temperatures are in the order of -30°C to -35°C. Snow is generally on the ground until the first week of June and ice does not leave the mid-sized lakes until the third week of June. Nearby Yathkyed Lake has ice cover usually until early or mid-July. Smaller lakes freeze over around the end of September. Therefore, most of the year the Angilak Project is covered with snow, except between June and the end of September. Permafrost is present from one metre to unknown depths in mid-summer. The thawed active layer is thick enough by mid to the end of June to allow till sampling and induced polarization surveys. Diamond drilling to 200 metres depths can be accomplished without salt or propane based upon experience.

5.3 Local Resources and Infrastructure

There is no permanent infrastructure on the Property, however the Nutaaq camp is a winterized semi-permanent camp that can operate most of the year. There is an esker airstrip located at the Nutaaq camp and a secondary esker airstrip located approximately 1.5 kilometres west of the Nutaaq camp. Exploration at the Property is typically conducted between the months of February and October. Local access to and around the Project site is by either helicopter, float plane or wheeled fixed wing aircraft such as a Single Otter. Due to the commercial-grade airport and the relatively close distance, Baker Lake, Rankin Inlet and/or Arviat are the logical mobilization points for all supplies and people. All required infrastructure for exploration can be easily brought in each field season as there is usually a Single Otter available in Baker Lake or Rankin Inlet. The gravel airstrip at Baker Lake is roughly 1,279 metres in length and is regularly serviced by commercial airlines. Most supplies and materials required to conduct basic exploration programs can be obtained in Baker Lake and what cannot be immediately procured can be brought in by barge or by cargo aircraft to Baker Lake. During the winter months a “cat train” service operating in Baker Lake offers overland freight haulage of bulk loads, fuel and equipment on cargo sleds.

Access to water for drilling and camp use is readily available across the Property from abundant glacial lakes and ponds. All required power for the Nutaaq camp and drilling is supplied by diesel generators. All drilling waste is stored onsite until it can be shipped out as backhaul loads to Baker Lake and then onward to Yellowknife or Quebec for proper collection and disposal. During the Author’s Property visit, the camp and drill sites, drill cuttings storage sites, and fuel storage sites were observed to be clean, properly bermed where required, and generally in an orderly state.

The Angilak Project lies approximately 225 kilometres southwest of Baker Lake and 325 km southwest of the tidewater of Rankin Inlet in the Kivalliq Region of Nunavut. Both Baker Lake and Rankin Inlet receive shipped and barged supplies during August through to the end of October once the sea is free of ice. Shipping is generally out of Montreal, QC or out of Churchill, MB. The deep-water port of Churchill is 260 kilometres to the southeast of Arviat and is connected to southern Canada via rail. Barging directly from Churchill, MB to Baker Lake, Rankin Inlet and Arviat can be conducted from July to October.

Most field exploration activities can be conducted year-round, although there may be periods from December to March, where snow conditions and temperatures may temporarily impede work. Sufficient water for exploration is available via local sources. The surface rights are a combination of Federal Government ownership and Inuit ownership.

There are no other significant factors or risks that the Author is aware of that would affect access or the ability to perform work on the Property.

5.4 Physiography

The Property is situated in the "barren lands," a large region of almost flat, treeless tundra characterized by poor bedrock exposure and extensive swampy areas with abundant small, shallow lakes. Elevation at the Property ranges from 150 to 250 metres above sea level (asl). Locally maximum relief ranges from 30 to 75 metres but is more commonly less than 20 metres. Glacial deposits in the area are extensive thus limiting rock exposure to less than a few percent of the total Property area.

6 HISTORY

Numerous polymetallic showings and one uranium deposit have been discovered in the project area by various exploration companies since the 1960's. To date, most showings occur close to the northern boundary of the Angikuni sub-basin, within both Archean basement and younger overlying basin-fill material. The high concentration of showings proximal to the unconformity between the basement and Proterozoic Angikuni sub-basin is partially due to a high volume of exploration targeting unconformity-related uranium, which is ideally applicable to this area (Jefferson et al., 2007). This was the model used by previous exploration companies in the late 1970's, and much of the mineralization noted to date, including the Lac 50 Deposit, relates to this model. However, many of the showings, particularly within the basin, have significant amounts of copper (Cu) and silver (Ag). Miller (1993) suggested a red bed copper mineralization model to explain this mineralization. More recently, companies such as Western Mining Corporation (WMC), Kaminak, Kivalliq Energy, and ValOre have suggested that the iron oxide copper gold (IOCG) deposit model is a possible explanation for some of the polymetallic showings. The historical claims of the area largely overlap with the ATHA claims, and, to the knowledge of the Author, the described work in this section only pertains to the claims currently held by ATHA.

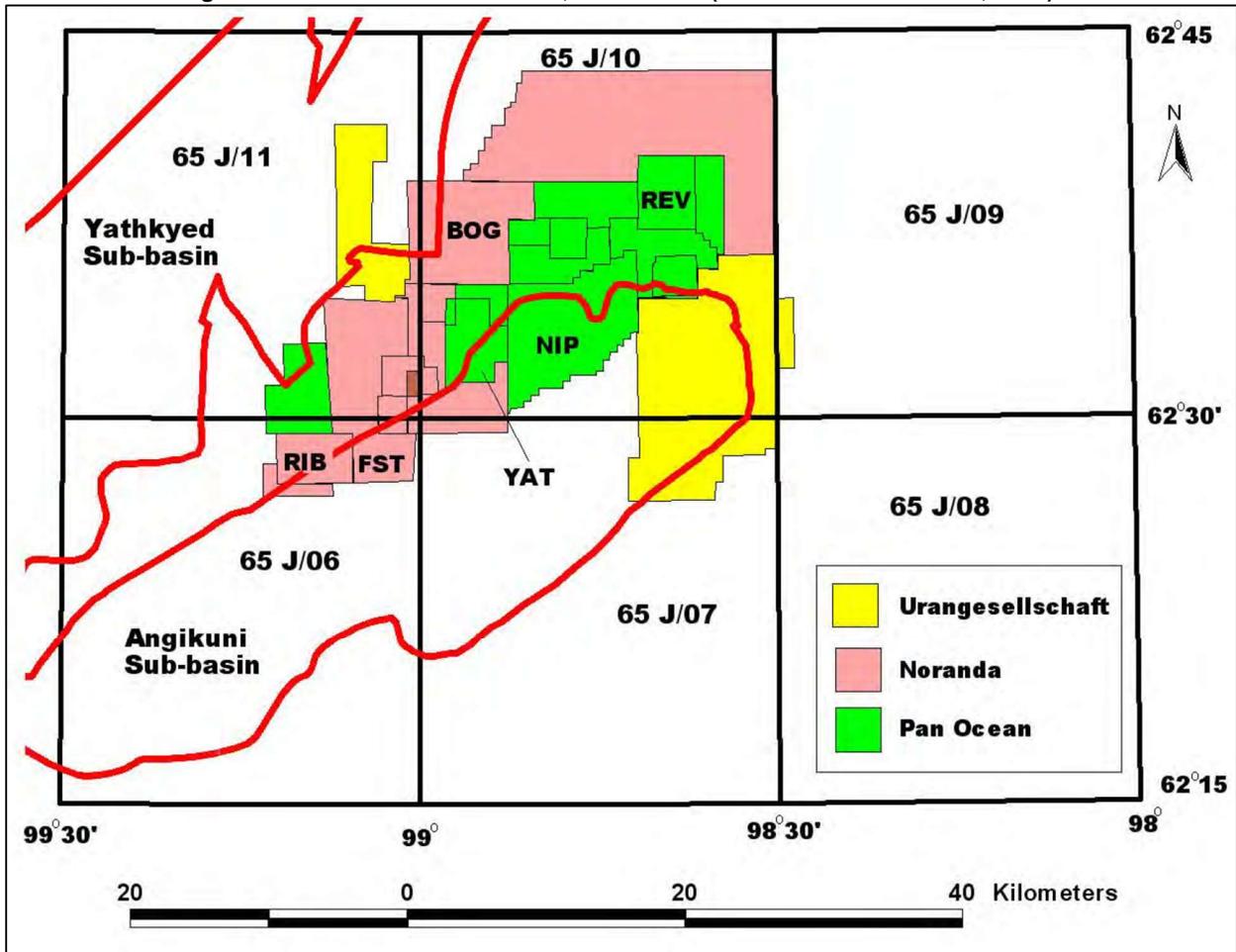
All drilling by previous companies on the Property from 1970's to 2023 is described in Section 10.

6.1 Early Exploration (1970-1982)

Previous exploration in the area is summarized below and in Table 6-1, highlighting the most relevant historical exploration, organized by company and year. Report numbers refer to numbers given to each assessment report by CIRNAC. The earliest historical exploration was completed between 1970 and 1981 and was concentrated along the northern margin of the Angikuni sub-basin as shown by the historical mineral claim position for the late 1970's (Figure 6-1). As historical property boundaries are not the same as the current property boundaries, some of the historical work completed during this timeframe may fall outside of the current margins of the Property. During this time period the most important exploration was completed by Urangesellschaft, Noranda and Pan Ocean (later Aberford Resources). The Lac 50 Uranium Deposit was discovered by Pan Ocean, but there is very little documentation or data that exists and is publicly available for this work completed on the deposit. Previous exploration by other companies in the region is described further in Dufresne and Shoeman (2024).

Exploration for uranium ceased abruptly at Lac 50 and the surrounding area when Pan Ocean divested its uranium projects in 1982. This was in large part due to accidents at the Three Mile Island Nuclear Power facility in 1979 and at Chernobyl in 1986 combined with the decline in oil prices during the mid 1980's. These events had a strong negative impact on uranium consumption and kept prices below US\$10 per pound throughout the 1980's, which curtailed global exploration and development.

Figure 6-1: Historical Land Tenure, Late 1970's (Dufresne and Shoeman, 2024)



Note: The sub-basins are demarcated by the red outline in the figure.

Table 6-1: Summary of Early Exploration (1970 - 1982)

Company	Years	Type of Work Conducted	Assessment Report #
Bluemont Minerals	1970	Airborne scintillometer survey, hydrogeochemical survey and minor mapping.	60294
Shell Minerals	1976	Prospecting.	80653
Comaplex Resources	1978	Regional prospecting, airborne radiometric survey, prospecting, mapping, VLF, lake bottom and water surveys.	81292
Essex Minerals	1976-1979	Geological, minor trenching, soil and water geochemical surveys and ground radiometric surveys. IP/EM/emanometer surveys. Mapping and diamond drilling.	080661, 081087
Urangesellschaft	1975-1981	Lake sediment and water survey, prospecting/mapping, soil sampling, scintillometer survey, chip sampling, trenching and ground magnetics. VLF, IP and Max-Min surveys. Diamond drilling and minor gravity surveying.	080810, 080619, 062011, 080977, 080981, 081091, 081451
Noranda Exploration	1975-1980	Airborne radiometric, magnetic and VLF-EM surveys. Mapping, prospecting, lake sediment sampling, soil sampling and radon emanometer surveys. Diamond drilling, ground magnetics, VLF and IP surveys.	080152, 080659, 080725, 080926, 080990, 081173, 081066
Pan Ocean	1975-1981	Airborne radiometric/magnetic/VLF survey, mapping, ground radiometric/magnetic/EM surveys, sampling, soil surveying, prospecting, diamond drilling, frost boil geochemistry survey, lake sediment sampling and water survey.	080598, 080597, 080618, 061692, 061562, 080714, 061814, 061815, 080945, 081075, 081072, 081082, 081368, 081358, 081387, 081433, 081453, 081361, 080715

6.2 Historical Exploration (1993-2007)

In 1993, Nunavut Tunngavik Incorporated (NTI) was formed to manage land and implement the Nunavut Land Claims Agreement (NLCA), which itself was established in 1993. Along with the formation of the territory of Nunavut in 1999, came the establishment of 37,000 km² of subsurface land parcels of Inuit Owned Land, including IOL Parcel RI30-001, which is situated over the historic Lac 50 Uranium Deposit.

In 1993 and 1994, Royal Bay/Leeward Capital/Taiga Consultants completed geological mapping, ground magnetics and heavy mineral sampling of areas targeted as possible kimberlite pipes (Assessment Report # - 083221, 083235, 083288, 083287).

In 1995, WMC completed mapping, ground magnetic/gravity surveys, diamond drilling and lakeshore/till/stream sediment sampling (Assessment Report # 083221, 083235, 083288, 083287).

In 2007, NTI announced its new pro-uranium policy and expressed interest in forming a partnership with exploration companies to conduct uranium exploration on IOL parcels in Nunavut. That same year, NTI and Kaminak signed a landmark uranium partnership to explore IOL parcel RI30-001 and Kaminak's surrounding federal mineral claims (Dufresne and Shoeman, 2024). This led to the creation of Kivalliq Energy (renamed ValOre in 2018) as a spin out company of Kaminak in 2008, formed with the express purpose to explore and advance the Angilak Project.

In 2007, Kaminak commissioned GeoVector to conduct a detailed compilation followed by a field program based on the results of this compilation (Setterfield, 2007). Kaminak's in-house technical team, along with GeoVector personnel, conducted geological mapping, prospecting and field verification of historical work, including verifying historical trench and drilling locations (Setterfield, 2007). APEX Geoscience Ltd. personnel were contracted by Kaminak and conducted a follow-up property visit later the same season, and between the two field programs, a total of 26 rock grab samples were collected from several historical showings (Dufresne and Shoeman, 2024).

Although the work completed by Kaminak personnel was reconnaissance in nature it confirmed and demonstrated the potential for several styles of uranium mineralization that could be related not only to unconformity and vein-type uranium models but potentially also to IOCG style mineralization. Rock grab samples collected by Kaminak personnel yielded assays of up to 0.87% U₃O₈, 2.45% Cu, 31.9 grams per tonne (g/t) gold (Au) and 1,170 g/t silver (Ag) within Angikuni sub-basin sedimentary rocks just above or adjacent to the basal unconformity along the northwestern margin of the Angikuni sub-basin. Kaminak personnel visited the historic Lac 50 Deposit area as well, where several outcrops were noted to yield significant radioactive readings.

6.3 Kivalliq Energy Corp. Exploration (2008 to 2018)

Between 2008 and 2016, exploration work on the Property by Kivalliq Energy was consistent and included multiple ground geophysical surveys (gravity, magnetics and VLF), airborne geophysical surveys (TDEM, magnetics, radiometrics, VLF-EM, and VTEM), diamond drilling, reverse circulation (RC) drilling, soil sampling, rock sampling, geological mapping, and prospecting. Exploration carried out is summarized below and described in more detail by Dufresne and Shoeman, 2024.

In 2008, exploration consisted of airborne and ground geophysical surveying, prospecting, rock sampling, and confirmation of historical drill collar locations. A combined magnetic, electromagnetic (EM) and radiometric AeroTEM III airborne geophysical survey was completed over the Property in May 2008. Magnetic (MAG), radiometric, and very low frequency electromagnetic (VLF-EM) ground geophysical surveys were completed on the Property. A field work program was completed with the objective of verifying and expanding information on several historical showings and drilling locations across the Property. During this program, 130 rock grab and historical drill core samples were collected, and the collar locations for 123 historical drillholes were verified.

The 2009 exploration program on the Property consisted of ground geophysical surveying, a diamond drill program, and the re-logging of historical drillholes. MAG and VLF-EM ground geophysical surveys were completed. The surveys resulted in the identification of a 9-kilometre-long trend of parallel VLF-EM conductors that are associated with the Lac 50 Uranium Deposit. The drill program results showed that the "Main Zone" of uranium mineralization is relatively predictable, dipping approximately 70° degrees to the south with a strike of 116°.

In 2010, exploration work completed included geochemical rock sampling, diamond drilling, environmental baseline monitoring, and the construction of the Nutaaq camp. The diamond drill program targeted the Lac 50 Deposit area.

During 2011, airborne and ground geophysical surveys, rock and soil sampling programs, diamond drilling, reconnaissance RC drilling, as well as continued environmental monitoring were completed on the Property. A helicopter mounted DIGHEM MAG, frequency domain EM, and radiometric survey was completed. The survey defined major conductive trends on the Property. A two-phase gravity ground survey program at seven major target areas on the Property was completed to aid drill planning. Weak to moderate gravity lows were observed at the VGR northeast, Yat and IM76 target areas, while the MM64 grid showed no anomalous results. The gravity results for the IM76 and VGR grids indicated potential for unconformity associated clay alteration and uranium mineralization. The Yat grid yielded a weak gravity anomaly associated with a conductive fault zone. Follow-up RC drilling on the “bullseye” gravity low at VGR proved that the anomaly was caused by clay alteration of bedrock. MAG and VLF-EM ground surveys at 24 target areas on the Property was conducted. Grids surveyed during the ground geophysical program yielded VLF-EM conductors of interest with at least minor uranium mineralization on surface apart from one or two conductors. The only new conductor identified by the survey was spatially associated with the AG Showing. The aim of the 2011 sampling program was to discover new mineral occurrences, to revisit areas of interest identified by the 2010 field program and to identify mineralization and geological trends on the Property. The rock sampling program identified the Nine Iron, Dipole, and Ag showings on the Property.

Exploration work completed during 2012 included geophysical surveys, prospecting, geological mapping, diamond and RC drilling, rock sampling, and continued environmental baseline monitoring. Condor Consulting Inc. (Condor) was contracted in spring 2012 to complete a review of all previous geophysical data to aid in the planning of additional geophysical surveys in 2012. The ground geophysical surveys completed on the Property included a gravity survey, capacitively coupled resistivity (OhmMapper), magnetics, VLF-EM, and multi-channel radiometric surveys. The gravity method was used to test the detection of anomalies due to density variations of rock types that contain uranium mineralization, clay alteration or fault zones. Weak gravity anomalies were identified at the Yat target grid that are associated with a conductive fault zone. The purpose of the magnetics, VLF-EM and resistivity (OhmMapper) surveys was to supplement previous work and better define subsurface conductors and magnetic bodies as priority target areas. Multi-channel ground radiometric surveys were completed as well with the purpose of testing the potential to identify subtly elevated background radioactivity. Overall, the results of the radiometric survey were deemed to be ineffective or inconclusive at most targets, with some moderate to strong anomalies being identified at the Forte and Nine Iron targets in association with known outcrops and structures with uranium mineralization.

A helicopter supported geological mapping and prospecting program was also carried out. The goal of the geological mapping program was to examine in detail the areas of interest identified during 2010 to 2011 prospecting programs, to follow up on geophysical anomalies (resistivity, VLF-EM, EM, gravity) identified by airborne and ground surveys completed between 2008 and 2012 and to produce a new geological map of the area relating known mineral showings to geological features such as faults, shear zones or specific rock units. The geological mapping program resulted in the identification of three structurally and lithologically distinct domains in the Property area: the Central/Western Gneissic Belt, the Volcanic Block, and the Southeastern Compressive Zone. Within the Central Gneissic Belt, mapping identified several slices of metavolcanic rocks, which are a part of the Archean Henik Group. The Dipole target occurs within one of these greenstone belts. The geological mapping program focused on the J4-Ray area of

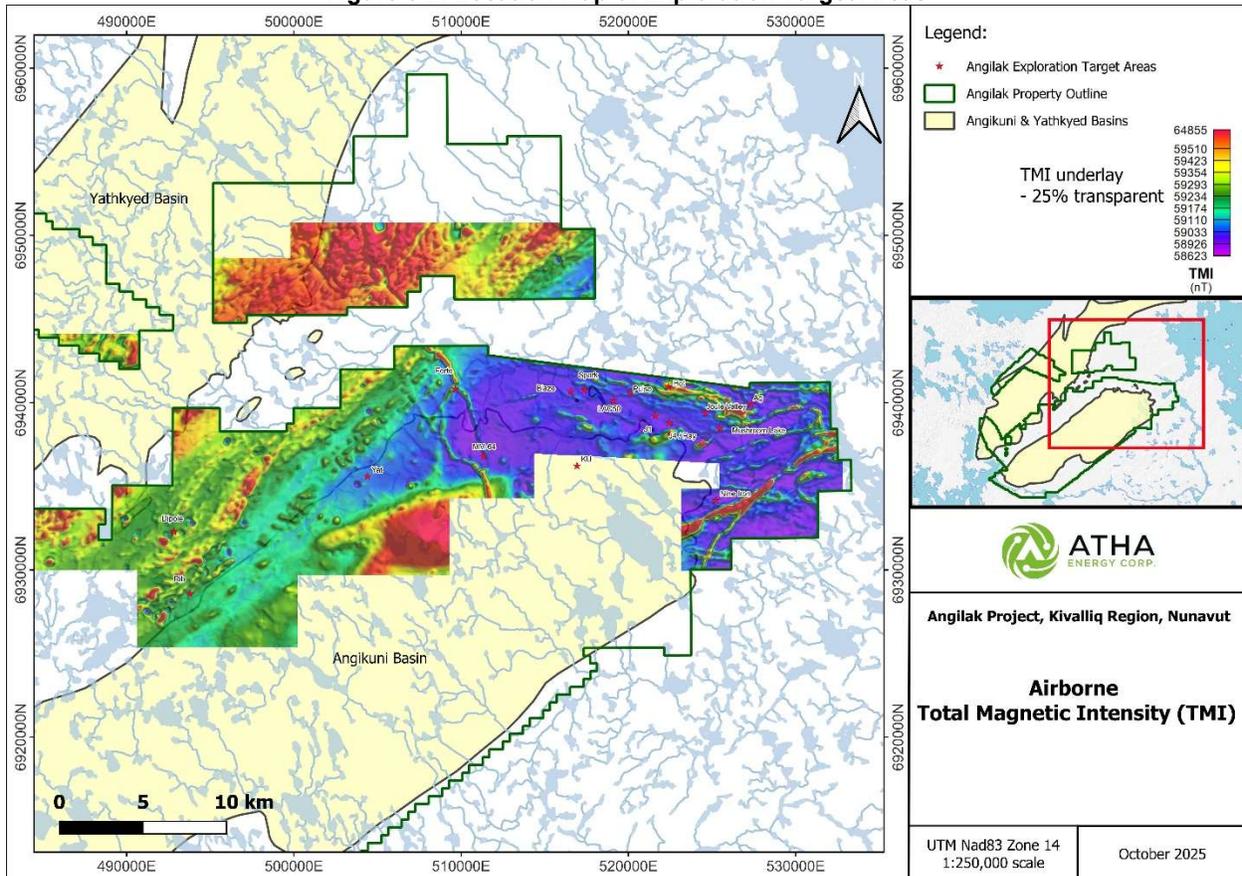
the Lac 50 Deposit within the Volcanic Block, which has relatively good outcrop exposure. This detailed mapping resulted in the better understanding of the structures and mineralized vein systems present at the J4-Ray area. The Southeastern Compression Zone, which hosts the Nine Iron showing is located to the southeast of the Volcanic Block. The geological mapping program identified strong compressional fabrics through the area, which are interpreted as being a result of Proterozoic deformation. At the Nine Iron showing, several rock grab samples returned significant Au values of up to 14.4 g/t Au. A total of 95 rock grab samples were collected from in-situ bedrock as well as from cobbles and boulders found in glacial till. Samples were sent for multi-element geochemical assay and whole rock lithochemical characterization. A total of 19 samples returned assays more than 0.1% U_3O_8 , with many yielding significant concentrations of Ag, Cu, Mo, Pb and Zn. Samples with assays more than 0.5% U_3O_8 were obtained from the target areas: J4, Nine Iron, and Yat. The diamond drill program targeted the Lac 50 Main Zone, the J4/Ray zone, the Pulse zone, and the Nine Iron zone, and RC drilling targeted areas with geophysical or geochemical anomalies identified in previous exploration programs.

Exploration during the 2013 field season included ground geophysical surveys, prospecting, soil sampling and diamond drilling. A soil sampling program was carried out with the objective of identifying surface anomalies relating to bedrock conductors to guide further drilling, as well as to test the effectiveness of the Enzyme Leach analytical method. Samples were collected for Enzyme Leach analysis, focusing on the Lac 50 Deposit area, the KU target, and the Nine Iron trend. The KU target is located within the Proterozoic Angikuni Basin south of the Lac 50 trend. The KU soil sampling grid identified an anomalous uranium in soil trend over an area of historical trenches that were dug in the 1980's. The soil sampling program at the Nine Iron trend identified multiple uranium-in-soil anomalies. Most of these uranium-in-soil anomalies are spatially correlated with the northeast-southwest oriented geophysical signature, identified in the April 2013 ground geophysical program, that strikes beneath the Angikuni basin in the Nine Iron trend area. At the KU target area sample highlights included two samples that returned 12,800 ppm U and 9,480 ppm U respectively. The 2013 soil sampling program primarily focused on the Lac 50 trend, where abundant uranium-in-soil anomalies were identified that correlate well with known mineral showings and associated geophysical signatures. At the Hot trend, a significant and broad uranium-in-soil anomaly of up to 2,880 ppb U was identified. In addition, the soil sampling program identified a uranium-in-soil anomaly that extends along the J4 VLF-EM northwest-southeast oriented conductor. The soil sampling program was successful in confirming the effectiveness of the enzyme leach method, as well defining anomalous uranium-in-soil trends associated with geophysical conductors in the Lac 50 deposit area. Several ground geophysical surveys to expand previous survey grids in areas of interest were completed. OhmMapper data was collected at the Lac 50 and KU grids. MAG and VLF-EM survey data were acquired from grids over the KU, Nine Iron and Dipole trends. A total of 9.5 kilometres of Extremely Low Frequency (ELF) data was collected in the Lac 50 trend area. The diamond drilling program targeted the J1 Zone (an approximate one-kilometre-long VLF-EM conductor that is located 800 metres to the west of the J4 deposit), Mushroom Lake (ML) including the ML "EM" conductor, and J4 West Zone

In 2014, a VTEM+ airborne geophysics survey and a soil sampling program was conducted (Figure 6-2). The goal of the soil sampling program was to identify anomalies below surface overburden using the enzyme leach analytical method. Soil samples were collected from sampling grids over multiple target areas with significant uranium-in-soil anomalies identified at the Dipole, RIB, Hot, KU, and Nine Iron trends. This sampling program successfully identified a several kilometre long uranium-in-soil trend over the Dipole target that coincides with a northeast-southwest trending electromagnetic (EM) conductor. A uranium-in-soil anomaly at the RIB trend was also confirmed to coincide with a linear EM conductor trend. Geotech Airborne Geophysical Surveys (Geotech) completed an airborne VTEM survey on two grids over the Dipole-RIB trend

and the KU-Nine Iron area. The survey successfully identified several large conductors and EM anomalous zones at the Dipole and RIB trend, which were subsequently confirmed by the enzyme leach soil sampling program that followed, and identified anomalies at the KU-Nine Iron trend zone. The results of the 2014 airborne VTEM survey were integrated with previous survey results obtained to produce a Total Magnetic Intensity Map covering most of the Property.

Figure 6-2: Location Map of Exploration Target Areas



Exploration work completed in 2015 included soil sampling and prospecting, and diamond drilling. A sampling program over the RIB and Yat target zones, which targeted historical showings, explored for new occurrences of uranium mineralization, and followed up on anomalies identified by previous geophysical surveys. The 2015 soil sampling program at RIB in-filled and extended the previous sampling grid from 2014. This expansion better delineated the several kilometre long uranium-in-soil anomalies spatially associated with the northeast-southwest oriented EM conductors identified by previous airborne and ground geophysical surveying. During the prospecting program at the Dipole-RIB trend, a rock grab sample from within the RIB soil grid returned 6.27% U_3O_8 , 0.26% Cu, 1.16% Mo, and 144 g/t Ag. Another sample returned 0.76% U_3O_8 , 0.30% Mo, and 14.9 g/t Ag and was sampled approximately 5 kilometres southwest along strike of the Dipole trend. Three boulders at Yat were sampled in 2015, and two returned significant polymetallic results including 1.82% U_3O_8 , 6.8 % Cu, 211 g/t Au, and 80,900 g/t A and 7.07 % U_3O_8 , 1.68 % Cu, 0.5 g/t Au, and 244 g/t Ag. A small enzyme leach soil sampling program was completed over the Yat area, successfully identifying a uranium-in-soil anomaly, confirming the mineralized grab sample from 2011. The 2015 diamond drill program focused on the Dipole target, with the objective of testing a prominent VLF-EM conductor and coincident uranium-in-soil anomaly, successfully delineating a wide area of steeply dipping zones of mineralization.

In 2016, exploration work included a soil sampling program, heavy mineral sampling, trenching, channel sampling, as well as rock sampling. A soil sampling program targeting the Yat and Dipole zones was conducted. The results from Yat greatly expanded on the previous sampling done in 2015 showing uranium-in-soil anomaly trends coincident with northeast-southwest oriented EM conductors that transect the Yat area, with enzyme leach samples returning up to 129 ppb U along with minor Ag anomaly trends also identified along the same conductor. Additional samples were collected from the Yat for conventional geochemical analysis, which also highlighted uranium-in-soil anomalies where soil samples returned up to 269 ppm U. Rock sampling within the soil sampling grid over the strong magnetic low zone returned multiple anomalous geochemical assay results for U and Ag including 26,000 ppm U and 3200 ppm Ag, as well as 201,000 ppm U and 358 ppm Ag.

The 2016 soil program at the Dipole trend was designed to extend upon the 2014 enzyme leach sampling grid. Results of the soil program identified uranium-in-soil anomalies northeast of the 2014 soil sampling grid, expanding the uranium-in-soil anomaly zone at Dipole to over approximately 3.5 kilometres. This new extended uranium anomaly overlays the central Dipole EM conductor, as well as overlaying a parallel EM conductor approximately 1.5 kilometres to the east. The trenching program involved the re-trenching of 3 historical trenches and the digging of eight new trenches in the Yat area. A total of 49 channel samples were also collected from the trenches for geochemical analysis. Radioactive, brecciated carbonate veining with sulphides, secondary yellow uranium staining and malachite was identified in several trench areas. Mineralization occurs as 1.0- to 1.5-metre-wide structural zones of narrow veins and stringers in sandstone, conglomerate and Christopher Island volcanics of the Proterozoic Angikuni Basin, striking northeast and parallel to the larger Yat EM conductor. Veining, adjacent wall rocks, and mineralized boulders encountered while excavating were sampled. In addition to the trenching program, rock and soil sampling were completed at the Yat target area. The heavy mineral sampling program ran concurrently with the trenching program, and 39 till samples were collected with the purpose of testing the heavy mineral expression in tills down ice of circular magnetic signatures to determine if they could be kimberlitic in origin.

No work was completed by Kivalliq Energy Corp. in 2017.

A summary of the specific work (excluding drilling) completed on the project by year can be found in Table 6-2.

Kivalliq Energy announced in 2018 that the company's name was changed to ValOre Metals Corp.

Table 6-2: Summary of Kivalliq Activities (excluding drilling) between 2008 and 2016

Year	Geophysics	Other
2008	5,620 line-km of airborne TDEM, magnetics and Radiometrics 140 line-km of ground magnetics, radiometrics and VLF-EM	Property wide prospecting and mapping
2009	621,2 line-km of magnetics and VLF-EM (over IOL parcel RI30-001)	Relogging of historic core
2010		290 outcrop and glacial float samples, baseline monitoring, Construction of Nutaaq camp
2011	5,470 line-km airborne EM-Mag 1,605 station ground gravity 1,597.5 line-km magnetics and VLF-EM	273 rock grab samples 348 soil samples
2012	309 line-km magnetics, VLF-EM and resistivity 2,556 station ground gravity 196 line-km ground radiometrics	Geophysics compilation Geological mapping 95 rock grab samples
2013	591.6 line-km resistivity 300.9 line-km of magnetics and VLF-EM 9.5 km of ELF-EM	1,538 soil samples
2014	1344.2 line-km VTEM+	1,514 soil samples
2015		Prospecting 408 soil samples
2016		504 soil samples trenching, till sampling

6.4 ValOre Metals Corp. Exploration (2018 to 2022)

No work was completed by ValOre between 2018 and 2021.

Exploration work completed on the Property in 2022 included geophysical surveys, a soil sampling program, RC drilling and diamond drilling.

A soil sample program was conducted across three priority targets: Lac 50 East grid, Dipole grid and the Noranda East traverse, totaling 896 samples. The goal of the soil geochemical survey was to classify and prioritize bedrock conductors for drilling by identifying conductors with associated surface geochemical anomalies. Results of the sampling program highlighted several uranium-in-soil anomalies within the Lac 50 and Dipole grids. Five samples collected from the far southwest extent of the Noranda East traverse line also returned anomalous uranium values. Ground magnetics and VLF-EM surveys over several priority grids was also conducted across the Property. The RC drilling program focused on the Dipole, J4 West and Yat zones, and the diamond drillhole program focused on the Dipole and J4 West zones. The 2022 diamond drilling results strengthened the interpretation that Dipole is geologically similar to the Lac 50 Deposit area (Section 10).

6.5 Latitude Uranium Inc. Exploration (2023)

Exploration work completed on the Property by Latitude Uranium Inc. (LUR) included a high resolution radiometric and aeromagnetic airborne survey during the spring of 2023 and diamond drilling completed in the summer of 2023 (Dufresne and Shoeman, 2024).

A low-level, tight drape, high resolution radiometric and aeromagnetic airborne survey was flown totalling 10,856-line kilometres over the portion of the Property covered by previous VLF-EM surveys (Figure 6-3). The goal of the survey was to identify new targets inside and outside the Lac 50 Trend and to assess the correlation with the existing soil sampling for future program planning.

Interpretation of the survey identified the most prospective conductors coincident with structures having the potential to host uranium mineralization both within and outside of the Lac 50 Deposit area (Figure 6-4). Numerous high priority areas were identified that showed good correlation with previous soil survey results and were deemed high priority drill targets. The survey also highlighted new prospective areas requiring follow up surface sampling and mapping.

Figure 6-3: Airborne Radiometrics and Magnetometer Survey Coverage 2023

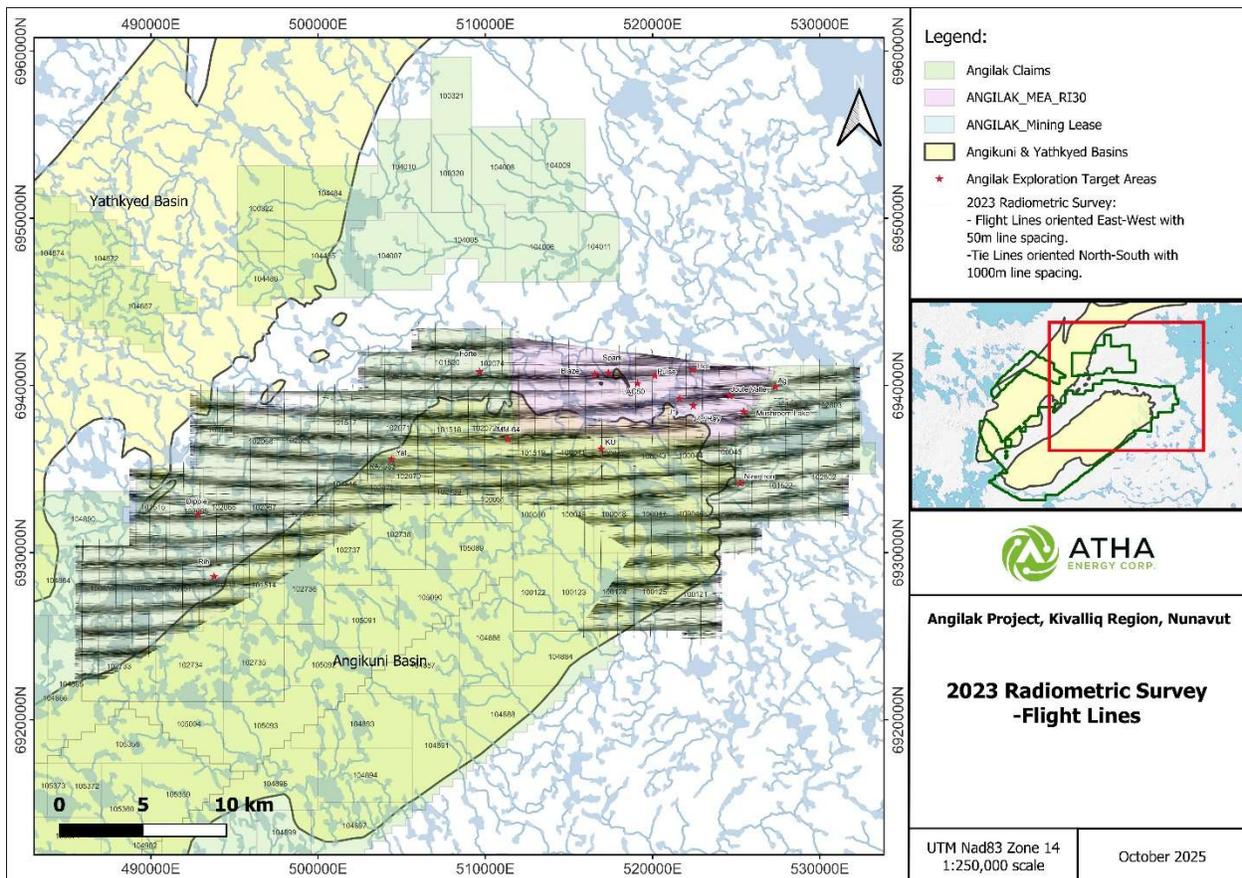
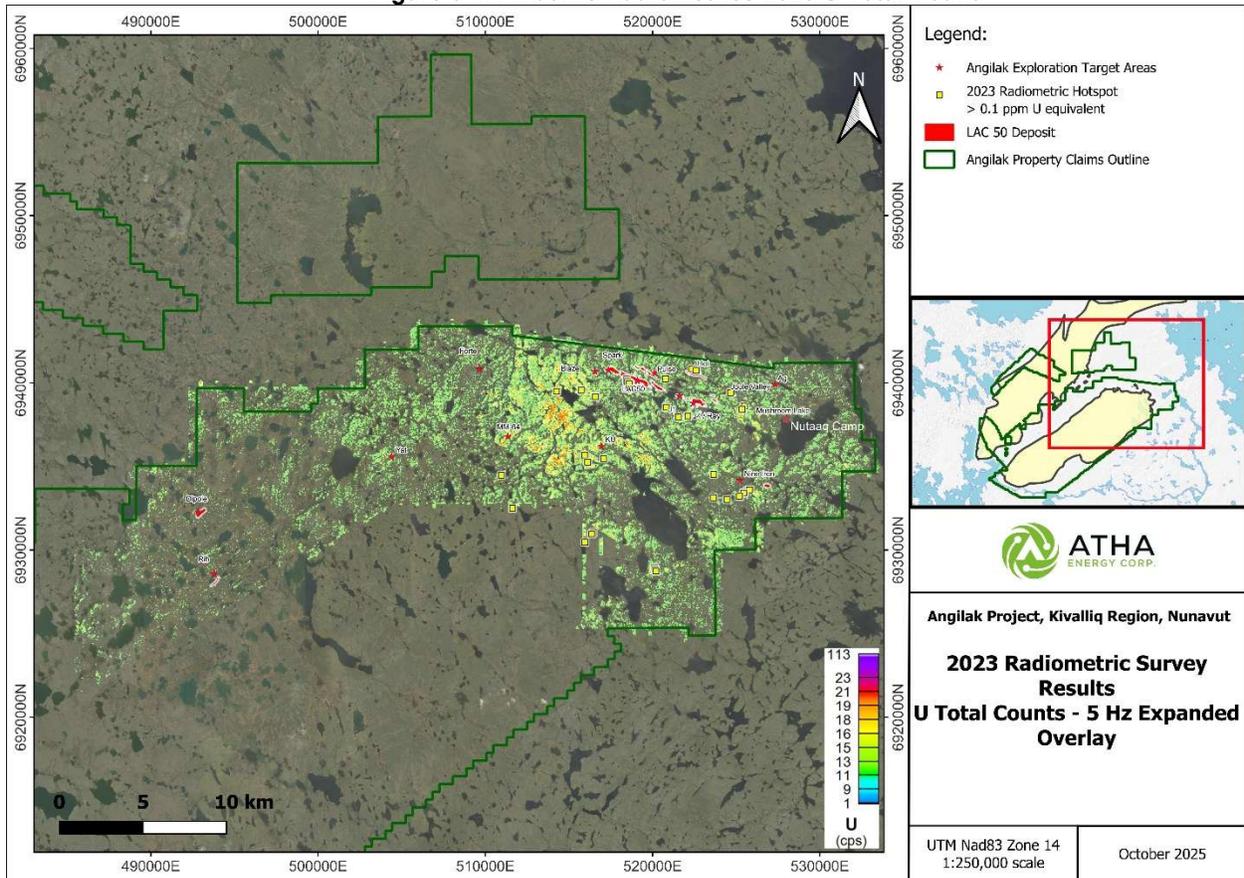


Figure 6-4: Airborne Radiometrics 2023 U Total Count



In 2023, LUR completed a diamond drill program in the Lac 50 Deposit area specifically targeting the Main Zone. The program successfully increased the extent of known mineralization and identified new mineralization horizons (Section 10).

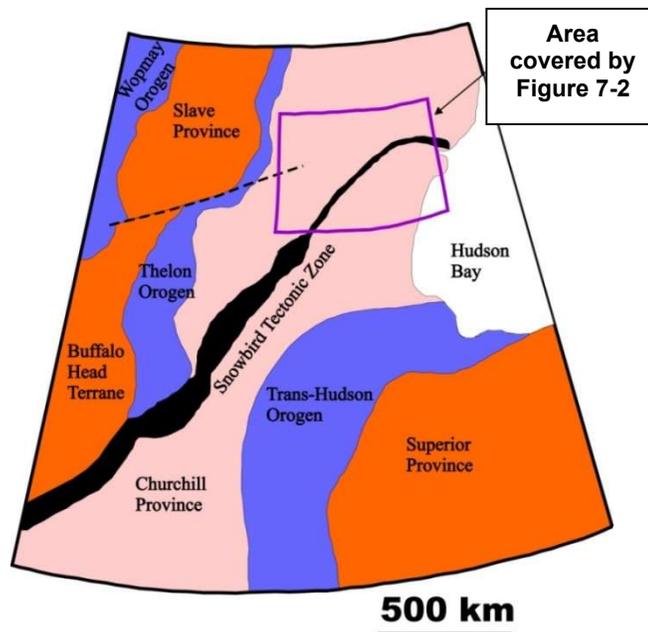
7 GEOLOGY AND MINERALIZATION

7.1 Regional Geology

The Angilak Project is within the Churchill province, a large Archean craton. The Churchill province is welded to the Superior province by the Trans-Hudson orogen, a northwest-dipping subduction zone and to the Slave province and Buffalo Head Terrane by the Thelon/Taltson orogen, an east-dipping subduction zone.

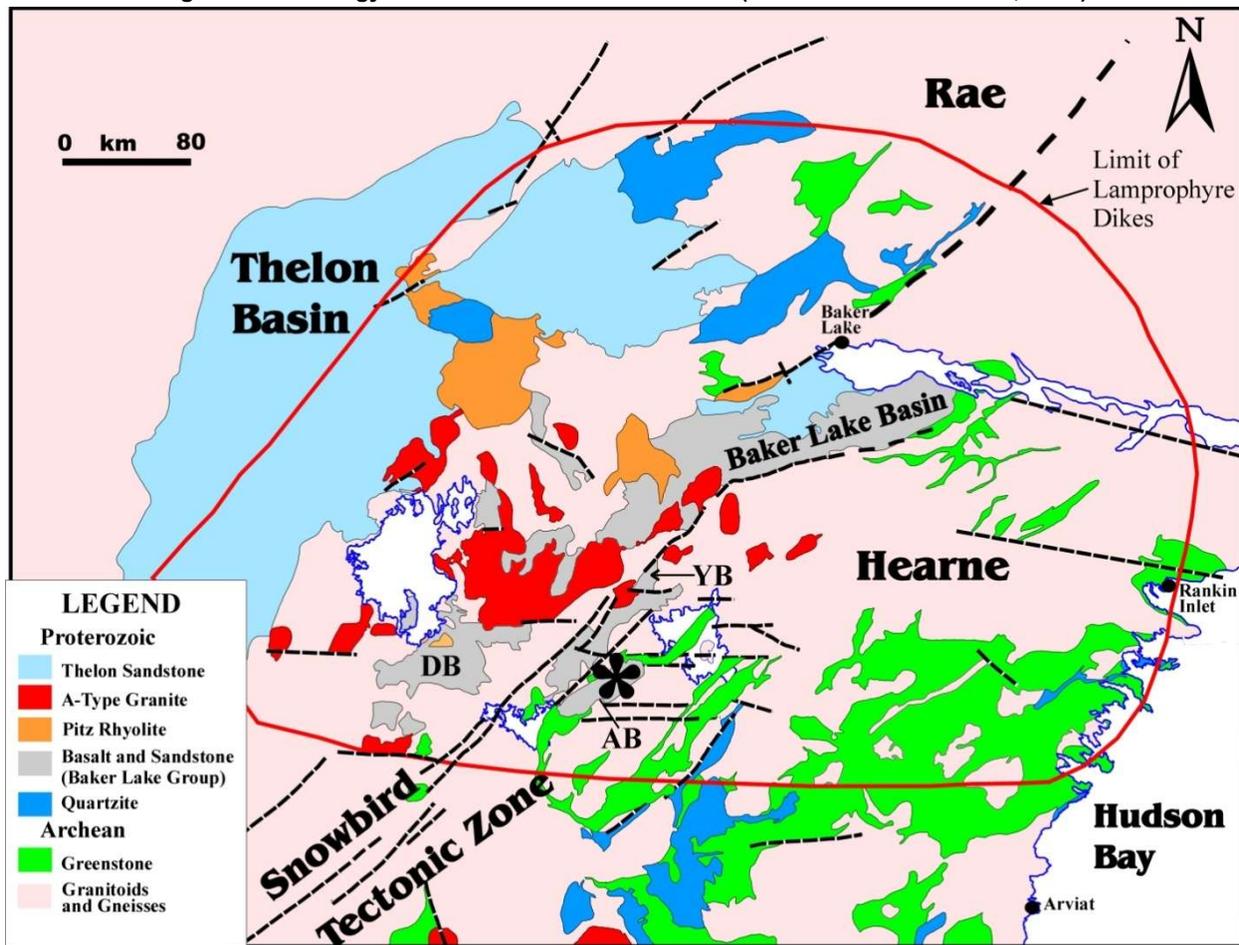
The Churchill Province is comprised of the Rae Domain to the northwest and the Hearne Domain to the southeast, sutured together along the northeast-trending Snowbird Tectonic Zone (Figure 7-1 and Figure 7-2) The Rae Domain is characterized by Mesoarchean basement upon which late Archean supracrustal rocks of the Prince Albert Group were deposited (Hoffman, 1990; Zaleski et al., 2000). While the Hearne Domain is composed mainly of late Archean juvenile tholeiitic greenstone belts with associated plutonic and sedimentary rocks (Sandeman et al., 2004). No in situ Mesoarchean crust has yet been identified in the Hearne Domain (MacLachlan et al., 2005), but inherited zircons (Henderson and Loveridge, 1990) and Nd isotopic signatures (Aspler et al., 2000; Sandeman et al., 1999) indicate at least some involvement of Mesoarchean crust in the vicinity of the Snowbird Tectonic Zone.

Figure 7-1: Simplified Tectonic Setting of the Slave, Churchill, and Superior Provinces (Dufresne and Shoeman, 2024)



**The Rae Domain is northwest of the Snowbird Tectonic Zone (STZ); the Hearne Domain is southeast of the STZ.*

Figure 7-2: Geology of the Thelon/Baker Lake Area (Dufresne and Shoeman, 2024)



*The star is the centre of the compilation area. Modified after Miller et al. (1987), Peterson and Rainbird (1990) and Gall et al. (1992). DB, YB and AB are Dubawnt, Yathkyed and Angikuni Sub-basins respectively.

The Snowbird Tectonic Zone is a major crustal feature that stretches over 3,000 kilometres from Hudson Bay to southern Alberta, and which has undergone a protracted, polyphase tectonic history (Mills et al., 2000). Various researchers have suggested that the Snowbird Zone is representative of an Archean intracontinental fault structure (Hanmer et al., 1994a, 1994b) while others maintain that it is a Proterozoic collisional suture (Hoffman, 1988). While the timing and tectonic significance of this structure are poorly understood, the fault zone likely played a major role in accommodating far-field stresses established by both the Thelon-Taltson and Trans-Hudson Orogeny's. During these orogenic events, the Churchill Province underwent significant crustal shortening and uplift, followed by northeast-directed "tectonic escape" and gravitational collapse (Peterson et al., 2002). This gravitational collapse led to the formation of the rift basins that host the Baker Lake Group (Rainbird et al., 2003) and may have had a significant influence on magmatic activity and metallic mineralization in the area.

In Nunavut, syn- to post-orogenic sedimentation occurred throughout the Thelon-Taltson/Trans-Hudson hinterland from approximately 1.83 to 1.75 billion years ago (Ga), beginning with deposition of the Baker Lake Group and culminating in the deposition of the Thelon Formation (Rainbird et al., 2003).

Volcanic and sedimentary rocks of the Thelon and Baker Lake basins have been assigned to the Dubawnt Supergroup, which has in turn been subdivided into the (oldest to youngest) Baker Lake, Wharton and Barrenslund groups (Table 7-1). Deposition of the Dubawnt Supergroup seems to have begun around 1.83 Ga and was probably completed by ca. 1.72 Ga (Peterson et al. 2002). Unconformities are present at the bases of all three formations of the Dubawnt Supergroup.

Table 7-1: Sequence and Timing of Regional Geology Events and Lithologies (Dufresne and Shoeman, 2024)

Age (Ma)	Group	Formation	Lithology
ca. 1270	MacKenzie Dykes		Diabase and gabbro dykes
ca. 1720	Barrenslund Group		
		Lookout Point	Dolostone
		Kuungmi	Subaerial Basalt
Minimum 1720		Thelon	Arenitic Pink Sandstone
ca. 1750	Nueltin Suite		Rapakivi A-Type Granite
ca. 1760	Wharton Group	Pitz	Fluorite-bearing Rhyolite
ca. 1830	Martell Syenite		Mafic Syenite; Carbonatite?
ca. 1830	Dyke Swarm	Christopher Island?	Lamprophyre & Minette
ca. 1850-1810	Hudson Suite		A-Type Granite
ca. 1840-1785	Baker Lake Group		
		Kunwak	Red-bed sandstone
		Christopher Island	Ultrapotassic minette lavas; volcaniclastics
		Kazan	Red-bed sandstone
		South Channel	Conglomerate, sandstone; regolith
Paleoproterozoic; >2100 Ma	Hurwitz and Amer Groups	Various	Quartzite, dolomite, arkose, iron-formation
	Tulemalu-MacQuoid		Gabbro and diabase dykes
Archean; >2500 Ma	Various	Various	Granitoid rocks (Snow Island Intrusive Suite)
			Greenstone Belts
			Gneissic granitoids

The Baker Lake Group, which is restricted to the Baker Lake basin system, consists of the South Channel, Kazan, Christopher Island and Kunwak formations. The ~1,800 metre thick South Channel formation consists of conglomerate with minor lenses of sandstone. The ~1,000-metre-thick Kazan Formation (locally called the Angikuni Formation) is dominated by red sandstones, with local mudstones, which commonly have desiccation cracks (Blake, 1980). The sandstone is geochemically similar to the overlying Christopher Island Formation, suggesting that early potassic volcanic rocks were eroded to form the lowermost sediments within the basin (Cousens, 1999). The Christopher Island Formation (CIF) is up to 2,500 metres thick, and is composed of potassic to ultrapotassic, dominantly subaerial lava flows with lesser pyroclastic rocks, debris flows and conglomerates (Peterson and Rainbird, 1990; Rainbird and Peterson, 1990). This formation is interpreted as the extrusive equivalent of the more widespread minette (a variety of lamprophyre) dykes (LeCheminant et al., 1987). A widespread suite of mafic syenitic plugs, the Martell Syenite, is also thought to feed the CIF (Smith et al., 1980). The Kunwak Formation (up to 2 km thick) is a coarse red-bed sequence with lesser interlayered debris flows and conglomerates (Rainbird and Peterson, 1990; Gall et al., 1992).

The Baker Lake group is unconformably overlain by the Wharton group, which consists principally of the Pitz Formation. This formation is up to 200 metres thick, erratically distributed between the

Thelon and Baker Lake basins and consists of grey to red rhyolite to dacite with lesser sedimentary rocks, typically red beds (Gall et al., 1992).

Rhyolites of the Pitz Formation are commonly ignimbritic and locally contain fluorite and/or topaz (LeCheminant et al., 1980). Widespread granites, which display rapakivi textures and contain fluorite (i.e., are A-type granites), are interpreted as intrusive equivalents to Pitz Formation volcanics (Gall et al., 1992). These granites have been assigned to the 1.76 Ga Nueltin Suite (Peterson and van Breeman, 1999; Peterson, 1996). Available ages for the Pitz Formation cluster in the 1.76 to 1.75 Ga range, almost 100 million years (Ma) later than CIF (Miller et al., 1989). The Barrenland Group overlies the Wharton Group and is mostly restricted to the Thelon Basin. The Amer/Hurwitz groups are early Proterozoic in age and were deposited prior to 1.83 Ga, when deposition of the Baker Lake Group commenced (Rainbird et al., 2003).

Uranium dominated polymetallic showings are abundant in the Baker Lake basin system. Mineralization including U-Cu ± Ag ± Au ± Pb ± Mo ± Zn occurs in fractures in Dubawnt Supergroup rocks or Archean basement, U-Cu-Ag ± Mo mineralization occurs in Kazan Formation red-beds adjacent to lamprophyre dykes, minor U-Cu-Ag-Au mineralization is associated with the unconformity at the base of the Thelon Basin, and minor U-Cu-Zn mineralization occurs associated with diatreme breccias (Miller, 1980; Miller et al., 1986).

The main diatreme breccia occurrence is east of Baker Lake and consists of angular, close-packed to sparse, clasts of Archean gneiss in a matrix of phlogopite-porphyrific, mafic "syenite" similar in appearance to flows of the CIF. The breccia cuts Archean gneiss and is variably carbonatized, chloritized and/or hematized, and contains a 10-metre-wide pod of pitchblende, chalcopyrite and minor sphalerite and pyrite (Miller, 1980). Similar breccias with no mineralization occur elsewhere. Red-bed copper mineralization is known in the Angikuni sub-basin at the base of the CIF (Miller, 1993).

Low grade REE-U-Th mineralization occurs near some of the alkalic dykes associated with the CIF (LeCheminant et al., 1987) and one syenite intrusion southwest of Dubawnt Lake contains up to 1% zirconium (Miller and Blackwell, 1992). Minor base metal (Pb-Cu ± Ag ± Zn) mineralization occurs in fluorite-bearing veins cutting the CIF spatially associated with a rapakivi granite (LeCheminant et al., 1980). Microdiamonds have been documented in minette dykes southeast of Baker Lake and have been reported from an interpreted diatreme near Dubawnt Lake.

7.2 Property Geology

The Lac 50 Uranium Deposit is located adjacent to the northeastern margin of the Angikuni Lake sub-basin and is hosted in Archean metasedimentary and metavolcanic rocks of the Henik Group (Figure 7-3). In the deposit area the dominant outcropping lithology is massive and pillowed propylitized metabasalt-meta andesite (Figure 7-4).

Figure 7-3: Geology of the Angilak Property (Modified after Stacey and Barker, 2013)

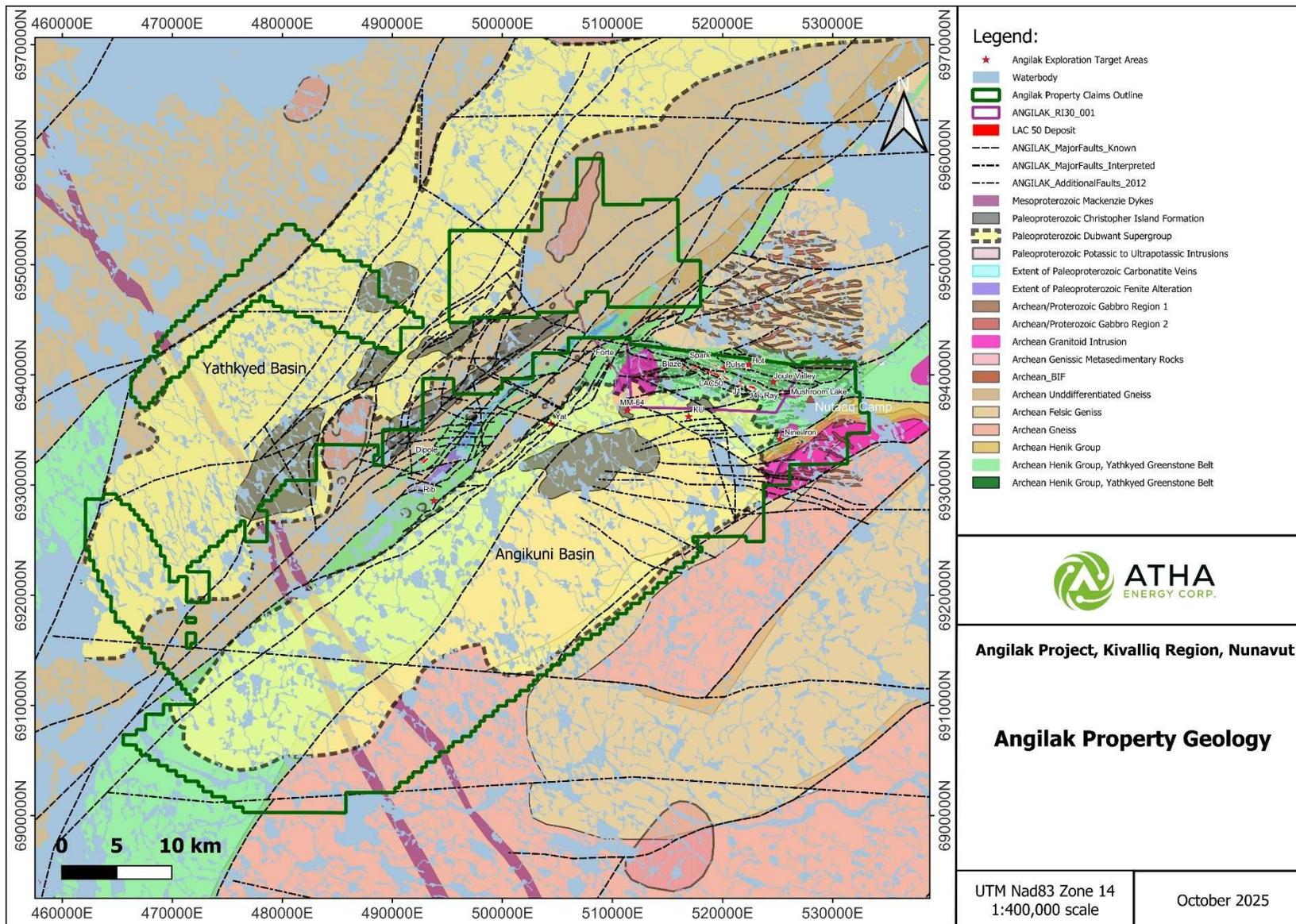
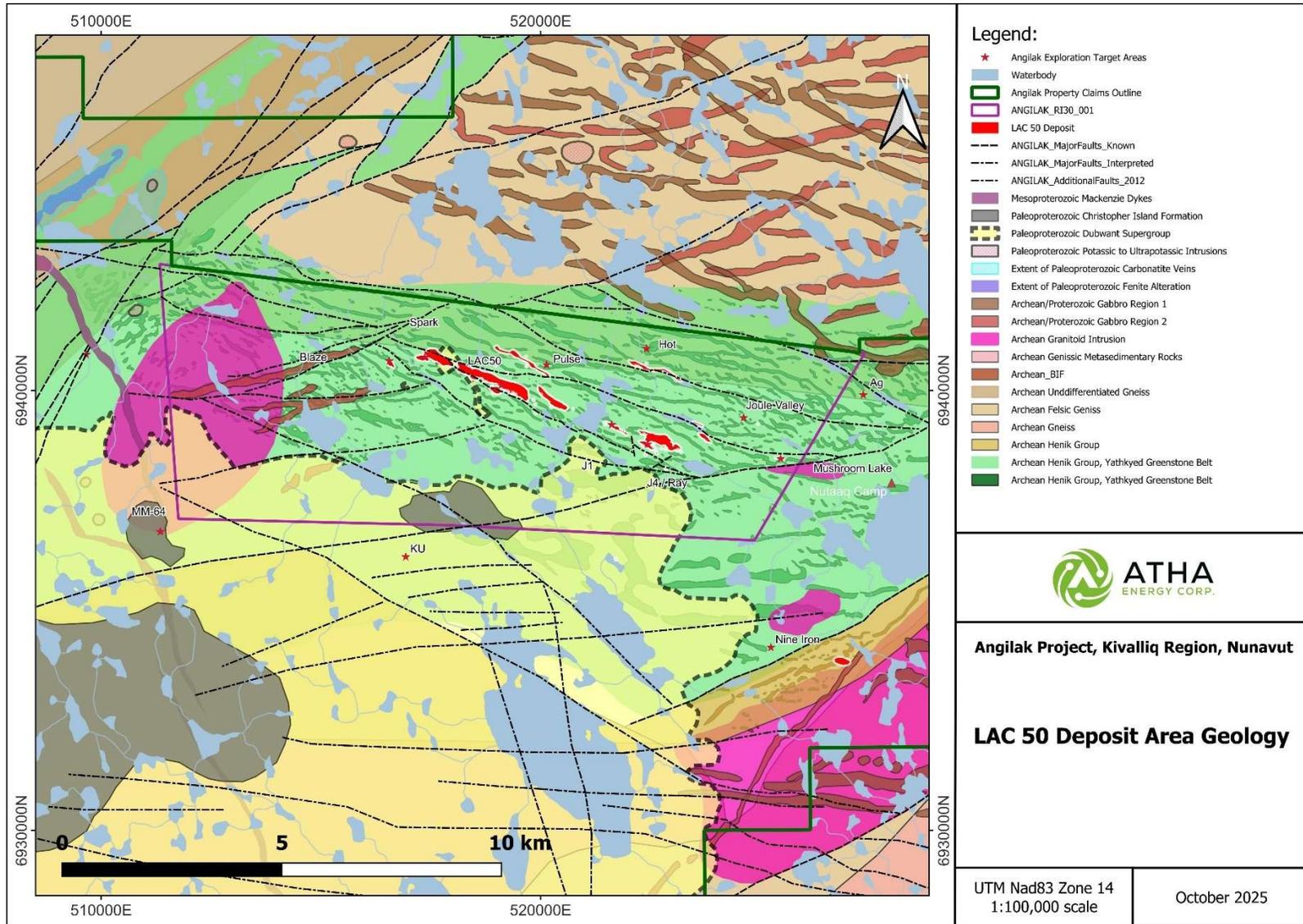
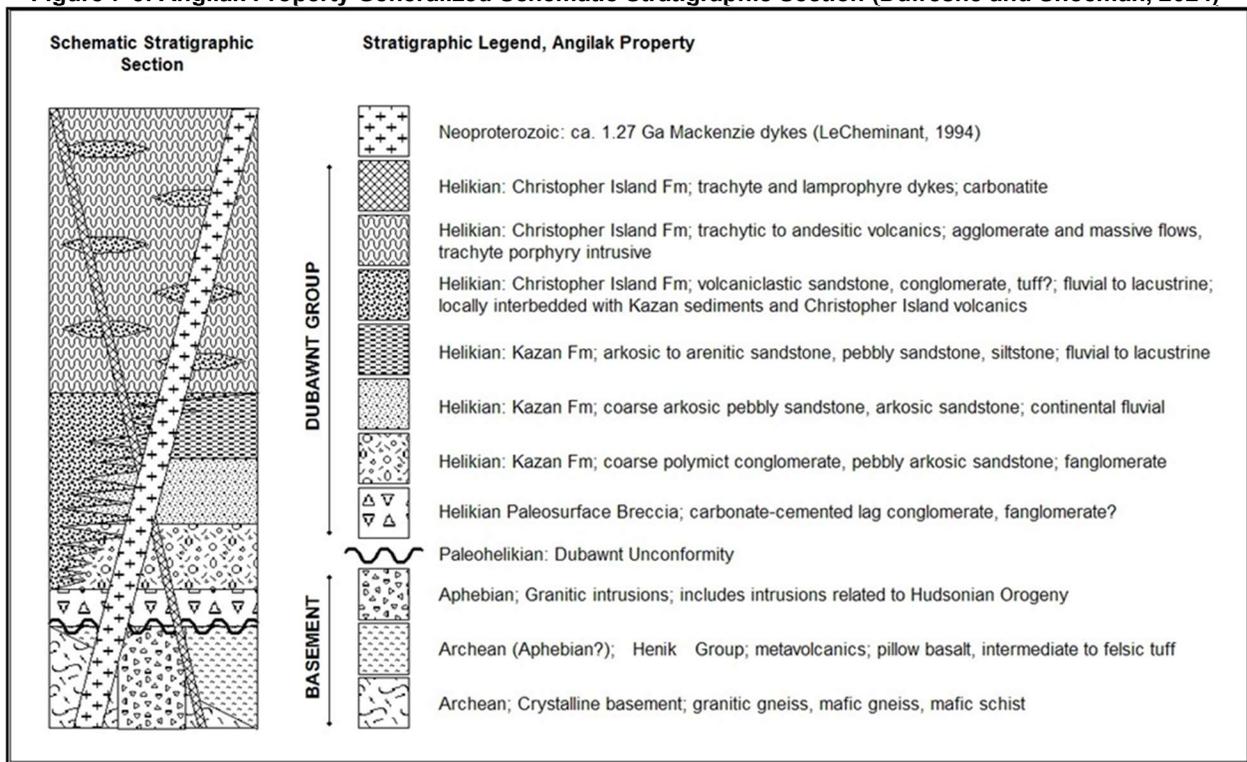


Figure 7-4: Geology of the Lac 50 Deposit Area (Modified After Stacey and Barker, 2013)



Prospecting and mapping completed by Bridge et al. (2010) around the Lac 50 Deposit has identified northeast striking fracture-controlled pitchblende-hematite-carbonate veins cutting east-southeast striking Archean metavolcanics that outcrop north and east of the overlying conglomerates of the Angikuni Sub-Basin. The geology of the Project area has been compiled from exploration geological mapping campaigns, historical assessment reports and regional mapping programs by the Geological Survey of Canada (Stacey and Barker, 2013). A schematic stratigraphic column for the Property is presented in Figure 7-5 with crosscutting relationships verified by field observations. Mapping by ValOre personnel took place during the summer field seasons of 2010 to 2012 and expanded on initial work performed by GeoVector in 2008 and 2009. ATHA completed an additional surficial mapping campaign in 2024. The programs were designed to validate existing maps and geological knowledge as well as providing a geological context for the various uranium showings on the Property (Stacey, 2010; Stacey and Barker, 2012 and 2013).

Figure 7-5: Angilak Property Generalized Schematic Stratigraphic Section (Dufresne and Shoeman, 2024)



The Angilak Project is situated between two very large fault systems: the Snowbird Tectonic Zone to the northwest, and the Tyrrell Shear Zone to the southeast. These fault zones initially formed during the assembly of the Archean Rae-Hearne sub-Provinces and were reactivated periodically in response to Proterozoic orogenic events. Transpressional tectonics between these two fault zones had a profound effect on the crustal geometry of the region, establishing an overall northeast-trending structural fabric defined by faults, isoclinal folds and shear zones. Many of these faults were reactivated with the initiation of extensional tectonics in the Mid Proterozoic, resulting in the northeast trending sedimentary basins of the Baker Lake Group. Archean basement rocks have undergone upper greenschist to lower amphibolite-facies metamorphism, while the sedimentary cover sequences are essentially unmetamorphosed.

Stacey and Barker (2013) have defined three structural domains within the boundaries of the Angilak Project based on evidence from field relationships, new geological mapping, and

geophysical surveys. These comprise the central/western gneissic belt, the Volcanic Block, and the southeastern compressive zone (Stacey and Barker, 2013). These three domains are structurally and lithologically distinct, having undergone related, but variable degrees of deformation and metamorphism.

The dominant structural fabric is defined by major 1st-order fault zones in the central/western gneissic belt and trends northeast-southwest (NE-SW). Regional mapping completed by the Geological Survey of Canada suggests that the largest of these structures root in the Snowbird Tectonic Zone near Angikuni Lake to the southwest (Tella et al., 2007). All rock fabrics in the gneissic basement trend NE-SW and dip steeply toward the NW or SE. Crystalline basement in this area is composed of granitoid gneiss, gabbro, and granitoid intrusions. Geological mapping in 2012 identified the presence of mafic volcanic rocks imbricated with gneissic basement in the central gneissic belt and was able to correlate these with Henik Group volcanics in the Volcanic Block. This correlation was previously unrecognized due to higher strain and metamorphic grade of the greenstones in the central gneissic belt (Stacey and Barker, 2013).

The eastern half of the Property is partially underlain by mafic to felsic volcanic rocks of the Yathkyed greenstone belt (termed the “Volcanic Block” by the Company). In contrast to the western part of the Property, this structural domain trends east-southeast and dips moderately (50°-70°) toward the south. The Volcanic Block is bounded by major fault zones: these faults are currently designated as “2nd-order” faults, but they may in fact be 1st-order faults that have been folded or faulted around a major synformal axis centered in the middle of the Property. If this were the case, then the southwest- and east-southeast -trending segments of the greenstone belt may define the limbs of a regional fold structure.

The geometry of greenstone packages in the central gneissic belt suggests that at least some of these rocks were imbricated with gneissic basement rocks during Archean and/or Proterozoic “thick-skinned” thrust faulting. It is therefore possible that the Volcanic Block started out as a northeast-trending thrust slice which was rotated around to an east-southeast orientation during Proterozoic dextral deformation, possibly related to Trans-Hudsonian orogenesis. It should be noted that the metamorphic grade of the Volcanic Block is somewhat lower than those observed in the western and far southeastern parts of the Property. Within this part of the belt, greenschist-grade mineral assemblages dominate, while the western half of the Property is more representative of lower to middle amphibolite-facies metamorphism. The far southeastern part of the Angilak Project is characterized by high-pressure, moderate-temperature metamorphism in the upper amphibolite facies. The mechanism responsible for this discrepancy in metamorphic grade is not well understood, but it is thought that the Volcanic Block occupied a higher structural position in the crust (i.e., closer to surface) than the surrounding higher-grade rocks during peak metamorphism (Stacey and Barker, 2013).

The third structural domain is located in the far southeastern part of the Property, in what is known as the Nine Iron (formerly BIF) area. In contrast to the Volcanic Block, this part of the Property is composed largely of metasedimentary rocks of turbiditic affinity, with very few mafic volcanic flows. Rock fabrics trend northeast and dip moderately (50° to 70°) toward the southeast. Metamorphic mineral assemblages and rock fabrics in this area indicate that this domain underwent extreme compressive deformation, largely unaccompanied by lateral shearing (Stacey and Barker, 2013). This is evidenced by the extreme flattening fabric visible in the rocks, as well as a general lack of lineations which would be apparent if strike-slip shearing had been a significant contributor to deformation in this zone. The presence of undeformed leucosomatic

partial melt material parallel to the flattening fabric is further evidence that lateral shearing did not occur during peak metamorphism in this domain (Stacey and Barker, 2013).

Within each of these structural domains, several orders of faults and shear zones are present, ranging from 1st order domain bounding faults to 4th and even 5th order structures (Stacey and Barker, 2013). Most of the higher-order structures can be deduced from geophysics and air photo lineaments, but many of the smaller lower-order faults are only observed in drill core. 1st and 2nd order faults may have originated in the Archean, and in most cases were reactivated as strike-slip faults during Proterozoic deformation. Late brittle faults (E-W to NW-SE-trending) transect and locally offset domain boundaries. Uranium mineralization can be correlated with fault zones at all scales, excepting the latest episodes of east-west brittle faulting. In the central/western gneissic belt, uranium mineralization seems to be associated with NE- to E-W-trending 1st to 2nd order faults. Within the Volcanic Block, uranium mineralization is exemplified by the Lac 50, Blaze and Joule (J4, Ray) deposits, which seem to be contained in 2nd to possibly 3rd order faults and breccia zones. In the southeastern compressive zone, uranium mineralization seems to be contained in narrow northeast-trending veins, which are parallel to 1st order fault structures and S1 foliations in this domain. However, the distribution of uranium mineralization in the Nine Iron area suggests that 3rd order faults at high angles to S1 may be a focus mechanism for mineralizing fluids, which then diffused into structures parallel to the foliation (Stacey and Barker, 2013).

The following description of the main lithologies encountered within the Angilak Project is summarized from Stacey and Barker (2012; 2013).

7.2.1 Archean Basement

The Archean component of the Property is dominated by felsic to intermediate gneiss, granitic to tonalitic intrusive rocks and gabbros, which extend northeast-southwest across the property. In general, basement rocks underlying the northwestern half of the property comprise granite and granitic gneisses, while those underlying the southeast half of the property are more granodioritic to tonalitic in composition and tend to be more massive rather than gneissic. The more massive granitoid rocks are interpreted to be younger than the gneisses, and have been assigned by Peterson (1994, 1996) to the ca. 2.6 Ga Snow Island Intrusive Suite. Migmatitic textures have been observed in basement gneisses at several locations on the property, indicating that metamorphic grades were locally high enough to induce at least some degree of partial melting.

Archean volcanic and metasedimentary rocks assigned to the Henik Group (Eade, 1986) are found in the eastern part of the property, where they underlie much of the northern part of the Angikuni Sub-basin. An Archean age of 2485 ± 62 Ma (K-Ar, hornblende) is indicated for the Henik Group in this area (Miller et al, 1986). Known collectively as the “Volcanic Block” or the “Yathkyed-Angikuni Greenstone Belt,” the lithological package extends southwestward beneath the sub-basin to Angikuni Lake. Immediately north of the central part of the Angikuni Sub-basin, mafic volcanic rocks are metamorphosed to amphibolite facies, while the main part of the Volcanic Block northeast of the sub-basin does not exceed greenschist facies metamorphism. Primary volcanic textures such as pillows, breccias, and lapilli are preserved at greenschist and lower amphibolite grades but are largely destroyed where metamorphic grades are higher and structural deformation is more severe. Deformation is strongest along the northwest and southeast margins of the greenstone belt, where mylonite zones separate metavolcanic rocks from adjacent gneissic and granitic basement.

The Henik Group in the project area is composed primarily of massive to pillowed basalt and subvolcanic gabbro, with local thin pyroclastic horizons comprising felsic to intermediate to mafic tuff. Fragmental, ashy, and water-lain tuffs can be interpreted where primary rock textures are preserved in outcrop and drill core. Basaltic sequences can be several tens to hundreds of metres thick, while tuff layers rarely exceed ten metres in thickness. All layers are transposed parallel to the steep regional foliation; possibly because of isoclinal folding associated with Archean tectonics and the Proterozoic Hudsonian Orogeny. Mineralogy in the basalt comprises chlorite + actinolite ± hornblende assemblages; garnet is locally found adjacent to quartz monzonitic dykes. The general absence of garnet and the prevalence of chlorite-actinolite assemblages indicate that metamorphic conditions less than the middle amphibolite facies were predominant. Sheared metasedimentary rocks, including psammite-semipelite, wacke, and iron-formation, are observed along the southeast flank of the Volcanic Block.

In the eastern part of the Angilak property, the east-southeast structural orientation of the Volcanic Block differs greatly from the regional northeast-southwest trend exhibited by most basement units. The exact mechanism by which the Volcanic Block has rotated is poorly understood.

7.2.2 Hudsonian Granitoid Intrusions

Though Hudsonian-aged intrusions are found throughout the Western Churchill Province, large expanses of this granite are not particularly common on the Angilak Project. However, the faulted northern boundary of the Volcanic Block and several large northeast-trending fault systems to the west seem to have been loci for sheet-like intrusion of pink, equigranular granite and rare pegmatite interpreted as being related to Hudsonian plutonism. Rather than forming discrete plutons, this granite has only been observed as dyke-like bodies, sometimes intruded in a stockwork fashion in proximity to major faults.

7.2.3 Helikian Paleosurface Breccia (Unconformity Surface)

The term Helikian Paleosurface Breccia (“Hpb”) was coined by Urangesellschaft personnel in the mid 1970’s to describe the strongly paleo weathered angular “lag conglomerate” locally exposed at the base of the Dubawnt Unconformity. The term is descriptive and highly appropriate, due to the fact that the horizon was developed in situ from the weathering of rocks directly below the unconformity. The Hpb has been observed on top of both mafic volcanic rocks of the Henik Group, and rare occurrences on top of basement gneisses are noted further to the west. Clast composition of the Hpb is highly dependent on the underlying lithology. A common feature of the Hpb, which is independent of clast composition, is a sandy matrix rich in iron carbonate and hematite. The matrix presumably formed during paleoweathering and is of a composition and texture which is unique to the Hpb. The carbonate-rich matrix may represent caliche-type evaporative cement and could be an indication of weathering in an arid environment.

The Paleosurface Breccia tends to have higher background radioactivity than the underlying basement (500 to 1000 counts per second) but is essentially unmineralized. Elevated background radioactivity of the Hpb is interpreted to be the result of uraniferous fluids migrating along the unconformity surface and precipitating minor amounts of uranium around clasts, in fractures, and in the matrix of the Hpb. This unit is not considered to be prospective for significant uranium mineralization.

The unit provides direct evidence of paleoweathering prior to deposition of the Dubawnt Supergroup and serves as a recognizable marker horizon within the overall stratigraphic sequence. In contrast to the Sub-Athabasca Unconformity in Saskatchewan, the Angilak Property did not undergo deep regolith weathering.

7.2.4 Baker Lake Group (Dubawnt Supergroup)

The Baker Lake Group is represented in the project area by the parallel Yathkyed (north) and Angikuni (south) Sub-basins, which extend northeast-southwest across the property. Though regional maps by Eade (1986), Peterson (1994), and Tella et al. (2007) all show the sub-basins to be comprised completely of volcanic rocks of the Christopher Island Formation (CIF), more detailed mapping by Miller (1993), Company personnel, and other exploration companies has proven that conglomerate and sandstone of the South Channel and Kazan Formations are present as well. The Late Proterozoic Thelon Formation is not found in the project area. Historically, the Helikian Paleosurface Breccia and the coarse-grained conglomeratic units directly above the unconformity are grouped with the South Channel Formation, while overlying finer-grained sandstone, siltstone, and mudstone units define the Kazan Formation. For the purposes of this report, the Paleosurface Breccia is defined as a separate entity, rather than being lumped with the South Channel Formation.

7.2.5 South Channel Formation (SCF)

The South Channel Formation (SCF) is the lowermost unit of the Baker Lake Group and directly overlies the Helikian Paleosurface Breccia. The transition from Hpb to South Channel rocks is quite sharp, though coarse clasts of re-sedimented Hpb can be found in the lowermost levels of the SCF. South Channel sediments mainly comprise poorly sorted, coarse to very coarse fluvial and fanglomerate-type conglomerates which display a wide variety of clast compositions. Clasts are rounded to subrounded granitic and gneissic rocks which have been transported a significant distance from their source. Rounded white vein quartz pebbles are also common. In proximity to Archean greenstone basement, a significant portion of the clasts (20 to 50%) comprise angular, hematite-altered volcanics, which suggests both distal and proximal sources of sedimentation for the SCF. Trachytic clasts are also observed in some areas, indicating that at least some local sedimentation was derived from the Christopher Island Formation. The matrix of the basal conglomerates is composed of angular, coarse to very coarse feldspathic sand and gravel containing up to 50% quartz grains. In other areas the matrix is mainly feldspathic.

The SCF varies between several metres and several tens of metres in thickness, and fines upwards into coarse pebbly sandstones with conglomeratic lenses or channels. Local siltstone and mudstone layers sandwiched between coarse-grained conglomerates are indications that parts of the SCF were deposited in a quiescent lacustrine to deltaic environment. The coarser-grained conglomerate was presumably laid down in a fluvial setting, suggesting subdued paleotopography crossed by relatively high-energy braided streams.

The boundary between the SCF and the overlying Kazan Formation is conformable and gradational and is typically defined where coarse conglomerate and poorly sorted coarse sandstone give way to well sorted, fine-grained arkosic sandstone, siltstone, and mudstone.

7.2.6 Kazan Formation (KF)

The Kazan Formation (KF) unit is composed primarily of fine to medium-grained, moderately to well sorted, pink to maroon sandstone, siltstone, and mudstone. Vein quartz pebbles persist in coarser pebbly sandstone layers, in contrast to quartz-poor Christopher Island Formation sediments. Siltstone layers commonly contain mud cracks, indicating periods of subaerial desiccation. Local finely interbedded sandstone, siltstone, and mudstone varves are indications of seasonally variable sedimentation in lacustrine settings.

Kazan sediments are flat lying to gently dipping (typically less than 5°), though rare fault blocks can be tilted as much as 30° and local warping has been observed in immediate proximity to fault zones. Bedding is typically massive, and channel-fill sedimentary structures are noted locally. Fault-related deformation within the Kazan Formation seems to have occurred almost entirely within the brittle strain field, leading to widespread fracturing and local brecciation around faults but almost no folding. In some cases, faults cutting through the Baker Lake Group may be related to the reactivation of pre-existing basement faults and as such present a highly attractive target for unconformity-style uranium mineralization.

Radiometrically, the Kazan Formation exhibits higher background radioactivity than the underlying basement rocks. Background levels of 250 to 350 counts per second (CPS) are the norm, though individual hematitic fractures and bedding planes can run as high as several thousand CPS. Hematite-altered radioactive fractures may have formed during the mobilization of uranium through the sedimentary package, whereas the origin of radioactive beds is more ambiguous. These beds may have been mineralized by uraniferous fluids percolating laterally along the unconformity (epigenetic) or through syngenetic deposition from uranium-rich source rocks. The widespread presence of red-bed-type copper mineralization may provide an indication that some uranium mineralization is epigenetic and possibly related to the fluid event(s) that deposited copper-bearing minerals in the sandstones.

7.2.7 Christopher Island Formation

The Christopher Island Formation (“CIF”) is composed primarily of trachytic to andesitic volcanic flows, pyroclastic fragmental volcanics and agglomerate, syenitic intrusions and volcanoclastic sedimentary rocks. Though the CIF largely overlies the Kazan Formation, significant overlaps of the depositional units exist, and in some areas CIF flows and sediments are complexly interfingering with Kazan-type sediments. A criterion for identification of parent lithology is the presence or absence of white vein quartz pebbles: quartz pebbles are not found in the CIF but may be present in rocks of Kazan parentage. In the absence of quartz pebbles, it can be very difficult to assign a specific parentage to sedimentary rocks which contain trachytic clasts; however, Kazan sediments typically contain at least some quartz in the matrix, while CIF sediments are primarily feldspathic. Trachytic agglomerates can be coarse to very coarse grained and, in some cases, clasts can exceed one m in diameter. Clasts are angular and supported by a trachytic microcrystalline to aphanitic groundmass. Typical CIF agglomerates have clast sizes on the order of 20 to 30 centimetres, composed primarily of trachyte with some andesitic clasts. Coarser-grained agglomerates may be associated with vent-proximal volcanic facies, though the relationship between texture and vent proximity is poorly understood.

In contrast to the volcanoclastic sediments and agglomerates, volcanic flows are easily identified by their composition and texture. Trachytes are pink to red and tan coloured and andesites are purplish brown to grey. Both are fine-grained and variably porphyritic: trachytes tend to contain K-feldspar phenocrysts and local biotite phenocrysts, whereas andesites are primarily biotite-phyric. Vesicular and/or amygdaloidal textures are commonly observed in andesitic rocks. Coarse K-feldspar-phyric syenite porphyry dykes are found throughout the property and are especially common in and around fault zones. Several U-Cu-Ag-Au showings may be hosted by or partially derived from trachytic bodies intruding CIF volcanics, CIF/Kazan sediments and gneissic basement, respectively. CIF dykes generally seem to be less than a few metres in width but can be much wider in places.

Uranium mineralization within the CIF has so far been limited to hematitic fracture fillings and occasional high-grade pitchblende ± hematite ± Cu-sulphide veins. Radiometrically, the CIF has the highest background signature of any rocks in the study area, commonly averaging 350 to 450 CPS in outcrop. Most of this background radioactivity is related to the highly potassic composition of the CIF, though background levels of uranium are slightly higher in the CIF than in the Kazan and South Channel Formations. Though the hydrothermal circulation system in the area is not fully understood, CIF volcanism may have been a significant contributor of fluid to the system and may also have been a source of uranium for remobilization to other areas on the property.

7.2.8 Syenite, Lamprophyre and Carbonatite (CIF)

Syenitic bodies throughout the property constitute the feeder system for Christopher Island volcanism. Dykes and stocks of syenitic composition are concentrated around major fault zones. Two conspicuously large intrusions occur on the northern and southern boundaries of the property and are interpreted as large, possibly zoned, alkalic complexes.

Lamprophyre dykes and stocks are common throughout the property and are related to CIF volcanism. The dykes are a distinctive brown colour and contain fine to coarse biotite and hornblende phenocrysts in a quartz-free, massive, fine-grained feldspathic matrix. Lamprophyric dykes were presumably emplaced during regional crustal extension and trend northeast-southwest throughout the property. To date, no significant uranium mineralization has been observed in proximity to lamprophyre dykes, though occasional radioactive, hematite-altered hairline fractures have been noted.

7.3 Mineralization

The Baker sequence records the initial and principal phases of development of the Baker Lake basin (Rainbird et al., 2003). Aspler et al. (2004) expanded on this idea and proposed that basin formation by strike-slip cannot be ruled out; however, a more appropriate model is likely regional uplift and extension within the west portion of the Western Churchill province due to terminal collision and post-collision convergence in the Trans-Hudson orogen. The base of the Baker Lake Group consists of coarse alluvial red beds from the South Channel Formation that are overlain by finer grained distal equivalents from the Kazan Formation (Donaldson, 1965; Rainbird et al., 2003). In the Angikuni sub-basin, the Kazan Formation is equivalent to a similar sedimentary succession called the Angikuni Formation (Blake, 1980). The Christopher Island Formation (CIF) is a suite of ultra-potassic lava flows and volcanoclastic deposits that have been found intercalated with overlying the strata of the South Channel and Angikuni Formations (Eade, 1986; Rainbird et al., 2003). Aspler et al. (2004) interpreted the conformable contact with the CIF and lack of volcanic detritus in the section to indicate that the Angikuni Formation was deposited between and during periods of active volcanism. SHRIMP U-Pb geochronological studies have yielded age groupings at 2.7 and 2.6 Ga for the 1.84 – 1.79 Ga Baker sequence (Rainbird and Davis, 2007). These ages are consistent with a proximal uplands source, and have been correlated to the northwestern Hearne domain (Rainbird and Davis, 2007)

Numerous mineral showings were discovered by various exploration companies during the late 1970's and early 1980's. Most of the showings occur close to the northern boundary of the Angikuni sub-basin, within both Archean basement and later basin-fill sedimentary and volcanoclastic material. A partial reason for the distribution of known mineralization could be that the most intense exploration effort was focused in this area, and it is likely the area of the unconformity with the most amount of outcrop. The important regional U-Cu-Au-Ag showings are summarized in the history section above and are discussed in more detail in Dufresne and Shoeman (2024).

The Lac 50 Uranium Deposit is structurally and stratigraphically controlled and is hosted within a graphite-chlorite tuffaceous metasediment interlayered within the Archean basement metavolcanics. Mineralization consists of disseminated pitchblende with sulphides and as fracture-controlled, brecciated hematite-pitchblende-quartz-carbonate veins within the tuff. Uranium and sulphides occur in widths up to 16.4 metres within a sheared tuffaceous host unit up to 17.4 metres wide. The deposit strikes southeast at 110 to 120° and dips south, variably between -45 and -80°. Mineralization occurs as southwest plunging shoots within the plane of the tuff unit and has been traced by drilling to a vertical depth of approximately 400 metres and along a strike length of 3.5 kilometres. Lac 50 Deposit is described as a basement hosted, vein-hydrothermal type, unconformity associated uranium deposit.

Mineralization within the Lac 50 Deposit occurs within or very proximal to graphite and sulphide bearing horizons. Generally, a number of sulphides are present within these horizons and may accompany uranium mineralization including pyrite, chalcopyrite, molybdenite, galena and sphalerite. Uranium mineralization generally consists of pitchblende (uraninite) and coffinite along with minor amounts of uranium oxide (U_3O_7), brannerite, uranophane, potassium uranyl fluoride hydrate [$K_3(UO_2)_2F_7 \cdot 2H_2O$] and richetite ($PbU_4O_{13} \cdot 4H_2O$) based on mineralogical work conducted by Morton and Grammatikopoulos (2011).

Mineralization at the Lac 50 Deposit and proximal showings can be divided into four types: (1) disseminated pitchblende with base metals in intensely fractured carbonaceous-sulphide-chert exhalite and adjacent tuffaceous metasediments; (2) carbonate + pitchblende + hematite ± chlorite breccias, in which pitchblende aggregates on clast and breccia margins; (3) discrete pitchblende veins that cut across exhalite tuff metasediments and; (4) quartz + carbonate + sulphides and pitchblende gash veins. The discrete pitchblende veins tend to be found throughout the hanging wall basalt and tuffs horizons. These “gash veins” range in size from a few millimetres to up to a metre across and can be almost barren to hosting several percent U_3O_8 . Some of the largest gash veins can be correlated between drillholes on the same drillhole fence.

The elemental signature of the Lac 50 Deposit is U+Ag+Mo+Cu+Pb+Zn. The mineralization is accompanied by complex alteration involving hematization, chloritization, carbonatization, silicification and albitization. The deposit is described as a vein-type hydrothermal derived deposit which resembles the classical uranium bearing veins of the Beaverlodge District in Saskatchewan (Miller et al., 1986; Setterfield, 2007). Banerjee et al. (2010) and Bridge et al. (2010), indicate that the alteration associated with the Lac 50 Deposit is low temperature hydrothermal and consists of widespread pervasive hematite - chlorite alteration in and around the deposit along with carbonate in and around veins within the main zone. Bridge et al. (2011) have dated the main Lac 50 uranium mineralization at $1,828 \pm 30$ Ma with slight resetting at $1,437 \pm 31$ Ma.

8 DEPOSIT TYPES

The following is reproduced, with minor formatting changes, from a previous Technical Report completed on the Property by Dufresne and Shoeman (2024) and summarizes the most likely mineral deposit types that might be encountered on the Angilak Project. These interpretations are based on examining historical assessment reports and field visits to key outcrops and mineral occurrences. The region is host to numerous polymetallic showings that contain variable amounts of U ± Cu ± Ag ± Au, which were discovered in the late 1970's but have received minimal attention since that time. The most important deposit type discovered to date and host to the Lac 50 Deposit is the Beaverlodge-type vein or structure hosted uranium deposit.

8.1 Beaverlodge-Type Uranium Deposits

The primary target of exploration on the Angilak Project is Precambrian Beaverlodge-type vein or structure hosted uranium deposits. The past-producing Beaverlodge uranium district is located in northern Saskatchewan and produced over 68 million pounds of uranium up until production ceased in 1982 (Beck, 1986). These types of deposits are commonly referred to as “vein-type” hydrothermal uranium deposits due to mineralization being hosted in near-vertical vein-like structures associated with faults and shear zones. Uranium ore minerals are typically pitchblende and uraninite and grades are typically on the order of 0.1 to 0.5% U₃O₈. Beaverlodge deposits were relatively small and low grade compared to the more prolific “unconformity-related” uranium deposits found in the Athabasca and Thelon Basins. For example, published resource estimates on the Kiggavik Deposit near Baker Lake are approximately 127.3 million pounds of U₃O₈ (Areva Resources Canada Inc., 2009).

A number of exploration companies and government scientists have compared the uranium occurrences in the Baker Lake and Angikuni Basins to the Beaverlodge examples and suggested they formed in similar environments. Al Miller of the Geological Survey of Canada described several uranium showings from IOL Parcel RI30-001 in a paper published in 1986, including the Lac 50 Uranium Deposit (Miller et al., 1986). Similarities between the classic Beaverlodge occurrences and Lac 50 include: 1) narrow, pod-like uranium shoots hosted in discrete fault zones, 2) age of host rocks and hydrothermal alteration assemblages, and 3) grade and distribution of uranium minerals. The overall characteristics of the Lac 50 Uranium Deposit appear similar to the Beaverlodge examples, however, when considered in a regional context the Lac 50 Deposit may represent just one of many mineralization styles in the area whose formation can be attributed to magmatic processes associated with iron oxide – copper – gold deposits, or a variant on high grade basement hosted deposits, similar to Eagle Point in the Athabasca region of Saskatchewan.

8.2 Iron Oxide Copper Gold (IOCG) Deposits

Historical uranium exploration in the Project area occurred prior to the development of IOCG deposit models. The best-known example of this class of ore deposit is the prolific Olympic Dam poly-metallic deposit located in Australia and discovered by Western Mining Corporation (WMC). The regional geology of the Yathkyed area shares many geological similarities with known IOCG districts, including age of the host rocks, the presence of an extensional tectonic regime that produced continental-derived mafic and felsic rocks, ultrapotassic magmatism and craton-scale structural breaks. WMC recognized these similarities and conducted an exploration program 10 kilometres south of IOL Parcel RI30-001 in 1995. However, WMC focused their efforts within the Angikuni basin itself and had purposely avoided uranium occurrences due to economic and political conditions at that time. Most if not all of these regional characteristics have been recognized in the Angilak Project as outlined by Dufresne and Shoeman (2024). On a deposit

scale there are many distinctive features of IOCG deposits however, there can be extreme variability in the presence or absence of key characteristics.

In 2007, Kaminak personnel conducted a one-week reconnaissance field program which covered RI30-001 and Archean basement rocks north and east of IOL Parcel RI30-001. At the outcrop scale, Kaminak recognized a number of key textural features of the IOCG deposit class: including the presence of brecciated and silicified felsic intrusive rocks displaying strong hematite and carbonate alteration. Overall, metal content of the mineralized zones (Au-Cu-U-Ag) and the composition of alteration assemblages (Si-Na-K-Ba-P) are consistent with accepted IOCG characteristics.

8.3 Unconformity-Related Uranium Deposits

The concentration of showings proximal to the unconformity between basement and the (Mid-Proterozoic) Angikuni sub-basin would suggest that an unconformity-related uranium deposit model (Jefferson et al., 2007) is applicable to this area. Indeed, this was the model used by previous exploration companies in the late 1970's, and much of the mineralization noted to date, including the Lac 50 Uranium Deposit, probably relates to this model. However, many of the showings, particularly within the basin, have significant amounts of Cu and Ag. Miller (1993) suggested a possible red bed Cu mineralization model to explain this mineralization.

Unconformity-related uranium deposits are characterized by small tonnage but very high-grade Uranium grades (sometimes over 25% U_3O_8). Some of the world's most prolific uranium deposits fall within this category of mineral deposits and include the Athabasca and Thelon Basins of northern Canada. A key factor in the formation of these deposits is the presence of the unconformity that separates Mid-Proterozoic clastic sandstone rocks from underlying Lower-Proterozoic graphitic pelites and associated Proterozoic "basement" rocks.

8.4 Unconformity-Related Banded Iron Formation Uranium Deposits

Since 2011, surface exploration work recognized a southwest uranium mineralized trend located about 10 kilometres southeast of the Lac 50 Deposit, referred to as the Nine Iron trend and formerly known as the "BIF Zone" (ValOre News Release, 2012). Unlike the Volcanic Block, the package of mafic igneous rocks hosting the Lac 50 Deposit, the Nine Iron Zone is predominantly hosted by intermediate to felsic tuff and volcanoclastic metasedimentary rock, with subordinate mafic volcanic flows (Stacey and Barker, 2012 and 2013). The Nine Iron trend is outlined by a distinct, 9-kilometre-long magnetic geophysical anomaly extending below the contact or 'unconformity' with the Angikuni sub-basin.

The uranium mineralization at Nine Iron trend is unconformity-related and associated with a banded iron formation (BIF). The emplacement of mineralization is structurally controlled and related to competency contrasts between the sedimentary and igneous layers. Uranium mineralization along the Nine Iron trend occurs over a 3-kilometre-long reactivated shear zone on the margin of the Yathkyed Greenstone Belt and within a package of mylonitized iron formation and tuffaceous volcano-sedimentary rock (Stacey and Barker, 2012 and 2013; ValOre News Release, 2012). Five surface samples have returned grades between 15% and 30.3% U_3O_8 . In keeping with the geochemical signature of uraniferous veins throughout the Property, strong uranium mineralization in the Nine Iron Zone is accompanied by significant Cu, Zn, Pb and Ag values (Stacey and Barker, 2012 and 2013).

8.5 Carbonatite-Related Deposits

In 2011, ValOre prospectors discovered a number of carbonatite occurrences in outcrop and float on the Angilak Project. Unlike hydrothermal carbonate veins, carbonatite bodies are emplaced in a molten or semi-molten state and have mineral assemblages that reflect their magmatic origin. Mineralogy can be highly variable, but is dominated by various carbonate minerals (calcite, ankerite, magnesite, etc.) with subordinate silicate minerals. Carbonatite bodies are typically associated with zoned alkalic intrusive complexes, though they are also found as veins, dykes, or small isolated plugs. Carbonatite is a very highly fractionated, late-stage magmatic phase, and as such tends to become enriched in incompatible elements. Notable carbonatite occurrences with economic concentrations of Rare Earth Elements (REEs), phosphates, copper, iron, precious metals, and/or other commodities include Oka, Québec; Mountain Pass, California; Jacupiranga, Brazil; and Palabora, South Africa (Verwoerd, 1986; Bell, 1998). In Canada, carbonatites are relatively common and have been mapped throughout the Canadian Shield and British Columbia.

The presence of carbonatite on the Angilak Project is not unusual, considering the enormous volume of alkalic magma that was produced during the Christopher Island volcanic event. In outcrop, carbonatite is spatially associated with subvolcanic syenite and lamprophyre and was probably emplaced in the waning stages of CIF volcanism. At this early stage of exploration, the size, distribution, and mineral tenor of carbonatites on the Property are poorly understood; however, the richness of some carbonatite deposits elsewhere in the world makes the Angilak occurrences an attractive exploration target. The association of carbonatite with zoned alkalic complexes is favourable from a geophysical standpoint, as they typically form concentric magnetic anomalies which are easily targeted for prospecting and drilling.

8.6 Red Bed Copper Deposits

Miller (1993) described a number of copper occurrences in the Angikuni Sub-basin which he attributed to red bed copper mineralization. These showings contain disseminated, stratiform and stratabound copper sulphide at or near the contact between the uppermost Kazan and lowermost Christopher Island Formations. Visually, copper-bearing strata are easily identified by their bleached grey to light pink colour, which contrasts sharply with orange-pink to maroon colours in unmineralized rock. This is characteristic of redox alteration: minerals associated with bleaching include chlorite, carbonate, and rare albite, formed when oxidized strata were invaded by copper-bearing reducing fluids. Elevated radioactivity locally accompanies copper mineralization, but most of these occurrences are non-radioactive, and spatially associated uranium may have formed through different processes than that which deposited copper in the rocks. This idea is reinforced by the fact that uranium tends to be concentrated in discontinuous fractures or veinlets, while copper sulphides are disseminated. If uranium and copper were deposited during the same fluid event, the uranium should be stratiform/stratabound and disseminated, rather than concentrated in discrete veinlets. However, the mechanisms of uranium emplacement in the sandstone packages are not well understood, and contemporaneous copper and uranium mineralization could have occurred on a local scale.

Though red bed copper occurrences on the Property are interesting and provide insight into the fluid history of the region, they are not considered a high-priority exploration target at this time. This may change if evidence for large-tonnage deposits is uncovered, but the showings described by Miller (1993) have so far proven to be of limited areal extent and the potential for large red bed copper occurrences is considered to be low.

8.7 Archean Mesothermal Gold and VMS Deposits

The potential for Archean mesothermal gold mineralization on the Property is considered low to moderate. The Kivalliq region is host to several significant gold deposits of this type, most notably Meadowbank and Meliadine. Portions of the Property are underlain by Archean pillowed mafic volcanic rocks that Eade (1986) has correlated with the Archean Henik Group. Similar rocks located 60 kilometres to the southeast are host to high grade (>10 g/t Au) surface occurrences known as the "SY" group of showings. Nonetheless, no significant shear zones or domains of high strain have been documented on the Property to date, and the observed mafic volcanic rocks are essentially devoid of important alteration minerals that are indicative of Archean mesothermal gold deposits (i.e., sulphides, quartz veining and carbonate). For these reasons the mesothermal gold potential is downgraded, however the presence of Archean metavolcanic sequences suggests gold may be present as a by-product in other deposit types.

As with mesothermal gold, the potential for volcanogenic massive sulphide (VMS) mineralization is considered low. These deposits are typically rich in copper, zinc and lead and are associated with bi-modal (mafic to felsic) volcanic centers. Important examples of this deposit type in Nunavut are the High Lake and Izok Lake deposits located in the central Kitikmeot. Occurrences of these types of deposits in the Kivalliq district are rare but small occurrences have been documented in the Kaminak Lake area approximately 150 kilometres east of the Property. However, no VMS-like known occurrences are known in the Property region and as a result the potential for this style of mineral deposit is considered low.

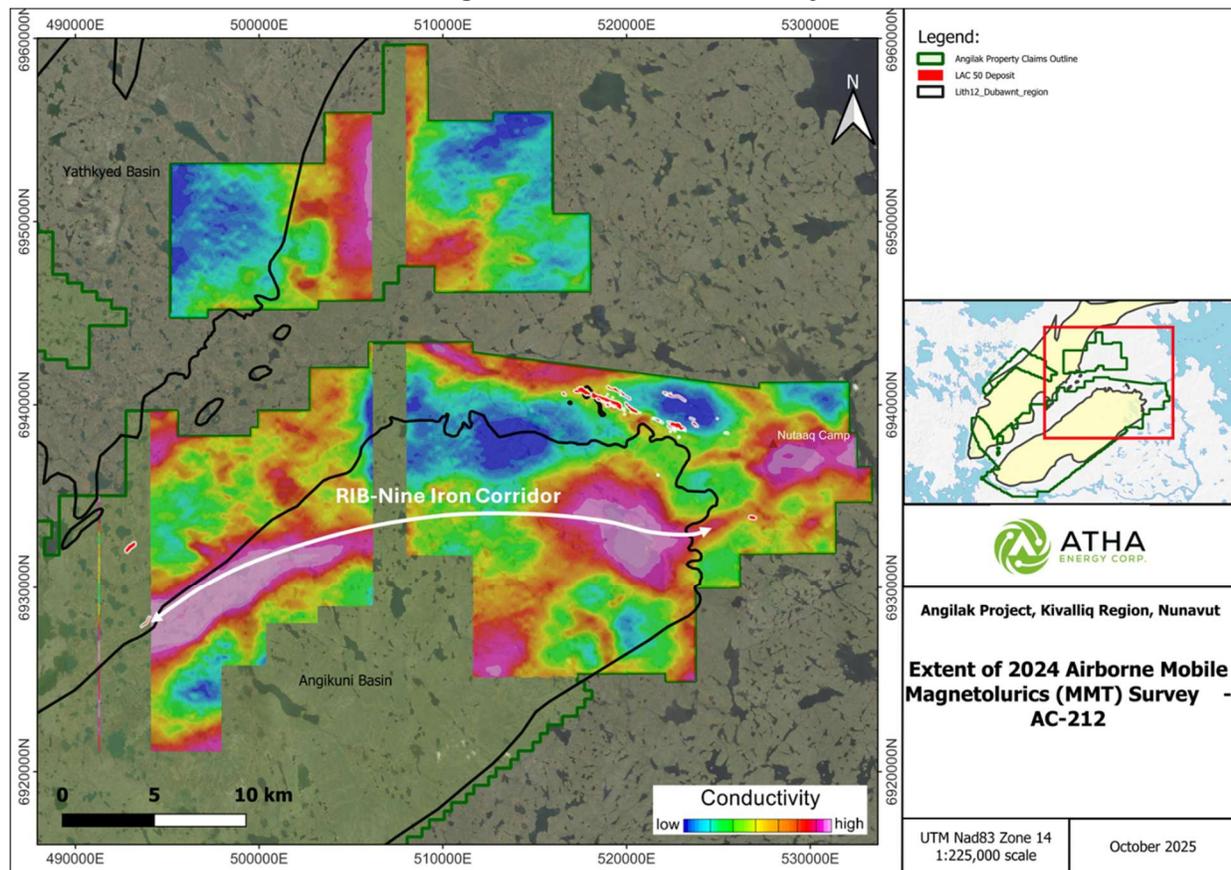
9 EXPLORATION

Exploration activities carried out on the Angilak Project by ATHA includes airborne Mobile MagnetoTellurics (MobileMT) and gravity surveys, local ground gravity and EM surveys, bedrock mapping and a soil sampling survey. Drilling completed in 2024 and 2025 is summarized in Section 10, and work prior to 2024 exploration activities is summarized in Section 6.

9.1 Airborne Geophysical Surveys

ATHA conducted a MobileMT survey between August 26th and September 27th, 2024. A total of 5,815 line-kilometres were flown by Expert Geophysics Limited at 150 metre line spacing and a tie line spacing of 1,500 metres. Terrain clearance for the helicopter and instrumentation was 150 to 160 metres. The data recording rate of ten times per second allows for magnetotelluric, radar, magnetic and GPS measurements to be acquired approximately every 2.2 metres along the survey lines. The results of the survey identified an approximate 31-kilometre prospective conductive corridor transecting the Angikuni Basin designated as the RIB-Nine Iron corridor (Figure 9-1).

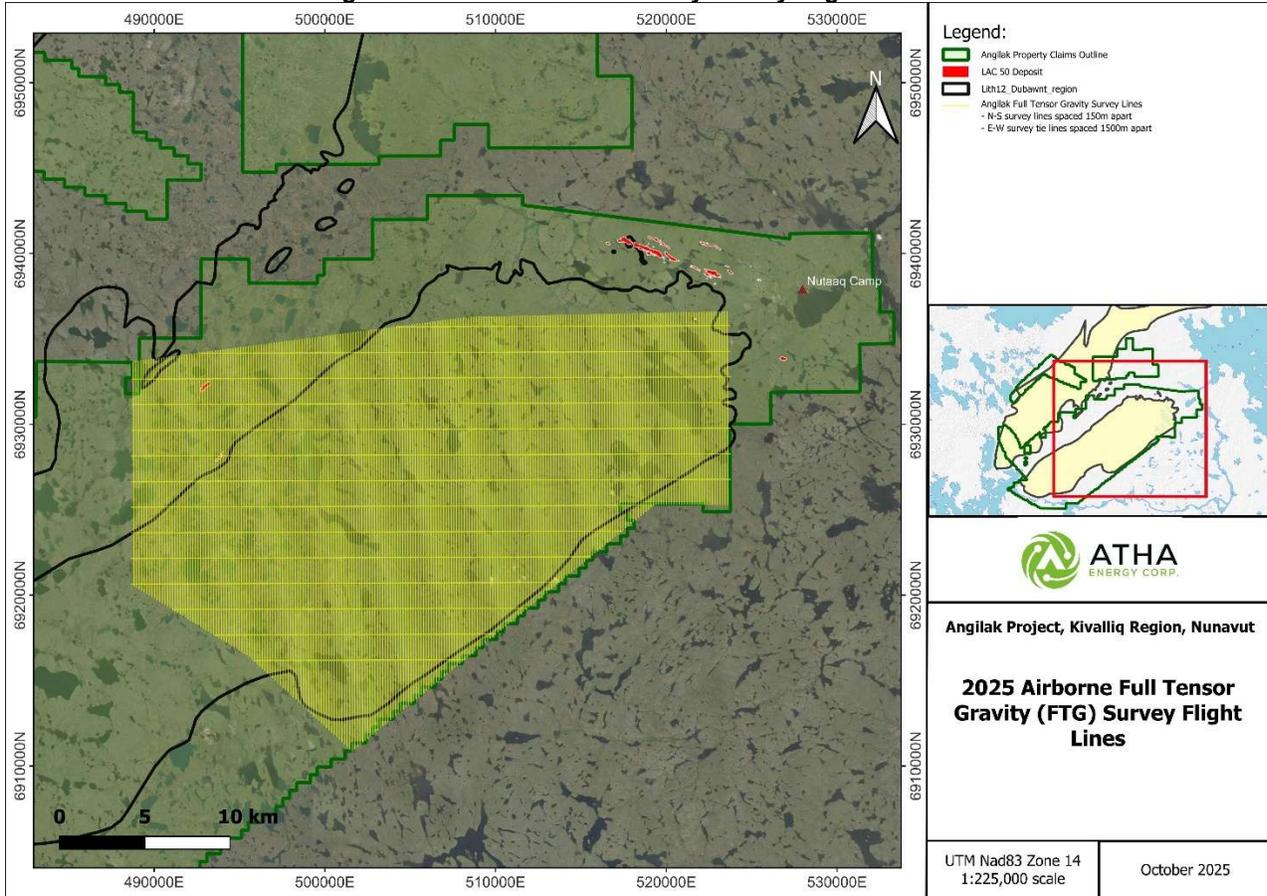
Figure 9-1: 2024 MobileMT Survey



A regional airborne gravity survey was completed between September 5th and September 16th, 2025. A total of 6,908 line-kilometres were flown by Bell Geospace Enterprises LLC at 150 metre line spacing and a tie line spacing of 1,500 metres. The survey acquired Air-Full Tensor Gravity (FTG) and high-resolution airborne magnetic data over the area shown in Figure 9-2. At the time of this report, final processed data from the 2025 airborne gravity survey have not been received

and therefore preliminary interpretation and recommendations are pending an internal review of the final dataset.

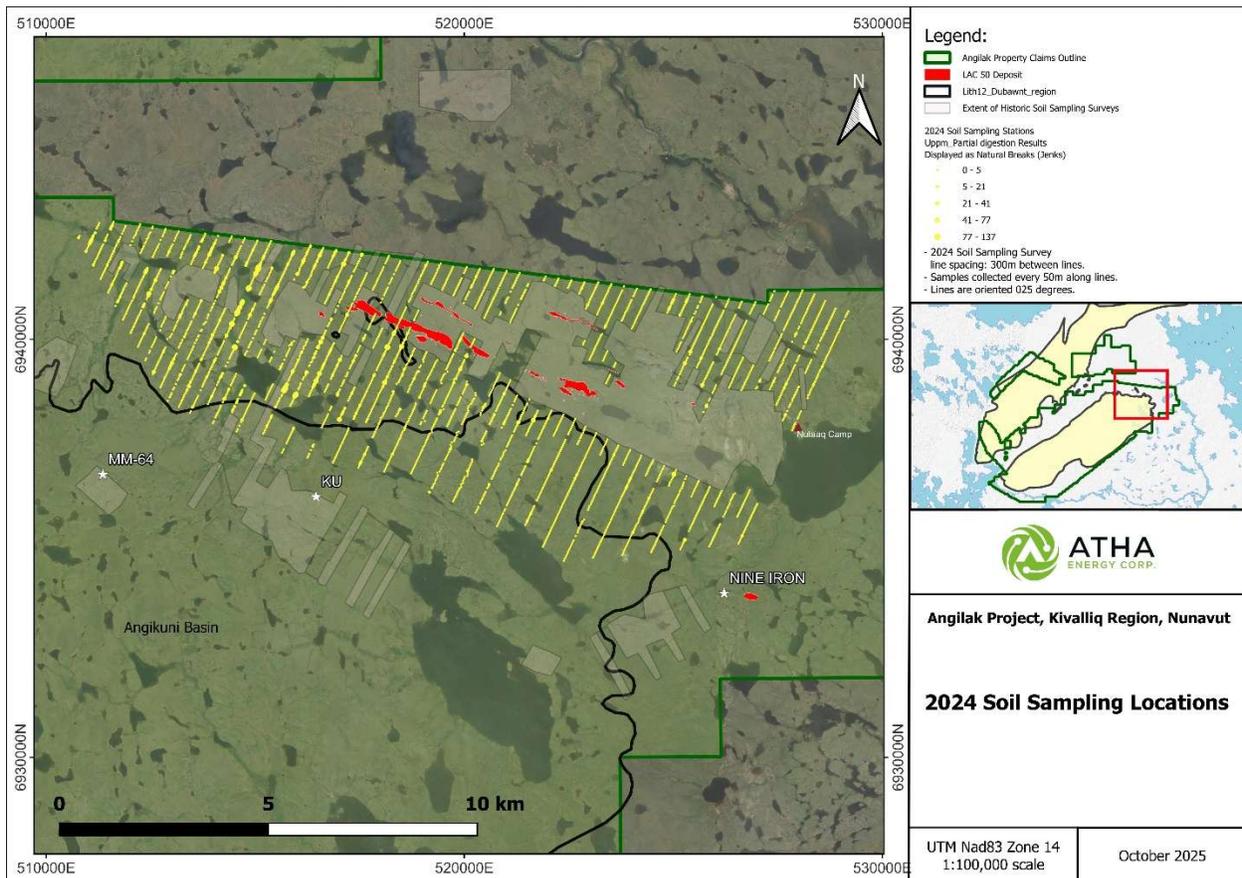
Figure 9-2: 2025 Airborne Gravity Survey Flight Lines



9.2 Geochemical Surveys

A helicopter-supported supplemental soil survey was conducted from August 2nd to August 25th, 2024, focused on completing gaps along the Lac 50 trend not captured in previous soil surveys executed by ValOre from 2008 to 2022. The geochemical survey anomalies coincident with known Very Low Frequency Electromagnetic (VLF) anomalies may highlight and focus future exploration targeting. A total of 3,584 sites were visited and a total of 3,291 samples of the A horizon were collected for analysis, including 130 duplicate samples; 293 sites did not yield any soil samples due to the presence of bedrock outcrop. All samples were submitted to the Saskatchewan Research Council (SRC) Geoanalytical Laboratory for processing. Results of the program are shown in Figure 9-3.

Figure 9-3: Angilak Property Soil Survey Grid Coverage



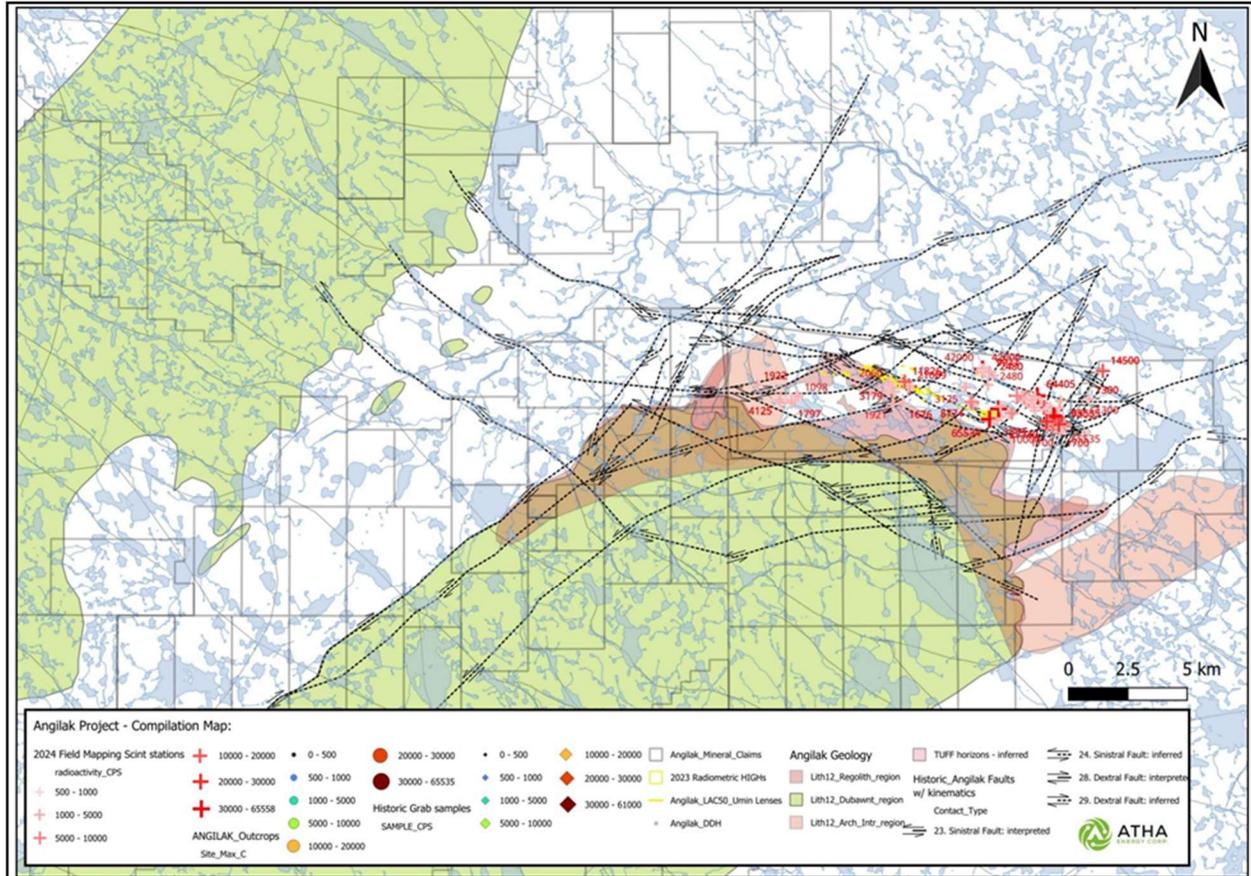
9.3 Geological Surveys

Concurrent with the soil sampling program, a helicopter-supported bedrock mapping program was also completed (Figure 9-4). The program aimed to define structures of interest, identify alteration zones, and verify radiometric anomalies identified in the 2023 airborne radiometric survey (low-level, tight drape, high resolution radiometric and aeromagnetic airborne survey was flown by Inertial, a division of Special Projects Inc. (SPI) out of Calgary, AB on behalf of LUR, from April 28 to May 8, 2023). Five property zones were a focus for the 2024 program:

- Mushroom Lake (ML)
- J4/Ray

- Hot
- Pulse
- Nine Iron

Figure 9-4: 2024 Bedrock Mapping Scintillometer Results (1000 to 65,535 cps)



Bedrock exposure was poor in most areas with the exception the ML Zone, which became the primary focus of the program. One day of mapping was allocated to the Nine Iron area. No rock samples were collected during this program.

The results of the ground program show that the fault system observed in the ML area aligns well with a strike-slip fault regime, consistent with the Riedel shear model. This likely has a significant influence on the control of uranium mineralization in this immediate area and the Lac 50 Deposit. High radioactive anomalies associated with pitchblende-bearing quartz-carbonate breccias are spatially correlated with the main ESE-WNW fault zones, especially where they intersect with northeast trending faults.

9.4 Regional Structural Study

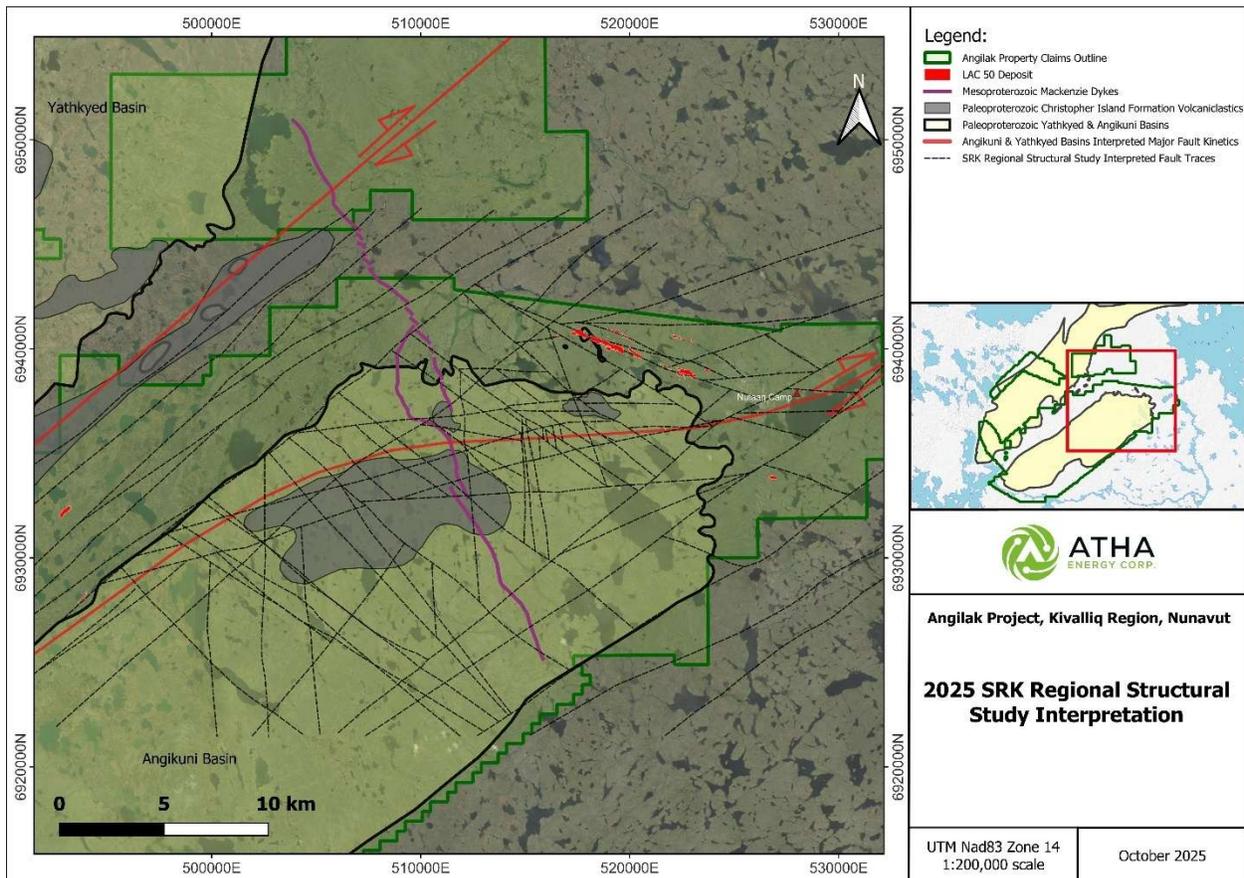
A preliminary regional structural study was initiated by SRK Consulting (Canada) Inc. in November 2024 focused on defining the regional structural architecture of the Angikuni Basin and controls on mineralization. The Angikuni Basin is bounded by dominant NE trending structures, the Ferguson and Angilak Shear zones respectively, which are connected by sigmoidal linkage

structures with EW to WSW trends. Structures show a strong relationship with the Angikuni Basin geometry defined by dominant NE trends and WNW Lac 50 trends (**Figure 9-5**).

Structures can be interpreted as forming an extensional pull apart basin complex, developed within a dextral strike slip system during NW-SE directed contraction. In this setting sigmoidal cross basin dextral transtensional strike-slip shear zones were likely active during basin development. Recent geochronology results suggest that U-mineralization was likely active during basin development and that the structural system identified likely also controlled fluid flow and mineralization, indicating the potential for unconformity and basement styles of mineralization to be found within the Angikuni Basin.

The study also illustrates that the RIB-Nine Iron trend is controlled by a large regional sigmoidal dextral relay structure that along strike may be prospective to host uranium mineralization. In addition, the WNW Lac 50 Deposit parallel trends, identified across the project area (outside of the Lac 50 area), are likely associated with extensional kinematics within this structural setting.

Figure 9-5: Angikuni Basin Preliminary Regional Structural Interpretation



9.5 Ground Geophysical Surveys

9.5.1 Ground TDEM

From April 28th to May 16th a ground, fixed-loop TDEM survey was conducted over the RIB area (**Figure 9-6**). A total of 67.5 line-kilometers were surveyed across 27 lines using 12 different 500m x 500m loops. Lines were spaced 100m and 200m apart, with stations spaced at 50m and 25m. A Geonics TEM57-MK2A transmitter was used to transmit a bipolar square wave at 15Hz. The EMIT fluxgate magnetometer was used to measure X, Y and Z components of the secondary B-field.

The survey was designed to test a broad conductive anomaly measured over the RIB area from the airborne Mobile MT survey conducted in 2024. This broad, conductive anomaly is considerably more conductive than other conductors measured on the Angilak property, and is similar to the conductivities seen in the Athabasca Basin. The goal of the ground TDEM survey was to enhance the resolution of this target and model the data to support drill targeting.

Overall, the survey was successful. The results characterized the conductivity of the RIB area. A total of 7 plates were modelled ranging from the weakly conductive 2.5S to the more strongly conductive 23.2S. Structural information was also interpreted from the plate modelling and were used for initial drill targeting.

9.5.2 Ground Gravity

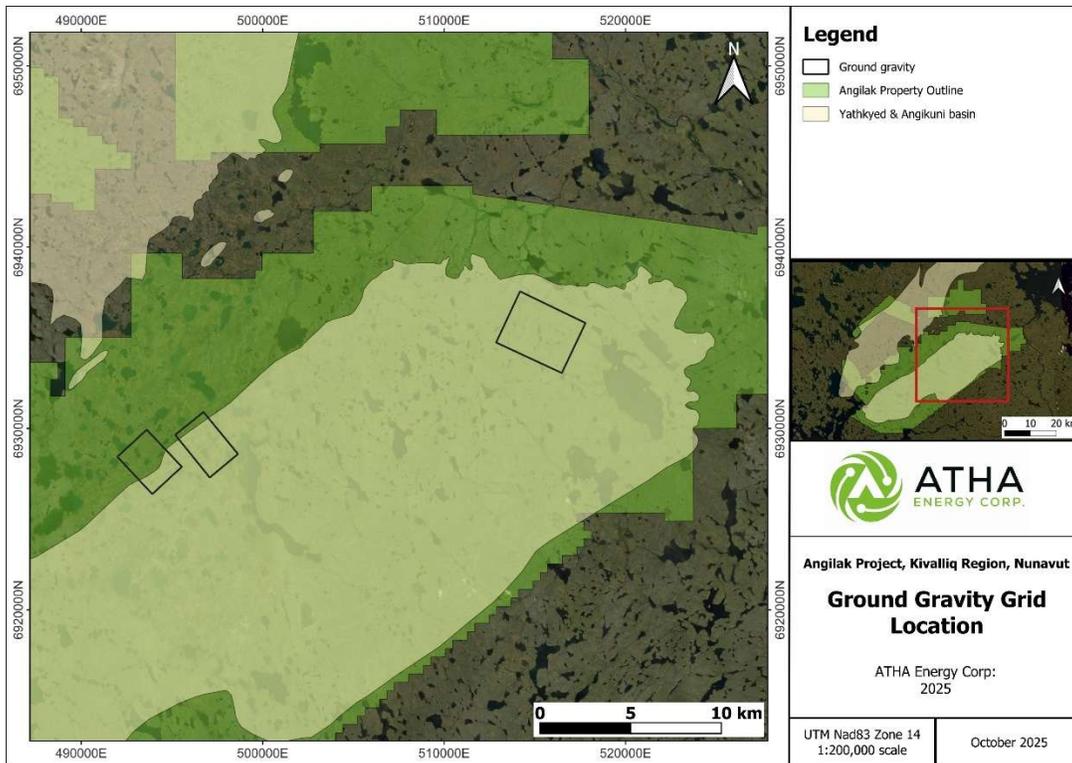
Between April 28th and May 23rd, 2025, ground gravity surveys were conducted at three sub-areas of the Angilak property: Rib, Rib east and KU (**Figure 9-7**). Line spacing was 200m and station spacing was 50m for each of the three areas. The Scintrex CG-6 Autograv was used to measure the vertical component of the gravity field and a Trimble R12i was used to take GPS measurements.

The purpose of the survey was to test the density variations across the respective areas and potentially characterize alteration zones that may host uranium mineralization. The results characterized the basin contact and dip in the Rib area, however, also indicated a low dynamic range of densities across the three areas.

Figure 9-6: 2025 Ground TDEM Grid Location Map



Figure 9-7: 2025 Ground Gravity Grid Survey Location Map



9.6 Exploration Target Model

The QP provided ranges for potential uranium quantity and grade as a target for further exploration on the Lac 50 Deposit. The ranges were derived from interpreted vein wireframes, drill core assays, grade interpolation/extrapolation, and an applied uncertainty range. The stated potential quantity and grade are conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.

The drillhole database provided to the QP contains RC and diamond (core) drilling. The RC drilling results were deemed to be imprecise relative to the validated core drilling results, and, thus, the RC drilling was not considered in the exploration target model. A total of 615 drillholes were considered for the exploration model including 2024 drill holes, representing 105,015 metres of drilling and 12,427 assay samples. Of the 615 drillholes, 453 drillholes were used to define the wireframes.

The wireframes were modelled in Seequent's Leapfrog Geo software (version 2024.1.1) using the assay results and a grade intercept limit equal to or greater than a minimum grade of 0.01 % U_3O_8 , although lower grades were incorporated in places to maintain continuity and represent the structural setting of the mineralized system. Extension distance for the mineralized wireframes was halfway to the next hole, or 200 metres in areas of no drilling, representing the potential at the deposit. No minimum thickness was used for modeling purposes. In total, 34 wireframes were created to represent the Lac 50 Deposit: 14 for Main Zone, 6 for Western Extension, 4 for J4 Zone, 3 for Blaze, 3 for Pulse, 2 for Mushroom Lake, 1 for Hot Zone, and 1 for the Eastern Extension (Figure 9-8 and Figure 9-9).

Figure 9-8: Plan View of Wireframes Underlain with Drillhole Traces

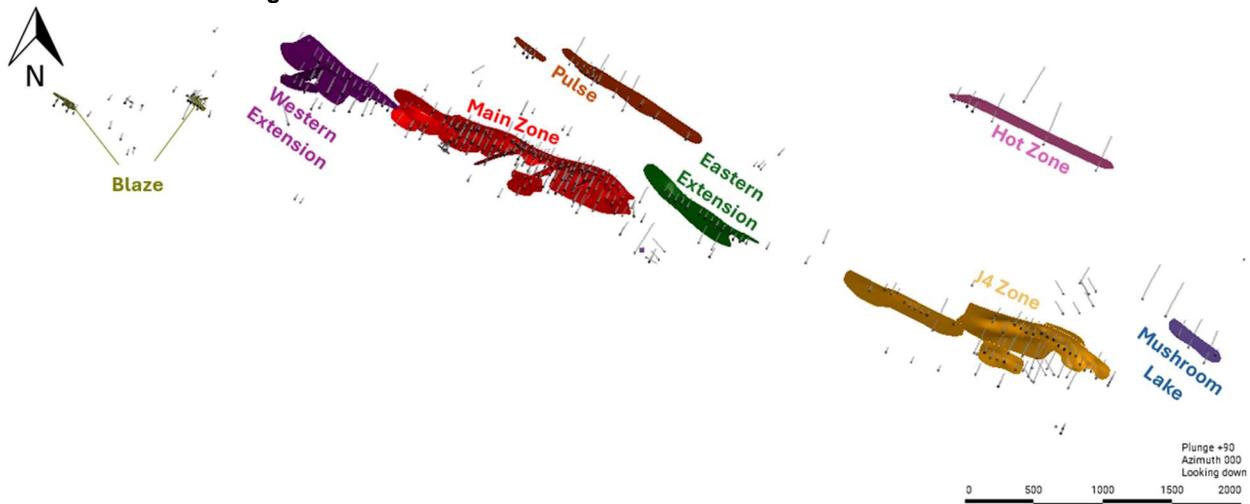
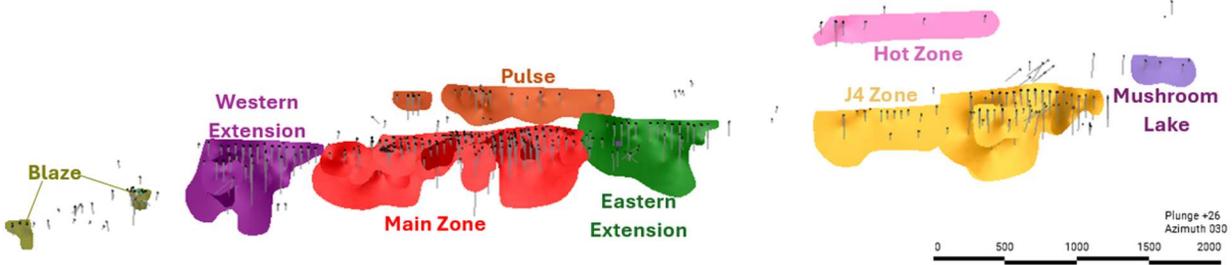


Figure 9-9: Oblique View (azimuth 30 and plunge of +26) of Wireframes Underlain with Drillhole Traces



The wireframes were exported from Leapfrog and imported into Maptek’s Vulcan software (version 2023.1). In Vulcan, uranium grade from drill core assays, after standardization and outlier management, was interpolated/extrapolated in the wireframes. The QP applied an uncertainty range to define a range for potential uranium using the interpolated grade and average density of the wireframes as the midpoint. The grade uncertainty was back calculated from the ranges in tonnes and metal content.

The exploration target is summarized in Table 9-1. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.

Table 9-1: Lac 50 Tabulated Exploration Target Ranges

Lac 50 Exploration Target			
Cutoff (% U₃O₈)	Tonnes (Mt)	Grade (% U₃O₈)	Metal Content (Mlbs U₃O₈)
0.1	7.4 - 9.3	0.37 - 0.48	60.8 - 98.2

Notes:

1. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.
2. The ranges were derived from interpreted vein wireframes, drill core assays, grade interpolation/extrapolation, and applied uncertainty ranges.
3. An assumed cut-off of 0.1% U₃O₈ was used for the tabulation of the exploration target model.

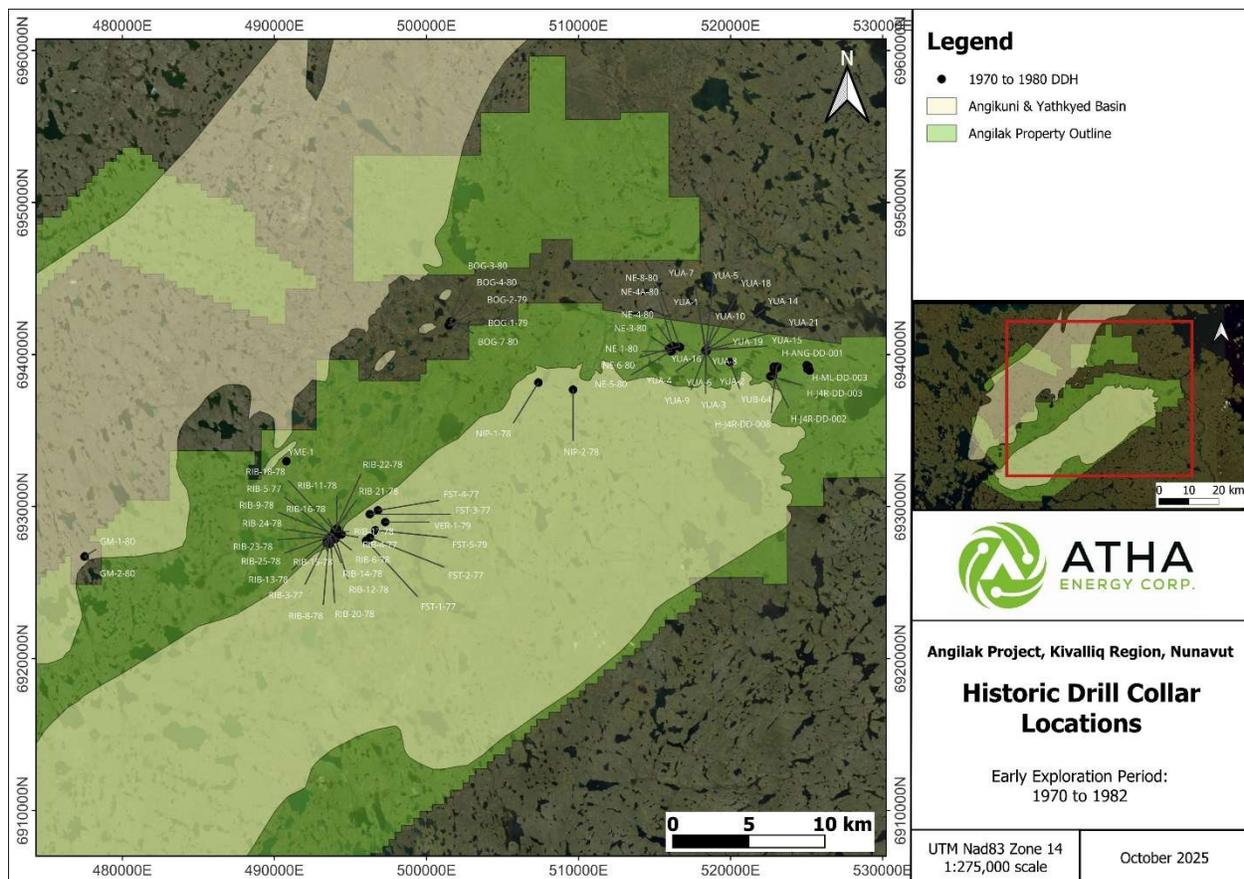
10 DRILLING

Drilling has been conducted on the Property by several previous exploration companies prior to ATHA acquiring the claims in 2024. The historical drilling and ATHA's drilling campaigns are summarized in this section. Only ValOre (Kivalliq) and LUR diamond drilling (2010 to 2023) and ATHA's 2024 drilling were considered in the target exploration model. The historical RC drilling was not used as the results were deemed unreliable.

10.1 Early Exploration Drilling (1970-1982)

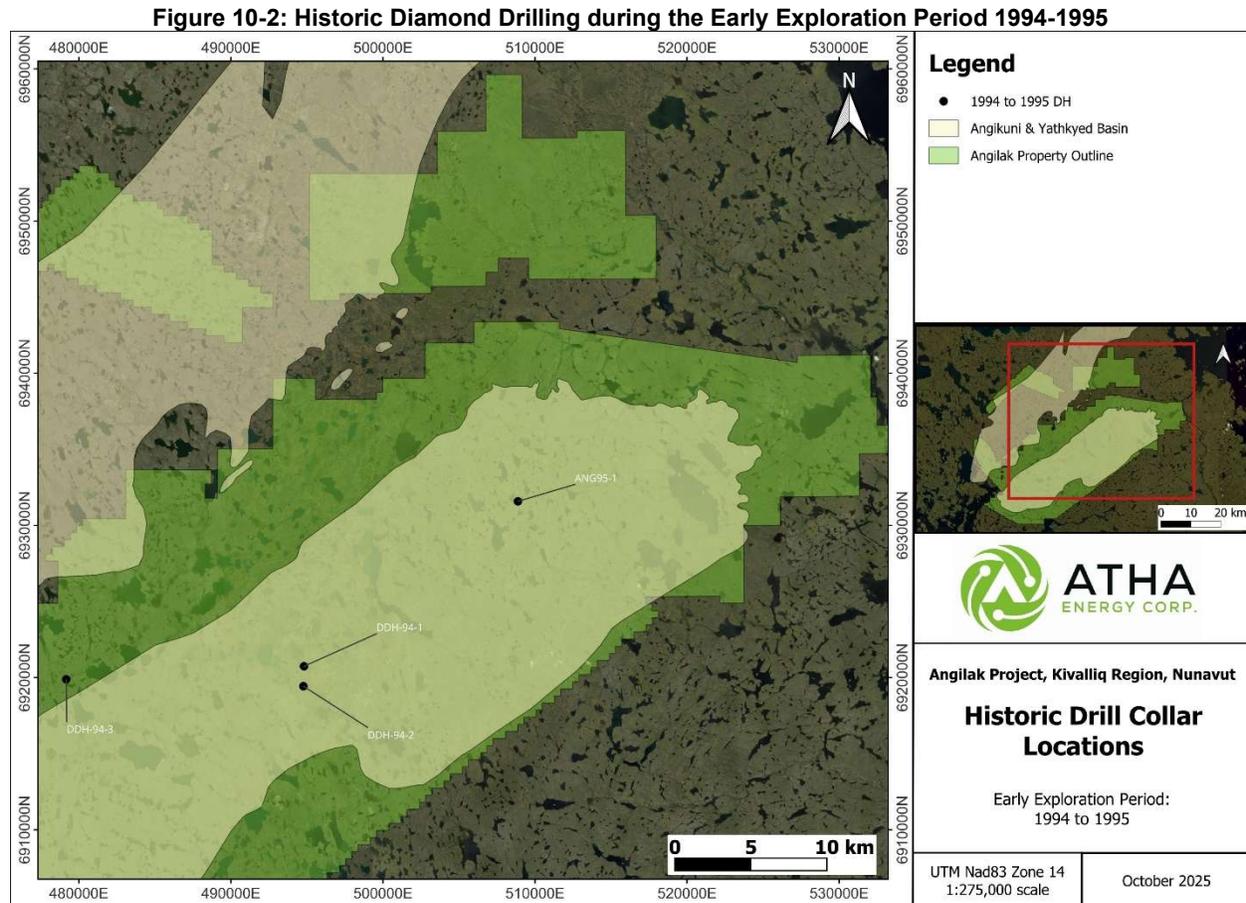
Essex Minerals, Urangesellschaft, Noranda Exploration and Pan Ocean conducted diamond drilling during the early exploration period 1970-1982 (Table 6-1; Figure 10-1). Documentation of drilling done by Pan Ocean (later Aberford Resources) in the late 1970's and early 1980's at the Lac 50 Deposit area is not available in government assessment reports. Miller et al. (1986) reported the presence of several high-grade uranium results from historical drillhole intersections over very narrow widths at the Lac 50 area with drill holes YUC24, YUC25 and YUC26 yielding 4.45% U_3O_8 over 0.62 m, 4.75% U_3O_8 over 0.72 m and 5.81% U_3O_8 over 0.62 m, respectively. The early exploration historical drilling is described further in Setterfield (2007) and Dufresne and Shoeman (2024).

Figure 10-1: Historic Diamond Drilling during the Early Exploration Period 1970-1982



10.2 Early Exploration Drilling (1994-1995)

Three of the four diamond drillholes completed in 1994-1995 by Western Mining Corporation fall partially or entirely with the current Property boundaries (Figure 10-2).



10.3 Kivalliq Energy Corp. Drilling (2008 to 2018)

10.3.1 Drilling Programs Summary

Historic drilling by Kivalliq Energy is summarized in Table 10-1 and the drillhole collars and traces are shown in Figure 10-3. Drillhole parameters and significant highlight sample intervals and results are provided in Appendix A and B, respectively.

The objective of the diamond drill program in 2009 was to verify and test the continuity of the Lac 50 Deposit. Of the 16 holes drilled, 15 drillholes targeted the Lac 50 Deposit, and 12 drillholes intersected intervals of significant uranium mineralization. The drill program results showed that the “Main Zone” of uranium mineralization is relatively predictable, dipping approximately 70° degrees to the south with a strike of 116°.

In 2010, 107 drillholes were completed for a total of 16,600 metres. Of these holes, 103 drillholes targeted the Lac 50 Deposit and 88 drillholes intersected anomalous uranium mineralization.

During 2011, a reconnaissance RC drill program was completed on the Property, where 88 RC holes were completed. Anomalous intersections in the Lac 50 Deposit area were followed-up with diamond drilling. A total of 153 diamond drillholes were completed targeting the Lac 50 Main Zone along with its eastern and western offset extensions (Figure 10-4), and reconnaissance drilling targeting the Blaze, Ag, J9, Joule-Mushroom Lake, Pulse and Spark prospect areas.

In 2012, the diamond drill program targeted the Lac 50 Main Zone, the J4/Ray zone, the Pulse zone, and the Nine Iron zone with a total of 172 drillholes completed. Thirty-eight RC drillholes were completed as an exploration tool to target areas with geophysical or geochemical anomalies identified in previous exploration programs.

During 2013, 12 drillholes targeting the J1 Zone, Mushroom Lake (ML), and J4 West Zone were completed. Four drillholes were completed at the ML Zone with two of the drillholes targeting the ML “EM” conductor intersecting intervals of anomalous uranium mineralization, including a 1.2 metre core-length interval of 1.42 % U₃O₈ in hole 13-ML-001. Seven diamond drillholes targeted the J1 zone, which is an approximate one-kilometre-long VLF-EM conductor located 800 metres to the west of the J4 deposit.

The 2015 diamond drill program focused on the Dipole target, with the objective of testing a prominent VLF-EM conductor and coincident uranium-in-soil anomaly. Nine drillholes were completed, successfully delineated a 25 to 48-metre-wide area of steeply dipping zones of mineralization that extend approximately 150 metres along strike, with multiple mineralized intervals being encountered in all holes. Hole 15-DP-009 returned the highest assay interval of 2.34 % U₃O₈, 1.14 % Mo, and 44 g/t Ag over 1.3 metres.

No drilling was conducted by Kivalliq in 2014 or 2016 to 2018.

Table 10-1: Summary of Kivalliq Drilling 2008 - 2018

Year	Diamond Drilling		RC Drilling	
	# Of Holes	Meterage (m)	# Of Holes	Meterage (m)
2008	0	0	0	
2009	16 (15)	1,745	0	
2010	107 (103)	16,600	0	
2011	153	23,849	88	6,411
2012	172	33,583	38	5,273
2013	14	2,100	0	
2014	0	0	0	
2015	9	958	0	
2016-2018	0	0	0	

Note: the number of holes that were drilled at Lac 50 Deposit is in brackets

Figure 10-3: Historic Drilling by Kivalliq Energy between 2008-2018

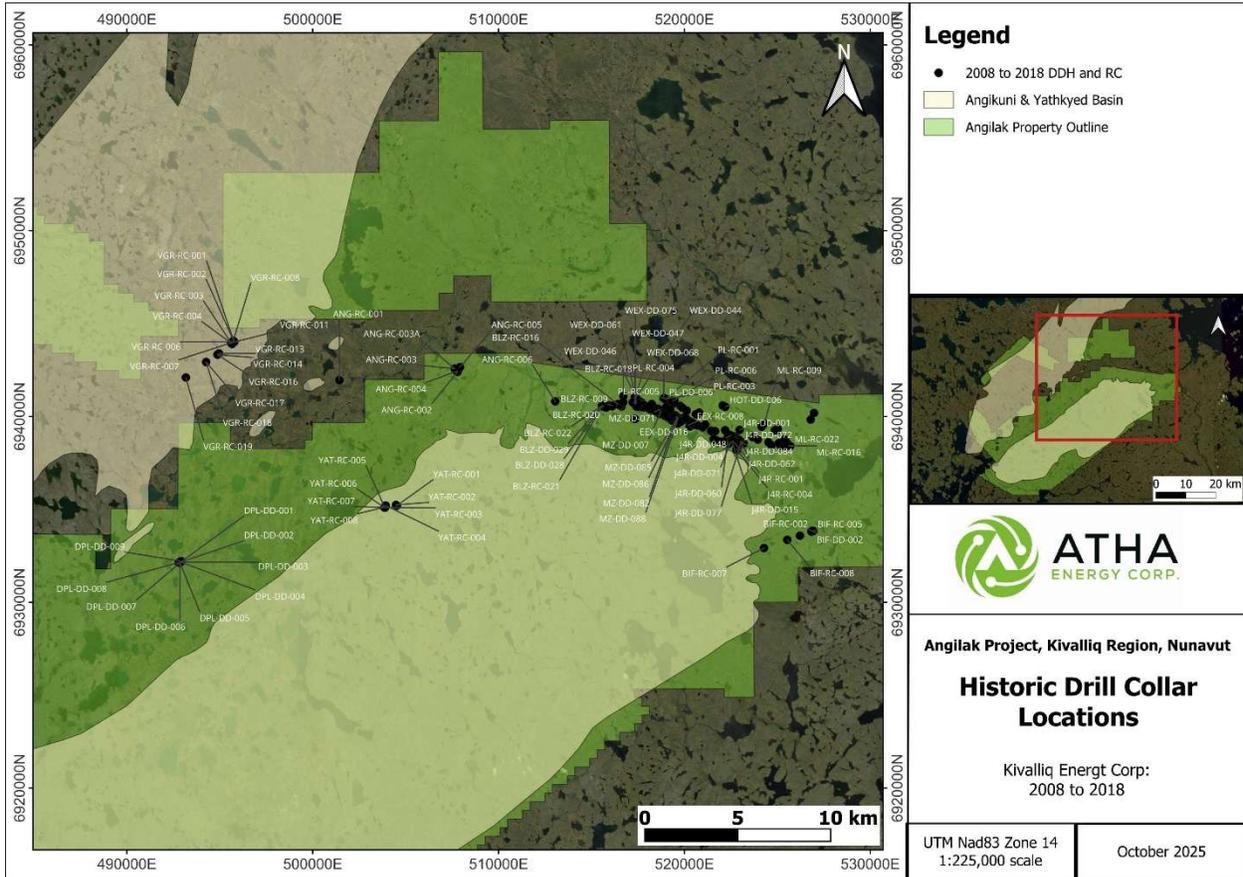
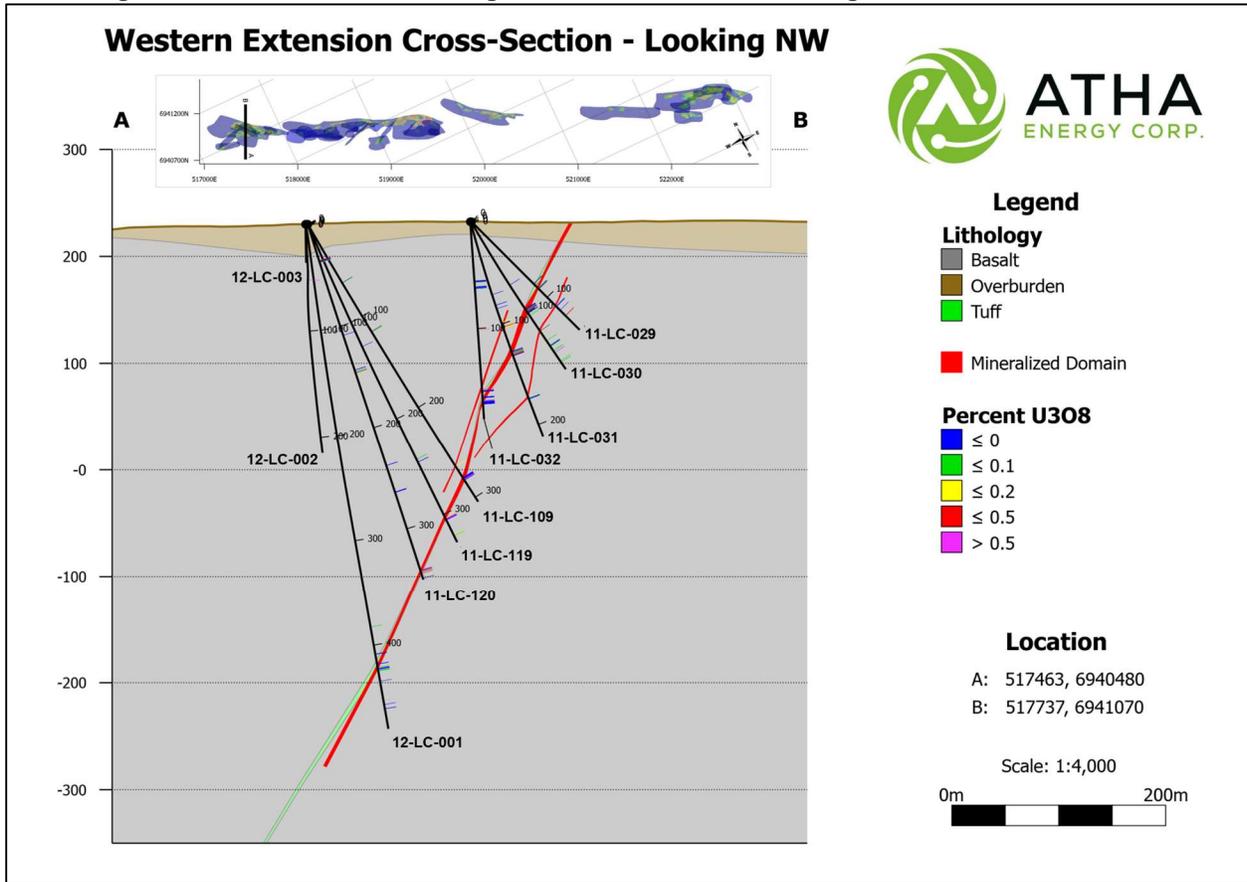


Figure 10-4: Cross-section showing 2011 and 2012 diamond drilling in the Western Extension



10.3.2 Drillhole Surveying

Drilling in 2009 and 2010 was conducted by Major Drilling using Boyles 37 and 17 heli-portable rigs under the supervision of Kivalliq Energy and APEX personnel. New drill holes were positioned using a handheld Garmin GPS unit in UTM NAD 83, zone 14 coordinates, on 25 m, 50 m and 100 m drill sections using a 116-degree historic baseline as a reference. At the end of the 2010 program, a legal land surveyor from Sub Arctic Surveyors based out of Yellowknife NT was flown to site to survey all marked collar locations.

Drilling in 2012 was also conducted by Major Drilling. Drill collars were spotted using a handheld GPS and a compass. Drillholes were surveyed down hole using a Reflex EZ-Shot at the end of each hole. On deeper holes, single shot survey readings were also collected every 100 m as the hole progressed. Downhole survey data were also plotted by field personnel in real time. Any downhole data that showed unrealistic hole orientations were considered suspect, and the data was not used. The drill collars were subsequently surveyed with a differentially corrected GPS by MEG Systems Ltd.

Survey procedures in 2013 and 2015 were consistent with the 2012 program.

10.4 ValOre Metals Corp. Drilling (2022)

10.4.1 Drilling Program Summary

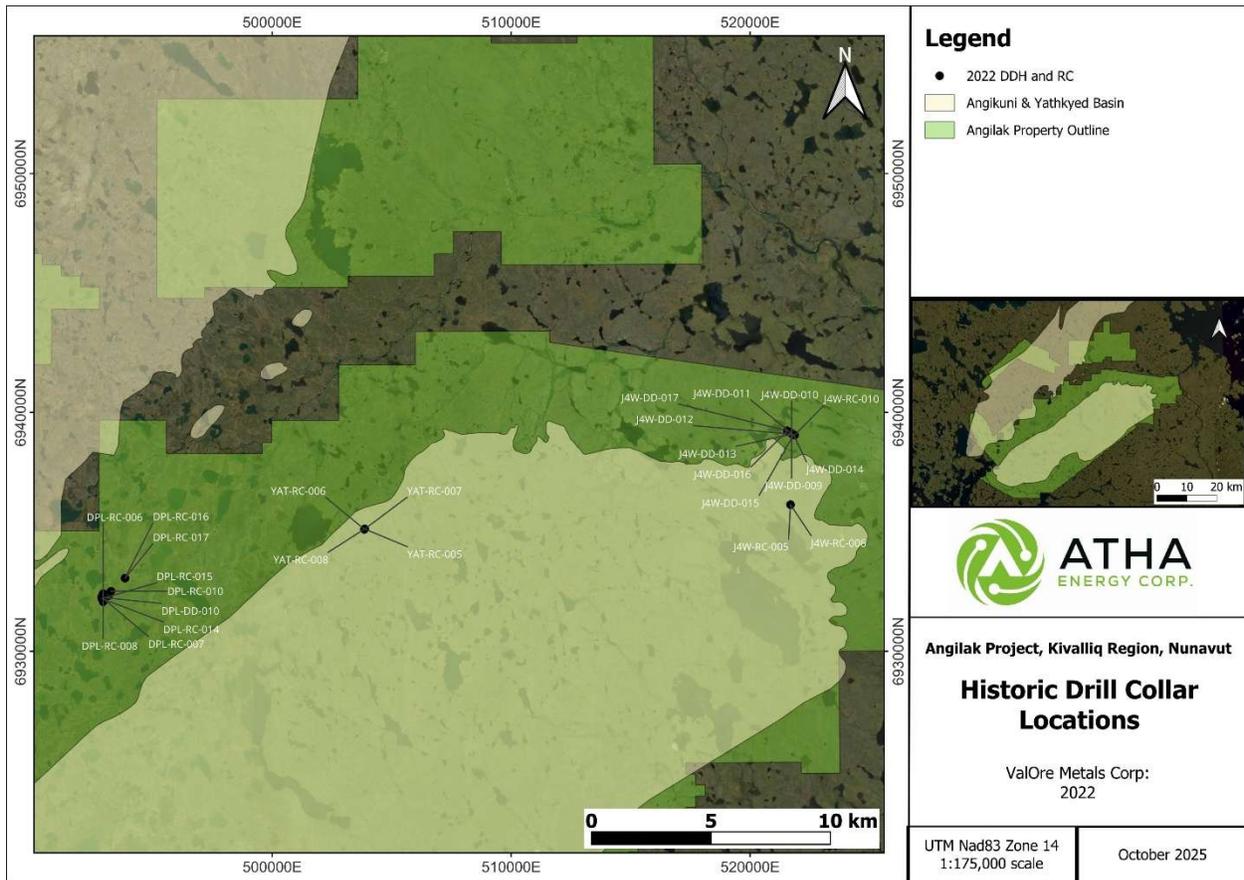
No drilling was conducted by ValOre between 2018 and 2021. Drilling by ValOre in 2022 is shown in Figure 10-5. Drillhole parameters and significant highlight sample intervals and results are included in Appendix A and B, and the significant results are summarized below from Dufresne and Schoeman (2024).

A twenty-seven hole, 3,165 metre RC program was conducted during the spring of 2022 focusing on Dipole, J4 West and Yat zones. A total of seventeen RC holes were completed at the Dipole target with the objective to test the extension of mineralization along strike to the northeast along the coinciding VLF-EM conductor and uranium-in-soil trends, as well as testing the down-dip extension of the shallow uranium mineralization encountered in the 2015 diamond drillhole program (ValOre News Release, 2022). Fourteen out of 17 holes drilled intersected shallow uranium mineralization ranging in interval widths of 1.5 to 22.9 metres along with wide zones of Ag-Mo-Cu mineralization in multiple holes. A total of six RC holes were completed at the J4 West target. Mineralization at J4 West was observed to be associated with a sheared section of hematite-altered, graphite and sulphide bearing tuff hosted within a foliated basalt and gabbro sequence (ValOre News Release, 2023a). Four of the six RC drillholes encountered anomalous uranium mineralization at the central and western zones, with two RC holes drilled at the eastern extent which did not intersect anomalous uranium mineralization. RC holes RC22-J4W-001 and RC22-J4W-002 intersected U_3O_8 intervals above 0.20%.

Four RC holes were completed at the Yat target. The objective of the Yat program was to test at depth the high-grade polymetallic Pd-Pt-Au-Ag-U results returned from the trench channel and boulder sampling program carried out in 2016. Three out of four drillholes intersected shallow zones of Cu-Ag mineralization as well as local zones of anomalous uranium mineralization (ValOre New Release, 2023a). The high-grade polymetallic mineralization encountered in the 2016 sampling program is interpreted to be confined to discrete, discontinuous veins hosted in the Proterozoic sedimentary and volcanic rocks of the Angikuni Basin.

A twenty-six hole, 3,590 metre diamond drillhole program was completed in the summer of 2022 focused on the Dipole and J4 West zones. At the Dipole target, a total of sixteen diamond drillholes were completed. The objective of the drill program was to test the extension potential northeast along strike of the drilling completed in 2015, as well as following up on the diamond drilling in 2015 and RC drilling in 2022 to test mineralization extension at depth. Fourteen holes encountered anomalous uranium mineralization ($>0.01\% U_3O_8$), while the remaining two holes were discontinued at 16 and 6 metres depth due to poor drilling conditions (ValOre New Release, 2023b). The 2022 diamond drilling results strengthened the interpretation that Dipole is geologically similar to the Lac 50 Deposit area, where the uranium mineralization is associated with sheared to brecciated pitchblende-sulphide bearing graphitic tuffs hosted within sequences of Archean mafic-intermediate volcanics (ValOre New Release, 2023b). Intervals of uranium mineralization were encountered at vertical depths of approximately 15 to 250 metres. At the J4 West target, ten diamond core holes were completed. The objective of the diamond drill program was to further test the potential for a sinistral off-set and continuation of mineralization to the southwest of the J4 deposit. Detailed logging of core from J4 West identified mineralization styles, alteration assemblages, and host lithologies bearing strong similarity to those observed at the J4 deposit.

Figure 10-5: Historic Drilling by ValOre in 2022



10.4.2 Drillhole Surveying

ValOre contracted 518 Drilling Ltd. from Woodlands, MB, to conduct the 2022 diamond drilling program using two Boyles 37 heli-portable drill rigs staged at the Property from previous drilling undertaken during 2009-2015. Drill pads were initially located with the use of handheld GPS and a Devico DeviSight with dual GPS to accurately measure and record location and azimuth. Once set up on the pad, the drill was aligned to the correct azimuth and dip by a geologist using a Devico DeviAligner followed by a check with a Brunton compass. After completion, drillholes were surveyed using a Stockholm Precision Tools (SPT) MagCruiser configured in a multi-shot setting. Survey data were recorded at 9 metre intervals. The MagCruiser records inclination, magnetic azimuth, magnetic field and temperature. Surveys were repeated or discarded if the down-hole data showed unrealistic hole orientations or a magnetic field more than 80,000 nT.

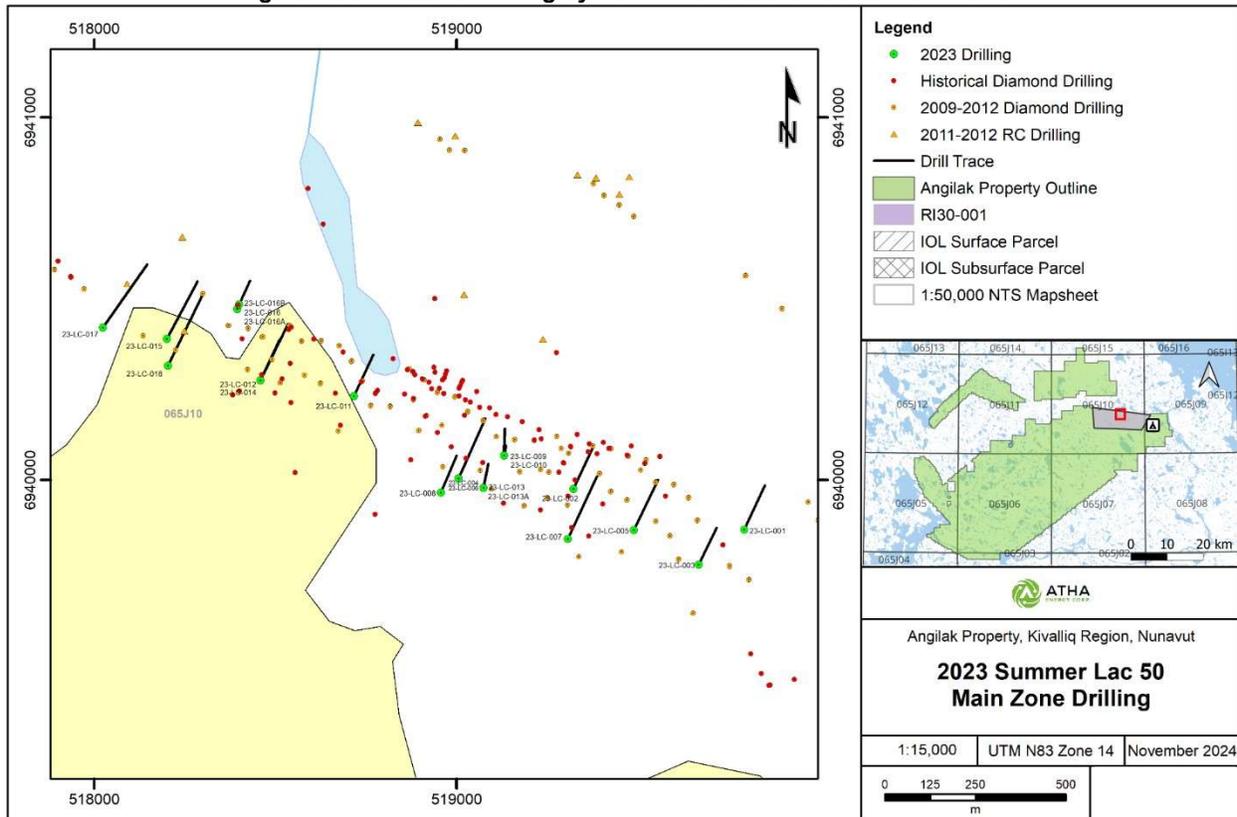
10.5 Latitude Uranium Inc. Drilling (2023)

10.5.1 Drilling Program Summary

Between July 4 and September 4, 2023, LUR completed a diamond drill program in the Lac 50 Deposit area specifically targeting the Main Zone. The 2023 drilling program successfully increased the extent of known mineralization and identified new mineralization horizons. A total of 18 diamond drillholes, with 3 restarts, were completed totalling 5,662 metres.

A location map of 2023 drillhole collars and traces relative to other historical drilling in the area is presented in Figure 10-6. Drillhole parameters and significant highlight sample intervals and results are included in Appendix A and B, and the significant results are summarized below from Dufresne and Schoeman (2024).

Figure 10-6: Historic Drilling by LUR in 2023 at the Lac 50 Main Zone



The east and central areas of the Main Zone were tested with eleven holes designed to test the continuity and extension of known mineralization.

- Drillholes 23-LC-001 and 23-LC-003 were drilled to test the potential continuity of mineralization from the easternmost end of the Main Zone to potential link with mineralization in the Eastern Zone. The holes targeted a coincident magnetic high with a low to moderate VLF anomaly. Drillhole 23-LC-003 was located approximately 60 metres from historical drillhole 11-LC-014 in the Main Tuff Horizon and was favorably associated with structure and alteration. Drillhole 23-LC-003 intersected weak uranium mineralization.
- The best result from the 2023 campaign was obtained from drillhole 23-LC-005, which tested mineralization continuity approximately 50 metres down-dip from historical hole 10-LC-089 and infilled a 100-metre gap in the historical drilling. Hole 23-LC-005 intersected 7.54% U_3O_8 over 1.6 metres starting at 218.0 m, demonstrating mineralization continuity down-dip.
- Drillholes 23-LC-002, 23-LC-007, 23-LC-009, and 23-LC-010 were designed as infill holes, step outs from known mineralization and to test the down-dip and up-dip potential of plunging mineralization trends within the Main Zone. All holes intersected uranium mineralization in either the tuff unit associated with the Main Zone or proximal parallel tuff horizons.

- Drillhole 23-LC-013 located within an east-northeasterly, cross-cutting structural corridor intersected mineralization within the Main tuff horizon at depth, as well as multiple shallow mineralized intervals within the Main Zone hanging wall.
- Drillholes 23-LC-004, 23-LC-006 and 23-LC-008 are located centrally along the southern side of the Main Zone within an east-northeasterly cross-cutting structural corridor. The holes were designed to target mineralization associated with a prominent tuff layer in the hanging wall of Main Zone and assess the influence of the east-northeast trending, cross-cutting structures. All three holes intersected shallow uranium mineralization and defined a new lens of near surface mineralization hosted within a parallel tuff horizon. Additional drilling is required to further delineate the parallel hanging-wall mineralized horizon.

The west side of the Main Zone was targeted with seven drillholes designed to test mineralization continuity along strike and at depth.

- Drillhole 23-LC-011 was drilled within a 100-metre gap along strike and 40 metres down-plunge of historical hole 09-LC-006, intersecting mineralization and confirming down-dip continuity.
- Drillholes 23-LC-012 and 23-LC-014 intersected multiple zones of mineralization at depth in the Main Tuff Horizon. The highest grades of coincident Cu and U₃O₈ in drillholes 23-LC-012 and 23-LC-014 were observed in the shallowest intervals associated with a fault zone at the base of a mappable conglomeritic unit.
- Four holes (23-LC-015, 23-LC-016, 23-LC-017 and 23-LC-018) located at the far west side of the Main Zone were all successful in expanding the footprint of mineralization intersecting anomalous mineralization at greater depths than historical drilling

10.5.2 Drillhole Surveying

Latitude contracted 518 Drilling Ltd. from Woodlands, MB, to conduct the 2023 drilling using two Boyles 17 heli-portable drill rigs. Drill collar and down-hole survey procedures followed were consistent with those applied during ValOre's 2022 program.

10.6 ATHA Drilling

10.6.1 2024 Drilling Program Summary

ATHA conducted a 10,052-metre helicopter-supported diamond drill program between June 4 to August 22, 2024. A total of twenty-five drillholes were completed, not including one lost drillhole. ATHA's drillhole targeting focused on expansion of the mineralization footprint of the Lac 50 Deposit and testing of high-priority targets on parallel structures to the Lac 50 trend that were previously identified as being prospective to host uranium mineralization. The project objectives are described below:

- Test along strike, down-dip and up-dip, and expand on existing mineralized lenses in the Western Extension, Eastern Extension and Main Zone area;
- Expand down-plunge and along strike of mineralized lenses in the J4 and Ray zone; and
- Test parallel mineralized corridors and VLF anomalies in close proximity to the main Lac 50 trend with limited historical drillholes (i.e. Hot, Pulse and Mushroom Lake Zones).

The project location, target areas, and 2024 drillhole traces are shown in Figure 10-7. Table 10-2 summarizes the drillhole parameters for the 2024 program and the drill collar locations are shown in Figure 10-8. Significant highlight sample intervals and results are included in Appendix B, and the significant results of the program are summarized below.

Cross-sections showing drilling results at the key zones/ trends and selected strip logs from the program are presented in the following sections. The cross-sections also show the results of historical drilling described above. All depths and sample intervals for the 2024 program are metres downhole; true thicknesses are yet to be determined.

Figure 10-7: 2024 Drill Target Area and Historical VLF Survey

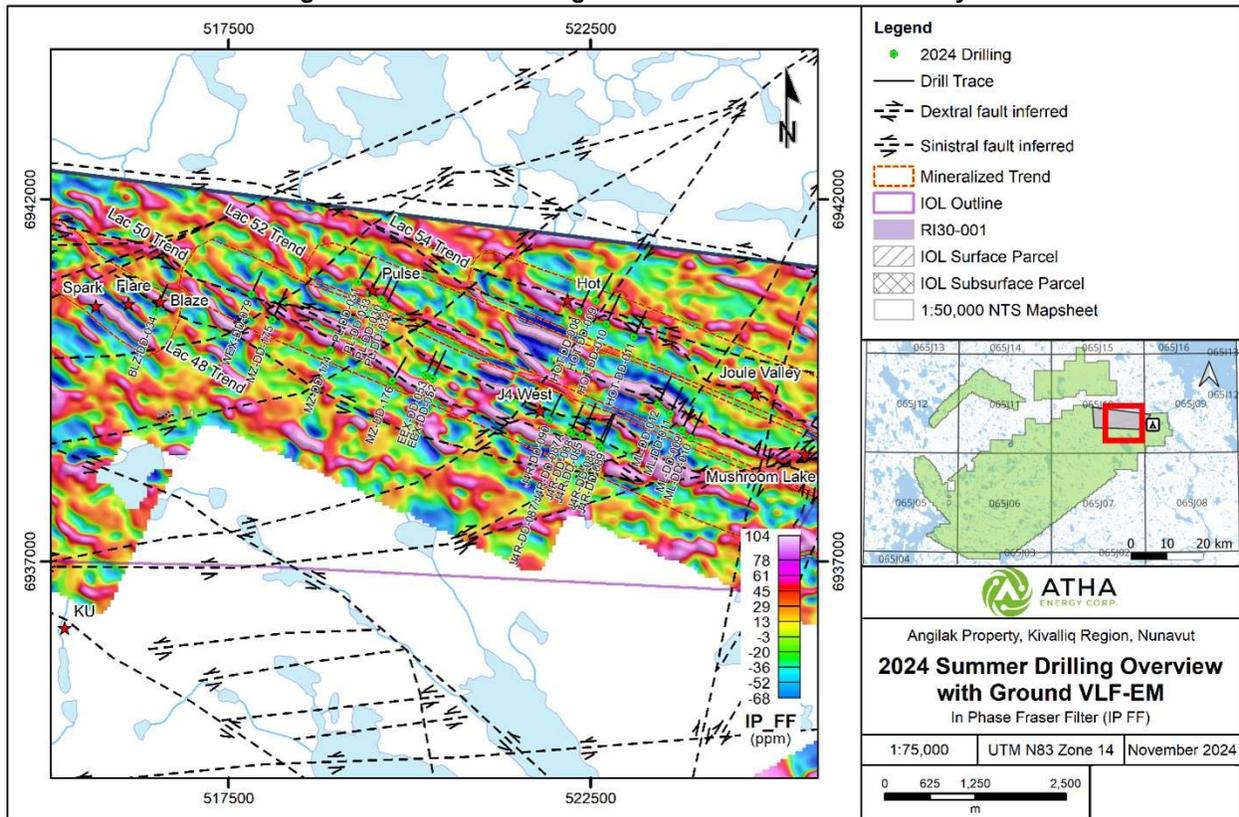
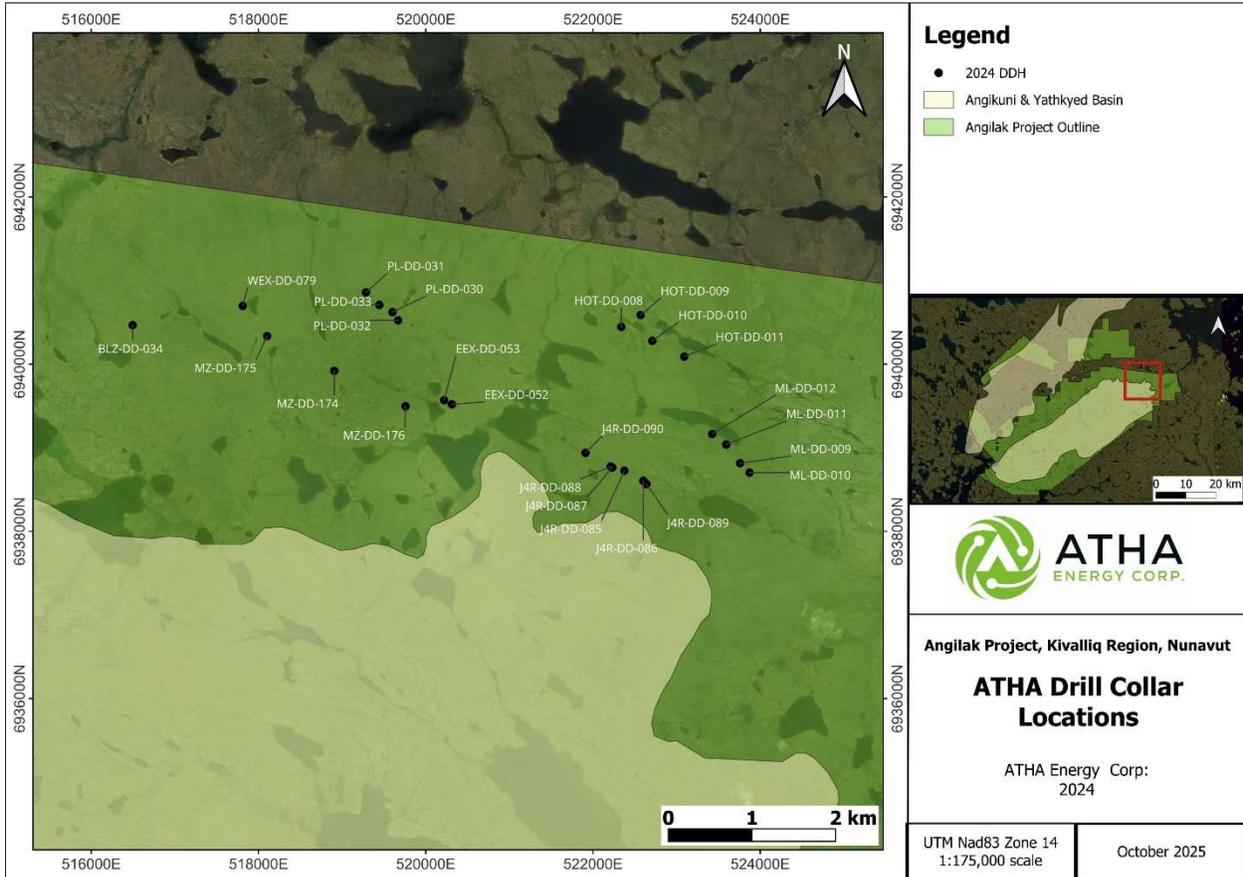


Table 10-2: Summary of ATHA 2024 Drillhole Details

Hole ID	Zone	Azi (°)	Dip (°)	Easting (m E)	Northing (m N)	Elev (m)	Final Depth (m)
MZ-DD-174	Main Zone	25	-79	518906.8	6939921.1	212.7	600.0
BLZ-DD-034	Blaze	24	-58	516495.6	6940468.9	226.4	390.0
MZ-DD-175	Main Zone	25	-50	518105.2	6940336.7	230.0	471.0
EEX-DD-052	Eastern Extension	25	-70	520313.3	6939521.3	194.9	427.3
J4R-DD-085	J4-Ray	23	-68	522376.8	6938726.7	216.7	490.8
EEX-DD-053	Eastern Extension	23	-65	520222.9	6939569.0	193.1	360.0
WEX-DD-079	Western Extension	25	-55	517815.1	6940690.9	236.1	334.8
J4R-DD-086	J4-Ray	25	-50	522603.5	6938605.8	207.6	468.5
J4R-DD-087/ J4R-DD-087A	J4-Ray	25	-55	522231.4	6938765.4	217.3	40.9
MZ-DD-176	Main Zone	30	-50	519760.1	6939494.0	196.3	373.8
J4R-DD-088	J4-Ray	25	-55	522212.1	6938776.4	217.0	419.4
PL-DD-030	Pulse Zone	25	-55	519604.9	6940622.6	213.2	360.2
HOT-DD-008	Hot Zone	25	-55	522339.2	6940447.7	184.6	355.8
PL-DD-031	Pulse Zone	30	-55	519286.2	6940859.7	225.3	380.1
J4R-DD-089	J4-Ray	22	-50	522640.9	6938566.4	204.0	502.9
PL-DD-032	Pulse Zone	25	-50	519670.5	6940525.5	212.5	416.0
HOT-DD-009	Hot Zone	30	-50	522570.4	6940589.5	183.8	450.0
PL-DD-033	Pulse Zone	30	-50	519445.8	6940710.4	220.9	392.0
ML-DD-009	ML Zone	25	-50	523760.1	6938816.2	211.4	351.1
J4R-DD-090	J4-Ray	25	-52	521922.6	6938952.3	203.2	395.0
ML-DD-010	ML Zone	25	-50	523873.3	6938704.2	209.4	389.1
HOT-DD-010	Hot Zone	25	-50	522710.6	6940280.4	185.7	452.0
ML-DD-011	ML Zone	25	-50	523594.8	6939040.6	209.8	400.5
HOT-DD-011	Hot Zone	25	-50	523094.6	6940095.4	199.1	400.0
ML-DD-012	ML Zone	25	-50	523426.3	6939165.7	207.9	430.3
					Total Metres		10,051.45

Figure 10-8: ATHA 2024 Diamond Drilling

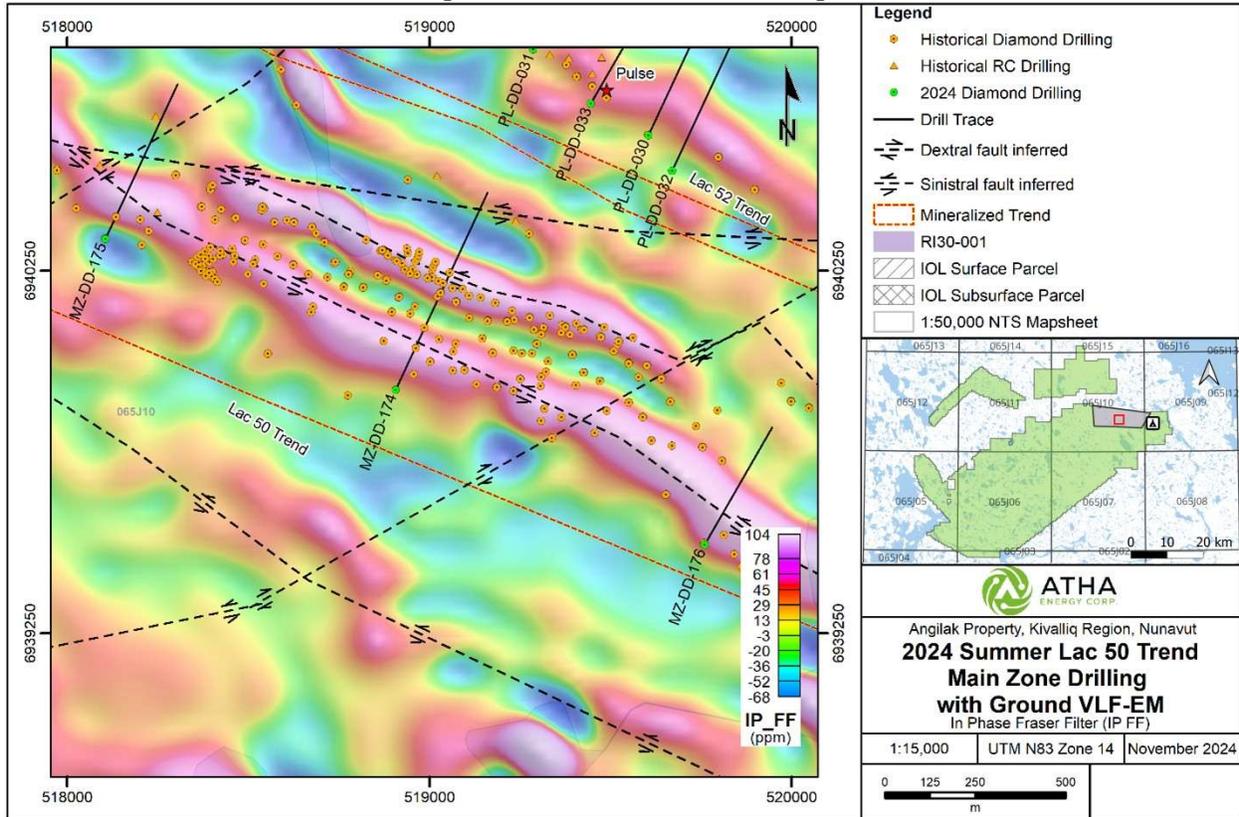


Main Zone Drilling

The Lac 50 Main Zone mineralization is structurally and stratigraphically controlled within a sulphidic-chloritic-graphitic tuffaceous metasediment and/or volcanoclastic interlayered within Archean basement metavolcanics, of which the protoliths are typically massive basalt, pillowed basalt and subvolcanic gabbro. These lithostructural characteristics are typical of all deposits and showings in the Lac 50 area. The objective was to evaluate along strike and downdip and expand the footprint of mineralization in the Main Zone area.

Three drillholes were completed in this area: MZ-DD-174, MZ-DD-175 and MZ-DD-176, for a total of 1,444.8 metres (Figure 10-9). Drilling results show narrow intercepts of veined and tuff-associated mineralization extending into the hanging wall and along strike of areas tested in the Main Zone.

Figure 10-9: 2024 Main Zone Drilling



MZ-DD-174 (Figure 10-10) tested the down-dip potential of hanging wall mineralization associated with a prominent tuff layer and possible north-east cross-cutting structures found in historical drillholes MZ-DD-159, MZ-DD-161, and MZ-DD-163; the latter contained several lenses of 0.04% - 0.25% U_3O_8 associated with the upper tuff layer. MZ-DD-174 intercepted 0.07% U_3O_8 over 0.5 metres from 30.6 metres, 0.10% U_3O_8 over 0.5 metres at 253.3 metres associated with increased fracturing and veining in basalt host rock, and 0.04% U_3O_8 over 0.5 metres from 466.6 metres. This is interpreted as resulting from enhanced fracturing caused by a cross-cutting structure.

MZ-DD-175 (Figure 10-11) and MZ-DD-176 tested the western and eastern boundary of the Main Zone respectively. MZ-DD-175 tested down-plunge and to the west of 2023 drillholes MZ-DD-173 and MZ-DD-170. These holes followed up high-grade mineralization in and near the footwall in MZ-DD-170 (up to 2.88% U_3O_8 over 0.5 metres at 299.1 metres) and structure suggesting influence from the north-east cross-cutting faults. MZ-DD-175 intercepted multiple mineralized lenses with a total composite thickness of 7.4 metres highlighted by intervals of 0.43% U_3O_8 over 0.5 metres from 74.6 metres and 0.46% U_3O_8 over 0.5 metres from 372.7 metres. MZ-DD-176 intersected multiple weakly mineralized lenses with a total composite thickness of 6.5 metres highlighted by 0.11% U_3O_8 over 0.5 metres from 331.0 metres.

Strip logs for MZ-DD-174 and MZ-DD-175 are shown in Figure 10-12 and Figure 10-13.

Figure 10-10: Main Zone Cross-Section on ATHA Drillhole MZ-DD-175 looking NW

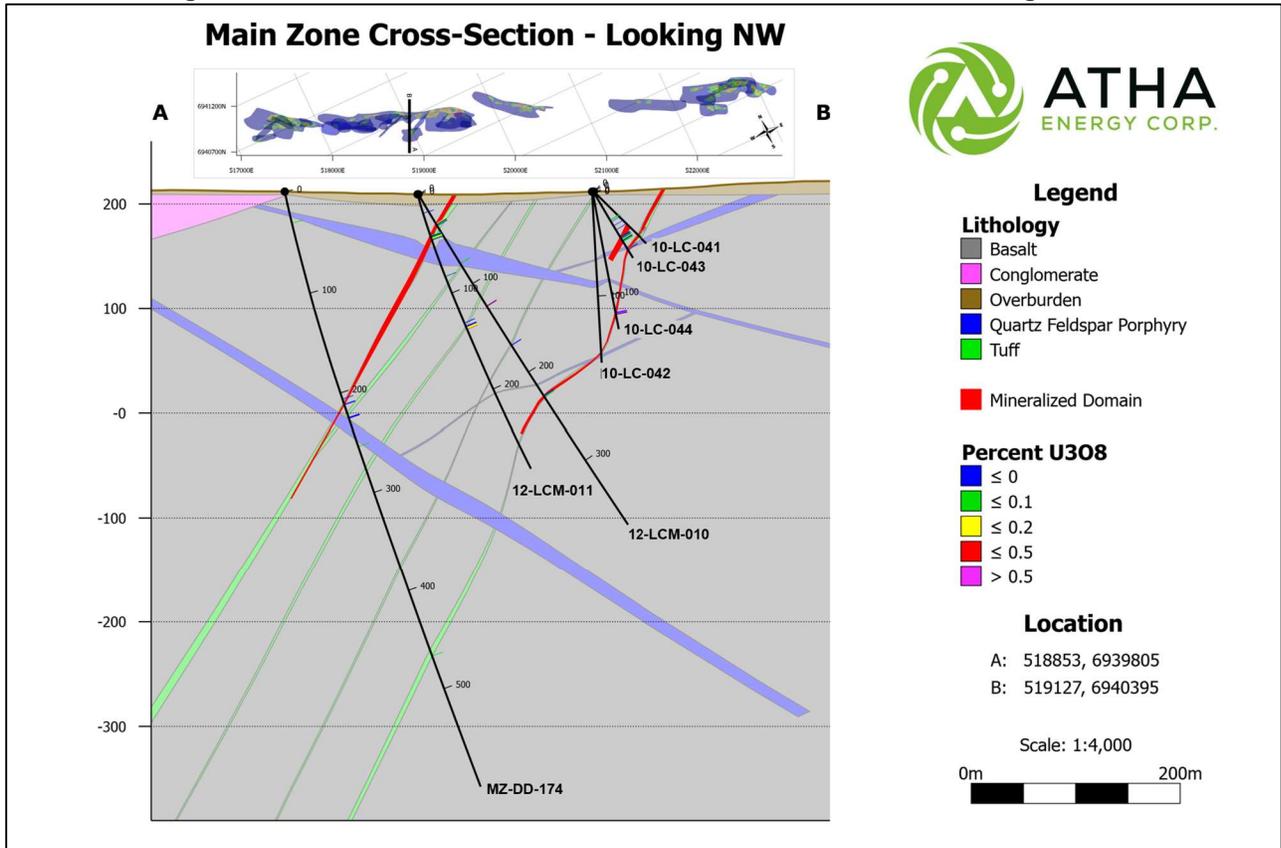


Figure 10-11: Main Zone Cross-Section on ATHA Drillhole MZ-DD-175 looking NW

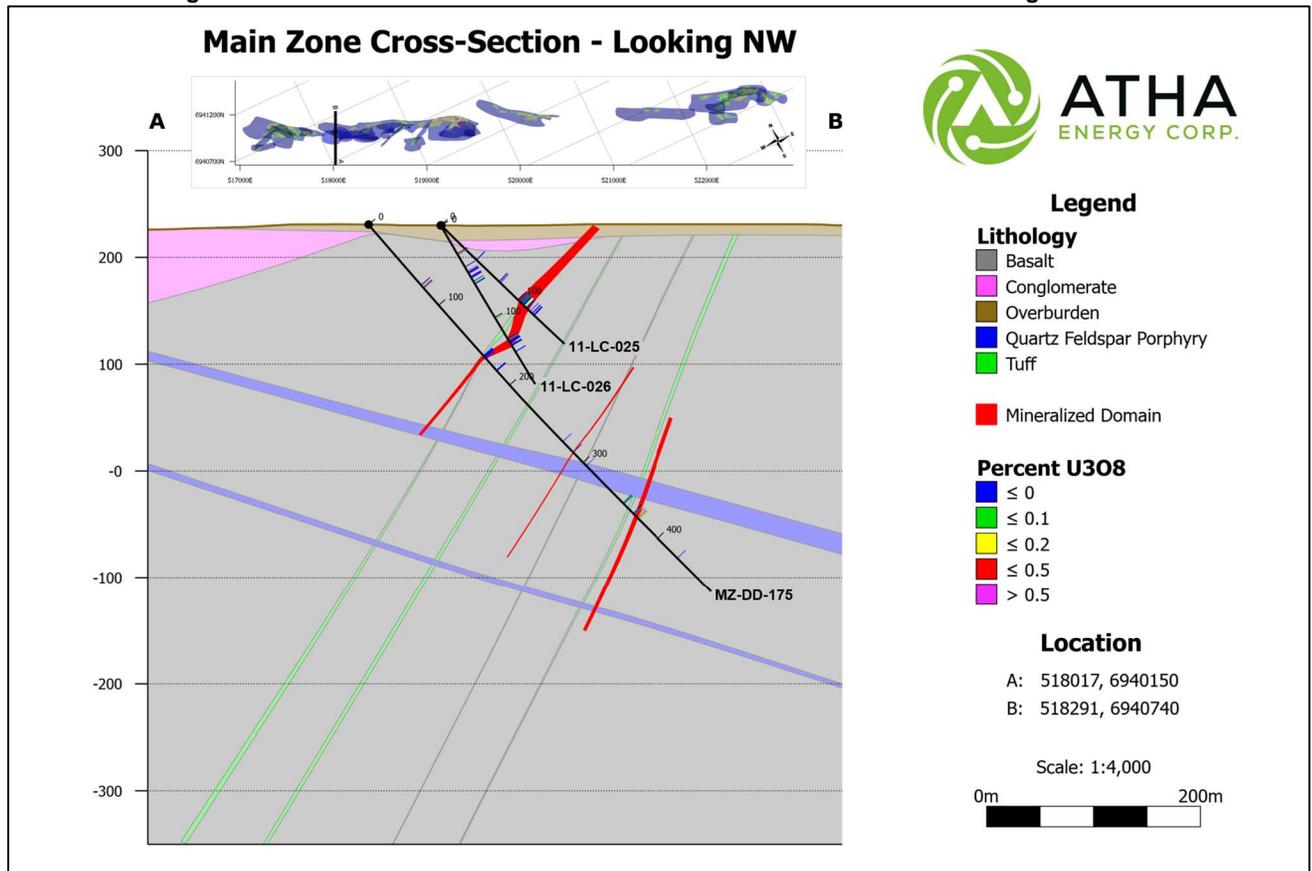
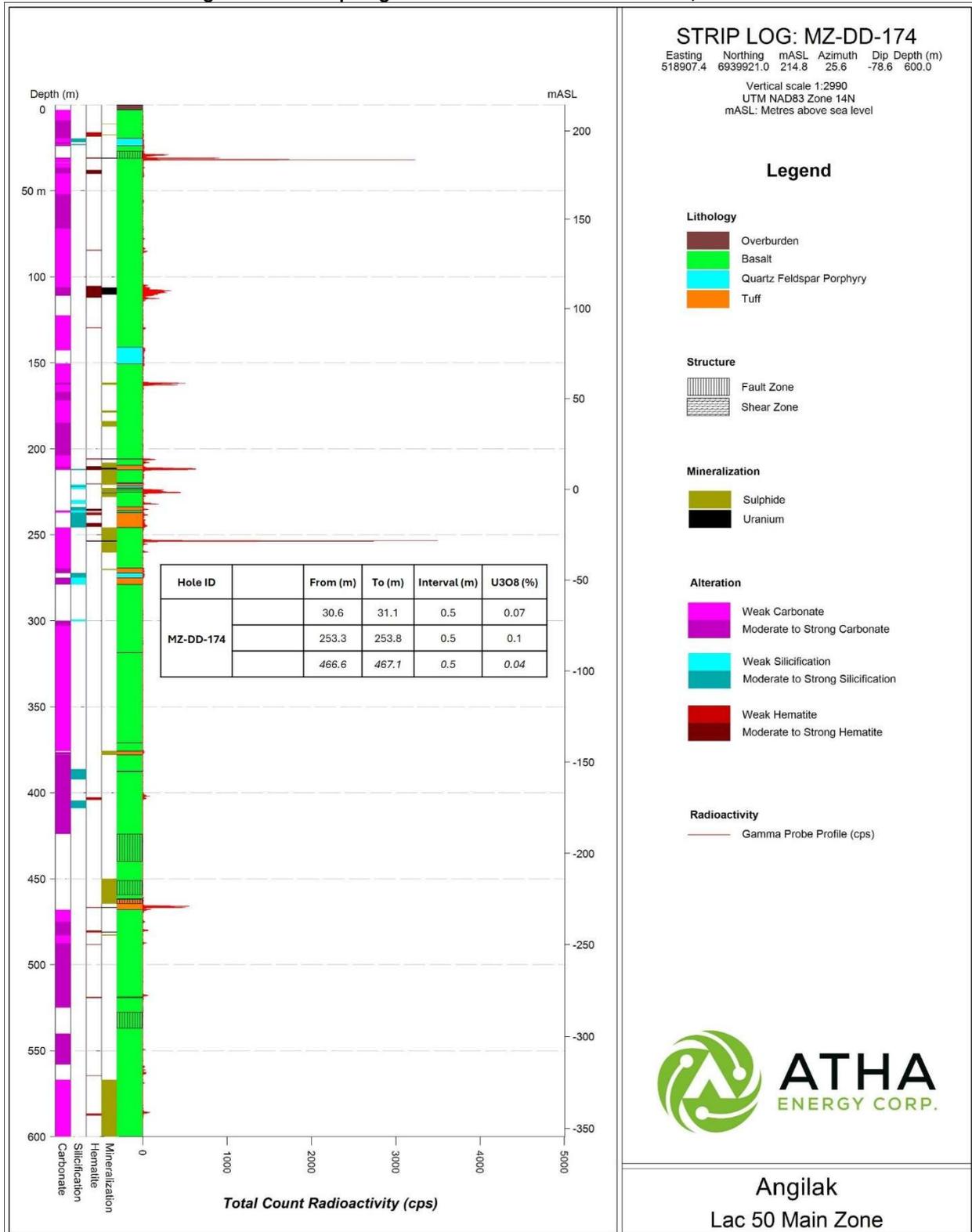
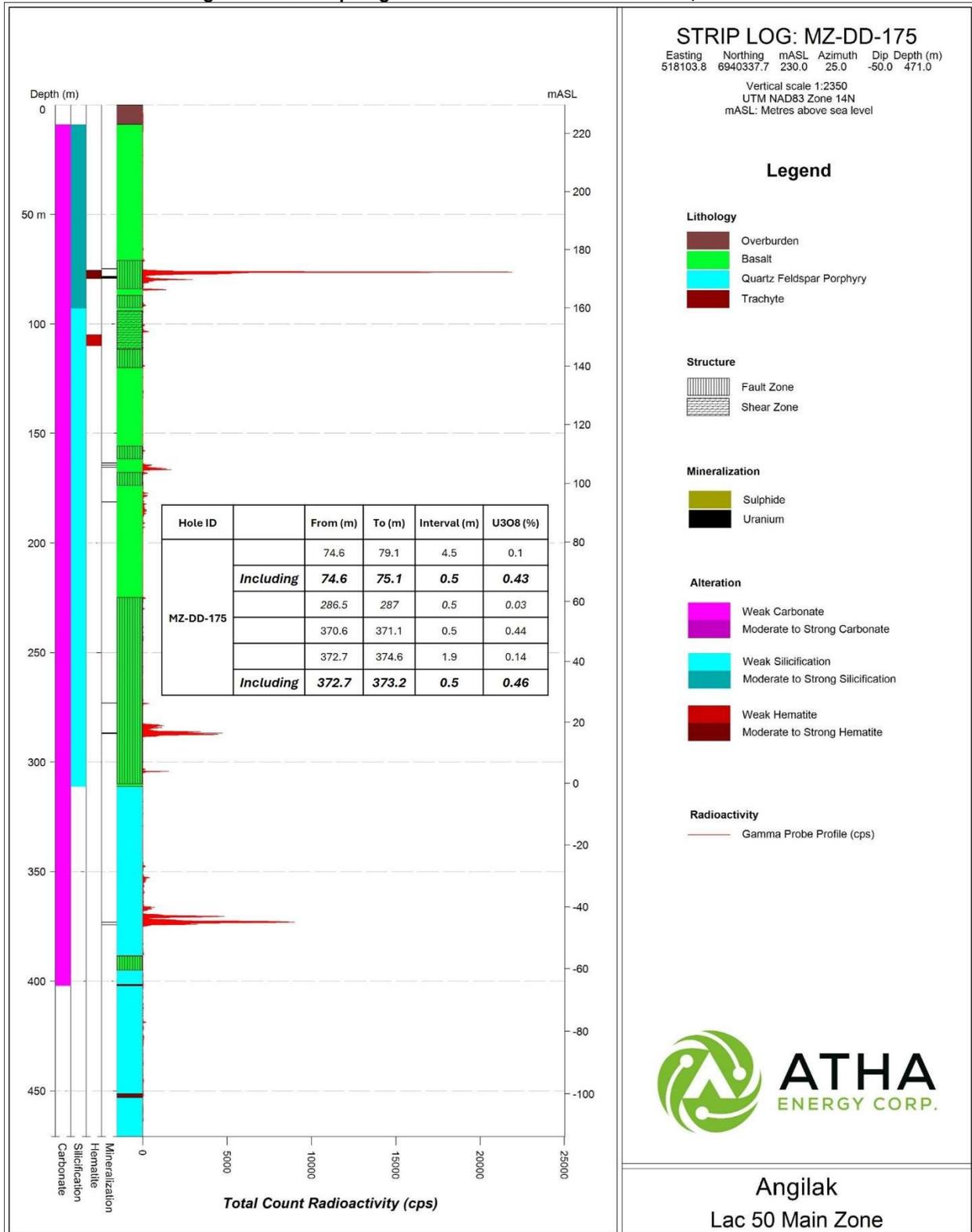


Figure 10-12: Strip Log of MZ-DD-174 at the Lac 50 Trend, Main Zone



Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

Figure 10-13: Strip Log of MZ-DD-175 at the Lac 50 Trend, Main Zone



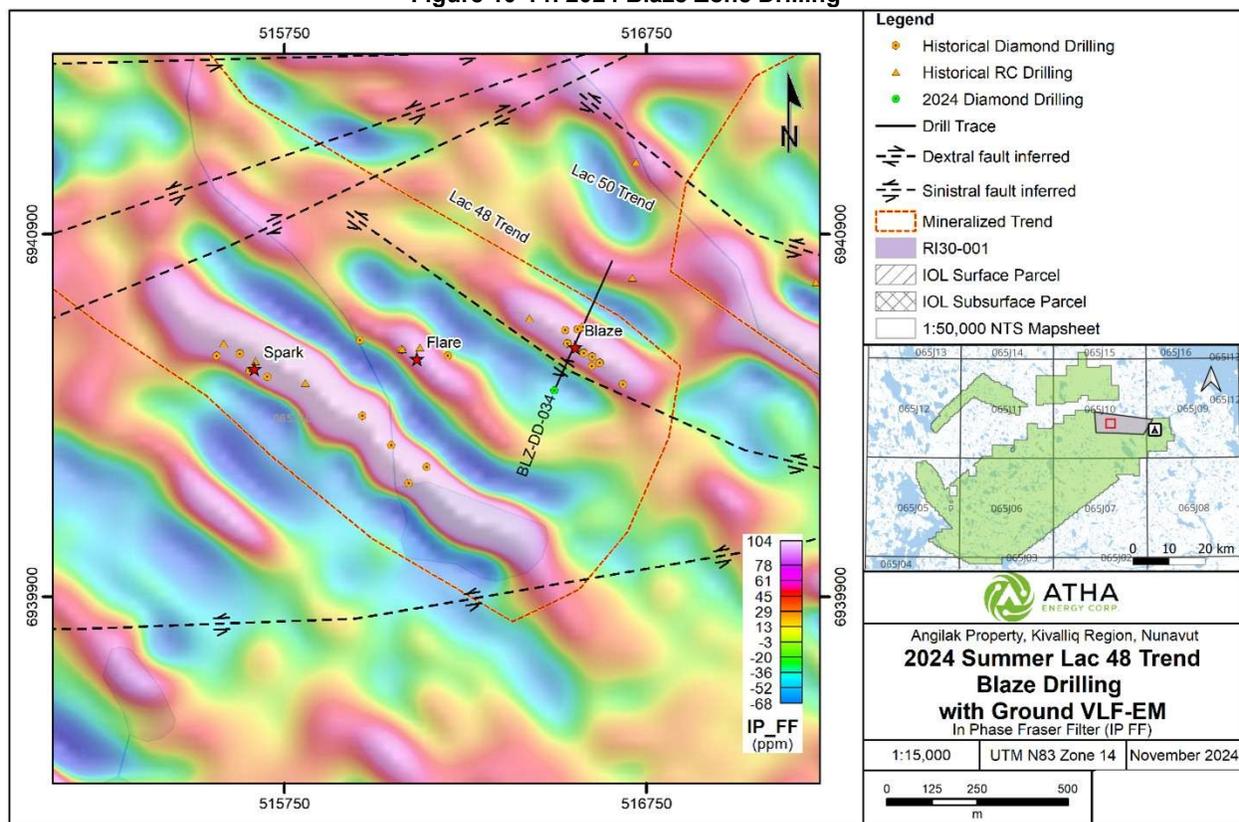
Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

Blaze Zone

Within the Blaze Zone, mineralization is associated with faults and breccia zones where the dominant northwest mineralization trend is cross-cut by a north-east trending structure. This may be the cause of significant normal faulting and large damage zone around mineralization.

One drillhole, BLZ-DD-034, tested the continuity of the mineralization at depth (Figure 10-14). Two zones of mineralization were intersected with grades of 0.37% U_3O_8 over 2.0 metres from 94.0 metres, including 0.99% U_3O_8 over 0.5 metres from 94.5 metres and 0.17% U_3O_8 over 0.5 metres from 164.0 metres. These intersections were identified as an extension of the main Blaze Zone. The mineralization is hosted within a quartz-carbonate-hematite breccia in the basalt host rock.

Figure 10-14: 2024 Blaze Zone Drilling



Eastern Extension

The Eastern Extension has similar geological characteristics and mineralization controls to the Main Zone. Mineralization is associated with structural disruption concentrated along tuff horizons cross-cut by more discrete north-east trending faults with associated quartz-carbonate veining and uranium mineralization. The objective of the 2024 drilling was to test the down-dip extension of mineralization at depth outside of the historical mineralized footprint.

Two drillholes, EEX-DD-052 and EEX-DD-053 for a total of 787.3 metres, targeted down-dip of historical drilling EEX-DD-023 to EEX-DD-030 (Figure 10-15). These drillholes exhibit grade intervals up to 2.66% U_3O_8 over 1.75 metres from 190.7 to 192.4 metres in EEX-DD-026 and 4.34% U_3O_8 over 0.72 metres from 88.8 to 89.5 metres in EEX-DD-030.

EEX-DD-052 (Figure 10-16) intercepted mineralization with 0.25% U_3O_8 over 0.5 metres in a quartz-carbonate-hematite vein at 346 metres. This occurred within a damage zone approaching the brecciated tuff horizon intersected from 365.6 to 366.1 metres. Another mineralized interval was intersected below the tuff unit within a quartz-carbonate-hematite veining with 0.1% U_3O_8 over 1.1 metres.

EEX-DD-053 (Figure 10-17) targeted approximately 110 metres to the southeast and at depth, along strike of the general mineralized trend. Mineralization was intercepted with 0.36% U_3O_8 and 0.27% Cu over 0.5 metres at 111.2 metres associated with a brecciated, hematized quartz-carbonate vein. Additionally, 0.04% U_3O_8 over 1.1 metres was intercepted at 305.3 metres. This mineralization is associated with bleached veins and shears within the tuff unit intersected between 304.6 and 307.9 metres. Overall, mineralization was successfully extended at depth within the Eastern Extension and remains open in all directions.

Figure 10-15: 2024 Eastern Extension Drilling

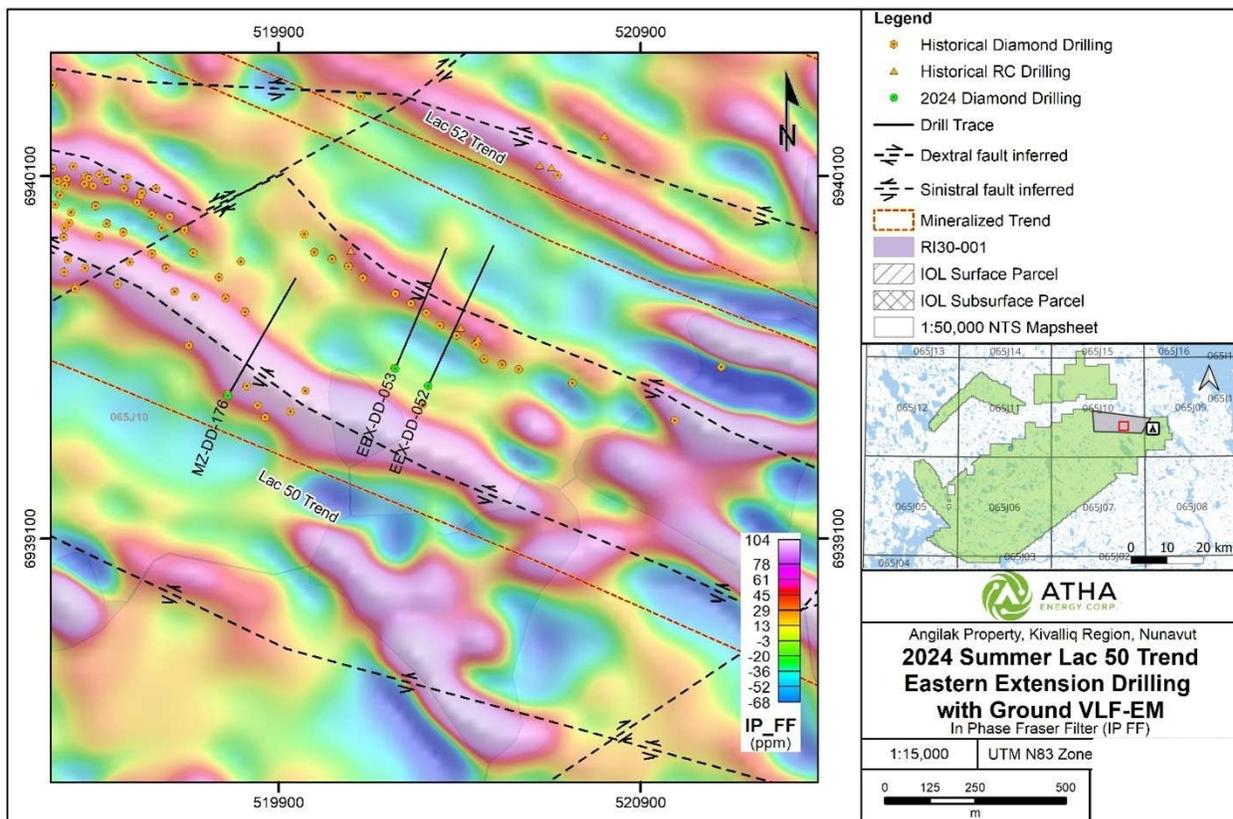


Figure 10-16: Eastern Extension Cross-Section on ATHA Drillhole EEX-DD-052 looking NW

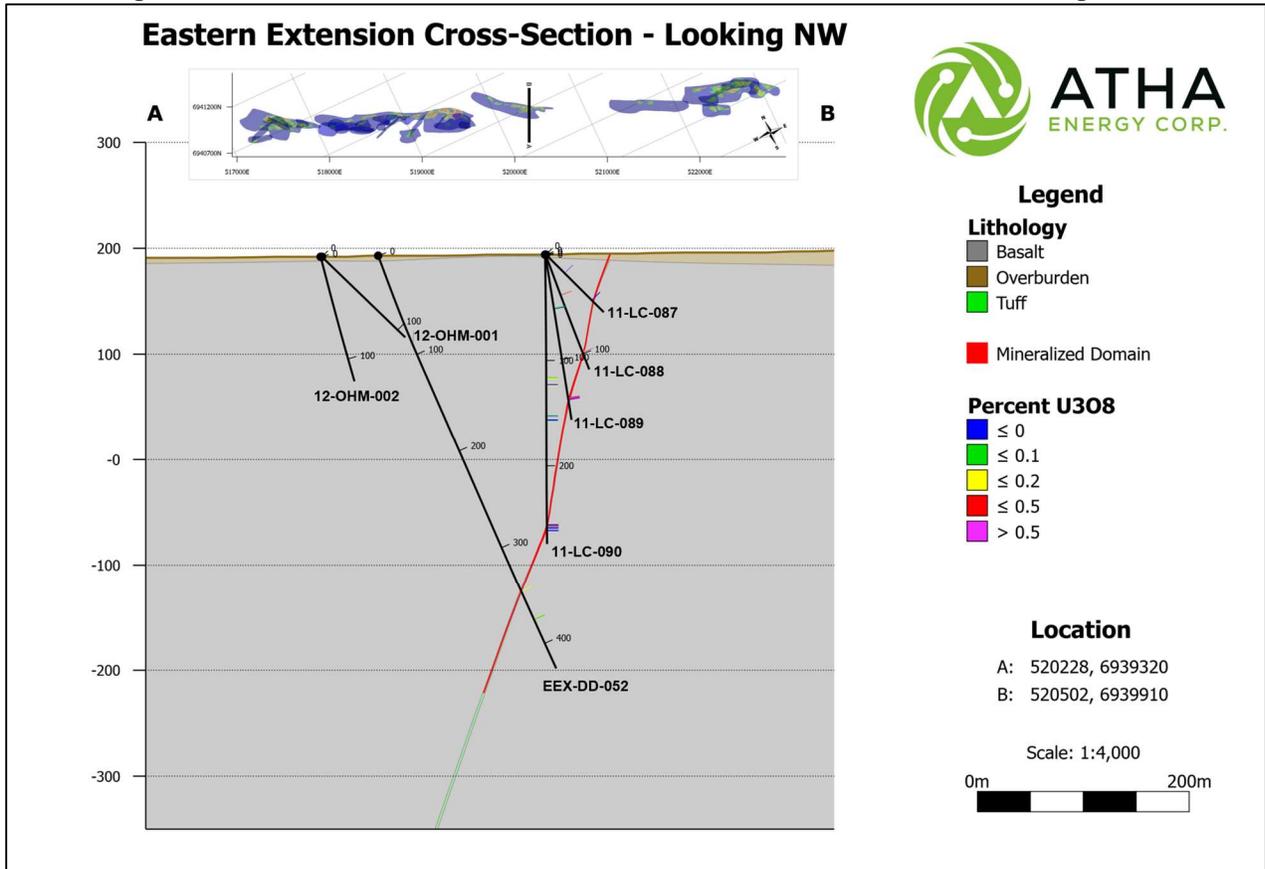
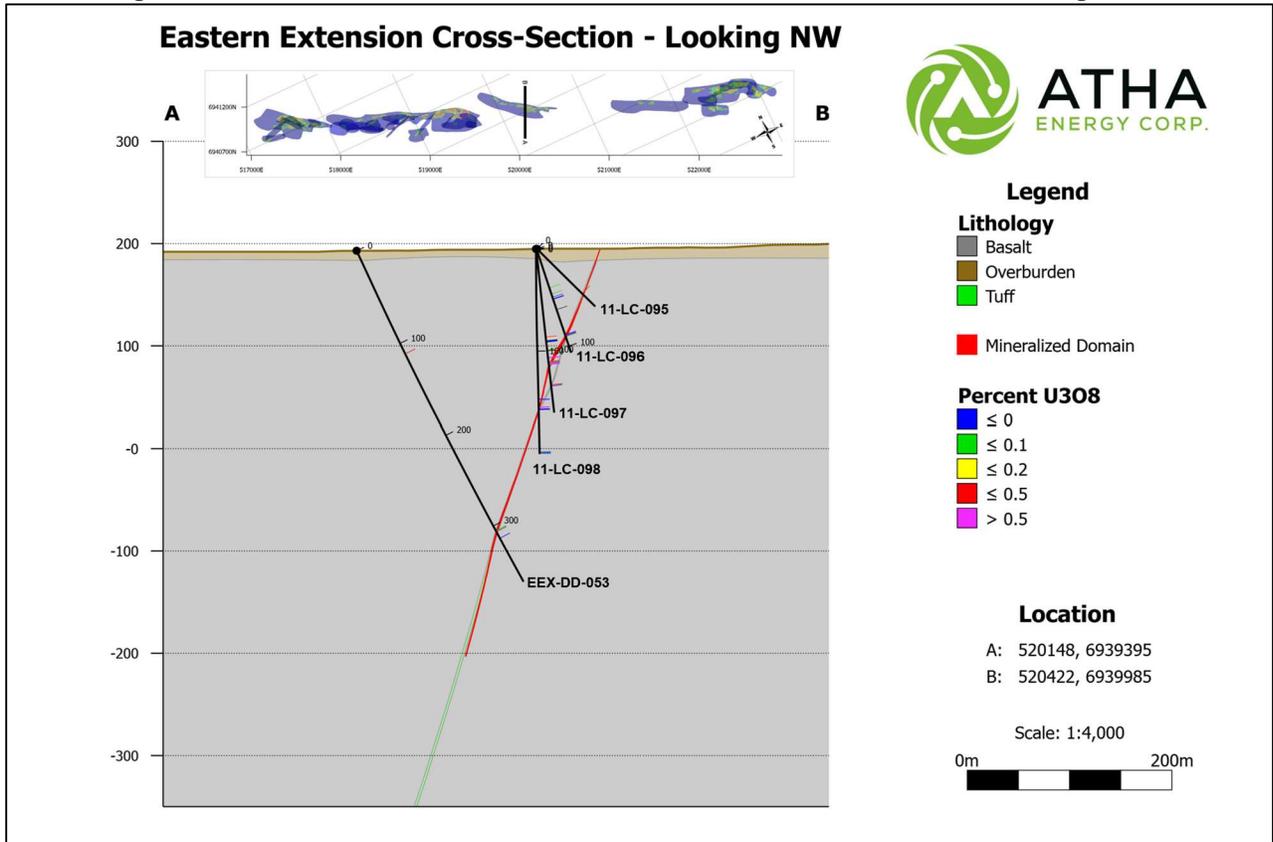


Figure 10-17: Eastern Extension Cross-Section on ATHA Drillhole EEX-DD-053 looking NW



Western Extension

One drillhole was completed in the Western Extension and was designed to test for potential mineralized structures within the footwall of the Western Extension (Figure 10-18). WEX-DD-079 (Figure 10-19) was drilled to 334.8 metres and intersected mineralization with 0.04% U_3O_8 over 1.7 metres from 39.8 metres, and 0.04% U_3O_8 over 0.5 metres from 60.6 metres.

Figure 10-18: 2024 Western Extension Drilling

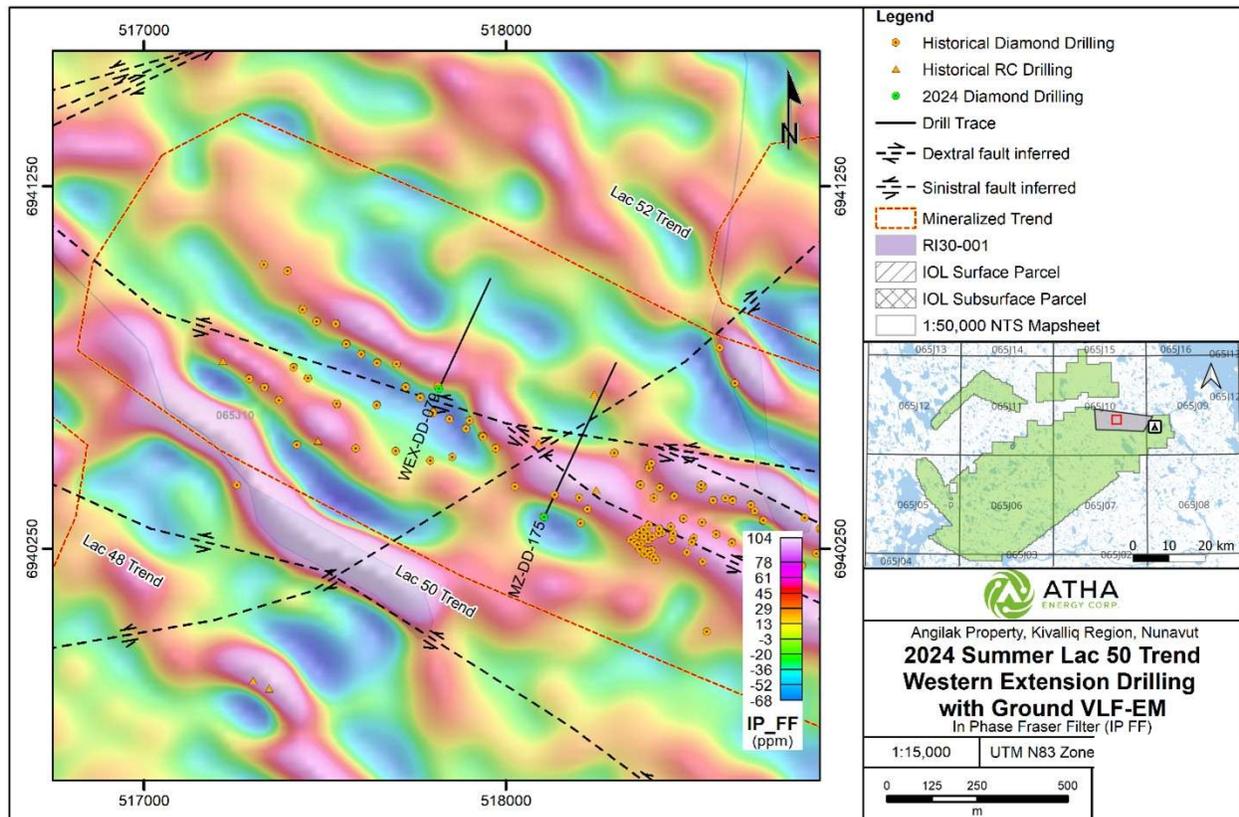
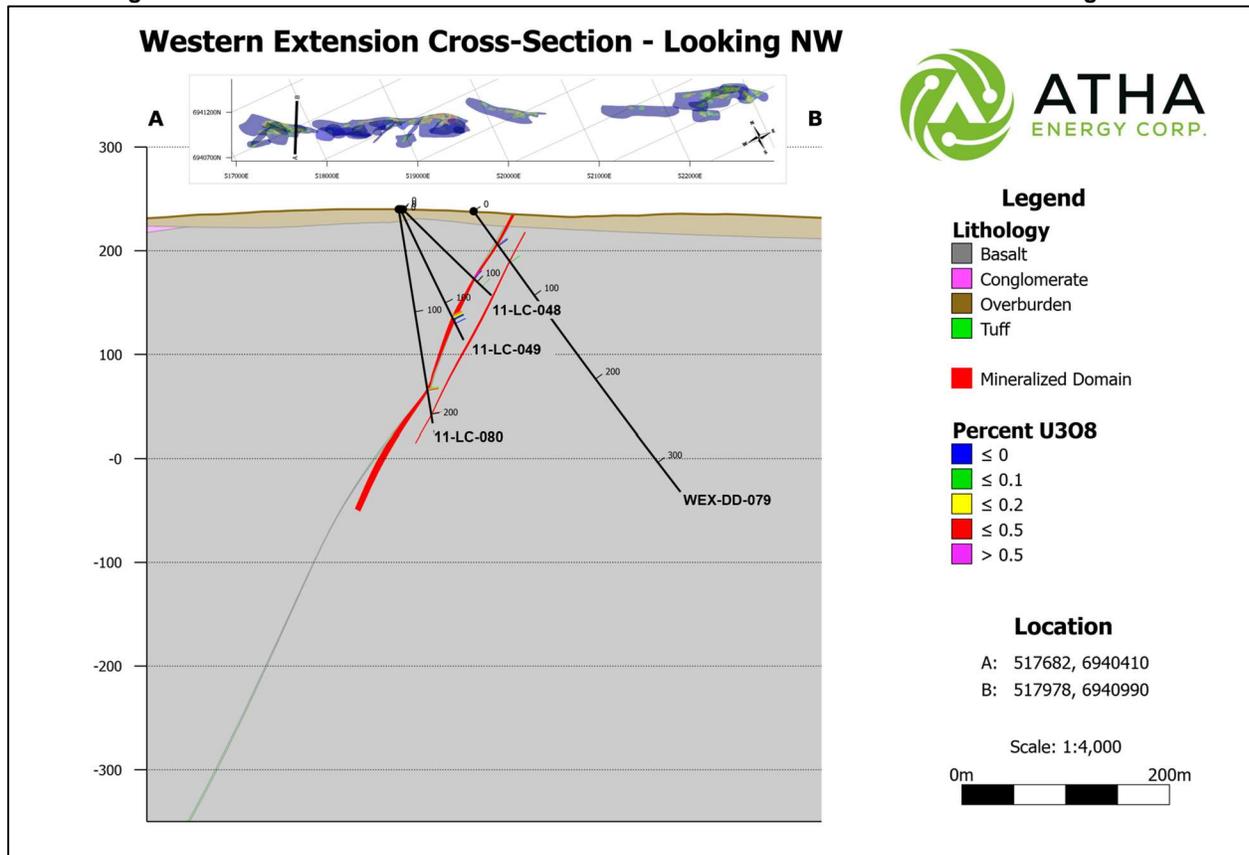


Figure 10-19: Western Extension Cross-Section on ATHA Drillhole WEX-DD-079 looking NW



Pulse Zone

Regional cross-cutting structures extend from the Blaze area through the Western Extension and into the Pulse area. VLF and magnetic signatures in the area indicate disrupted lithologies and presumably favorable conditions for structural traps to mineralization. Historical drilling in the Pulse area previously identified several intercepts of uranium mineralization within a fence of the drillholes PL-DD-10 and PL-DD-13 ranging from 0.17% U₃O₈ over 0.4 metres to 0.63% U₃O₈ over 0.63 metres. In another section approximately 50 metres to the east-southeast, historical drillhole PL-DD-16 intersected 0.4% U₃O₈ over 0.92 metres, thus showing continuity between the aforementioned drill sections. Four drillholes were completed in the Pulse area during the 2024 drill campaign (Figure 10-20), designed to test the continuity of mineralization and potential for a mineralized shoot related to the cross-cutting structures.

PL-DD-031 and PL-DD-033 tested the western and eastern end, respectively, of a structurally bound tuff horizon. PL-DD-031 verified continuity of the mineralized lens to the west intersecting 1.06 % U₃O₈ over 0.5 metres at 215.0 metres. PL-DD-033 intersected several zones with elevated radioactivity above background levels and one mineralized interval from 150.8 to 151.3 metres with 0.03% U₃O₈.

PL-DD-030 and PL-DD-032 tested the VLF response in an approximate 340 metre gap where both VLF and magnetic data suggest the mineralized horizon is proximal to an off-setting northeast structure. PL-DD-030 (Figure 10-21) intersected mineralization of 0.26% U₃O₈ over 5.0 metres from 262.5 metres, including 1.29 % U₃O₈ over 1.0 metre 262.5 to 263.5 metres, with an

additional upper zone of Cu mineralization encountered from 261.0 t 262.5 metres grading 1.0% Cu. Drillhole PL-DD-032 encountered mineralization of 0.08% U₃O₈ from 28.8 to 29.3 metres, and 0.12% U₃O₈ from 63.3 to 63.8 metres. All mineralization is associated with an increase in fracturing and veining and in general these results suggest potential for multiple mineralized, parallel horizons within the Pulse Zone.

Figure 10-20: 2024 Pulse Zone Drilling

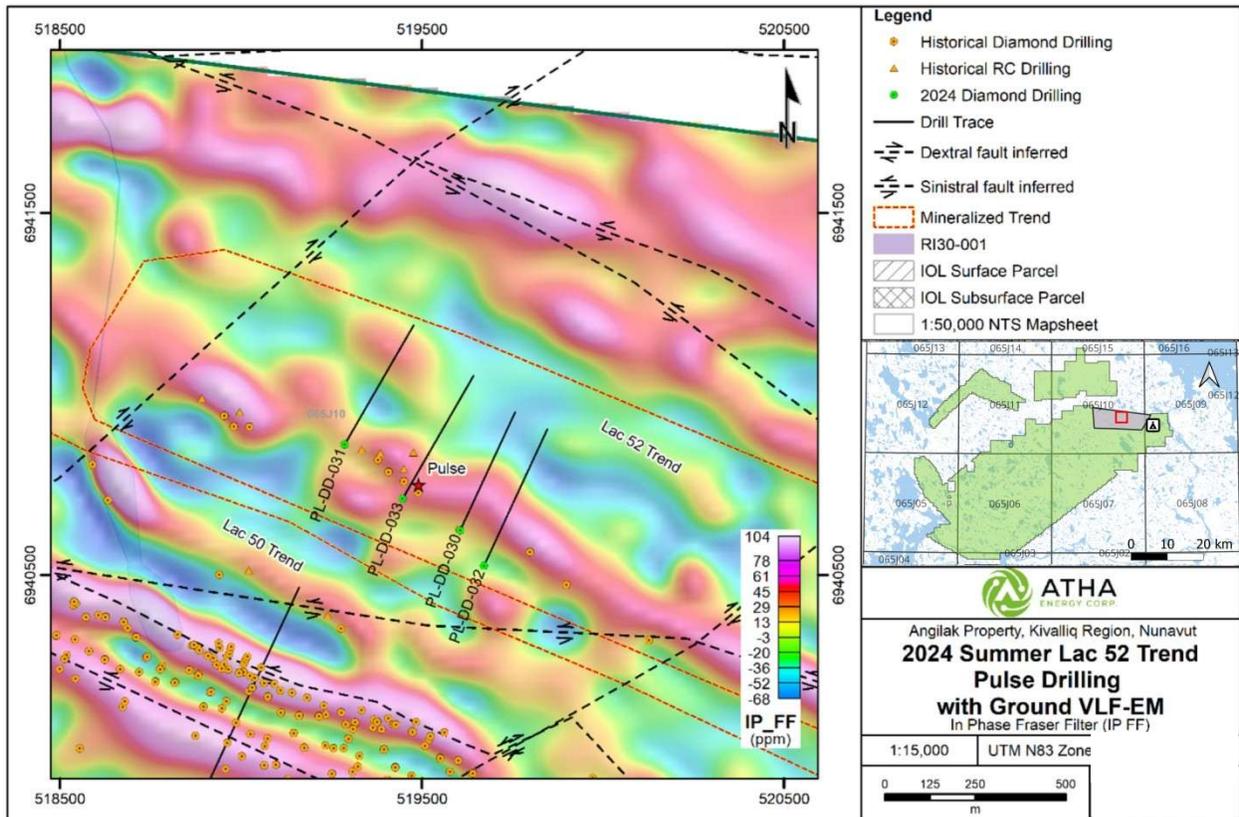
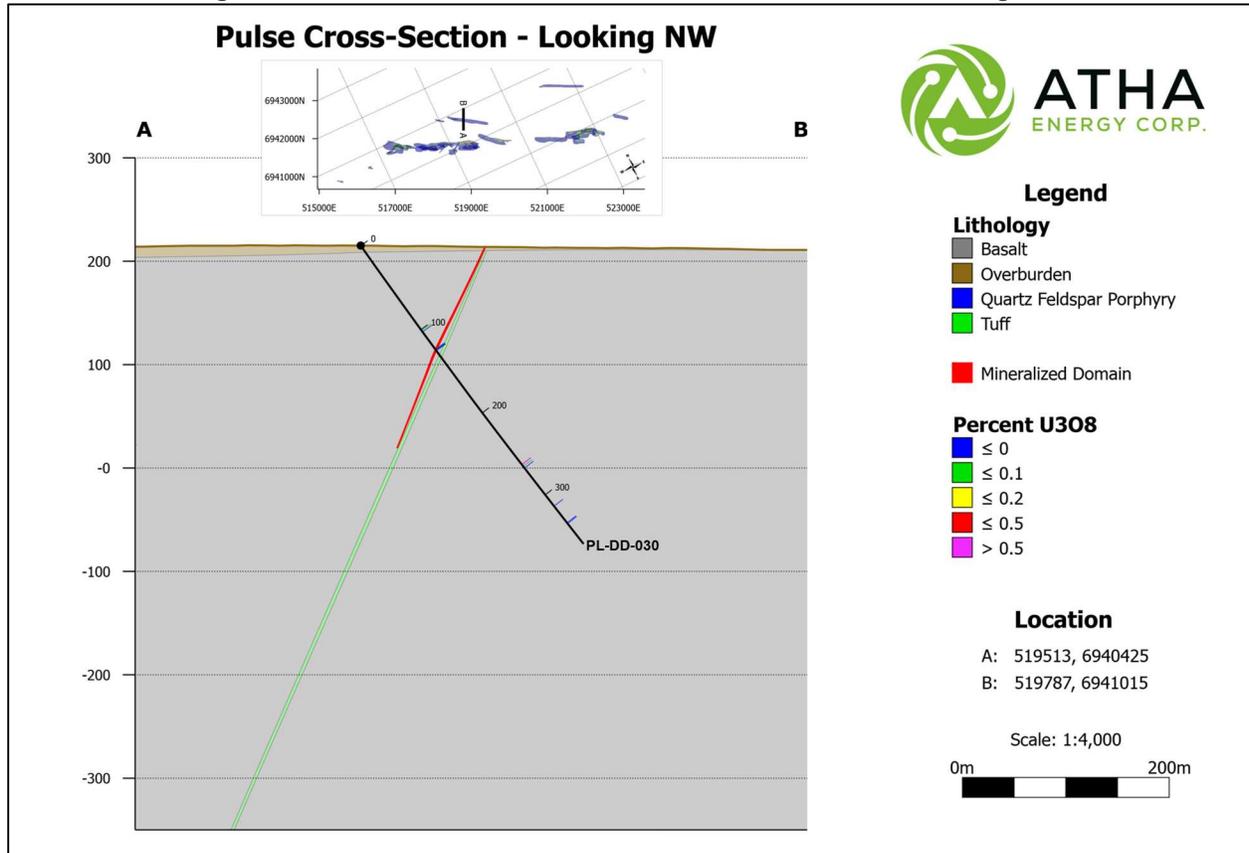


Figure 10-21: Pulse Cross-Section on ATHA Drillhole PL-DD-030 looking NW



Mushroom Lake

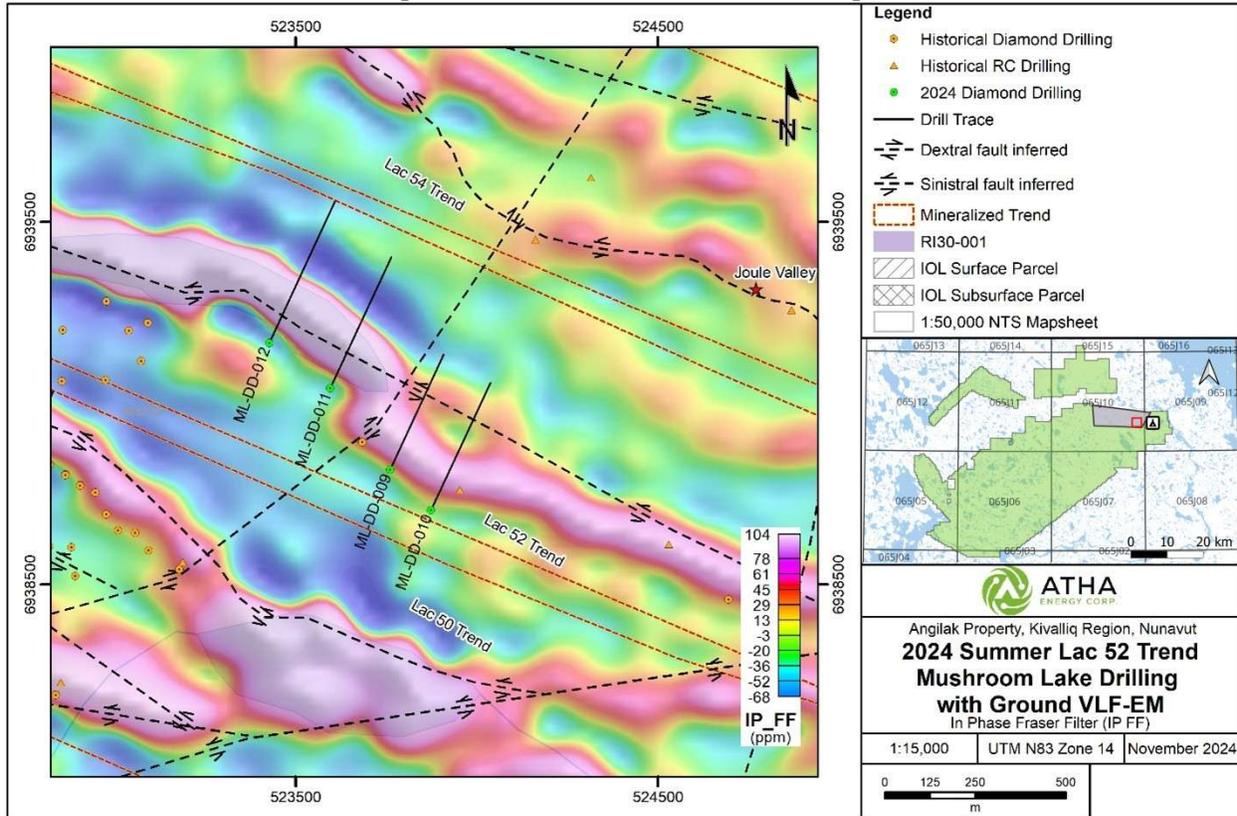
The ML Zone has seen limited historical drill testing to date, however mineralized intercepts have been recorded in this area up to 0.45% U_3O_8 over 4.35 metres (ML-DD-005). The mineralization is structurally controlled, occurring within sheared graphitic tuffaceous metasediments with carbonate infill interlayered with Archean basement metavolcanics. The protolith of the metavolcanics are predominantly massive basalt. The objective of the 2024 drilling program was to test the continuity of mineralization along strike, following the VLF-EM conductor and stepping east and west from ML-DD-005.

Four drillholes were completed in this area: ML-DD-009, ML-DD-010, ML-DD-011, and ML-DD-012, for a total of 1,571 metres (Figure 10-22). Drilling results show intercepts of mineralized veins and hydrothermal breccias along the Mushroom Lake VLF-EM anomaly and along strike from the mineralization intersected in ML-DD-005.

ML-DD-009 and ML-DD-010 tested the continuity of the mineralized structure eastward from ML-DD-005 and along the VLF-EM conductor. ML-DD-009 intersected mineralized lenses with grades of 0.08% U_3O_8 over 1.0 metre from 22.9 metres, 0.08% U_3O_8 over 0.5 metres from 89.1 metres, 0.14% U_3O_8 over 0.5 metres from 96.1 metres and 0.77% U_3O_8 over 0.5 metres from 102.5 metres. These intervals are within a hydrothermal breccia hosted within the main basalt package. Drillhole ML-DD-010 intersected several zones of elevated radioactivity compared to background, including 0.01% U_3O_8 from 127.1 to 128.1 metres.

ML-DD-011 and ML-DD-012 tested the continuity of the mineralized structure westward from ML-DD-005 and along the VLF-EM conductor. ML-DD-011 intercepted a mineralized interval of 0.05% U_3O_8 over 1.0 metres at 252.1 metres, and ML-DD-012 intercepted a mineralized interval of 0.19% U_3O_8 over 0.5 metres at 190.3 metres. Both mineralized intervals are associated with hydrothermal breccias containing carbonate veins.

Figure 10-22: 2024 Mushroom Lake Drilling.



Hot Zone

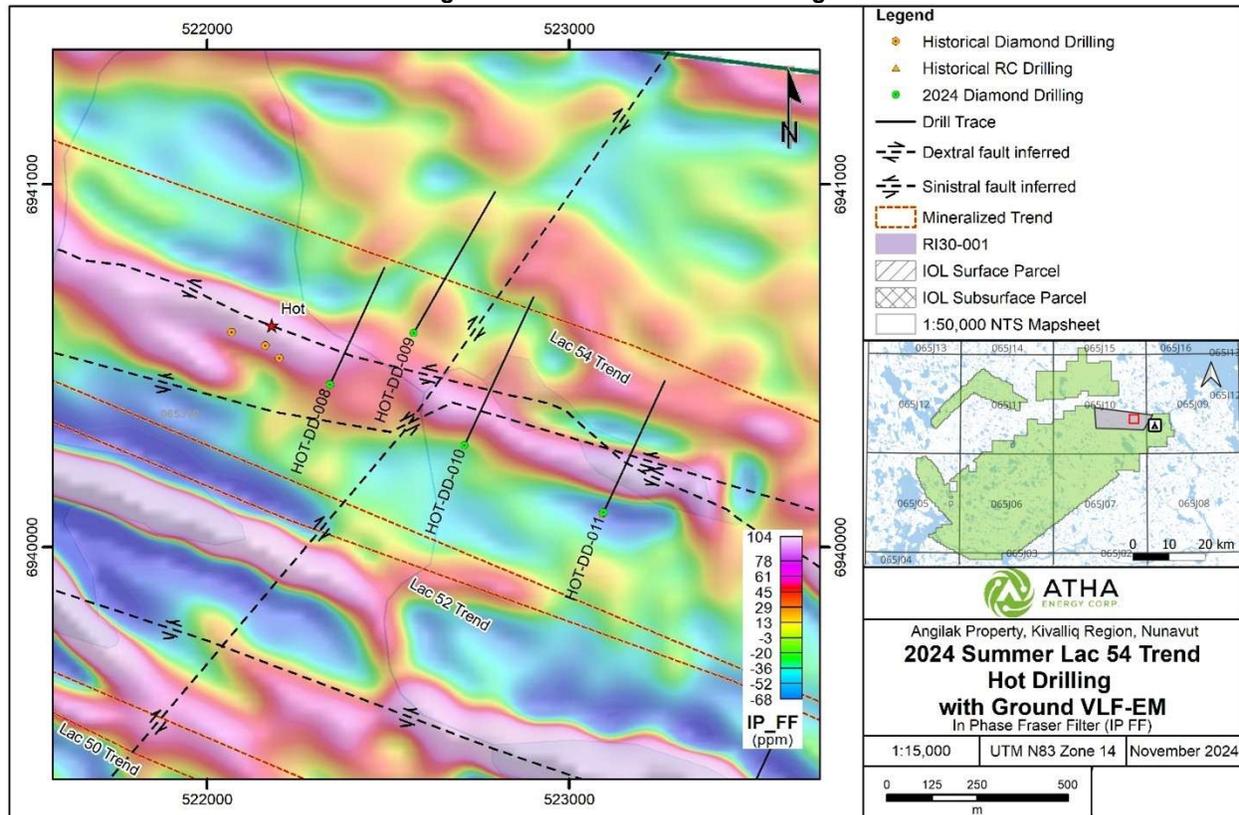
The Hot zone was drilled by ValOre in 2012, intercepting mineralization of up to 0.84% U_3O_8 over 3.0 metres in historical drillhole HOT-DD-004. Mineralization in this zone is structurally controlled and hosted in veins within Archean basement metavolcanics, with protoliths typically consisting of basalt. Four drillholes were completed within the Hot Zone in 2024: HOT-DD-008, HOT-DD-009, HOT-DD-010, and HOT-DD-011, for a total of 1,657.8 metres (Figure 10-23). These holes targeted the extension of mineralization along the VLF-EM anomaly on strike from historical hole HOT-DD-004.

HOT-DD-008, HOT-DD-010, and HOT-DD-011 tested the continuity of the mineralized structure to the southeast from historical drillhole HOT-DD-005. HOT-DD-008 intercepted a mineralized interval with 0.4% U_3O_8 and 0.22% Cu over 0.5 metres at 123.2 metres, 0.08% U_3O_8 over 0.5 metres from 127.7 metres and 0.06% U_3O_8 over 0.5 metres from 294.6 metres. This mineralized interval is associated with increased veining and hydrothermal breccia. HOT-DD-010 intercepted mineralized lenses with the highest grade of 0.20% U_3O_8 and 0.19% Cu over 0.4 metres at 290.2 metres. HOT-DD-011 intercepted a mineralized lens of 0.03% U_3O_8 over 0.5 metres at 113.3

metres. The mineralized intercepts in both drillholes are associated with hydrothermal breccias within deformed structural intervals similar with features observed in HOT-DD-008.

HOT-DD-009 tested a coincident magnetic tilt with a VLF anomaly following an east-west mineralized structure located at the surface to the north-northeast. Mineralized intervals were intercepted, including 0.02% U_3O_8 over 3.0 metres metres at 171 metres, and 0.04% U_3O_8 over 0.5 metres at 210.7 metres. These intervals are associated with carbonate veins and hydrothermal breccias with hematite alteration halos.

Figure 10-23: 2024 Hot Zone Drilling.



J4/Ray Zone

Mineralization within the J4/Ray Zone (Figure 10-24) is structurally and lithologically controlled, occurring within sulphidic-graphitic tuffaceous metasediments and/or volcanoclastic rocks interlayered with Archean basement metavolcanics. The protoliths of the metavolcanics are predominantly massive basalt, pillowed basalt, and subvolcanic gabbro. The objective of the 2024 drilling was to test along strike and downdip to expand the footprint of mineralization in the southern part of the J4/Ray area.

Five drillholes were completed in this area: J4R-DD-085, J4R-DD-086, J4R-DD-088, J4R-DD-089, and J4R-DD-090, for a total of 2,276.55 metres. Drillhole J4R-DD-087 was moved and re-started as J4R-DD-088 due to drilling-related issues with the original setup which necessitated restarting the drillhole.

J4R-DD-085 (Figure 10-25 and Figure 10-26) tested the down-dip continuity of a mineralized northeast trending structure associated with a graphitic-sulphidic tuff layer proximal to historic holes J4R-DD-012 and J4R-DD-060. J4R-DD-085 intersected 5.85% U_3O_8 over 0.5 metres at 111.8 metres within the upper graphitic tuff layer. A second mineralized horizon intersected mineralization of 0.56% U_3O_8 over 2.0 metres from 428.1 metres, including 1.52% U_3O_8 over 0.5 metres at 429.1 metres. A final mineralized horizon of 0.2% U_3O_8 was intersected from 456.9 to 458.9 metres including 0.63% U_3O_8 over 0.5 from 457.4 metres. Mineralization is associated with a hematized carbonate breccia cross-cutting a graphitic tuff layer.

J4R-DD-086 and J4R-DD-089 tested mineralization continuity up to 115 metres to the southeast from historical drillholes J4R-DD-015 and J4R-DD-073. J4R-DD-015 returned up to 0.29% U_3O_8 over 0.4 metres at 57.0 metres. Drillhole J4R-DD-086 intersected a mineralized zone grading 0.09% U_3O_8 from 34.0 to 37.5 metres, including 0.62% U_3O_8 from 35.5 to 36.0 metres. A second mineralized interval was encountered from 92.5 to 93.0 metres grading 0.15% U_3O_8 . The main J4 Zone mineralized zone was intersected from 393.0 to 395.5 metres with an average grade of 0.87% U_3O_8 , including 3.92% U_3O_8 from 393.5 to 394.0 metres. Drillhole J4R-DD-089 intersected mineralization from 9.2 to 9.7 metres grading 0.13% U_3O_8 , from 403.6 to 404.4 metres with an average grade of 1.24% U_3O_8 , from 410.4 to 410.9 metres with a grade of 0.1% U_3O_8 , and from 422.0 to 422.5 metres with a grade of 0.17% U_3O_8 .

J4R-DD-088 tested the continuity of the mineralization 80 metres to the west from historical holes J4R-DD-007, J4R-DD-008, J4R-DD-009 and J4R-DD-010. J4R-DD-088 intersected a mineralized interval with 0.03% U_3O_8 over 0.5 metres at 331.0 metres.

J4R-DD-090 tested the continuity of the mineralization in the hanging wall 350 metres to the northwest from J4R-DD-088, and 120 metres to the southeast from historical RC holes J4W-RC-010 and J4W-RC-011. J4R-DD-090 intercepted 0.45 % U_3O_8 over 0.4 metres at 132.3 metres.

Figure 10-24: 2024 Jay 4 Ray Drilling.

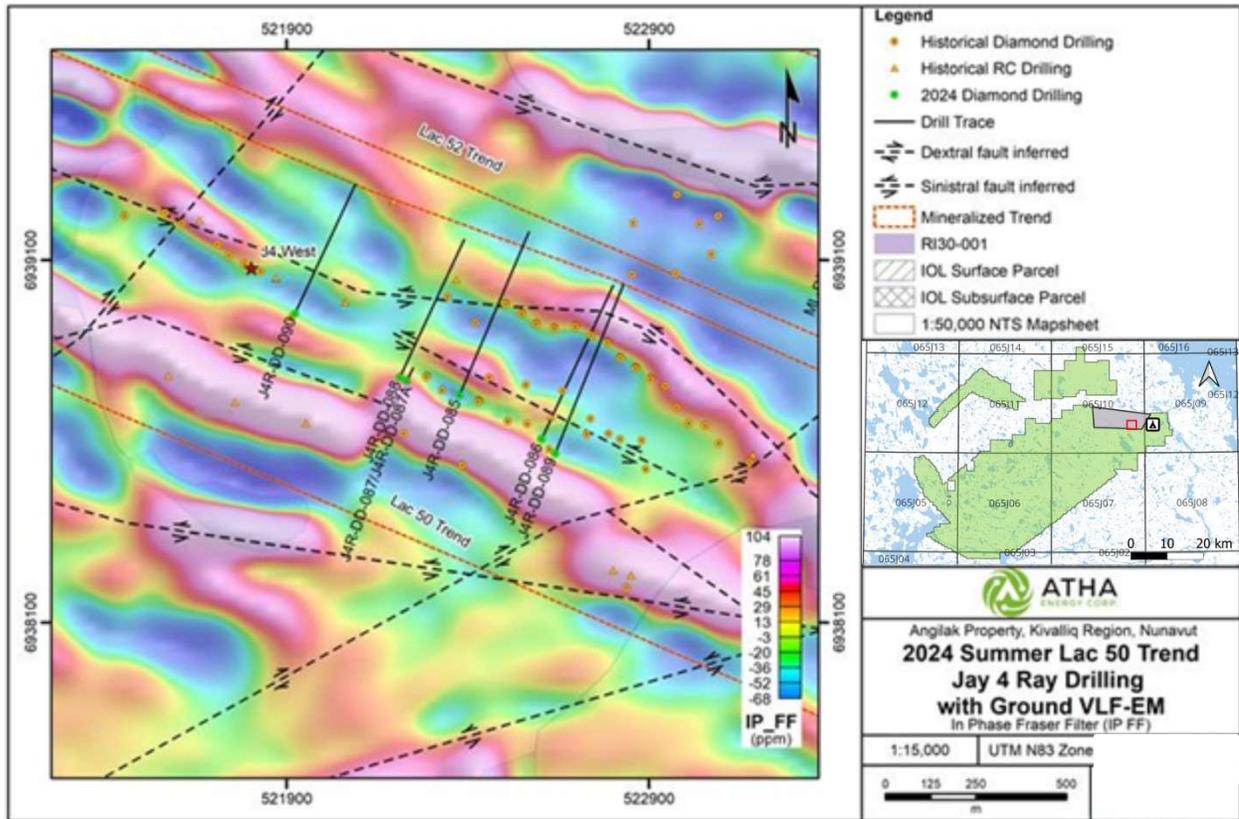


Figure 10-25: J4/Ray Zone Cross-Section on ATHA Drillhole J4R-DD-085 looking NW

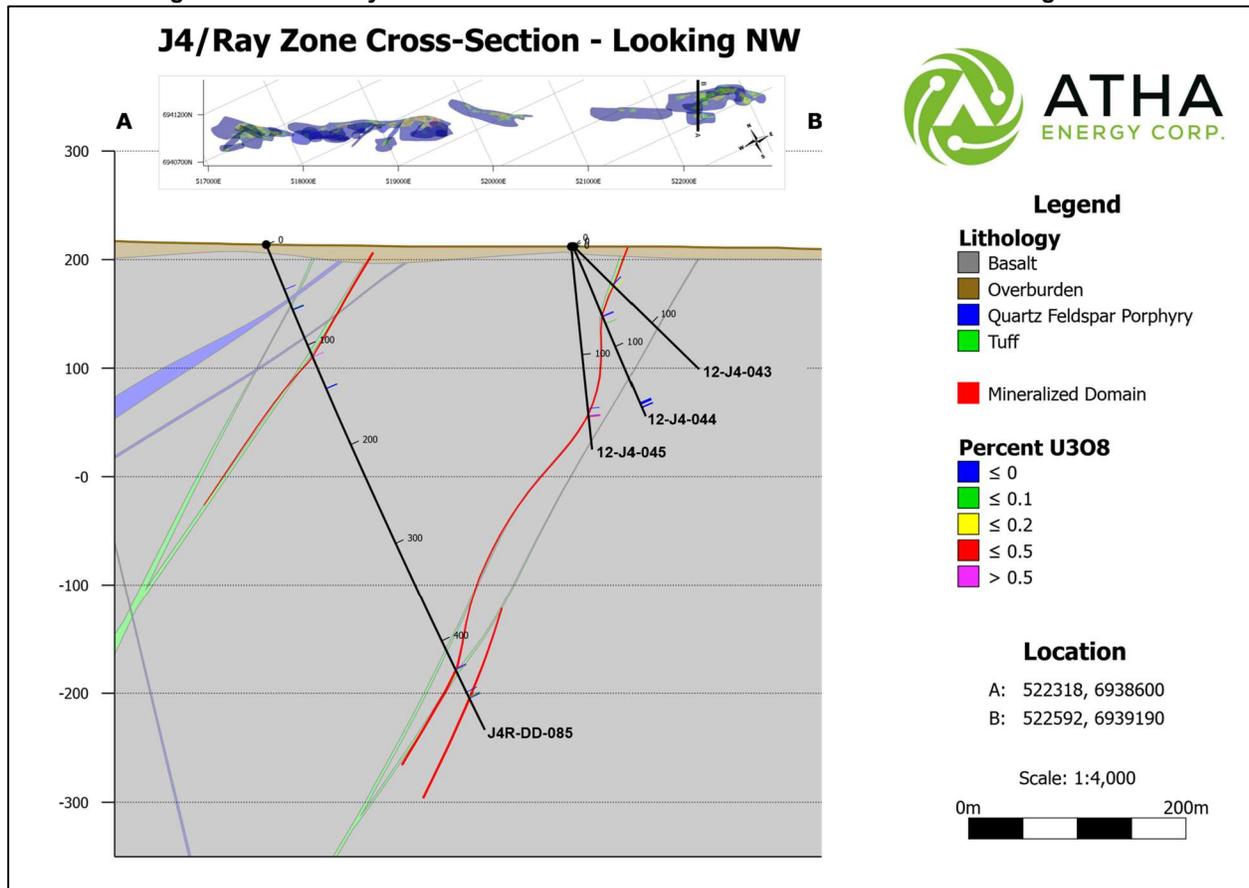
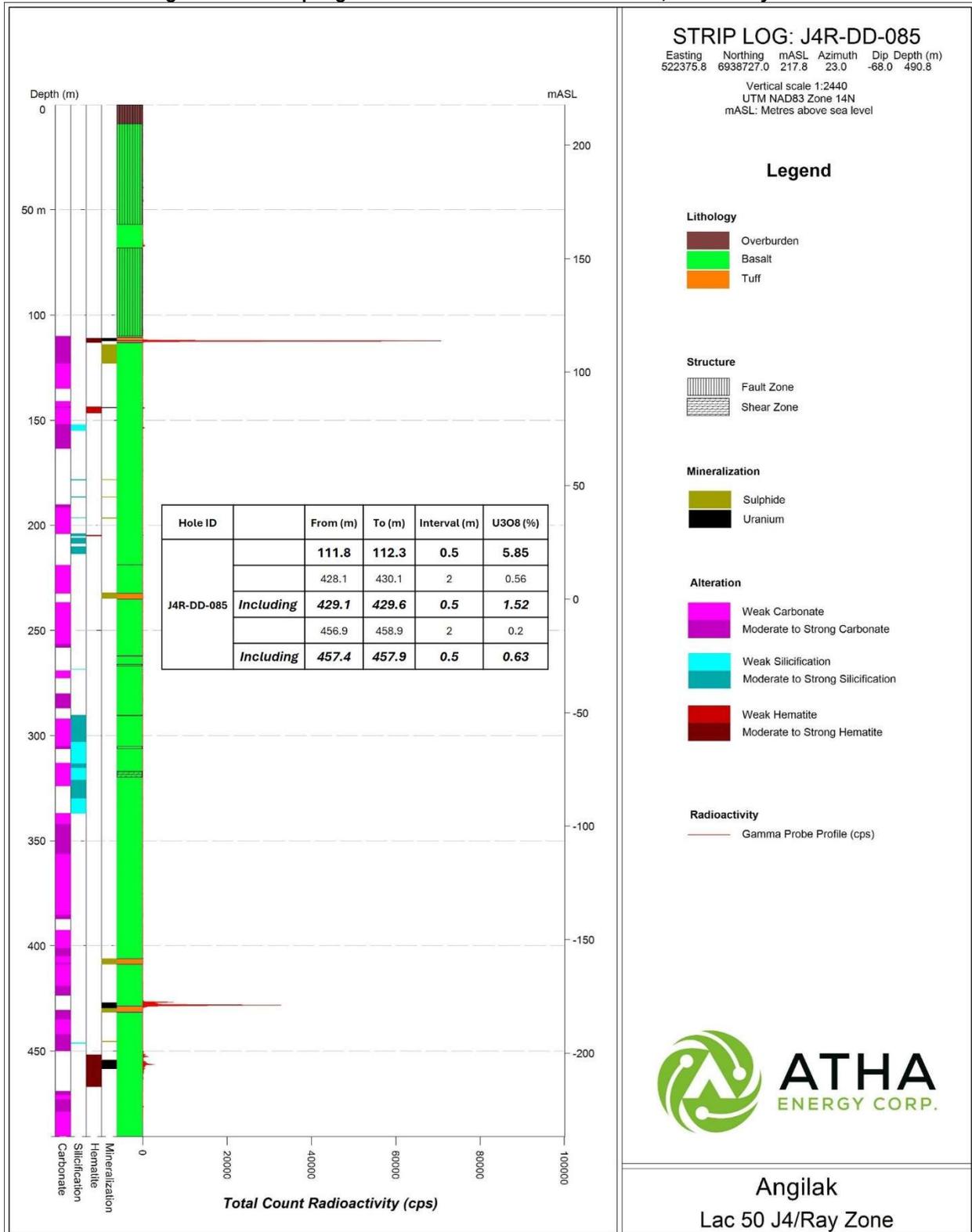


Figure 10-26: Strip log for J4R-DD-085 at the Lac 50 Trend, J4 and Ray Zones



Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

10.6.2 2025 Drilling Program Summary

ATHA conducted a 10,774-metre helicopter-supported diamond drill program between May 31 and September 6, 2025. A total of twenty-two drillholes were completed, excluding one lost drillhole (Table 10-3 and Figure 10-27). The drilling program had several key objectives; to expand the mineralization footprint within the Lac 50 trend, specifically targeting the down-dip extension of the J4 and Ray Zones, and down-dip extension of surficial mineralization identified at the Mushroom Lake zone. Additionally, the 2025 campaign was designed to test high priorities targets at the KU and RIB regional areas located within the 31-km RIB-Nine Iron structural corridor.

To date, final drill core sample assays have not yet been received and all results are based on preliminary down-hole gamma probe results obtained from a Mount Sopris 40TGU-1000 Triple Gamma Geiger probe. The Company considers high-grade mineralization to be any interval with radioactivity derived from downhole gamma probe >10,000 counts per second (CPS). The total gamma results provided were selected using an average cutoff of 500 CPS over intervals of 0.1 metre width. All drill intercepts are core width and true thickness is yet to be determined. All mineralized drill core samples have been submitted to Saskatchewan Research Council (SRC) for assay analysis.

Figure 10-27: 2025 Drill Collar Location Map



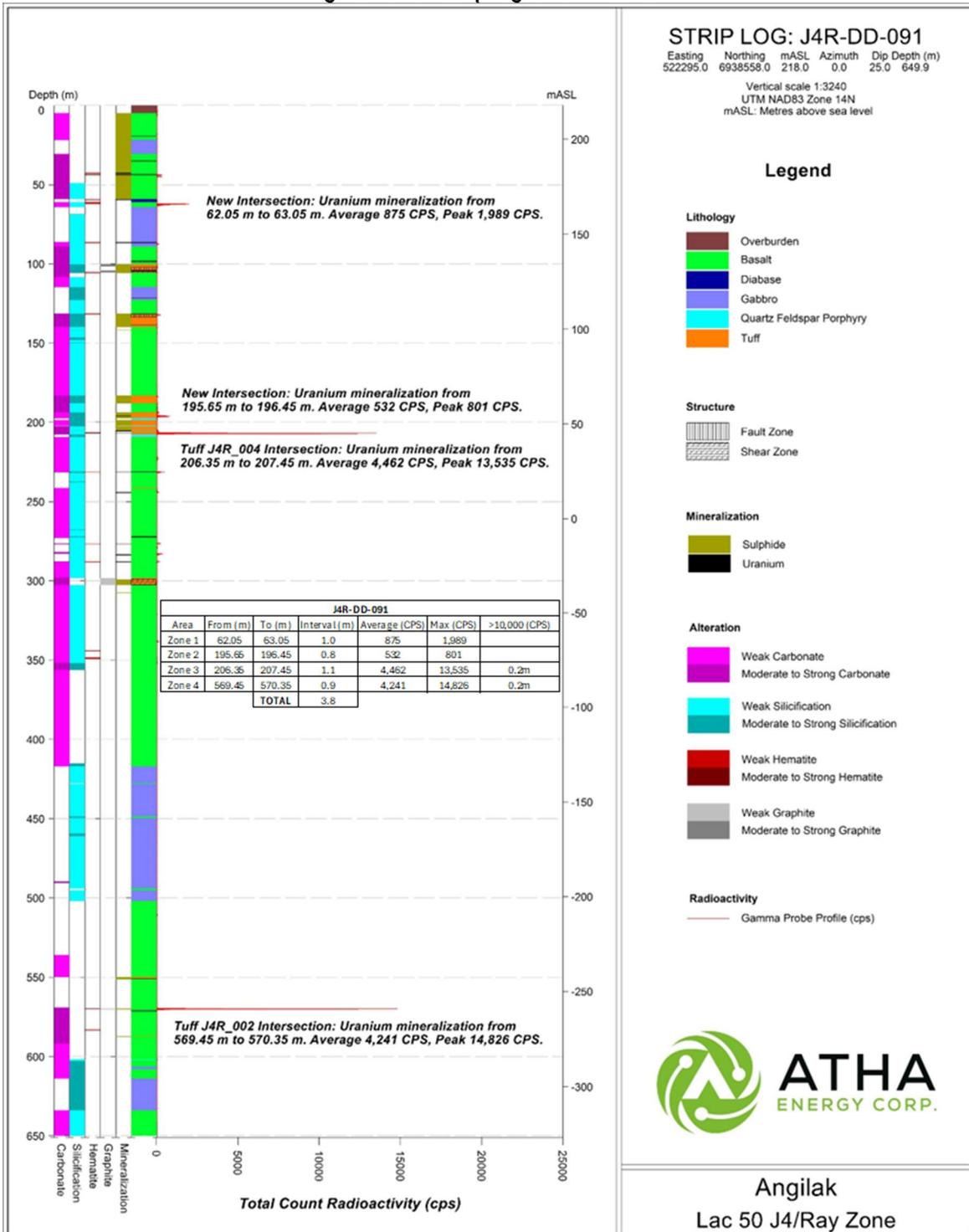
Table 10-3: Summary of ATHA 2025 Drillhole Details

Hole ID	Zone	Azi (°)	Dip (°)	Easting (m E)	Northing (m N)	Elev (m)	Final Depth (m)
J4R-DD-091	J4/Ray	25	-57	522295	6938558	218	650
RIBE-DD-001	RIB East	145	-55	497928	6929449	270	443
RIBE-DD-002	RIB East	145	-55	497766	6929322	271	345
RIBE-DD-003	RIB East	145	-63	497524	6929337	271	398
RIBE-DD-004	RIB East	145	-60	497404	6920180	271	428
RIBE-DD-005	RIB East	155	-65	497530	6929401	270	472
RIBE-DD-006	RIB East	145	-60	497670	6929501	273	491
RIBE-DD-007	RIB East	325	-50	497798	6929101	274	467
RIBE-DD-008	RIB East	325	-55	498284	6929287	264	464
RIBW-DD-001	RIB West	150	-50	495831	6929490	274	503
RIBW-DD-002	RIB West	145	-55	497766	6929322	271	380
RIBW-DD-003	RIB West	325	-55	497645	6930031	275	347
RIBN-DD-001	RIB North	300	-65	499574	6929887	261	623
RIBS-DD-001	RIB South	150	-50	495747	6927640	277.5	377
KU-DD-001	KU Target	30	-70	515830	6936190	256.5	599
KU-DD-002	KU Target	30	-70	515525	6936210	251	616
KU-DD-003	KU Target	30	-70	515758	6936059	268.5	56
KU-DD-003A	KU Target	30	-68	515758	6936059	268.5	605
KU-DD-004	KU Target	30	-60	515757	695641	255	602
KU-DD-005	KU Target	210	-70	515980	6935734	256	302
KU-DD-006	KU Target	30	-70	514794	6935805	275	647
ML-DD-013	ML Target	25	-50	523968	6939404	215	551
ML-DD-014	ML Target	25	-50	524869	6939109	206	407
Total Metres							10,774

In the Lac 50 deposit area a total of three drill holes and 1,608 metres were completed at the J4/Ray and Mushroom Lake zones. At the J4/Ray zone, drilling successfully extended mineralization down-dip by approximately 100 metres and remains open both along strike and at depth (**Figure 10-28**). At Mushroom Lake, two drill holes were completed to test the depth extent of uranium mineralization identified in outcrop. Both drill holes successfully intersected uranium mineralization as identified by preliminary down-hole radiometric probe results and remains open at depth and along strike.

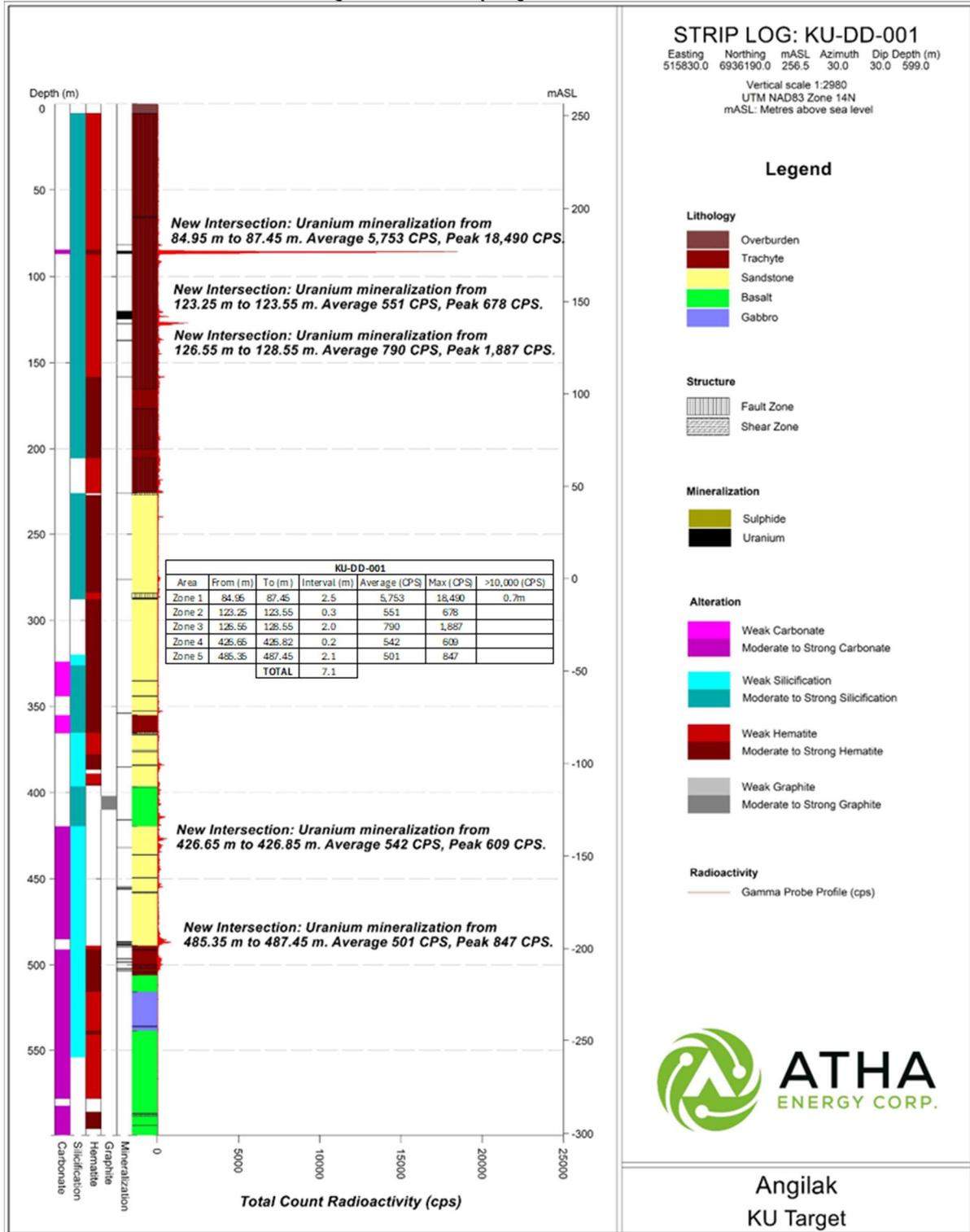
At the KU regional target area, a total of six drill holes and 3,427 metres of drilling were designed to test stacked gravity and EM geophysical anomalies coincident with interpreted NW-SE and E-W trending structures within the 31-km RIB-Nine Iron corridor. All drill holes successfully intersected mineralized structures with elevated radioactivity and is highlighted by KU-DD-001 which successfully targeted stacked gravity and structural anomalies. KU-DD-001 intersected shallow high-grade uranium mineralization as well as numerous zones of lower-grade mineralization throughout the hole as shown **Figure 10-29**. The hole intersected total composite mineralization of 7.1 m, including a continuous shallow zone of mineralization between 84.95 m and 87.45 m with 0.7 m of high-grade mineralization with radioactivity between 10,297 to 18,490 CPS.

Figure 10-28: Strip log for J4R-DD-091



Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

Figure 10-29: Strip log for KU-DD-001



Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

Within the RIB regional target area, a total of thirteen drill holes and 5,739 metres were completed to test stacked gravity and EM anomalies coincident with interpreted NE-SW, E-W and N-S trending structures along the regional RIB-Nine-Iron corridor. All drill holes successfully intersected mineralized structures associated with graphitic shears zones over an interpreted 12-kilometre conductive EM trend within the RIB area (**Figure 10-30**), with the most significant result coming from the RIB North area. Drill hole RIBN-DD-001 intersected 26.3 m of total composite uranium mineralization over 10 intervals from 345.55 m to 460.05 m (**Figure 10-31**). The hole intersected 13.6 m of continuous mineralization from 426.25 m to 439.85 m including 1.7 m of high-grade mineralization with radioactivity from 10,614 to 55,730 CPS.

Figure 10-30: Plan Map of RIB Conductive Trend Derived from 3D Inversion Model of 2024 MMT Survey

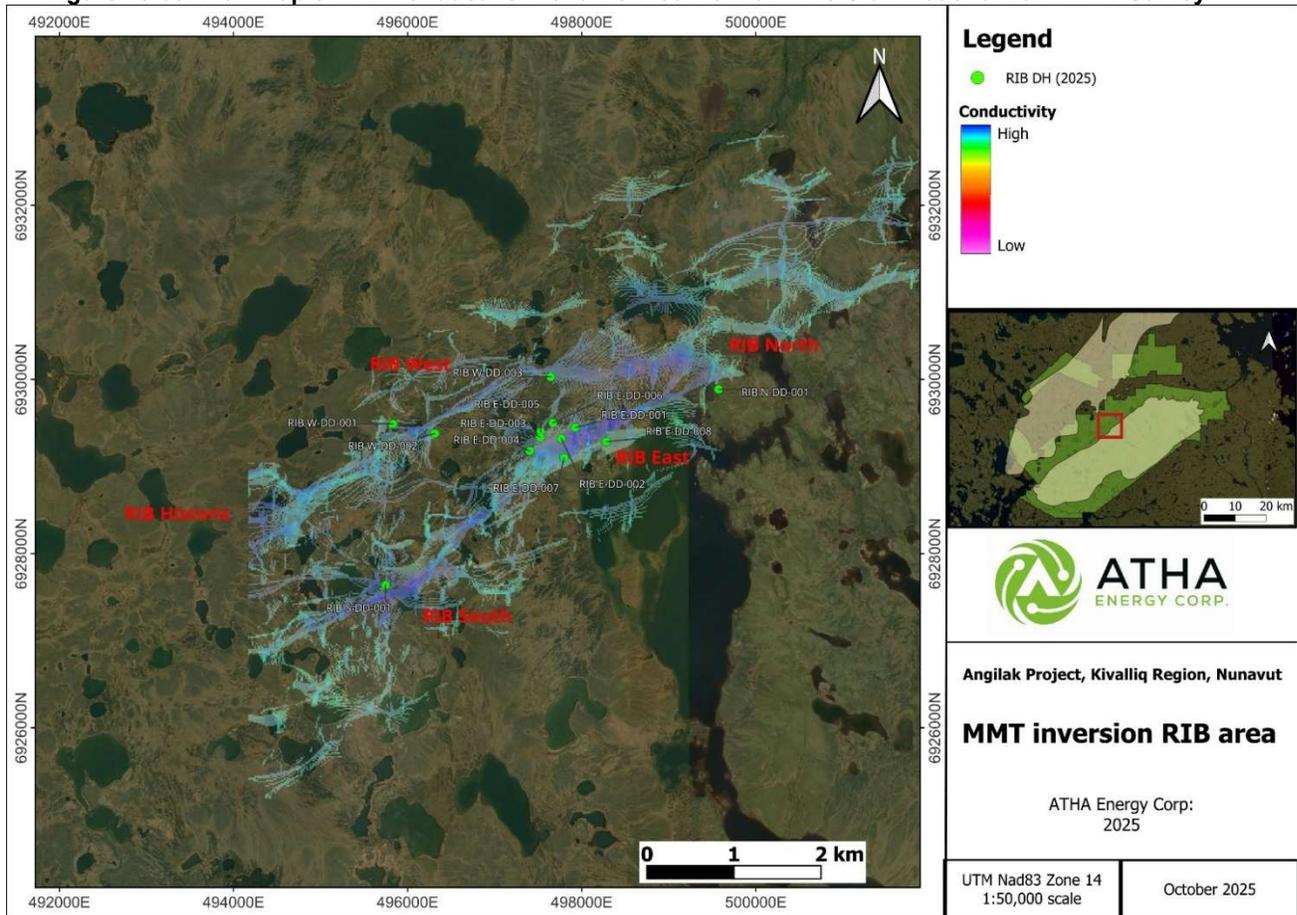
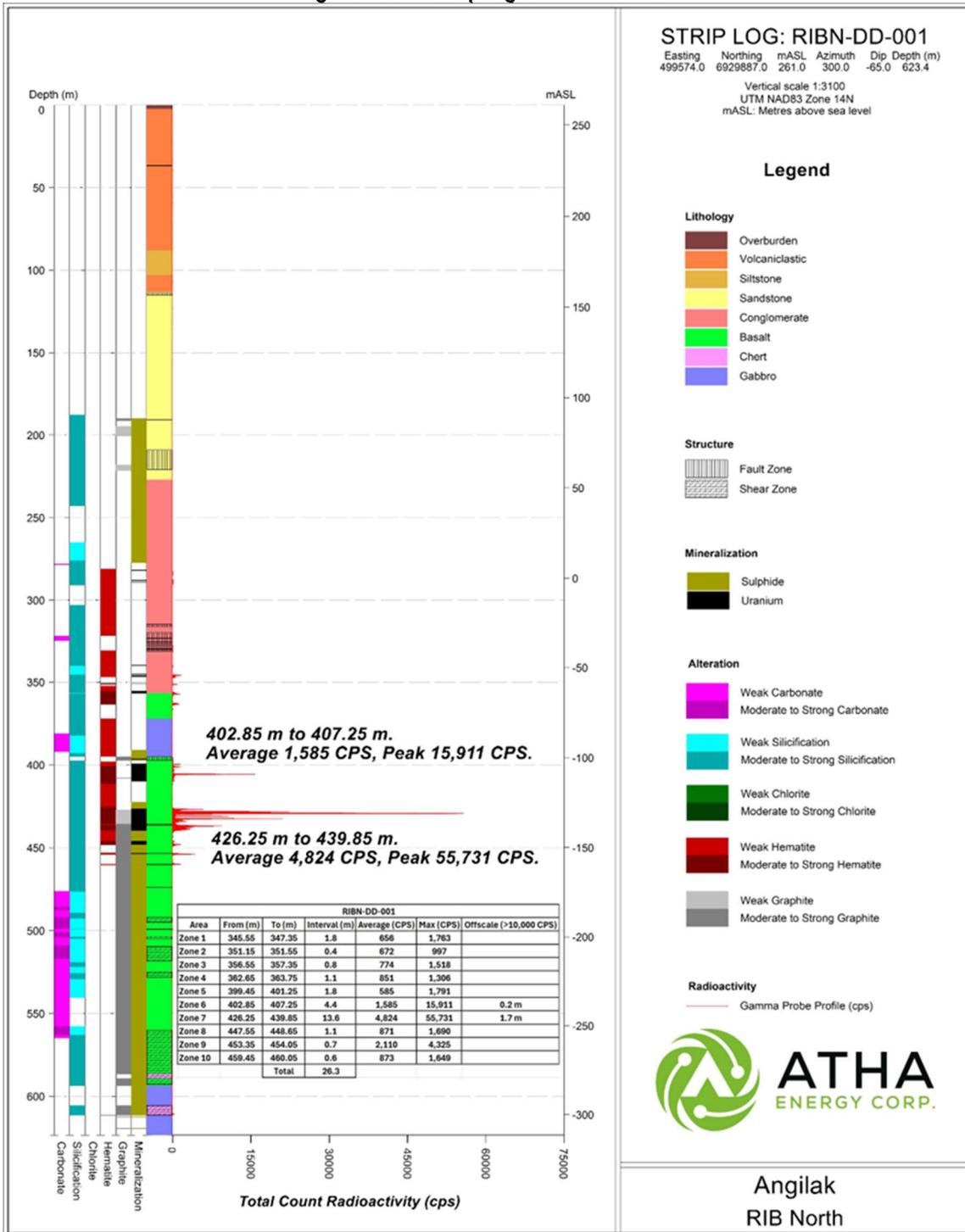


Figure 10-31: Strip log for RIBN-DD-001



Note: All depths and intervals are metres downhole, true thicknesses are yet to be determined.

10.6.3 Drill Contractor and Equipment

The 2024 and 2025 drilling campaigns were completed by Base Diamond Drilling using two X10 Diamond Drills mobilized to the Angilak Project in April 2024, staffed with a standard drill crew consisting of a supervisor, four drillers, four helpers and a 5th man/pad builder. Drilling was performed to the end of the hole with NQ rods (480 mm core diameter) and a 4.2 metre core (hexagonal) barrel. All drill casing and drill anchors were removed after final drilling on drill pad.

10.6.4 Drillhole Surveying

Each drill was aligned using the Devialigner, a north-seeking gyro system designed to accurately measure relative to true north, inclination of the drillhole, and the roll angle of the drill. The system includes a smartphone connection which displays the drill set-up information and is operated by the driller, helper and/or foreman.

During drilling, the drillhole deviation was recorded using the DeviGyro OX MINI and DeviCounter. The DeviGyro OX Mini is an overshoot surveying tool used to track the orientation of the drill hole every 50 metres, with one continuous survey conducted at the end of the hole. For continuous surveys, the DeviGyro OX Mini and the DeviCounter is connected to the DeviGyro app which provides representative depth measurements. The tool collects azimuth and dip data while descending and ascending the hole. This information is then correlated with the depths recorded by the DeviCounter. Finally, the data is displayed on the smartphone app to visualize the deviation, azimuth average, inclination average, survey parameters, and to confirm whether the survey has been properly completed. Each drill was equipped its own DeviGyro, DeviCounter, and two smartphones to connect to the tools. The equipment was operated by the driller and helper and the information sent via an Excel file to the project geologist for approval and uploading to the database.

Core orientation was measured using the CHAMPORI NQ equipment. This equipment consists of four components: a smartphone with the CHAMPORI app, two magnetically activated tools that link to the smartphone, and a level with a magnet to activate the tools. Each tool is attached to the end of the core barrel, activated, and linked to the smartphone to record the original orientation. Once each drill run is completed and the core barrel extracted, the drill helper rotates the barrel to align the core until the smartphone confirms the correct orientation, after which the bottom of the hole is marked. Each drill was equipped with its own core orientation kit, with an additional spare system located in camp.

Once drilling was complete, holes were probed using a 40TGU-1000 Triple Gamma Geiger probe. The probe was connected to an MX winch with a cable length of either 500 or 1000 metres. The winch system connects to a SCOUT PRO acquisition system and laptop running the Logger suite software. The system collects real-time data to ensure proper functionality at the start of each test. Down hole surveys were completed by the geological team, logging both down and up hole at a speed of 4-10 metres per minute, ensuring slower speeds were used over the mineralized intervals. Upon completion of the surveys, logging files were converted into industry standard. LAS file formats and reviewed by the geological team to ensure probe peaks aligned with radioactivity scintillometer profiles obtained directly from the drill core.

10.6.5 Drill Core Handling and Logging Procedures

Upon completion of a drill run, the core was removed from the core tube by the drill contractors and placed directly into an NQ-sized wooden core box which holds approximately 4.5 metres of continuous drill core. Each drill run was separated by a wooden block labeled with the appropriate

depth meterage. Intervals of lost core were identified with separate block indicators providing the interval from and to position of the lost core interval, however core recovery at the Angilak Project is very good due to the competent nature of the host rocks in the Lac 50 Deposit area. At the end of each 12-hour drill shift the core was transported via helicopter to the core shack located at Nutaaq Camp.

Once core was received at the logging facilities, the geologist and geological technician organize the boxes on the core tables and verify the depths recorded on the wooden blocks at the end of each run. Each drill run is then oriented by projecting the orientation mark from the CHAMPORI equipment to the remaining core. The technician typically marks the core at one-metre intervals and records geotechnical data such as core recovery, fracture breaks, and Rock Quality Designation (RQD). Magnetic susceptibility and conductivity are measured every three metres using a KT-20 physical property measuring system, and radioactivity is assessed with a handheld scintillometer (RS-120). The geotechnician takes scintillometer readings throughout the drillhole at 3 metre intervals in non-mineralized areas and values are averaged for each interval. In mineralized zones where scintillometer values exceed five times the background (approximately 500 counts per second with RS-120), readings are recorded at 10-centimetre intervals. The Geotechnician also labels the core trays with aluminum tags and felt markers.

The geologist logs the core for lithology, alteration, mineralization, structural intervals, point structures, defines sample intervals, and photographs the core. Core is stored in core racks in a designated, non-fenced core storage area away from the main camp site.

The sampling of the drillholes is divided into four categories: Composite, Reflectance, Density and Assay samples. Composite and Reflectance samples are collected from the same depth intervals and represent typical background (non-mineralized) core. Composite samples are sent to the SRC for a comprehensive geochemical analysis; Reflectance samples are collected and sent off-site for spectral analysis to identify the mineralogy of the sampled rock type. Assay sample intervals are determined based on the observed geology and scintillometer readings and sent to the SRC for geochemical analysis and U₃O₈ Wt % assay. For more details concerning sampling procedures, see Sample Methods below.

The logging and sampling data is entered into the MXDeposit™ online database platform where it can be viewed in real-time by the project geologist, who ensures that the information is complete and accurately recorded in the database.

Sample Method: Composite samples

Composite geochemical samples are taken throughout selected drillholes, excluding areas where assay or density samples are collected. Each sample covers an interval of approximately 10 metres. The samples consist of 1 to 2 centimetre discs of core collected from the bottom of each row in the core box over the specified interval. No lithological contacts or boundaries are crossed within a single sample. The samples are bagged and sealed in plastic bags with the corresponding sample tag, then sealed for shipment in plastic pails.

Sample Method: Assay samples

ATHA submits assay samples for mineralized core intervals where core recovery allows. Mineralized core is scanned using a handheld scintillometer (model RS-120). To minimize contamination from ambient background radiation, the core is removed from the core box for scanning, and the results are recorded in counts per second (cps). After scanning, the core is returned to its original location and orientation in the core box. Core registering values over 500

cps are marked as anomalous and later split for laboratory assay. Samples collected in 2023 by LUR had variable intervals ranging from 0.3 to 1.0 metres in length, while samples collected by ATHA were predominately collected at 0.5 metre intervals. Additionally, barren samples of 0.5 metres in length were taken at both ends of the mineralized intervals, extending to cover a total length of 2 metres beyond the mineralized interval.

All assay samples are split using a hydraulic core splitter according to the sample intervals previously marked on the core by the geologist. One half of the core is returned to the core box for reference, while the other half is bagged and sealed in a plastic bag with the corresponding sample tag. Samples are sealed for shipment in plastic or metal pails, depending on the radioactivity level. The samples were shipped to SRC via air transport, accompanied by Transport of Dangerous Goods (TDG) documentation completed by qualified personnel.

Sample Method: Samples for Spectra Analysis

Rock chip samples are collected at the same intervals as the composite samples. Each sample consists of a 1 to 2-centimetre core disk with a fresh surface. The samples are bagged, tagged, and sealed in small plastic bags for analysis. They are then analyzed for mineral identification using reflectance spectroscopy at the Nutaaq camp. The samples are organized and stored in plastic pails at the logging facility on-site.

Sample Method: Density samples

Density samples are primarily taken from mineralized intervals and regionally from unaltered and unmineralized rock. Core samples approximately 10 centimetre in length are marked, tagged, and collected for density measurement. Field density measurements are determined using the water immersion method, weighing the dry core in air and then when immersed in water. The measurements are entered into MXDeposit™, and the sample is dried and sealed in a plastic bag with the corresponding tag. Depending on the radioactivity level, density samples are stored in plastic or metal pails and shipped to SRC via air transport, accompanied by TDG documentation completed by qualified personnel.

11 SAMPLE PREPARATION, ANALYSES, AND SECURITY

11.1 Sample Preparation

11.1.1 ValOre (Kavalliq Energy) Diamond Drilling (2009 to 2022)

Core samples collected during the 2009 to 2012 diamond drilling programs as well as the 2013, 2015 and 2022 drilling campaigns, comprised half split NQ drill core. Drill core from these programs were logged, sampled and stored at the Nutaaq logging facilities. Sample intervals were selected based upon both lithology and radiometrics. Sample thickness ranged from 0.5 to 1.5 metres. Mineralized zones were completely sampled along with one or more 0.5 to 1.0 metre wall rock buffer samples usually collected on either side of intersected mineralized zones.

The samples were accepted in Yellowknife by Discovery Mining Services and then loaded onto trucks for transportation to the SRC Laboratory in Saskatoon, Saskatchewan. The samples were first analyzed by SRC's inductive coupled plasma-optical emission spectroscopy (ICP-OES) multi-element uranium exploration ICP1 method. The method analyzed for multi-elements include Ag, Mo, Cu, Pb, Zn and a suite of rare earth elements. ICP results U>1000 parts per million (ppm) were analyzed using SRC's U₃O₈ Assay method.

11.1.2 ValOre Soil Sampling (2022)

Soil samples were collected from the B horizon within 3 metres of the proposed GPS coordinates. Samples were described, photographed and recorded at each site. A total of 926 samples were sent for analysis during the program; 880 of which were soil samples, along with 16 duplicates and 30 QA/QC samples. Batches of samples were placed within 20-litre pails and sealed with a tamper proof lid. All samples were sent to Activation Laboratories Ltd. (ActLabs) in Ancaster, Ontario for Enzyme Leach Analysis.

11.1.3 ValOre Reverse Circulation Drilling (2022)

Geological samples were collected over five-foot drill runs with a small portion used for logging in Nutaaq camp. A total of 401 samples were collected in plastic pails at the drill and 21 QA/QC samples were added to the sample batches. All sample pails were flown to Baker Lake and forwarded on cargo planes to Yellowknife. From Yellowknife, the sample pails were transported by a contracted carrier, Manitoulin, from Discovery Mining's warehouse to Vancouver and Saskatoon, respectively.

A total of 135 RC samples were sent to ALS Laboratories (ALS) in North Vancouver, BC, and 266 RC samples were sent to the SRC in Saskatoon, SK.

11.1.4 LUR Diamond Drilling (2023)

Core samples collected during LUR's 2023 diamond drilling program comprised half split NQ drill core. Intervals were selected based upon mineralization, radiation, lithology and structure. Sample lengths ranged from 0.2 to 1.2 metres, and where radioactivity was present buffer samples of 0.2 to 1.2 metres in length were taken above and below the radioactive samples. Core was split using a hydraulic core splitter and half the core was collected for sampling.

Samples were placed in plastic bags with identification tags, sealed with secure plastic ties and subsequently packed into plastic pails sealed with tamper proof lids. If the outside surface of the plastic pail measured greater than 5,000 CPS, the core was packed into an IP3 steel drum for shipping. The IP3 drums were put into crates in Baker Lake to facilitate further transport. Radioactive core was packed into the center of the drum surrounded by non-radioactive core on

all sides. Sample submittal forms were filled out to include shipment numbers along with sample sequences and total numbers of samples.

A total of 838 core samples were flown to Baker Lake and forwarded on cargo planes to either Yellowknife or Winnipeg and then road transported to the SRC laboratory in Saskatoon, Saskatchewan.

There were no significant issues identified concerning sample shipments or sample security during 2023 drilling program.

11.1.5 ATHA Diamond Drilling

Core samples collected during ATHA's diamond drilling programs include; 1) half-split NQ drill core for assay samples, 2) 10-metre composite samples comprised of 1 to 2 centimetre discs of core taken from the bottom of each row in the core box over 10-metre intervals, and 3) 10 centimetre-core lengths for density samples. All core was logged, sampled, and stored at the Nutaaq Camp logging facilities. Splitting areas at the camp were thoroughly cleaned of dust and rock chips between samples to prevent cross-contamination during the splitting process. The samples were placed in sample bags and sealed in plastic or metal pails, depending on the radioactivity level.

All samples were flown to Baker Lake, then transported by cargo planes to Winnipeg before being road transported to the SRC. Sample shipments were accompanied by Transport of Dangerous Goods (TDG) documentation completed by qualified personnel. A request for analysis form was prepared prior to shipment, detailing each batch of samples, sample types, preparation codes, and analysis codes.

All samples for assay, density, or geochemical analysis were submitted to the SRC Laboratory in Saskatoon, Saskatchewan.

There were no significant issues identified concerning sample shipments or sample security during 2024 drilling program.

11.2 Analyses

11.2.1 Drill Core Geochemical Analyses and Assay

ICP1 (Uranium multi-element exploration analysis by ICP-OES)

In ICP-OES analysis, the atomized sample material is ionized, and the ions then emit light (photons) of a characteristic wavelength for each element, which is recorded by optical spectrometers. Calibrations against standard materials allow this technique to provide a quantitative geochemical analysis.

The analytical package includes 63 analytes (47 total digestion, 16 partial digestion), with nine elements analyzed for partial and total digestions (Ag, Co, Cu, Mo, Ni, Pb, U, V, and Zn). Samples may also be analyzed for gold by fire assay, upon request. The ICP1 analytical package includes the following analytes.

Total Digestion: Ag, Al₂O₃, Ba, Be, Cd, CaO, Ce, Cr, Co, Cu, Dy, Er, Eu, Fe₂O₃, Gd, Ga, Hf, Ho, K₂O, La, Li, MgO, MnO, Mo, Na₂O, Nb, Nd, Ni, P₂O₅, Pb, Pr, S, Sc, Sm, Sn, Sr, Ta, Tb, Th, TiO₂, W, U, V, Yb, Y, Zn, Zr.

For total digestion analysis, an aliquot of pulp is digested in a hot block digester system using a mixture of concentrated HF, HNO₃, and HClO₄. The dried residue is then dissolved in 15 mL of dilute HNO₃ and analyzed using the same instrument(s) as for partial digestion.

Partial Digestion: Ag, As, Bi, Co, Cu, Ge, Hg, Mo, Ni, Pb, Sb, Se, Te, U, V, Zn.

For partial digestion, an aliquot of pulp is digested in a digestion tube using a mixture of HNO₃ and HCl in a hot water bath for approximately one hour, then diluted to 15 mL with deionized water. The samples are then analyzed using a Perkin Elmer ICP-OES instrument.

In addition, boron is determined by ICP-OES analysis after fusion with NaO₂/Na₂CO₃.

ICP-MS Exploration Package

This analytical package includes the analysis of 54 elements and oxides using a three-acid total digestion method (HF: HNO₃: HClO₄) and a suite of 44 elements using a two-acid partial digestion method (HNO₃: HCl). The package also includes the analysis of lead isotopes (²⁰⁴Pb, ²⁰⁶Pb, ²⁰⁷Pb, and ²⁰⁸Pb). PerkinElmer instruments are currently used. The samples analyzed by this package are generally non-radioactive, non-mineralized sandstones and basement rocks with low uranium concentrations (<100 ppm).

The package consists of three separate analyses:

1. ICP-MS analysis on the partial digestion includes the following elements: Ag, As, Be, Bi, Cd, Co, Cs, Cu, Dy, Er, Eu, Ga, Gd, Ge, Hf, Hg, Ho, Mo, Nb, Nd, Ni, ²⁰⁴Pb, ²⁰⁶Pb, ²⁰⁷Pb, ²⁰⁸Pb, Pb, Pr, Rb, Sb, Sc, Se, Sm, Sn, Ta, Tb, Te, Th, U, V, W, Y, Yb, Zn, and Zr.),
2. ICP-OES analysis for major and minor elements on the total digestion (Al₂O₃, CaO, Fe₂O₃, K₂O, MgO, MnO, Na₂O, P₂O₅, TiO₂, Ba, Ce, Cr, La, Li, Sr, S, V, and Zr),
3. ICP-MS analysis for trace elements on the total digestion ((Ag, Be, Bi, Cd, Co, Cs, Cu, Dy, Er, Eu, Ga, Gd, Hf, Ho, Mo, Nb, Nd, Ni, ²⁰⁴Pb, ²⁰⁶Pb, ²⁰⁷Pb, ²⁰⁸Pb, Pb, Pr, Rb, Sc, Sm, Sn, Ta, Tb, Th, U, W, Y, Yb, and Zn.

For partial digestions, an aliquot of pulp is digested in a mixture of ultra-pure concentrated nitric acid (HNO₃:HCl) in a digestion tube placed in a hot water bath. The solution is then diluted to 15 mL with de-ionized water prior to analysis.

For total digestion, an aliquot of pulp is digested in a hot block digestion system using a mixture of ultra-pure concentrated acids (HF:HNO₃:HClO₄). The dried residue is dissolved in 15 mL of 5% nitric acid (HNO₃) and brought to volume with de-ionized water prior to analysis.

U₃O₈ wt% Assay by ICP-OES

When the uranium partial values from ICP1 are ≥500 ppm, sample pulps are re-assayed for U₃O₈ using SRC's ISO/IEC 17025:2017-accredited method for determining U₃O₈ (wt%). In the case of uranium assays conducted by ICP-OES, a pulp is already generated during the initial phase of sample preparation and assaying. For analysis by aqua regia, an aliquot of the sample pulp is digested in a 100 mL volumetric flask using a mixture of HCl:HNO₃ in a 3:1 ratio on a hot plate for approximately one hour. The solution is then diluted to volume with de-ionized water for analysis by ICP-OES.

Fire Assay

During fire assay an aliquot of sample pulp was mixed with standard fire assay flux in a clay crucible and a silver inquart was added. The mixture was fused in a fire assay oven. The fusion melt was poured into a metal form and cooled. The lead bead was recovered and put into the oven for cupellation until only the precious metal bead remained. The bead was parted in a solution heated in a boiling water bath until the silver dissolved. The solution containing the silver was decanted, leaving the gold in the test tube. Aqua Regia was added to the gold in the test tube and heated in the boiling water bath until the gold dissolved. The sample was diluted to volume and analyzed by ICP-OES. This method is suitable for all pulverized and core samples for the determination of gold. The detection limit for Au using this method is 1 ppb.

Density by Dry Bulk Method

Drill core samples collected for dry bulk density measurements are sent to SRC. Upon receipt, the samples are first weighed and then submerged in de-ionized water for re-weighing. Afterward, the samples are dried until a constant weight is achieved. They are then coated with an impermeable layer of wax and weighed again while submerged in de-ionized water. The weights are entered into a database, and the bulk density of each sample is calculated. Additionally, the water temperature at the time of weighing is recorded and used in the bulk density calculation.

11.2.2 ValOre Soil Sampling Analysis (2022)

All samples were analyzed using Enzyme Selective Extraction (ESE) and analyzed by inductively coupled plasma mass spectrometry (ICP-MS).

11.2.3 ValOre Reverse Circulation Sampling Analysis (2022)

The reverse circulation samples were prepared and analyzed for partial digestion (ICP1), total digestion (ICP1), Au fire assay (Au 2), and U_3O_8 assay. The partial digestion, total digestion, U_3O_8 assay and Au fire assay analyses are the same as described in the section 11.2.1 Drill Core Geochemical Analyses and Assay.

11.3 Security

In 2024 and 2025, as each hole was drilled, drilling contractor personnel placed the core in boxes at the drill site and secured core boxes with lids screwed on, tied, or nailed to the box. Core was then delivered to the core processing facility via helicopter daily, twice a day, or when weather conditions permit. All core was logged, sampled and stored at the Nutaaq Camp logging facilities. On site sample preparation consists of core splitting by geological technicians under the supervision of geologists. One half of the core is placed in sample bags with a sample number tag and the other half is returned to the core box, which is later stored at the core storage area located near the logging facility. The bags containing the split samples are then placed in buckets with lids for transport.

All samples were flown to Baker Lake, then transported by cargo planes to Winnipeg, before being road transported to SRC. The samples were accompanied by Transport of Dangerous Goods (TDG) documentation completed by qualified personnel. A request for analysis form was prepared prior to shipment, detailing each batch of samples, sample types, preparation codes, and analysis codes.

Samples were received at SRC either as dangerous goods requiring appropriate TDG documentation or as exclusive-use samples (with no radioactivity documentation attached). Upon arrival, all information pertaining to a received shipment of samples is verified by sample receiving personnel, including sample numbers, number of pails, sample type/matrix, condition of samples,

and request for analysis. After the completion of analyses, data are sent securely via electronic transmission to ATHA. These results are provided as a series of PDFs and an Excel spreadsheet.

SRC places a large emphasis on confidentiality and data security. Appropriate steps are taken to protect the integrity of samples at all processing stages. Access to the SRC premises is restricted and monitored. SRC is an ISO/IEC 17025/2005 and Standards Council of Canada certified analytical laboratory and is independent of the Author and the issuers.

In reviewing previous documentation on the project, the Author is of the opinion that sample handling, shipment, and security for samples collected between 2009 and 2023 was completed in a similar manner to that of the ATHA procedure outlined above.

11.4 Quality Assurance and Quality Control

11.4.1 Protocols

Historical Drill Core QA/QC Protocols (2009 to 2022)

The 2009 and 2010 core drilling programs used barren gabbro from hole DDH 09-775-01 as blank material and inserted it into the sample stream. During the 2011 drill program additional non-mineralized gabbro drill core was sourced from DDH 10-LC-061 and inserted as blank material. Blank material during the 2012 program comprised of non-mineralized gabbro or basalt from hole DDH 10-LC-061 or DDH 11-LC-006. Similarly, blank material for the 2013 and 2015 drill seasons was sourced from non-mineralized gabbro or basalt drill core in drillholes DDH 10-LC-061 (2013) or DDH 11-LC-112 (2015). The 2022 diamond drilling program used certified coarse blank material from OREAS and inserted randomly using a pre-assigned tag number at the rate of one in every 50 samples.

ValOre (Kivalliq) purchased certified reference material (CRM or standard) for insertion into the sample stream during 2011 and 2012 from the Canada Centre for Mineral and Energy Technology in Ottawa, Ontario. Four certified uranium CRMs were used: BL2-A, BL4-A, BL5, and CUP 1. For the 2013-2015 drill programs, three certified uranium CRMs were used: BL4-A, BL5, and CUP 1. In 2022, ValOre purchased certified reference material for insertion into the sample stream from OREAS. Four certified uranium CRMs were used: Oreas 120, Oreas 122, Oreas 123 and Oreas 124.

LUR Drill Core QA/QC Protocols (2023)

Quality control samples were inserted into the core sample stream as CRMs and certified coarse blanks. Duplicate samples were split from half split core with a hydraulic splitter. Blanks, certified reference materials, and repeats were inserted into the sample stream at regular intervals by LUR and the SRC in accordance with quality assurance/quality control (QA/QC) procedures. Geochemical assay data were subject to verification procedures by qualified persons employed by LUR prior to disclosure.

LUR purchased certified reference material for insertion into the sample stream from OREAS North America, Sudbury, Ontario, Canada. Seven certified uranium CRMs were used: OREAS 23b, OREAS 90, OREAS 120, OREAS 121, OREAS 122, OREAS 123 and OREAS 124. The performance of the standards was evaluated using the criterion that assay results fell within 3 standard deviations from the certified value based on the standard deviation reported by the manufacturer. Results for all standards fall within control limits. There is no indication of systematic analytical errors in the uranium or copper assays. For detailed results, refer to Dufresne and Shoeman, 2024.

A total of 86 duplicate core samples were collected to assess sample preparation bias. Duplicate core samples were taken at random approximately every 25th sample by splitting the remaining core in half, leaving one quarter core for reference in the core box. The comparison returned a correlation coefficient of 0.7624 which is considered low. In the QP's opinion, the low correlation is due to the exceedingly low grades of the samples selected for duplicate analysis. The dataset has an approximate range of 0.1 to 300 ppm U and a median of ~8 ppm U. A sample measured at 8 ppm can have a 1 ppm change in the duplicate yet represent a 12.5% change in U.

ATHA Drill Core QA/QC Protocols

Standard Operating Procedures (SOPs) ensure consistency and quality control in the selection and preparation of core samples, safeguarding data integrity at the Angilak Project. The sampling procedures define the frequency at which control samples are inserted into the core sample stream by ATHA personnel. Control samples include blanks, standards, and duplicates. Duplicate samples are split from half-split core using a hydraulic splitter. Blanks and standards are certified reference materials (CRM).

ATHA purchased certified reference materials (CRMs) for insertion into the sample stream from OREAS North America, located in Sudbury, Ontario, Canada. Five certified uranium CRMs were used: OREAS 120, OREAS 121, OREAS 122, OREAS 123, and OREAS 124 (Table 11-1). The performance of these standards was evaluated based on the criterion that assay results fall within three standard deviations of the certified value, as reported by the manufacturer. For CRMs where total digestion techniques (i.e. 4-acid digestion) used for certified values vary from SRC's total digestion technique (i.e. 3-acid digestion), CRM mean and standard deviations have been determined directly from SRC analytical results and used for quality control purposes typically resulting in more stringent control criteria. Additionally, three certified blank CRMs were used: OREAS coarse sand (silica), OREAS 22h, and OREAS 90.

11.4.2 QA/QC Results

Results of the QA/QC program have been well documented by ATHA, LUR, and ValOre. The QP reviewed the documentation provided by ATHA in addition to reviewing the QA/QC data.

Results from the QA/QC samples are continually tracked by ATHA as certificates for each sample batch are received, checking for batches that exceed the failure criteria. Standard reference materials fail when results are more than three standard deviations from the expected value. Blank samples fail when results are greater than 10 times the lower detection limit. If QA/QC samples of a sample batch pass within acceptable limits, the results of the sample batch are imported into the master database. If the QA/QC sample fails, the entire batch is reanalyzed.

Historical Drill Core QA/QC Results (2009 to 2022)

All historical certificates were reimported into ATHA's database, with various blanks and standards verified for any failures. Any minor errors were corrected, and the correction was recorded in the sample table. Lab standards used were BL2, BL2-A, BL4A, BL5, CAR110, CAR218, CG51509, and SRCUO2. Field standards used were BL2, BL4A, BL5, CUP 1, MPb1b, OREAS 120, OREAS 122, OREAS 123, and OREAS 124. Z-scores of these field standards are shown in Figure 11-1. Of the 1181 of the historic standards measured, only four fell outside of the acceptable range of the mean plus or minus 3 standard deviations. Of these, three of the standards were from the 2012 drilling season and one was from the 2022 season. Three of the failed standards are considered low grade (BL2A and OREAS122) and one is considered high grade (BL5). ATHA and the QP consider this to be acceptable for historical data, but further investigation will be conducted to identify the cause of the failures. Note that some standards

have no record in the database to their source. Work is currently in progress to source these standards.

Table 11-1: Certified Reference Material Details

CRM Code	Sample Decomposition	Analytical Method	Element	Unit	Certified Values	Standard Deviation	Certified Value Source
BL2A	HF:HNO3:HClO4	ICP1-OES2	U	ppm	4264	32.5	Lab
BL2A	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.502	0.0031	Lab
BL3	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	1.21	0.0067	Lab
BL4A	HF:HNO3:HClO4	ICP1-OES2	U, ICP	ppm	1260	20	Lab
BL4A	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.147	0.001	Lab
BL5	HF:HNO3:HClO4	ICP1-OES2	U	ppm	71200	350	Lab
BL5	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	8.36	0.0133	Lab
CAR110	HF:HNO3:HClO4	ICP1-OES2	U, ICP	ppm	3350	85	Lab
CAR218	HF:HNO3:HClO4	ICP1-OES2	U	ppm	3014	36	Lab
CG51509	HF:HNO3:HClO4	ICP1-OES2	U, ICP	ppm	2	0.6667	Lab
CUP 1	HF:HNO3:HClO4	ICP1-OES2	U, ICP	ppm	1272	19	Company
CUP 1	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.149	0.0012	Company
DCB01	HF:HNO3:HClO4	ICP1-MS2	U	ppm	124	3.7415	Lab
KEC Blank	HF:HNO3:HClO4	ICP1-OES2	U, ICP	ppm	4	5	Company
OREAS 120	HF:HNO3:HClO4	ICP1-OES2	U	ppm	39.82	0.64	Company
OREAS 120	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.005	0.0005	Company
OREAS 121	HF:HNO3:HClO4	ICP1-OES2	U	ppm	208.3	1.53	Company
OREAS 121	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.0245	0.0005	Company
OREAS 122	HF:HNO3:HClO4	ICP1-OES2	U	ppm	415.8	4.61	Company
OREAS 122	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.048	0.001	Company
OREAS 123	HF:HNO3:HClO4	ICP1-OES2	U	ppm	849	5.66	Company
OREAS 123	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.0987	0.001	Company
OREAS 124	HF:HNO3:HClO4	ICP1-OES2	U	ppm	1796	14.46	Company
OREAS 124	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.212	0.0025	Company
OREAS 22h	HF:HNO3:HClO4	ICP1-OES2	U	ppm	1	0.02	Company
OREAS 23b	HF:HNO3:HClO4	ICP1-MS2	U	ppm	6.28	0.08	Company
OREAS 90	HF:HNO3:HClO4	ICP1-MS2	U	ppm	3.5	0.1	Company
OREAS CS Blank	HF:HNO3:HClO4	ICP1-MS2	U	ppm	0.79	0.71	Company
OREAS CS Blank	HF:HNO3:HClO4	ICP1-OES2	U	ppm	1.14	0.65	Company
OREAS CS Blank	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	0.0006	0.0002	Company
SRCU02	HCl:HNO3	U3O8-ASSAY	U3O8	wt %	1.58	0.03	Lab

Blank material (Table 11-1) from 2009 to 2015 was sourced from non-mineralized gabbro or basalt from drillholes completed on site. The follow up drill program in 2022 used certified coarse blank material from OREAS. Of the 518 blank samples measured historically (Figure 11-2), three blanks fall as outliers. The reason for these could potentially be due to cross contamination of lab equipment or because the blank sources historically were field samples assumed to be non-mineralized but potentially could have been weakly mineralized. Further investigation will continue relating to these blanks, but ATHA and the QP consider this dataset to be acceptable.

Drill Core QA/QC Results (2023 – 2024)

In 2023 and 2024, LUR and ATHA utilized a range of standard reference materials sourced from OREAS to ensure the quality and accuracy of analytical results. As the SRC laboratory does not provide certified values for its analytical methods in conjunction with these standards, in-house certified values are being established. Field control limits were calculated from the 2023 and 2024 assay results and applied retroactively; any significant deviations from the expected values are promptly flagged and addressed. Figure 11-3 to Figure 11-6 illustrate the performance of all field standards and blanks from the 2023 and 2024 field seasons. Of the 391 standards measured in 2023-2024, only three standards fall outside of the acceptable range of mean plus three standard deviations (SD). Two of the standards are related to Oreas 121 and one is related to Oreas 120, which are both low grade standards. ATHA and the QP consider these results acceptable. ATHA has communicated that re-analysis will be completed on the failed standards. Of the 193 blanks measured in 2023-2024 no samples fall outside of the accepted range.

SRC Lab Standards

SRC completes their own QA/QC before returning results, and they are verified a second time by ATHA. Any certificates that include a standard sample that fall outside of 3 SD is returned for re-analysis. As well, any certificate that includes two successive samples that fall outside 2 SD are also returned for re-analysis. All laboratory control samples fall within control limits.

SRC Lab Duplicates

One in every 40 samples is analyzed in duplicate by the laboratory. The 2023 data (Figure 11-7) show a high reproducibility of lab duplicates within the acceptable 10% tolerance, with one lab repeat falling outside the 10% tolerance range which requires further follow-up. The 2024 results (Figure 11-8) also show the same high reproducibility, with one lab repeat plotting outside of the 10% tolerance range. However, it should be noted the analytical result for this specific lab repeat is still pending from the laboratory. The QP believes the laboratory's reproducibility meets or exceeds industry standards.

LUR and ATHA Field Duplicates

Core duplicates are prepared by collecting a second sample of the same interval, through splitting the original sample (quarter core samples), and are submitted as an independent sample. Duplicates are typically submitted at a minimum rate of one per 25 samples. Variability observed when comparing field duplicates to original assay results (Figure 11-9) is attributed to several factors, including mineralization heterogeneity associated with narrow-vein style mineralization, sample size reduction (i.e. quarter core samples from NQ diameter drill core) and analytical error. As well, the 2023 data was predominately collected from low grade (less than 100 ppm) samples which also contributes to the observed variability within the field duplicate results.

Figure 11-1: Z-Scores of all field standards used historically (2009-2023) for ICP1 OES, Total Digestion

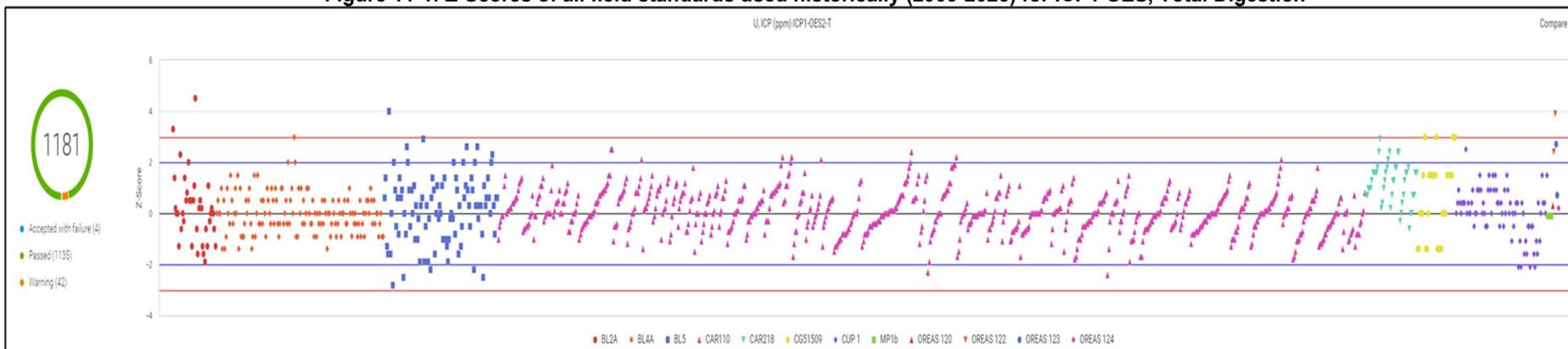


Figure 11-2: Historical (2009-2023) Blank Results (ICP1 OES, Total Digestion) using Non-Mineralized Gabbro or Basalt Drill Core

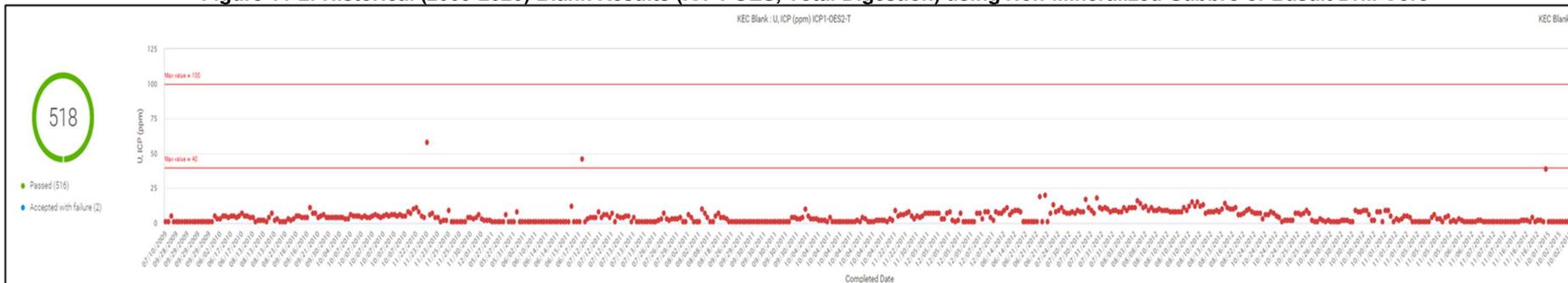


Figure 11-3: Z-Scores of all Field Standards used in 2024, for ICP1 OES Total Digestion



Figure 11-4: Z-Scores of all Field Standards used in 2023, for ICP1 MS Total Digestion

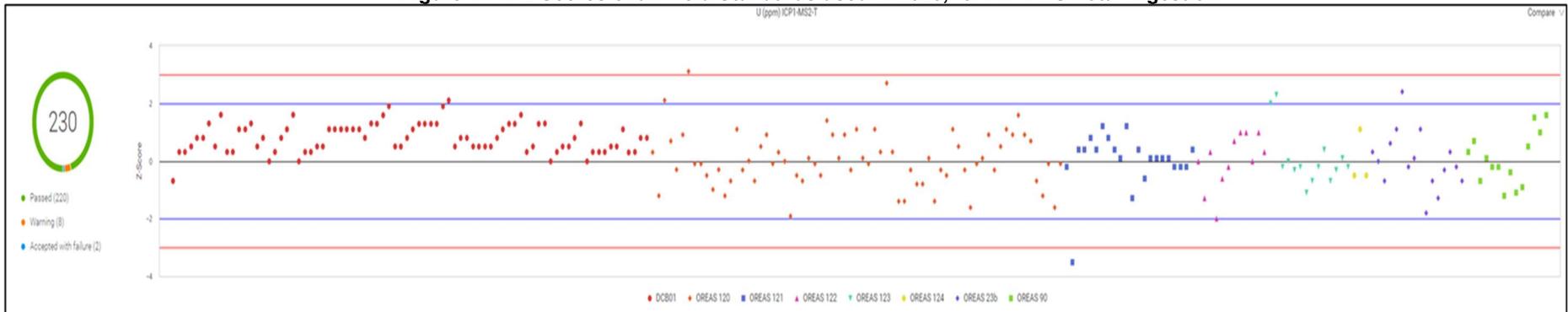


Figure 11-5: 2024 Blank Results (ICP1 OES, Total Digestion) using the CS OREAS Standard

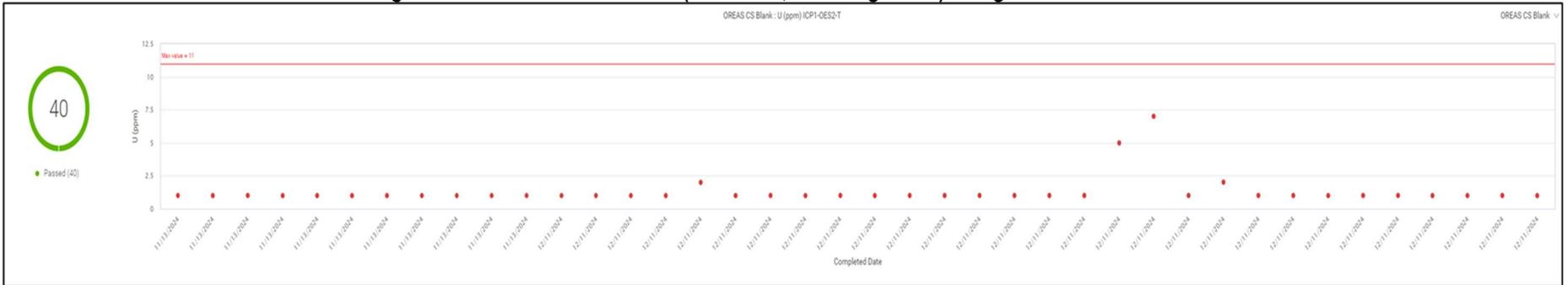


Figure 11-6: 2023 Blank Results (ICP1 MS, Total Digestion) using the CS OREAS Standard

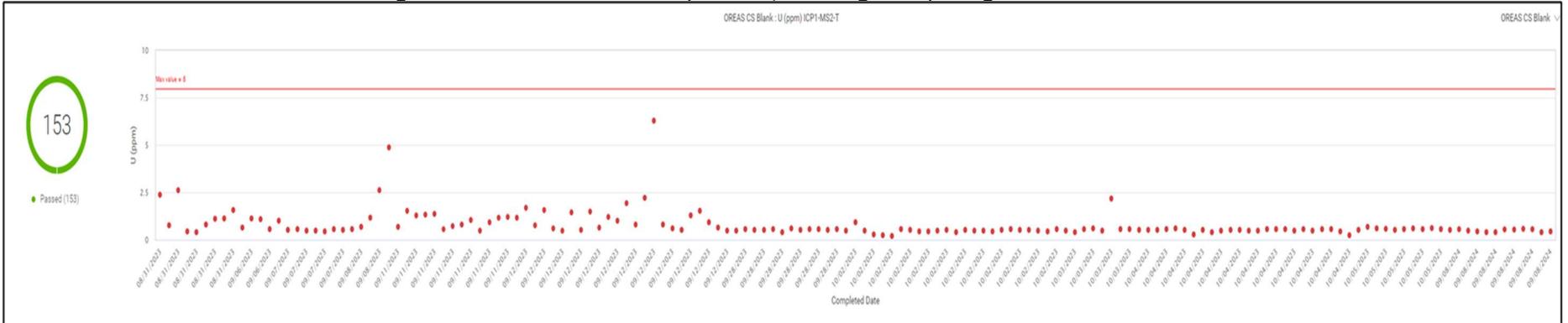


Figure 11-7: Lab Repeats from 2023 Show an Acceptable 10% Tolerance for ICP1 OES Total Digestion

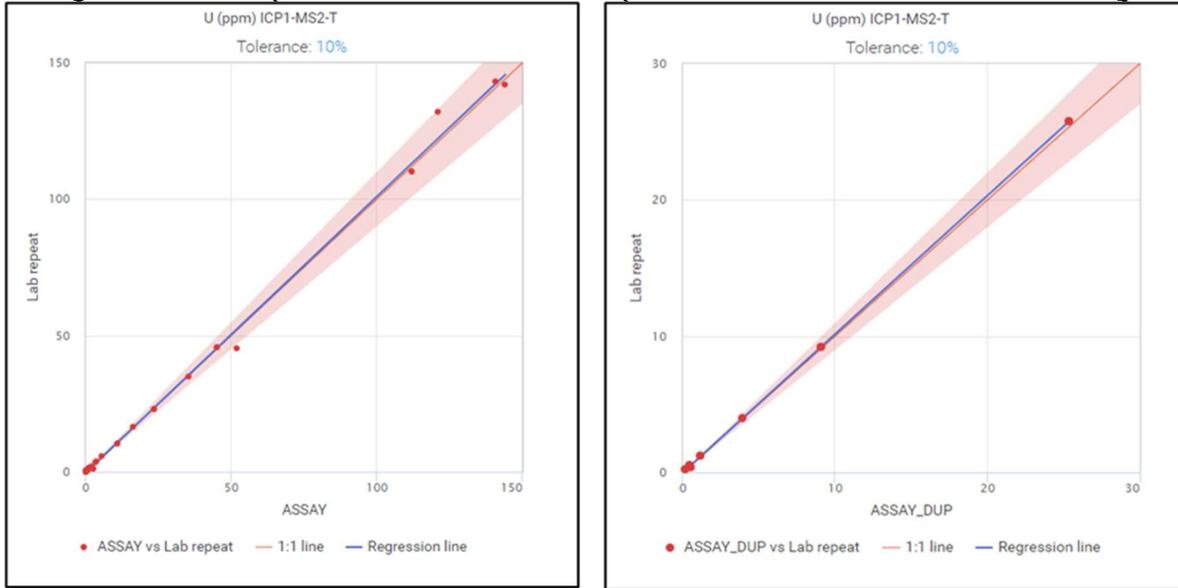


Figure 11-8: Lab Repeats from 2024 Fall Within an Acceptable 10% Tolerance for ICP1 OES Total Digestion

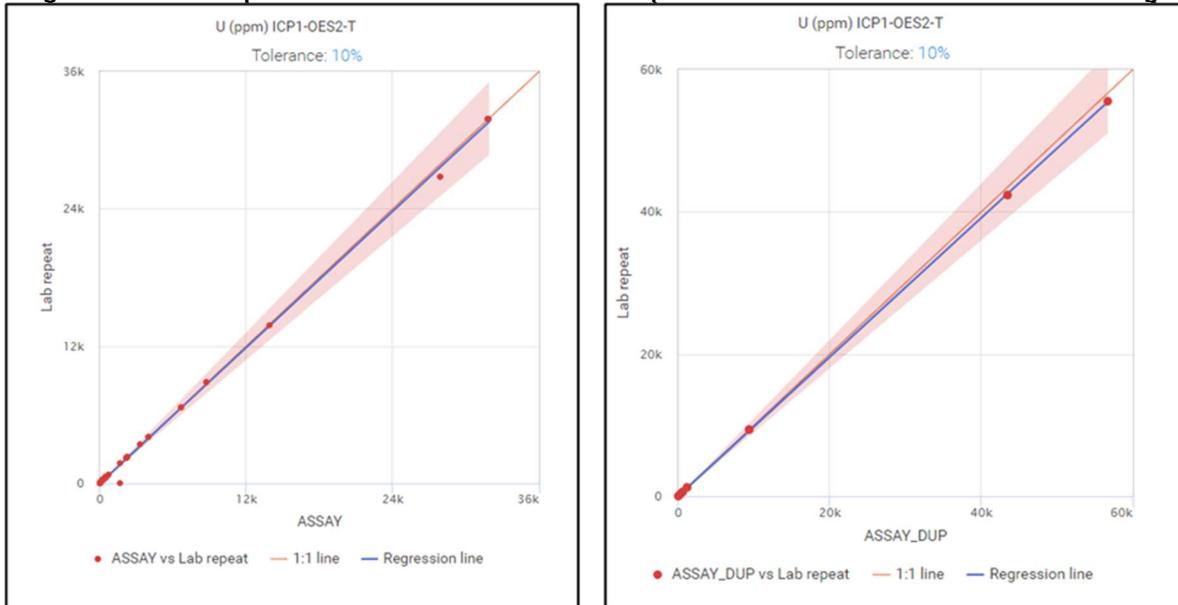
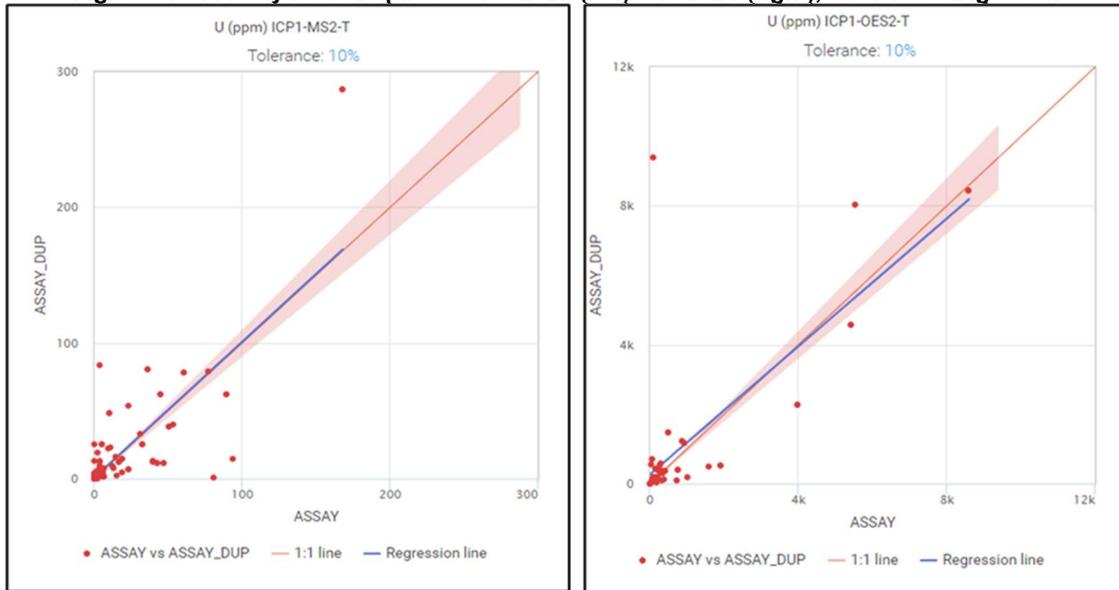


Figure 11-9: Assay Field Duplicates for 2023 (left) and 2024 (right), ICP1 Total Digestion



11.5 QP Comment on Section 11

The QP has reviewed the 2009 to 2024 data and is of the opinion that the procedures and systems employed to collect and manage this information meet or exceed industry standard practice, and that the QA/QC results demonstrate acceptable levels of accuracy and precision at the laboratories.

No sample assay or geochemical results were available for the 2025 drilling campaign as of the effective date of this technical report.

12 DATA VERIFICATION

12.1 Site Visit

A site visit to the Angilak Property was carried out August 14-15, 2024, by UMR's QP, Matt Batty, MSc, P. Geo. The two-day site visit included:

- Review of drill core from an ongoing drillhole,
- Review of mineralized drill core from eleven historic and recent drillholes,
- Confirmation of three drillhole collar locations,
- Review and verification of the geological setting / environment of the Project,
- Review of drilling, logging, sampling, analytical and QA/QC procedures, and
- Review of overall site facilities.

Mr. Batty reviewed (1) the entirety of available core from ML-DD-010 (0 to 329.7 m), which was being drilled at the time of the visit, and (2) mineralized intervals from drillholes 11-LC-97, 23-LC-004, MZ-DD-175, EEX-DD-053, 11-LC-083, 12-J4-030, 12-774-011, 23-LC-005, J4R-DD085, and J4R-DD-086 (Figure 12-1). The selected drillholes provided examples of low- and high-grade uranium mineralization, an overall sense of the Property's geology, spatial representation, and different drill programs. A comparison of the drill logs and assay results with the drill core showed that the information recorded in the drill database matched well with the drill core. As part of the review, the QP verified the occurrences of mineralization visually and by way of a hand-held scintillometer (Figure 12-2).

Figure 12-1: Angilak Core Review



Figure 12-2: Confirmation of Mineralization via a RS 121 Scintillometer



The locations of three drillhole collars were confirmed visually and with a handheld Garmin GPS, inclusive to MZ-DD-176, J4R-DD-087, and DDH 774-003. The database records were within 3 metres of the less accurate handheld measurements; and therefore, were deemed acceptable. The collar locations for the holes were demarked with tree branches or timbers inserted into the ground near the drill collar (Figure 12-3).

Figure 12-3: Drill Collar MZ-DD-176



12.2 Database Validation

ATHA verified the available exploration data for the Angilak Project, including soil and rock geochemical data along with airborne, ground magnetics, VLF-EM and radiometric geophysical data and all drilling data including work conducted by ValOre (formerly Kivalliq) from 2008 to 2022, LUR (formerly Labrador) in 2023, and ATHA in 2024 and 2025.

The soil and rock sampling data collected by ValOre were provided in Excel spreadsheets and ESRI shapefile formats. Data was imported into ArcGIS software to check for any obvious geospatial errors. All sample sites appeared to be correctly located. The soil and rock datasets were compared against copies of the laboratory certificates and found to be free of errors.

Airborne and ground geophysical data from work conducted between 2008 and 2016 were provided as either Geosoft Montaj™ databases or as ASCII line data. All data were reviewed for completeness. The airborne and ground geophysical images from the various surveys completed over the years were all brought into ArcGIS software for review and verification. The 2022 ground magnetics and VLF-EM geophysical data were provided as line data and were processed by APEX personnel and brought into ArcGIS software for review and verification. Similarly, the 2023 airborne radiometric survey data was imported into ArcGIS and plotted by means of the Geosoft extension for review and verification. The QA/QC procedures applied during the processing were deemed sufficient to provide quality data.

Drilling data collected by ValOre was originally compiled in digital format as a Microsoft Access database in 2017. This database contained a combination of historical data compilations from Kivalliq and ValOre, as well as original assay certificate data and geological logs from the 2009 to 2015 drilling programs. The drillhole database included collar coordinates, downhole survey information, geological interval data, and assay information. A total of 471 drillholes for 78,806 metres of diamond drilling were identified in the database. All of the 2022 drilling data collected by ValOre at the end of the 2022 season was captured in raw Excel and pdf formats.

Data acquired in 2023 by LUR was provided in Excel spreadsheet format, Access Databases, and ESRI shapefiles. Data was imported into MXDeposit™ and ArcGIS software was used to check for geospatial errors. Some drillholes with erroneous elevation values were corrected using an accurate topographic analogy and this correction was recorded in MXDeposit™.

ATHA personnel designed and oversaw the import of previous geological data into MXDeposit™. This included database constraints to ensure proper data entry, identification and correction of errors in data from previous drilling campaigns, and developing workflows to ensure both field, and laboratory control samples were properly verified for importing geochemical certificates. All drilling and sampling data collected by ATHA during the 2024 and 2025 drilling campaigns were logged directly into the MXDeposit™ database.

ATHA personnel completed an internal audit of the Angilak Project drillhole database as part of the 2024 exploration program. All data collected by LUR in 2023 was checked and validated against pdf hard copy assay certificates and geological logs. Data collected by ValOre from 2009 and 2022 was verified by comparing 10% of the database entries to original hardcopy drill logs, assay certificates and collar coordinate survey information. Minor typos and column mismatches were found and rectified, but overall, the data integrity met or exceeded industry standards. However, the reverse circulation drilling results were deemed to be imprecise relative to the validated core drilling results, and possibly inaccurate; thus, the reverse circulation drilling was not considered in ATHA's evaluations. The QP agrees with this conclusion and did not use the reverse circulation information in the exploration target model.

12.2.1 Additional Database Validation by the QP

Mr. Batty further validated the diamond drilling database via the following digital queries:

- Header table: searched for incorrect or duplicate collar coordinates and duplicate hole IDs.
- Survey table: searched for duplicate entries, survey points past the specified maximum depth in the collar table, and abnormal dips and azimuths.
- Lithology, alteration, and structure tables: searched for duplicate entries, intervals past the specified maximum depth in the collar table, overlapping intervals, negative lengths, missing collar data, missing intervals, and incorrect logging codes.

- Geochemical, density, and assay tables: searched for duplicate entries, sample intervals past the specified maximum depth, negative lengths, overlapping intervals, sampling lengths exceeding tolerance levels, missing collar data, missing intervals, and duplicated sample IDs.

No significant issues were identified.

12.2.2 Validation Limitations and Adequacy of the Data

The QP reviewed the adequacy of the exploration information and the visual, physical, and geological characteristics of the mineralization of the Property and found no significant issues or inconsistencies that would cause one to question the validity of the data provided by ATHA.

In the QP's opinion, the Angilak Project exploration data are free of any material or systematic errors, well validated and of sufficient quality for use in this Technical Report.

Based upon the QP's evaluation of the drilling, sampling and QA/QC programs completed by previous operators and ATHA, which meet or exceed industry standards, and based on the QP's own validation of the data, it is Mr. Batty's opinion that the Angilak drill and assay data are appropriate for use as presented in this Technical Report.

13 MINERAL PROCESSING AND METALLURGICAL TESTING

13.1 SGS Mineralogy Analysis

In February 2013, SGS provided ValOre a mineralogical characterization of 14 samples (Grammatikopoulos and Morton, 2013). Ten samples were collected from radioactive mineralized intersections representative of mineralization of the Lac 50 Deposit, in addition to four samples from the Blaze Zone (Table 13-1). The purpose of the investigation was to determine the overall mineral assemblage with an emphasis on the characterization of uranium minerals and their associated minerals. The mineralogical investigation included analyses with QEMSCAN™ technology (Quantitative Evaluation of Materials by Scanning Electron Microscopy), Scanning Electron Microscope equipped with an Energy Dispersive Spectrometer (SEM-EDS), optical microscopy, X-ray Diffraction (XRD) and Electron Microprobe Analysis (EMPA).

The mineralogical investigation revealed that the samples consist mainly of carbonates (calcite, ankerite and dolomite), feldspars (plagioclase and K-feldspars), quartz, chlorite, hematite, mica, apatite, zircon, barite and kaolinite (Table 13-2). Sulphides included pyrite, chalcopyrite, galena, molybdenite, bornite and covellite; although sulphides show an erratic distribution, it was shown that carbonate rich rocks have very low sulphide content (Grammatikopoulos and Morton, 2013). The overall mineral abundances determined from the mineralogical work are provided in Table 13-3 below with a picture of their spatial distribution provided in the QEMSCAN™ as Figure 13-1.

Table 13-1: Samples Collected for Mineralogical Analysis Conducted at SGS

Sample #	Hole ID	From (m)	To (m)	Interval (m)	Sample Type	Description
90001	11-LC-036	185.5	185.6	0.1	Petrograph	Hematite altered U-carbonate veining within moderate to strongly altered fine grained basalt with trace sulphides.
90002	11-LC-075	103.3	103.36	0.06	Petrograph	Sheared, brecciated basalt; silica-carbonate-hematite alteration
90003	11-LC-102	92.64	92.7	0.06	Petrograph	Mafic tuff; chlorite-albite-quartz-epidote alteration; trace sulphides; hematite-altered U mineralization
90004	11-BZ-005	52.82	52.89	0.07	Petrograph	Fine grained pillowed, amygdaloidal basalt; moderate hematite-carbonate-graphite alteration; 3% fine grained pitchblende within veinlets
90005	11-BZ-010	49.8	49.88	0.08	Petrograph	Hematite-altered, oxidized, U-mineralized basalt; quartz-carbonate-graphite veining and brecciation
90006	11-BZ-019	99.65	99.7	0.05	Petrograph	Fine grained, moderately hematite altered basalt; quartz-carbonate stringers- minor U-minerals
90007	11-BZ-017	68.6	68.68	0.08	Petrograph	Hematite-altered basalt with sulphides-carbonate-quartz-hematite alteration
90008	11-LC-030	99.15	99.2	0.05	Petrograph	Quartz-carbonate-hematite altered basalt with quartz-carbonate-sulphide-U veining
90009	11-LC-043	112.9	112.97	0.07	Petrograph	Brecciated and sheared basalt; quartz-carbonate-hematite-sulphide alteration associated with U veining
90010	11-LC-056	100.6	100.66	0.06	Petrograph	Pitchblende bearing veinlet within weakly hematized, foliated fine grained basalt
90011	11-LC-083	127.11	127.18	0.07	Petrograph	Brecciated and sheared basalt; silica-hematite-sulphide alteration with fracture-controlled pitchblende stringers
90012	11-LC-066	92.06	92.12	0.06	Petrograph	Sheared and brecciated basalt/tuff; strong hematite-iron carbonate-chlorite alteration associated with U mineralization
90013	11-LC-094	191.13	191.2	0.07	Petrograph	Brecciated, foliated mafic tuff; quartz-carbonate-epidote-pyrite-graphite-albite alteration; U minerals
90014	11-LC-116	297.7	297.75	0.05	Petrograph	Shear zone; hematite-carbonate-sulphide alteration; 80% carbonate veining

Table 13-2: Summary of Modal Mineralogy

Sample ID	90001	90002	90003	90004	90005	90006	90007	90008	90009	90010	90011	90012	90013	90014
Sulphides	2.9	8.2	10.8	2.2	5	0.6	16	1.1	0.3	0.2	2.9	0.1	15.2	0.2
U-Minerals	58.1	0.6	25.9	0.2	8.4	0.4	21.1	18.2	8.9	12.4	2.9	0.1	8.8	0.6
Feldspars	19.7	42.9	6.8	10.7	38.4	24.6	2.1	3.3	0.4	1.1	17.3	27.4	5.4	0.3
Quartz	2.5	4.5	21.3	8.9	1.9	11.6	1.9	9.5	2.8	2.1	13.6	10.1	51.1	5.6
Micas/Clay	5.3	13.1	3.6	16.8	11.3	17.9	2.9	2.3	3.4	6.8	8.9	14.1	5.2	3.7
Chlorite	0.7	0.9	0.2	39.3	17.8	28.4	7.9	2.3	5.3	17.7	20.1	6.3	0	5.1
Carbonates	6.7	23.1	30.2	13.2	13.8	6.6	32	53.9	77.7	58.3	30.1	31.8	13.4	82.8
Fe-(Ti)-Oxides	0.5	3.1	0.3	2.7	1.7	3.5	13.2	8.2	0.3	0.2	0.2	2.7	0.1	0.2
Apatite	1.2	0.2	0	0.3	0.5	0.2	0.6	0	0.1	0.2	2.3	1.1	0	0
Other	2.4	3.5	0.8	5.6	1.4	6.3	2.4	1.3	0.8	1.1	1.7	6.2	0.7	1.4

Figure 13-1: QEMSCAN™ Pseudo Image of Sample 90001 Illustrates Structural Control of Uranium Mineralization among Silicates and Carbonates.

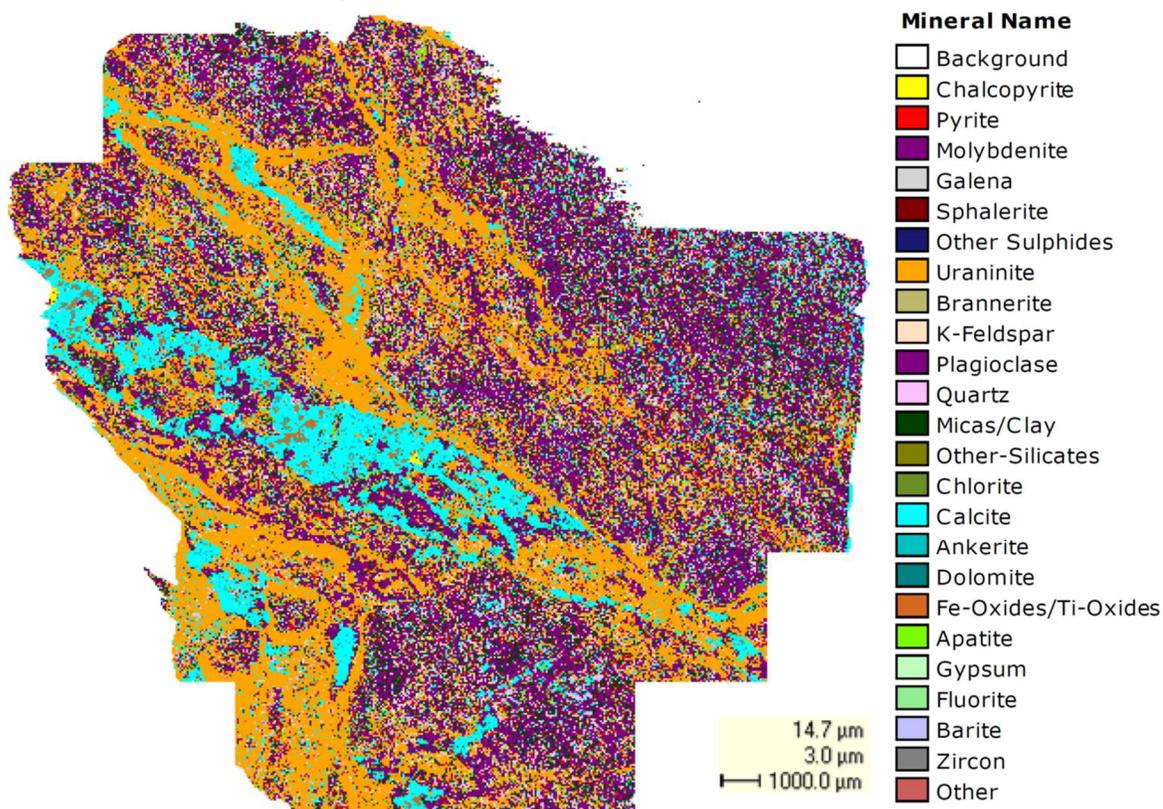


Table 13-3: Mineral Abundance (wt. %) for Each Sample

Sample		90001	90002	90003	90004	90005	90006	90007	90008	90009	90010	90011	90012	90013	90014
Fraction		-1000/+3um													
Mass Size Distribution (%)		100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0	100.0
Calculated ESD Particle Size		14642	9741	6196	4302	7228	13851	16002	15205	15746	11597	9359	15353	12855	10569
		Sample													
Mineral Mass (%)	Chalcopyrite	0.0	0.0	8.4	0.7	0.1	0.0	0.5	0.8	0.0	0.0	2.6	0.0	0.1	0.0
	Pyrite	2.2	7.9	1.5	1.2	2.4	0.5	8.3	0.0	0.0	0.0	0.1	0.1	13.0	0.0
	Molybdenite	0.2	0.0	0.2	0.0	2.4	0.0	7.0	0.0	0.0	0.0	0.0	0.0	1.2	0.0
	Galena Sphalerite	0.5	0.0	0.3	0.0	0.1	0.0	0.1	0.0	0.1	0.0	0.0	0.0	0.7	0.1
	Other Sulphides	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
	Uraninite	0.0	0.3	0.4	0.2	0.0	0.0	0.1	0.2	0.2	0.1	0.2	0.0	0.2	0.0
	Brannerite	57.8	0.6	25.8	0.2	7.1	0.3	20.9	18.1	8.7	12.2	2.2	0.1	8.6	0.6
	Coffinite	0.2	0.0	0.1	0.0	1.2	0.0	0.2	0.1	0.3	0.2	0.7	0.0	0.2	0.0
	K-Feldspar	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0	0.0
	Plagioclase	2.0	5.1	2.5	1.8	0.5	2.1	0.1	0.0	0.0	0.0	0.1	8.4	3.3	0.0
	Quartz	17.7	37.8	4.3	9.0	37.8	22.5	2.0	3.2	0.4	1.1	17.2	19.0	2.1	0.3
	Micas/Clay Other-Silicates Chlorite	2.5	4.5	21.3	8.9	1.9	11.6	1.9	9.5	2.8	2.1	13.6	10.1	51.1	5.6
	Calcite Ankerite	5.3	13.1	3.6	16.8	11.3	17.9	2.9	2.3	3.4	6.8	8.9	14.1	5.2	3.7
	Dolomite	1.0	3.4	0.4	5.2	1.1	6.1	0.8	1.0	0.7	0.9	1.1	5.8	0.5	1.1
	Fe-(Ti)-Oxides	0.7	0.9	0.2	39.3	17.8	28.4	7.9	2.3	5.3	17.7	20.1	6.3	0.0	5.1
	Apatite Gypsum	6.3	19.1	30.1	12.4	13.6	6.5	12.5	50.8	75.8	57.6	30.0	22.6	10.0	81.7
	Fluorite	0.4	2.1	0.0	0.3	0.2	0.1	5.1	3.1	1.8	0.7	0.0	1.0	0.9	1.1
	Barite Zircon	0.0	1.9	0.0	0.5	0.0	0.0	0.0	14.4	0.0	0.0	0.0	8.2	2.5	0.0
	Other														
Mean Grain Size by Frequency (µm)	Chalcopyrite	71	22	131	30	27	24	42	42	26	23	50	30	28	22
	Pyrite	25	55	44	39	26	44	30	22	22	22	22	33	117	23
	Molybdenite	24	22	32	23	38	24	51	22	23	22	23	24	29	25
	Galena Sphalerite	23	22	23	22	22	23	23	24	31	27	24	23	23	40
	Other Sulphides	23	33	22	37	0	22	22	0	0	0	0	22	28	23
	Uraninite	22	23	23	23	22	23	23	23	49	33	23	24	22	22
	Brannerite	96	24	60	22	37	29	59	52	58	58	27	25	35	39
	Coffinite	24	23	26	0	24	24	25	27	25	25	24	22	24	25
	K-Feldspar	0	0	0	0	0	0	0	0	0	0	0	0	0	0
	Plagioclase	26	27	28	32	25	33	34	24	22	22	24	29	29	22
	Quartz	57	70	31	42	65	48	35	40	29	46	65	35	29	40
	Micas/Clay Other-Silicates Chlorite	27	26	73	36	26	40	26	35	30	31	68	31	213	33
	Calcite Ankerite	27	30	33	32	27	33	27	27	30	29	29	29	34	33
	Dolomite	23	24	25	25	23	25	25	24	24	23	23	24	23	24
	Fe-Oxides/Ti-Oxides	24	24	24	71	37	50	64	38	53	78	64	30	23	68
	Apatite Gypsum	51	47	115	39	69	38	34	117	283	189	82	41	54	263
	Fluorite Barite	27	25	29	23	23	22	27	29	28	25	22	25	26	26
	Zircon Other	24	32	31	29	22	22	41	22	22	22	25	44	40	22

The detailed analyses determined that the most abundant uranium minerals in the Lac 50 Deposit are uraninite (commonly known as pitchblende) and coffinite, with trace amounts of brannerite and uranophane (Grammatikopoulos and Morton, 2013). Uranium mineralization is closely associated with mainly carbonates, chlorite and sulphides (particularly pyrite, chalcopyrite and galena).

The occurrence of uranium is complex and shows dissolution and re-crystallization textures. Uranium mineral grains exhibit rugged outlines, irregular grain boundaries and form fine grained outliers within the associated gangue minerals. Uranium minerals are generally fine grained but form coarse polycrystalline aggregates, layers or distinct domains. The mesoscopic appearance of the uranium minerals is characterized as patchy and disseminated. Microscopically, uranium minerals reveal net veining, discontinuous thin (micrometre in nature) layers that are clearly secondary in nature. Other textures include discontinuous rims and fine-grained inclusions in micro-fractures (Grammatikopoulos and Morton, 2013).

13.2 SRC Metallurgical Test Work

In June 2012, ValOre engaged the SRC to perform a second phase alkaline leaching program for the Lac 50 Deposit using sulphide flotation to optimize the alkaline leach (Zhang, 2013). The SRC program was intended to follow up on first phase metallurgical testing initiated in 2010 by SGS Mineral Services (SGS), a division of SGS Canada Inc. of Lakefield, Ontario. SGS was engaged to examine uranium recovery from a composite of laboratory pulp rejects from drillcore submitted to SRC for geochemical analysis during ValOre's 2009 drilling program (Brown and Todd, 2011). SGS examined a variety of leach conditions and sample grinds. Uranium extraction results were good, with up to 98% dissolution from acid leach tests and up to 94.7% dissolution from alkaline leach tests. Acid consumption, attributed to a high carbonate content in the Lac 50 composite, with rates up to 489 kg/t was considered high.

Alkaline leaching is typically preferred for high carbonate content uranium deposits. The 2012 SRC metallurgical testing program was designed to investigate uranium alkaline leaching optimization after the removal of sulphide minerals by flotation (Zhang, 2013). The testing was expanded in late 2012 to include a preliminary evaluation of the purity levels of the yellowcake product. A summary of the work conducted by the SRC is provided below and is taken from Zhang (2013).

There are two reasons to float the sulphide minerals. First, the sulphide minerals consume reagents during the alkaline uranium leaching. The removal of sulphides from the alkaline leach feed will reduce reagent consumption. In addition to uranium, the Lac 50 Deposit contains elevated contents of Ag, Mo, Cu, Zn and Pb. The majority of these metals occur as sulphide minerals, from which the metals are not extracted by either alkaline leaching or atmospheric acid leaching.

The objectives of the 2012 SRC tests were to maximize uranium extraction through optimizing the alkaline leaching process for flotation tailings; maximize the recovery of sulphides through flotation and compare yellowcake product purity levels to ASTM C967-13 uranium concentrate specifications.

13.2.1 Sample Receiving and Preparation

The SRC mineral processing group received from SRC Geoanalytical Labs, 166 crushed quarter split and half split pulp reject samples weighing approximately 60 kg. The samples were derived from core submitted to SRC from 51 drillholes for geochemical analysis. The holes were part of ValOre's 2010 and 2011 diamond drilling programs on the Lac 50 Main Zone, Western Extension

and Eastern Extension uranium deposits. A master composite sample was made by aggregating, blending and homogenizing the crushed drill core sample pulp rejects. The composite sample was split into two individual samples of approximately 30 kg each. The first of these was ground to 100% passing 200 mesh (74 µm) using a ball mill. A head grade sample was taken from the resulting composite and analyzed by SRC's ICP 1 total digestion method. It contained 0.737% U, 0.217% Mo, 0.667% Cu, 0.221% Zn, 0.231% Pb and 26.7 g/tonne Ag. The SRC assay certificate is included as Table 13-4 below (SRC Report No: G-12-2325).

Table 13-4: SRC Assay Certificate for Report No. G-12-2325.

SRC Innovation Place		SRC Geoanalytical Laboratories				Report No: G-12-2325	
Attention: Jack Zhang		125 - 15 Innovation Blvd., Saskatoon, Saskatchewan, S7N 2X8					
PO #/Project: 13427		Tel: (306) 933-8118 Fax: (306) 933-5656 Email: geolab@src.sk.ca					
Samples: 3		ICP1 Total Digestion				Date of Report: December 05, 2012	
Column Header Details							
Silver in ppm (Ag)							
Copper in ppm (Cu)							
Iron in wt % (Fe2O3)							
Molybdenum in ppm (Mo)							
Lead in ppm (Pb)							
Uranium in ppm (U, ICP)							
Zinc in ppm (Zn)							
Sample Number	Ag ppm	Cu ppm	Fe2O3 wt %	Mo ppm	Pb ppm	U, ICP ppm	Zn ppm
CAR110	3.6	239	4.46	67	452	3450	126
KI215	26.7	6670	12.1	2170	2310	7370	2210
KI215 R	26.0	6690	12.2	2130	2280	7280	2250
Total Digestion: A 0.125 g pulp is gently heated in a mixture of HF/HNO3/HClO4 until dry and the residue is dissolved in dilute HNO3. The standard is CAR110.							

13.2.2 Mineralogical Analysis

A quantitative mineralogical microprobe scan was performed on a sample of the homogenized composite ground to 100% passing 20 µm to get good liberation of the sulphide minerals. As shown in Figure 13-2, the results of the scan indicate that the composite sample is dominated by carbonate minerals, primarily calcite and dolomite, with subordinate quartz and other gangue silicates. Pyrite is the dominant sulphide mineral present, but chalcopyrite is also observed in the samples. Three uranium-bearing minerals are present in the sample: uraninite, coffinite and trace amounts of uranophane.

Sulphide flotation is performed to remove the sulphide minerals which consume sodium carbonate and oxygen in an alkaline uranium (U) leach circuit. Test charges were ground to 100% passing 200 mesh (74 µm). Several different xanthate collectors and hydroxamate acid were tested. Flotation tests were performed at the same flotation conditions except that one stage cleaner flotation was conducted when the hydroxamate acid was used as collector. A schematic flotation process is shown in Figure 13-3.

Figure 13-2: Quantitative Mineral Abundances

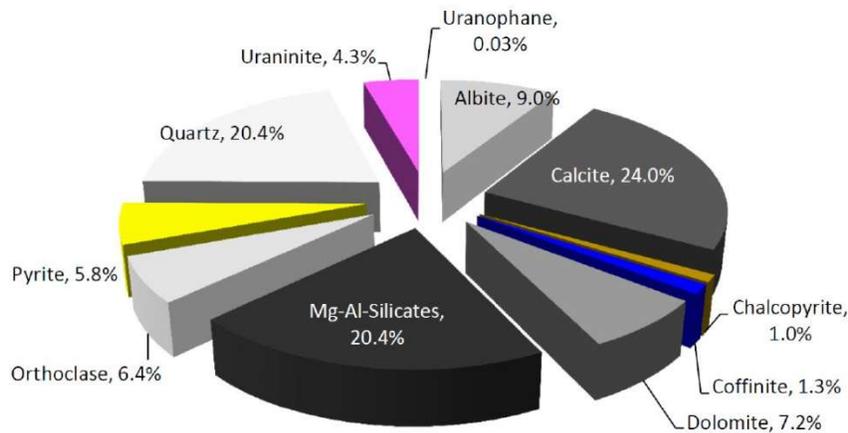
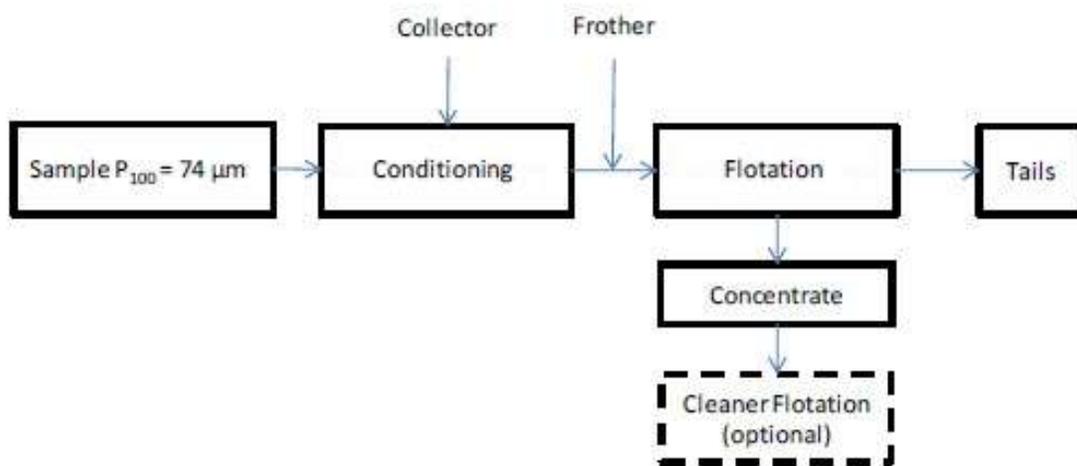


Figure 13-3: Schematic Flotation Process



The target of the flotation optimization is to maximize sulphide recovery to the float concentrate. Greater than 95% of the uranium can be recovered through alkaline leaching of flotation tails. A flotation test using a mixed collector made from KAX 51 and a butyldithiophosphate at the ratio of 2/1 at a pH of 10.5 yielded good flotation results. The flotation conditions are summarized in Table 13-5. The collector conditioning time, collector dosage, flotation temperature, feed size and pH were investigated.

Table 13-5: Flotation Conditions

Test	Conditions							
	Mixed Collector		MIBC		Feed Size (mesh)	pH	Temp. (°C)	Flot. Time (mins)
	Dosage (kg/tonne)	Cond. Time (mins)	Dosage (kg/tonne)	Cond. Time (mins)				
1	0.03	5	0.17	0.5	-200	10.5	65	5

The flotation results are shown in Table 13-6. The results indicate that the mixed collector was able to recover 70.4% of Cu, 50.2% of Ag, 86.1% of Zn, 37.6% of Pb, and 80.5% of total S and 94.6% sulphide. The consumption of collector was low at 0.03 kg/tonne. Frother (MIBC) consumption was 0.17 kg/tonne. The sulphide flotation results remain subject to further improvement by optimization.

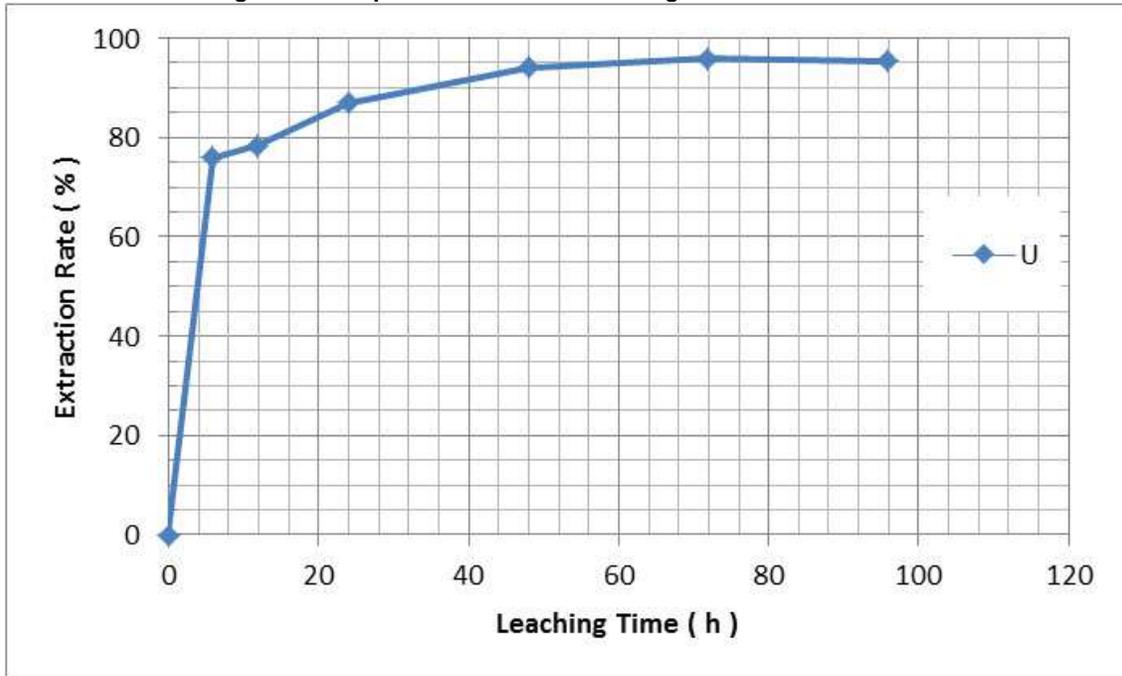
Table 13-6: Flotation Results Using a Mixed Collector at pH of 10.5

	Feed		Concentrate	Tails	Recovery (%)
	Direct Assay	Calculated Assay			
Mass, g	200	197.7	15.6	182.1	7.8
Ag, ppm	27.2	25.15	160	13.6	50.2
Cu, ppm	6520	6196.56	55300	1990	70.4
Mo, ppm	2320	1611.77	9290	954	45.5
Pb, ppm	2360	2348.30	11200	1590	37.6
U, ppm	7140	7253.07	9390	7070	10.2
Zn, ppm	2260	2199.58	24000	332	86.1
C, %	3.99	4.05	3.13	4.13	6.1
S, %	2.93	2.50	25.5	0.53	80.5
Sulfide, %	1.81	1.87	22.4	0.11	94.6

13.2.3 Alkaline Leaching

Due to the high carbonate content of the composite feed, alkaline leaching is considered to represent a viable extraction process for the Lac 50 Deposit uranium mineralization. Alkaline leaching optimization tests have been highly encouraging. Optimized results, as shown on Figure 13-4 indicate that at 70°C, atmospheric pressure, 50% pulp density, sufficient oxidation and a reagent addition rate of 70 kg/t (50 kg Na₂CO₃ and 20 Kg NaHCO₃), 94.1% of the uranium was extracted in 48 hours and 95.9% of the uranium was extracted in 72 hours from the composite sample. An advantage of alkaline leaching for the Lac 50 Deposit mineralization is low reagent consumption. At this stage of bench testing, consumption rates have not yet been accurately determined. A second advantage of alkaline leaching is that the process is very selective resulting in a pregnant leaching solution that is clean with low impurity levels.

Figure 13-4: Optimized Alkaline Leaching Kinetics for Uranium



The high selectivity of alkaline leaching has at least three benefits: 1) simple subsequent processes to produce yellowcake; 2) unlike the raffinate handling from acid leaching circuits, no complicated effluent treatment processes are needed; 3) simplified tailings handling with the ability to utilize tailings for backfill during mining.

13.2.4 Comparative Whole Ore and Float Tails

As a first step toward optimization, a series of alkaline leaching tests were performed using whole ore and flotation tails at various temperatures. Tests demonstrate that 50-60% of the uranium from whole ore samples can be extracted in the first 6 hours. After 6 hours, the leaching rate slows but uranium extraction continues to increase with leaching time. As shown on Figure 13-5 for the whole ore sample, the highest final uranium extraction (94.9%) was achieved at 70°C and the lowest final uranium extraction (75.0%) was at 90°C. Alkaline leaching was conducted using solution containing 50 g/L Na_2CO_3 and 20 g/l NaHCO_3 .

Figure 13-6 shows the leaching of the flotation tails sample. In the flotation tails sample, the sulphide minerals are partially removed. The leaching of the flotation tails sample showed the same pattern as the whole ore sample. Over 50% of the uranium was extracted in the first 6 hours. After 6 hours the leaching rate slows but uranium extraction continues to increase with leaching time. In comparison to the whole ore leaching, higher final extraction rates are generally achieved with the flotation tails. The uranium extraction was 83.4% at 60°C, 94.4% at 70°C, 91.0% at 80°C, and 80.6% at 90°C, respectively.

Figure 13-5: Whole Ore Uranium Alkaline Leach at Variable Temperatures

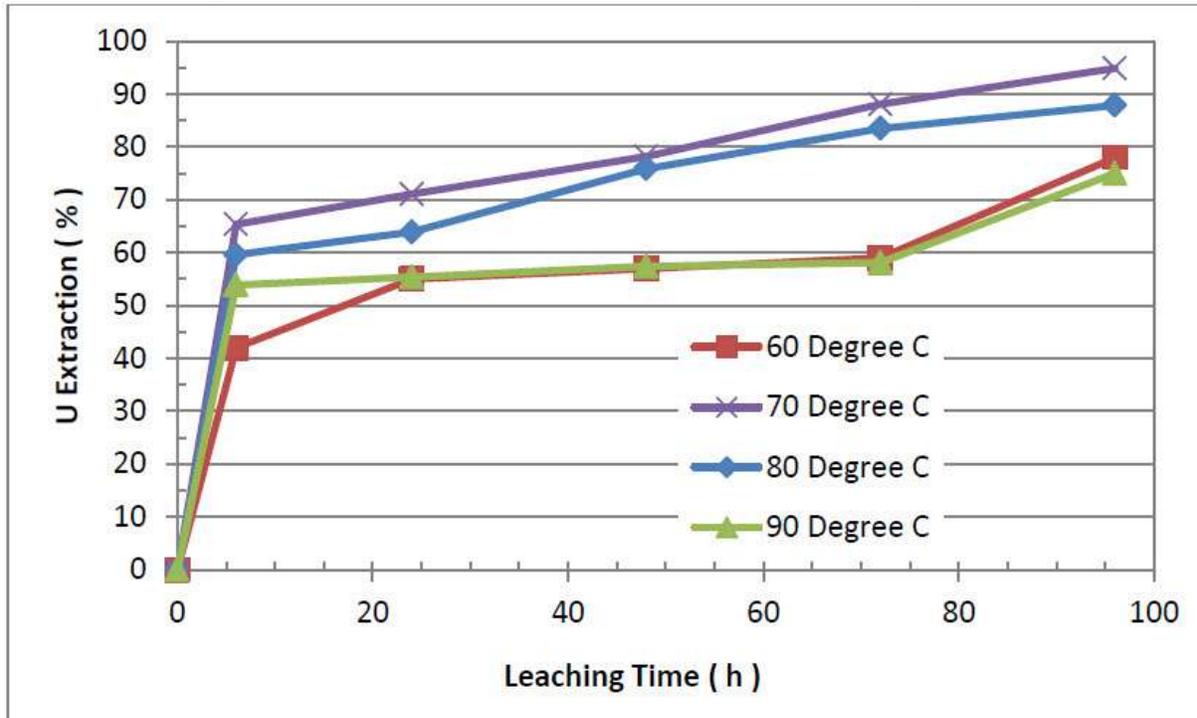
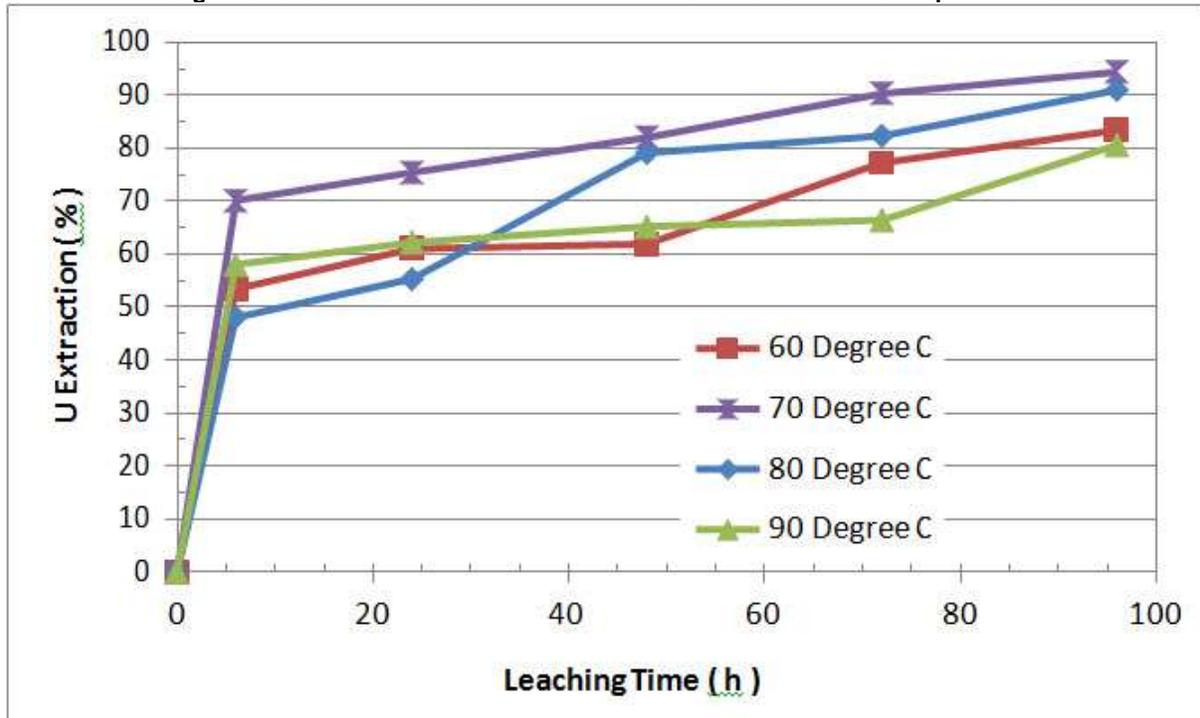


Figure 13-6: Flotation Tails Uranium Alkaline Leach at Variable Temperatures



The leaching results of both the whole ore sample and flotation tails sample showed a leaching temperature of 70°C gave optimum uranium extraction rates of approximately 95%. In an alkaline leach operation, alkaline leach solution is recycled for re-use. If too much sulphide is present in

the feed material, reagent consumption is excessive and therefore an initial sulphide flotation is recommended.

13.2.5 Effects of Oxidation

Hydrogen peroxide was used as the oxidant in alkaline leach tests. With alkaline leaching optimization tests (the temperature variation tests) hydrogen peroxide was added from the second hour of leaching. In a plant operation, pressurized oxygen will be supplied continuously during the leaching process. To assess hydrogen peroxide utilization more fully, batch addition of hydrogen peroxide was compared to continuous addition. Significant improvement of leaching kinetics was achieved by adding hydrogen peroxide slowly but continuously. Figure 13-7 shows the comparison of leaching kinetics at 70°C using batch and continuous addition of hydrogen peroxide. When the hydrogen peroxide was added continuously, leaching completion was almost reached in 48 hours. Only slight improvement was observed when the leaching time increased from 48 hours to 72 hours and 96 hours. The continuous addition of hydrogen peroxide, or continuous oxidation, more accurately simulates the oxidation of field operations. Oxidation will play a critical role in optimizing leaching kinetics. The reduction of leaching time from 96 hours to 48 hours has the potential to reduce operating costs significantly.

13.2.6 Effects of Feed Size

The sulphide flotation tails using different feed grind sizes were alkaline leached as well to investigate the effects of grind size on leaching kinetics and uranium extraction. Figure 13-8 shows the leaching kinetics of uranium utilizing different size fractions. Oxidant, hydrogen peroxide, was added continuously in all of the tests. It is interesting to see that very similar leaching kinetics and uranium extraction were achieved with the various size feeds. The -200 mesh feed and the -400 mesh had almost identical leaching kinetics and final uranium extraction. However, the -635 mesh feed had slightly slower leaching kinetics and final uranium extraction. This indicates that feed with size smaller than -200 mesh has very little effect on the leaching kinetics.

Figure 13-7: Leaching Kinetics with Different Oxidation

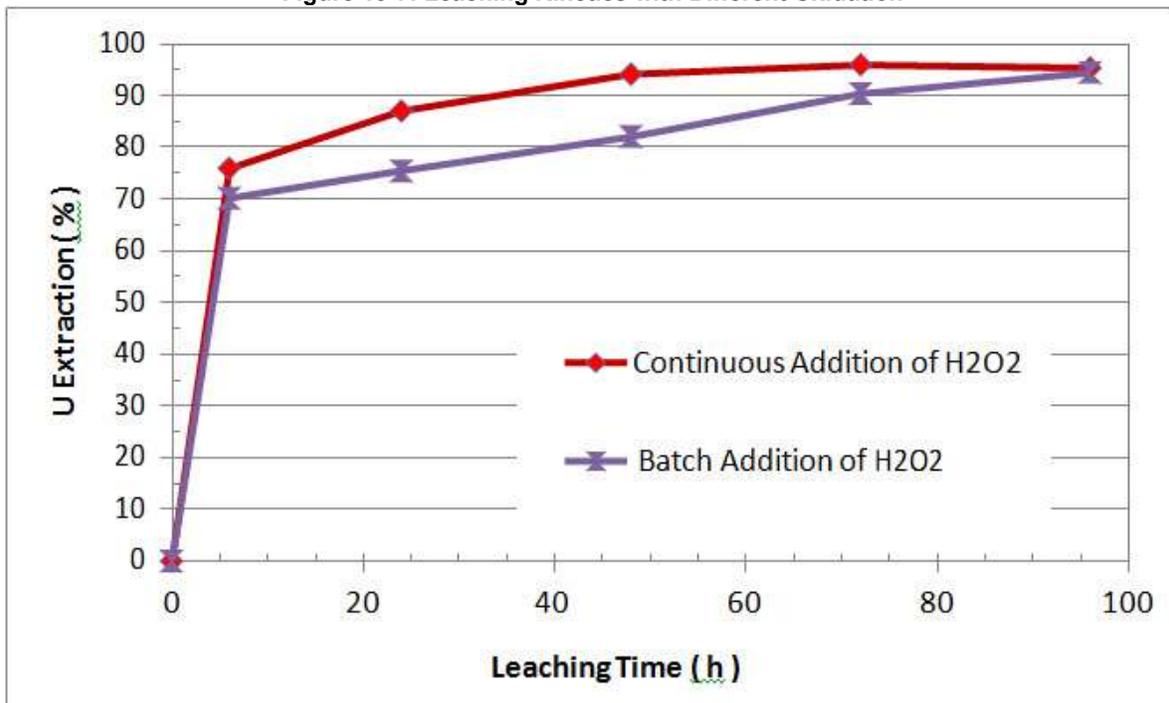
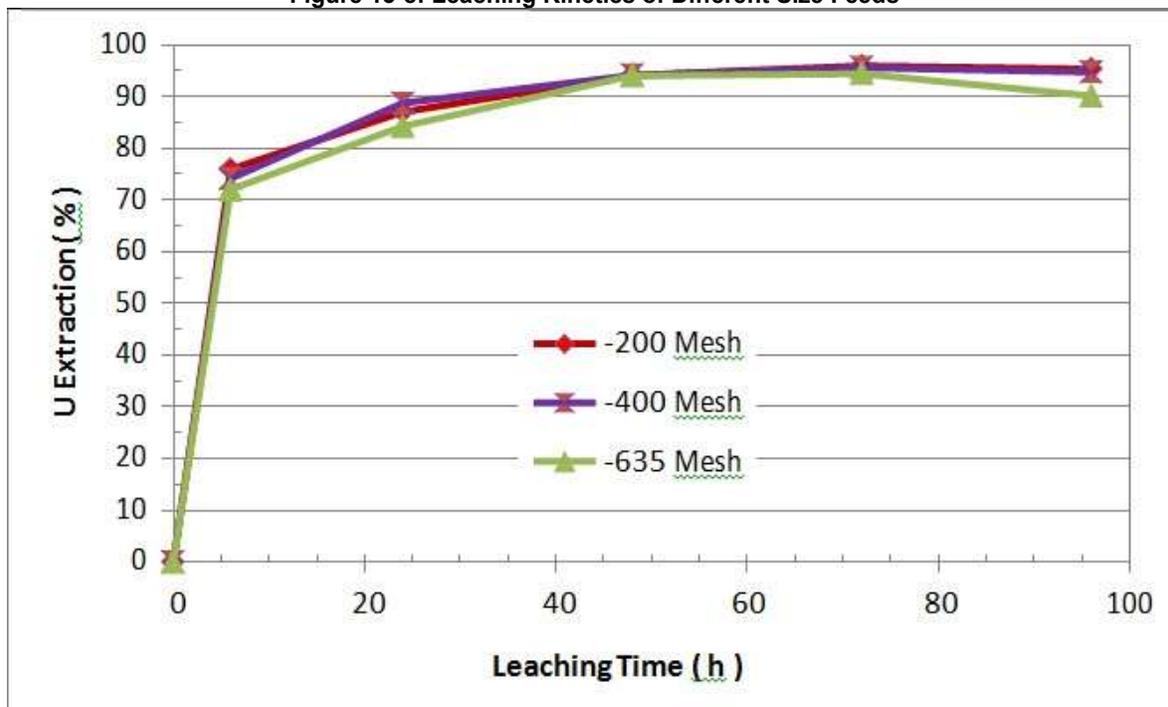


Figure 13-8: Leaching Kinetics of Different Size Feeds



13.2.7 Yellowcake Production Test

With the encouraging results from the alkaline leaching tests, a decision was made to investigate the purity of a yellowcake product from the Lac 50 Deposit composite. A preliminary yellowcake precipitation was performed. Direct sodium hydroxide precipitation was performed first to produce sodium diuranate ($\text{Na}_2\text{U}_2\text{O}_7$). The sodium hydroxide precipitation was conducted at 70°C for 6 hours. Over 99% of uranium in the pregnant solution was precipitated as sodium diuranate. The sodium diuranate was then purified through acidification and hydrogen peroxide (H_2O_2) precipitation. The uranium value attained was 71.9% for a final yellowcake product.

Both the sodium diuranate and final yellowcake samples were analysed for several impurities and uranium, the results for which are shown compared with Impurity Maximum Concentration Limits from ASTM C967-123 specifications in Table 13-7. Assayed impurities fell below the Maximum Concentration Limit Without Penalty standard specifications for uranium ore concentrate. Low impurity levels achieved in preliminary yellowcake tests are very encouraging at this early stage of testing.

Table 13-7: Impurity of the Preliminary Angilak Yellowcake Product

Specifications	ASTM C967-13 (Mass%, Uranium Basis)		ValOre (Mass%, Uranium Basis)
	Limit without Penalty	Limit without Rejection	YC Product
Uranium (U)	N/A	65% min.	71.9%
Arsenic (As)	0.05%	0.1%	0.0009%
Barium (Ba)	N/A	N/A	0.0001%
Boron (B)	0.005%	0.1%	N/A
Cadmium (Cd)	N/A	N/A	0.00006%

Specifications	ASTM C967-13 (Mass%, Uranium Basis)		ValOre (Mass%, Uranium Basis)
Calcium (Ca)	0.05%	1%	0.02%
Carbonate (CO ₃)	0.2%	0.5%	0.069%
Chromium (Cr)	N/A	N/A	0.018%
Fluoride (F)	0.01%	0.1%	N/A
Halides (Br, Cl, I)	0.05%	0.1%	N/A
Iron (Fe)	0.15%	1%	<0.01%
Lead (Pb)	N/A	N/A	0.007%
Magnesium (Mg)	0.02%	0.5%	N/A
Mercury (Hg)	N/A	N/A	N/A
Moisture (H ₂ O)	2%	5%	N/A
Molybdenum (Mo)	0.1%	0.3%	0.0004%
Phosphorus (PO ₄)	0.1%	0.7%	0.03%
Potassium (K)	0.2%	3%	<0.002%
Selenium (Se)	N/A	N/A	<0.0001
Silica (SiO ₂)	0.5%	2.5%	N/A
Silver (Ag)	N/A	N/A	0.0003%
Sodium (Na)	1%	7.5%	<0.01%
Sulfur (S)	1%	4%	0.125%
Thorium	0.1%	2.5%	0.00006%
Titanium	0.01%	0.05%	<0.002%
²³⁴ U	56 µg/gU	62 µg/gU	N/A
Vanadium (V)	0.06	0.3%	<0.0001%
Zirconium (Zr)	0.01%	0.1%	N/A

13.2.8 SRC Recommendations

Based upon the results of the SRC's metallurgical test work and specifically the alkaline leaching program for the Lac 50 Deposit, the SRC provided a number of recommendations for further studies going forward to assist with future process engineering and economic studies:

- Continue sulphide flotation tests to maximize sulphide recovery to flotation concentrate,
- Continue sulphide flotation concentrate acid leaching tests to maximize uranium dissolution,
- Additional alkaline leach tests to maximize uranium recovery,
- Initiate yellowcake precipitation tests using dilute sodium hydroxide solution for pH control to minimize reagent cost,
- Initiate testing of a composite from the Lac 50 J4 deposit, discovered in 2012,
- Continue processing tests of the leached sulphide flotation concentrate to produce a potentially marketable by-product, and
- Initiate a bench-scale pilot plant test of the optimized unit operations to optimize the integrated process.

14 MINERAL RESOURCE ESTIMATE

There is no current Mineral Resource estimate for the Angilak Project.

15 MINERAL RESERVE ESTIMATE

There is no current Mineral Reserve estimate for the Angilak Project.

16 MINING METHODS

This section is not applicable.

17 RECOVERY METHODS

This section is not applicable.

18 PROJECT INFRASTRUCTURE

This section is not applicable.

19 MARKET STUDIES AND CONTRACTS

This section is not applicable.

20 ENVIRONMENTAL STUDIES, PERMITTING AND SOCIAL OR COMMUNITY IMPACT

This section is not applicable.

21 CAPITAL AND OPERATING COSTS

This section is not applicable.

22 ECONOMIC ANALYSIS

This section is not applicable.

23 ADJACENT PROPERTIES

The tenure ownership of the area surrounding the Angilak project is presented in Figure 23-1. The QP has not verified the information on the adjacent properties presented in this section and notes that the information is not necessarily indicative of the mineralization on the Angiluk Property.

The Kiggavik Project uranium project, jointly owned by Orano Canada Inc. (66.2%), Denison Mines (16.9%) and Uranium Energy Corp. (16.9%), is located in the Kivalliq region of Nunavut, approximately 210 kilometres northeast of the Angilak Property and 90 kilometres west of Baker Lake. The Kiggavik Project is operated by Orano and has a reported historical Mineral Resource as presented in Table 23-1 (Denison Mines, 2023; Orano 2022). Cut-off grades and other assumptions, parameters and methods used to estimate the mineral resources are unknown. The historical mineral resources for Kiggavik are from Denison's website.

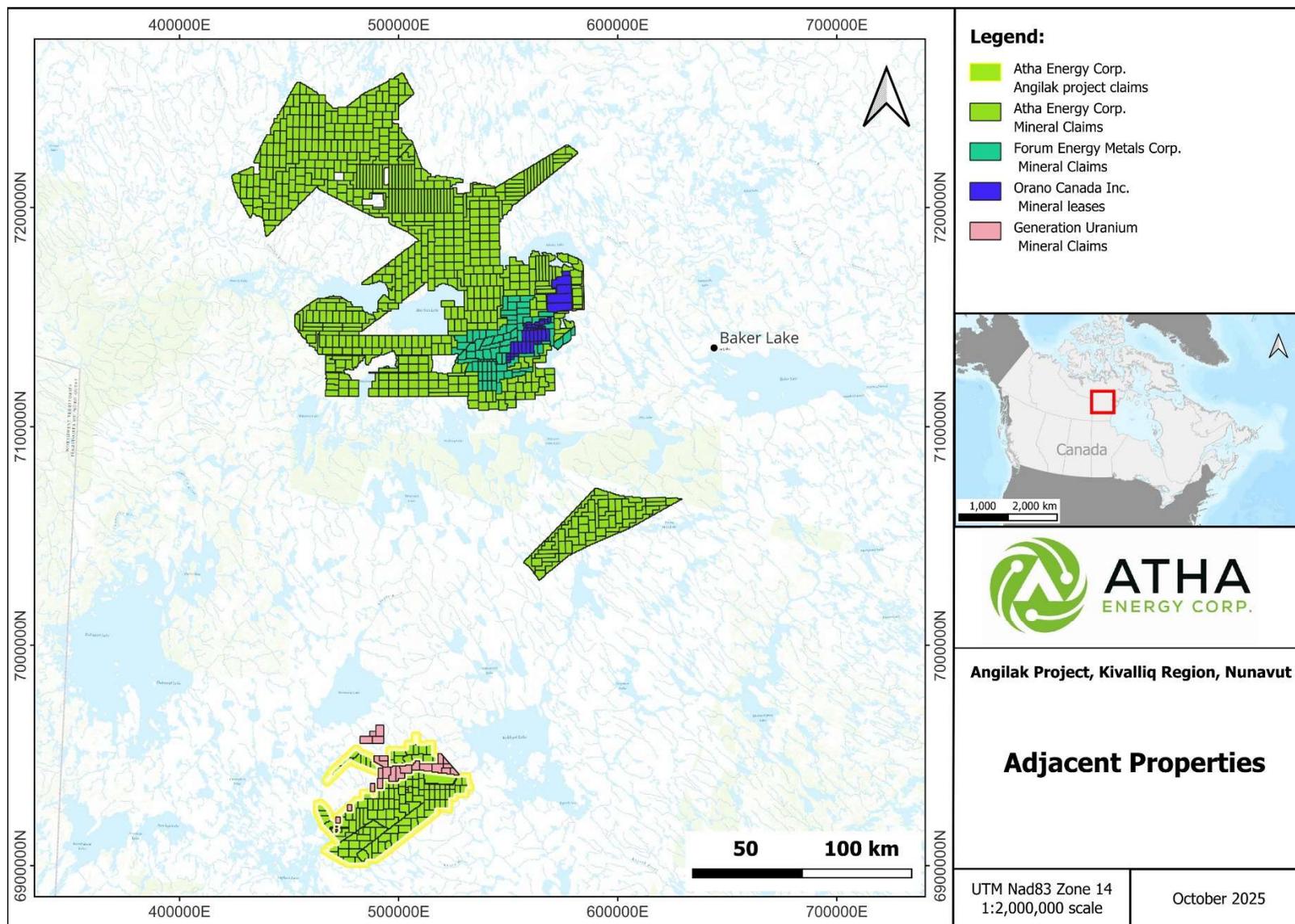
Table 23-1: Kiggavik Historical Mineral Resource (Denison, 2023)

Kiggavik	Indicated			Inferred		
	Tonnes	Grade U ₃ O ₈ (%)	Lbs U ₃ O ₈ (,000)	Tonnes	Grade U ₃ O ₈ (%)	Lbs U ₃ O ₈ (,000)
	10,418,000	0.47	127,300	731,000	0.28	5,400

Areva Canada Inc. (now Orano Canada Inc.) completed an initial feasibility study and submitted a Draft Environmental Assessment Study to the Nunavut Impact Review Board in 2007 (Areva, 2008). Following public hearings in March 2015, the Nunavut Impact Review Board (NIRB) recommended Kiggavik not be approved at that time. NIRB stated it does not intend for the project not to proceed at any time, but that it should be resubmitted when a project start date and development schedule can be provided. The federal government supported NIRB's decision (NIRB website).

In 2022, Forum Energy Metals Corp. (Forum) expanded their land position around the Orano leases to encompass 95,518 ha of prospective land (Forum's website). Forum's Nunavut Uranium Project (located approximately 195 kilometres north of the Angilak Project) covers two high-grade unconformity style uranium deposits – Tatiggaq and Qavvik, and the Ayra uranium showing (Forum's website).

Figure 23-1: Tenure Ownership of Area Surrounding the Angilak Property



24 OTHER RELEVANT DATA AND INFORMATION

No additional information or explanation is necessary to make this Technical Report understandable and not misleading.

25 INTERPRETATION AND CONCLUSIONS

The Angilak Project is located 350 kilometres west of Kangiqliniq (Rankin Inlet) and 225 kilometres southwest of Baker Lake in the Kivalliq Region of Nunavut. The Angilak Property hosts the Lac 50 Deposit and is 192,913 hectares in size.

The Angilak Project is located within the Western Churchill Province, a large Archean craton that experienced significant crustal shortening and uplift during the Proterozoic, where the subsequent gravitational collapse led to the deposition of several rift basins, including the Baker Lake Basin.

Two major structural corridors surround the Property: the Snowbird Tectonic Zone to the northwest, and the Tyrrell Shear Zone to the southeast. These corridors formed because of the assembly of the Churchill Province and were later reactivated by tectonic activity in the Proterozoic. The Archean basement rocks underlying the Property consist of tonalite-granodiorite gneisses and granitoids, as well as the metasedimentary and metavolcanic greenstone belt rocks of the Henik Group. These are unconformably overlain by the Angikuni and Yathkyed sub-basins (Baker Lake Group). The Baker Lake Basin and the associated Angikuni and Yathkyed sub-basins were formed because of these tectonic processes. The contact between these Proterozoic basins and the Archean represents an unconformity that has been targeted globally for uranium, a deposit type termed “unconformity style uranium”. The most prolific occurrences of this deposit type are found in the Athabasca basin in northern Saskatchewan.

Although historical exploration in the Yathkyed Lake area targeted unconformity style uranium, a vein-type hydrothermal uranium deposit, the Lac 50 Deposit, was found on IOL Parcel RI30-001. The Lac 50 Deposit lies within the Property and is located adjacent to the northeastern margin of the Angikuni sub-basin. It is hosted in Archean metasedimentary and metavolcanic rocks of the Henik Group. Mineralization at the Lac 50 Deposit is structurally and stratigraphically controlled and bears similarities to Beaverlodge-type vein or structural uranium deposits.

25.1 Previous Exploration

Previous exploration by a variety of companies during the late 1970's and early 1980's in the Yathkyed Lake region resulted in the discovery of numerous uranium ± base metals ± silver showings and the Lac 50 Deposit, a Beaverlodge style, vein-type uranium deposit. Most of the showings occur close to the western, northern and northeastern boundary of the Angikuni sedimentary sub-basin, within both Archean basement and later basin-fill sedimentary and volcanoclastic material and were the product of exploration for unconformity style uranium mineralization as the main target.

The exploration season of 2008 marked the first work program in over 25 years at the Angilak Property. The 2008 exploration program completed by ValOre (formerly known as Kivalliq Energy) included 5,620 line-kilometres of airborne TDEM, magnetics, radiometrics and Property wide prospecting and mapping.

In 2009, ValOre completed ground VLF-EM survey over IOL RI30-001 and identified a 9 km-long conductive trend hosting the historical Lac 50 Deposit. This was followed up with an initial 1,745 metres drill program at the Lac 50 Main Zone that successfully intersected U₃O₈ mineralization in 13 of 14 drillholes.

ValOre drilled over 16,600 metres at the Lac 50 Main Zone and surrounding geophysical targets in 2010. In 2011, 30,500 metres were drilled, 5,470 line-kilometres of EM-magnetics were flown, and ground geophysical surveys were completed. New zones of uranium mineralization were discovered and drilled, including: Western Extension, Eastern Extension, Blaze, Pulse and Spark.

The largest exploration program in ValOre's history (\$20M) was conducted in 2012, with a focus on mineralized footprint expansion and new discoveries. In total, 38,856 metres were drilled in conjunction with extensive ground geophysical surveys. New zones of uranium mineralization were discovered which included: J4, Ray, Hot, Flare, Southwest and Nine Iron. ValOre also expanded the Angilak land position by 32,375 hectares.

Exploration in 2013 consisted of 2,100 metres of drilling and ground geophysical surveying. New mineralized zones discovered included J1 and Mushroom Lake.

In 2014, 963 soil samples and 1,078 line-kilometres of airborne TDEM and magnetics geophysical surveys were completed. In 2015, 958 metres were drilled at the Dipole target, resulting in the first significant uranium discovery outside of the Lac 50 Deposit area. Additional soil results confirmed kilometre-scale uranium anomalies along the Dipole and RIB geophysical trends.

Soil sampling in 2016 expanded the area of uranium anomalism, extending the Dipole uranium signature to over 3.5 kilometres. Trenching at the Yat target confirmed the presence of a high-grade polymetallic zone in bedrock and uranium-in-soil anomaly along a 1.6 kilometre-long EM conductor.

In the spring of 2022, ValOre conducted ground magnetics and VLF-EM surveys covering 1,547.62 line-kilometres with 80,329 VLF-EM measurements collected over 3 priority grids in the Lac 50 East area, an area straddling the RIB and Dipole targets and further southwestward to the Property boundary. A soil sampling program was conducted in the summer of 2022, where 880 soil samples were collected and submitted for Enzyme Leach analysis.

An RC drill program was conducted during spring 2022 with 3,165.35 metres drilled in 27 holes on the Dipole (17 holes), Yat (4 holes) and J4 West (6 holes) targets. The RC drilling was used to follow up on core drilling results at Dipole from 2015, historical drilling at Yat and core and RC drilling at J4 West from 2013. A diamond drilling program was conducted during summer 2022 with 3,590 metres drilled in 26 holes at the Dipole (16 holes) and J4 West (10 holes) targets. Diamond drilling at the Dipole target tested the extension potential northeast along strike of the drilling completed in 2015, as well as following up on the diamond drilling in 2015 and RC drilling in 2022, to test mineralization extension with depth. Diamond drilling at the J4 West tested the potential for a sinistral off-set and continuation of mineralization to the southwest of the J4 deposit.

In 2023, LUR completed a low-level, high resolution radiometric and aeromagnetic airborne survey totaling 10,856 line-kilometres over areas previously covered by VLF-EM surveys. This was followed by an 18-hole drill campaign totalling 5,662 meters of drilling in the Lac 50 Deposit area, primarily focused on the Main Zone. The program successfully increased the extent of known mineralization and identified new mineralized horizons within the hanging wall of the Main Zone

25.2 Exploration Conducted by ATHA

The maiden 2024 Angilak Exploration Program concluded after twenty-five diamond drillholes were completed between early June and late August for a total of ~10,051 metres. The program was highly successful and on budget. All objectives were successfully achieved with the expansion of the historic footprint of mineralization along the Lac 50 Trend and the identification of new parallel mineralized trends called the Lac 48, Lac 52, and Lac 54 Trends. The Lac 48, Lac 50 (host to the Lac 50 Deposit), Lac 52, and Lac 54 Trends remain open in all directions with high prospectivity for further discovery and expansion of uranium mineralization. Additionally, untested areas between the newly identified trends are also prospective for discovery of new mineralized trends.

Within the Lac 50 Trend twelve holes were drilled for a total of 4,884 metres, targeting expansion of uranium mineralization beyond the historic mineralized extents. All holes achieved the objective of intersecting uranium mineralization outside of the historic mineralized domains and expanding the footprint of mineralization of the known zones. The footprint of mineralization extends along the Lac 50 Trend over a strike length of ~3.9 kilometres and remains open along strike and at depth.

Thirteen additional holes for a total of 5,167 metres were completed at prospective targets within the Lac 48, Lac 52 and Lac 54 Trends. All holes discovered new lenses of uranium mineralization, expanded on previously discovered showings, or identified prospective structures. Mineralization within the Lac 48, Lac 52 and Lac 54 Trends remain open along strike and at depth.

In 2025, ATHA completed its second exploration campaign on the Angilak Project which was focused on continued expansion of the Lac 50 deposit mineralization footprint at the J4/Ray Zone, testing the down-dip continuity of mineralization identified on surface at the Mushroom Lake zone, and testing of high-priority regional targets within the KU and RIB areas located within the prospective RIB-Nine Iron corridor. As part of the 2025 campaign, a total of twenty-two diamond drill holes for a total 10,774-metres of drilling was completed. To date, final drill core sample assays have not yet been received and all results are based on preliminary down-hole radiometric probe results obtained from a Mount Sopris 40TGU-1000 Triple Gamma Geiger probe.

In the Lac 50 deposit area a total of three drill holes and 1,608 metres were completed at the J4/Ray and Mushroom Lake zones. At the J4/Ray zone, drilling successfully extended mineralization down-dip by approximately 100 metres and remains open both along strike and at depth. At Mushroom Lake, two drill holes were completed to test the depth extent of uranium mineralization identified in outcrop. Both drill holes successfully intersected uranium mineralization as identified by preliminary down-hole radiometric probe results and remains open at depth and along strike.

At the KU regional target area, a total of six drill holes and 3,427 metres of drilling were designed to test stacked gravity and EM geophysical anomalies coincident with interpreted NW-SE and E-W trending structures within the 31-km RIB-Nine Iron corridor. All drill holes successfully intersected mineralized structures with elevated radioactivity and highlighted by drill hole KU-DD-001.

Within the RIB regional target area, a total of thirteen drill holes and 5,739 metres were completed to test stacked gravity and EM anomalies coincident with interpreted NE-SW, E-W and N-S trending structures along the regional RIB-Nine-Iron corridor. All drill holes successfully intersected mineralized structures associated with graphitic shears zones over an interpreted 12-kilometre conductive EM trend within the RIB area, with the most significant intersection coming from the RIB North area in drill hole RIBN-DD-001.

25.3 Lac 50 Deposit Exploration Target Model

Mr Batty, the QP for this technical report provided ATHA with ranges for potential uranium quantity and grade as a target for further exploration on Angilak's Lac 50 Deposit (Table 25-1). The ranges were derived using interpreted vein wireframes, drill core assays, grade interpolation, and an applied uncertainty range.

Table 25-1: Lac 50 Tabulated Exploration Target Model Ranges

Lac 50 Exploration Target			
Cutoff (% U₃O₈)	Tonnes (Mt)	Grade (% U₃O₈)	Metal Content (Mlbs U₃O₈)
0.1	7.4 - 9.3	0.37 - 0.48	60.8 - 98.2

Notes:

1. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.
2. The ranges were derived from interpreted vein wireframes, drill core assays, grade interpolation/extrapolation, and applied uncertainty range.
3. An assumed cut-off of 0.1% U₃O₈ was used for the tabulation of the exploration target model.

The wireframes were modelled using a grade intercept limit equal to or greater than a minimum grade of 0.01 % U₃O₈, although lower grades were incorporated in places to maintain continuity and represent the structural setting and continuity of the mineralized system. Extension distance for the mineralized wireframes was halfway to the next hole, or 200 metres in areas of no drilling, representing the potential at the deposit.

Uranium grade from drill core assays, after standardization and outlier management, was interpolated/extrapolated in the wireframes.

The QP applied an uncertainty range to define a range for potential uranium using the interpolated uranium grade as the midpoint. The stated potential quantity and grade is conceptual in nature, and there has not been sufficient exploration to define a mineral resource, and it is uncertain if further exploration will result in the target being delineated as a mineral resource.

25.4 Metallurgical Work to Date

In June 2012, the SRC commenced a metallurgical testing program that built on first pass work completed in 2010. The initial 2010 results indicated alkaline leaching as the most effective extraction process for the Lac 50 Deposit. The objective of the 2012 program was to investigate uranium alkaline leaching optimization and perform a preliminary evaluation of the purity levels of a final yellowcake product. The SRC aggregated a master composite sample weighing approximately 60 kilograms by blending and homogenizing 166 quarter-split and half-split pulp reject samples from 51 core holes. The sampled 2010 and 2011 core holes represent 3.2 km of strike length of uranium mineralization along the Lac 50 Main Zone, Western Extension and Eastern Extension. A head grade sample from the 2012 composite assayed 0.737 % U, 0.217% Mo, 0.667% Cu, 0.221% Zn, 0.231% Pb and 26.7 g/t Ag. Optimized results from alkaline leaching indicate that 94.1% of uranium can be extracted in 48 hours and 95.9% of the uranium extracted in 72 hours with a final yellowcake product that contained 71.9% uranium. It is encouraging at this early stage that the assayed impurities in the yellowcake product are below the maximum allowable concentration limits without penalty for uranium ore concentrate specifications. Additional metallurgical work is warranted.

26 RECOMMENDATIONS

Based on the historical exploration work discussed in this Technical Report and the exploration program completed by ATHA, it is the opinion of the Author of this Technical Report that the Angilak Property warrants further exploration work.

Based upon the results of exploration conducted to date, the Author recommends that the following work be completed at the Angilak Property:

- 1) Additional drilling along the Lac 48, 50, 52 and 54 Trends to further expand mineralization immediately along strike, and at depth, and along parallel and cross-cutting mineralized structural corridors identified by previous drilling;
- 2) Additional drilling within the RIB target area to continue to test the prospective 12-kilometre structural corridor following on drill results from the 2025 exploration campaign,
- 3) Exploration drilling including:
 - testing of geophysical conductors proximal to the Lac 50 Deposit, including conductors along strike that could represent extensions and parallel trends prospective to host uranium mineralization;
 - further drill testing at the KU, Nine Iron, and Dipole showings; and
 - reconnaissance drilling of additional exploration targets within the Angikuni Basin;
- 4) Further airborne geophysical surveys to help characterize, de-risk and prioritize regional targets across the Property;
- 5) Baseline environmental monitoring in support of future project evaluation studies; and
- 6) Ongoing community consultation.

A preliminary cost estimate for the recommended work to be carried out in 2026 is provided in Table 26-1.

Item	Cost Estimate (CDN\$M)
Geophysical Surveys	\$2.0
Drilling (10,000m) and Logistical Support	\$9.0
Baseline Environmental Monitoring	\$0.5
Community Consultation	\$0.1
TOTAL	\$11.6

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28 DATE AND SIGNATURE PAGE

This report titled "Technical Report on the Angilak Property, Nunavut, Canada" with an effective date of October 14, 2025, was prepared and signed by the following Author:

Dated in Saskatoon, SK
October 14, 2025

Matthew Batty, MSc, P.Geo.
Geostatistician and Owner of UMR
(Signed and Sealed) "M.D. Batty"

29 CERTIFICATE OF QUALIFIED PERSON

I, Matt Batty, MSc, P.Geo., as the Author of this report entitled “Technical Report on the Angilak Property, Nunavut, Canada”, prepared for ATHA Energy Corp. with an effective date of October 14, 2025 and dated October 14, 2025:

1. I am a Geologist with and owner of Understood Mineral Resources Ltd. of 22 Middleton Crescent, Saskatoon, Canada.
2. I am a graduate of the University of Saskatchewan in 2012 with a B.Sc. degree in Geology and a graduate of the University of Alberta in 2022 with a M.Sc. degree in Mining Engineering (Geostatistics).
3. I am a Registered Professional Geologist (Member No. 25595) with the Association of Professional Engineers and Geoscientists of Saskatchewan (APEGS). I have worked as a geologist for a total of 12 years since my graduation. My relevant experience for the purpose of the Technical Report is:
 - Mineral Resource estimation and preparation of NI 43-101 Technical Reports.
 - Resource and Geology Lead, with NexGen Energy Ltd., responsible for resource evaluation and reporting for uranium projects in Canada.
 - Mine Geologist with Cameco Corporation at the Eagle Point Mine.
4. I have read the definition of "qualified person" set out in National Instrument 43-101 (NI 43-101) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I am a "qualified person" for the purposes of NI 43-101.
5. I visited the Angilak Property from August 14 to 15, 2024.
6. I am responsible for the entirety of the Angilak Technical Report.
7. I am independent of ATHA Energy Corp.
8. I have had no prior involvement with the property that is the subject of the Technical Report.
9. I have read NI 43-101, and the Technical Report has been prepared in compliance with NI 43-101 and Form 43-101F1.
10. At the effective date of the Technical Report, to the best of my knowledge, information, and belief, the Technical Report contains all scientific and technical information that is required to be disclosed to make the Technical Report not misleading.

Dated this 14th day of October 2025

(Signed & Sealed) “M.D. Batty”

Matt Batty

**APPENDIX A:
HISTORICAL DIAMOND DRILLHOLE PARAMETERS**

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2009	09-775-001	NAD83 / UTM zone 14N	522980.0	6939281.0	213.26	30.0	-45.0	194.00	30-Aug-09	02-Sep-09
Diamond	2009	09-LC-001	NAD83 / UTM zone 14N	519310.0	6940075.0	210.49	26.0	-55.0	125.00	09-Aug-09	10-Aug-09
Diamond	2009	09-LC-002	NAD83 / UTM zone 14N	519309.0	6940074.0	210.61	26.0	-67.0	143.00	11-Aug-09	12-Aug-09
Diamond	2009	09-LC-003	NAD83 / UTM zone 14N	518910.0	6940277.0	211.25	26.0	-45.0	62.00	13-Aug-09	14-Aug-09
Diamond	2009	09-LC-004	NAD83 / UTM zone 14N	518910.0	6940277.0	211.26	26.0	-63.0	74.00	13-Aug-09	14-Aug-09
Diamond	2009	09-LC-005	NAD83 / UTM zone 14N	518910.0	6940276.0	211.07	26.0	-80.0	92.00	14-Aug-09	15-Aug-09
Diamond	2009	09-LC-006	NAD83 / UTM zone 14N	518740.0	6940274.0	209.61	26.0	-65.0	149.00	15-Aug-09	17-Aug-09
Diamond	2009	09-LC-007	NAD83 / UTM zone 14N	518741.0	6940274.0	209.56	26.0	-55.0	149.00	17-Aug-09	19-Aug-09
Diamond	2009	09-LC-008	NAD83 / UTM zone 14N	518917.0	6940178.0	210.74	26.0	-53.0	152.00	19-Aug-09	21-Aug-09
Diamond	2009	09-LC-009	NAD83 / UTM zone 14N	518917.0	6940177.0	210.85	26.0	-62.0	170.00	21-Aug-09	23-Aug-09
Diamond	2009	09-LC-010	NAD83 / UTM zone 14N	518869.0	6940306.0	211.37	26.0	-70.0	71.00	23-Aug-09	24-Aug-09
Diamond	2009	09-LC-011	NAD83 / UTM zone 14N	518870.0	6940307.0	211.61	26.0	-45.0	23.00	24-Aug-09	25-Aug-09
Diamond	2009	09-LC-012	NAD83 / UTM zone 14N	518572.0	6940384.0	219.63	26.0	-70.0	110.00	25-Aug-09	26-Aug-09
Diamond	2009	09-LC-013	NAD83 / UTM zone 14N	518572.0	6940383.0	219.66	26.0	-50.0	92.00	27-Aug-09	28-Aug-09
Diamond	2009	09-LC-014	NAD83 / UTM zone 14N	519426.0	6940088.0	209.08	26.0	-60.0	80.00	27-Aug-09	28-Aug-09
Diamond	2009	09-LC-015	NAD83 / UTM zone 14N	519427.0	6940088.0	209.25	26.0	-45.0	62.00	29-Aug-09	30-Aug-09
Diamond	2010	10-LC-001	NAD83 / UTM zone 14N	519509.0	6940027.0	209.03	26.0	-45.0	104.00	24-Apr-10	26-Apr-10
Diamond	2010	10-LC-002	NAD83 / UTM zone 14N	519289.0	6940087.0	211.87	26.0	-71.0	134.00	27-Apr-10	29-Apr-10
Diamond	2010	10-LC-003	NAD83 / UTM zone 14N	519289.0	6940087.0	211.85	26.0	-78.0	164.00	29-Apr-10	01-May-10
Diamond	2010	10-LC-004	NAD83 / UTM zone 14N	519256.0	6940022.0	207.56	26.0	-60.0	197.00	01-May-10	04-May-10
Diamond	2010	10-LC-005	NAD83 / UTM zone 14N	519255.0	6940022.0	207.58	9.0	-61.0	194.00	05-May-10	07-May-10
Diamond	2010	10-LC-006	NAD83 / UTM zone 14N	519232.0	6940029.0	207.61	26.0	-68.0	218.00	08-May-10	10-May-10
Diamond	2010	10-LC-007	NAD83 / UTM zone 14N	519232.0	6940029.0	207.63	26.0	-54.0	185.00	10-May-10	12-May-10
Diamond	2010	10-LC-008	NAD83 / UTM zone 14N	519174.0	6940024.0	206.24	26.0	-60.0	215.00	13-May-10	15-May-10
Diamond	2010	10-LC-009	NAD83 / UTM zone 14N	519174.0	6940024.0	206.30	26.0	-72.0	254.00	15-May-11	17-May-10
Diamond	2010	10-LC-010	NAD83 / UTM zone 14N	518490.0	6940332.0	225.03	26.0	-63.0	215.00	18-May-10	20-May-10
Diamond	2010	10-LC-011	NAD83 / UTM zone 14N	518490.0	6940332.0	224.95	26.0	-55.0	188.00	20-May-10	21-May-10
Diamond	2010	10-LC-012	NAD83 / UTM zone 14N	518490.0	6940333.0	224.91	26.0	-49.0	167.00	22-May-10	23-May-10
Diamond	2010	10-LC-013	NAD83 / UTM zone 14N	517890.0	6940581.0	237.55	26.0	-69.0	143.00	24-May-10	26-May-10
Diamond	2010	10-LC-014	NAD83 / UTM zone 14N	517889.0	6940581.0	237.66	26.0	-79.0	137.00	22-Jun-10	24-Jun-10
Diamond	2010	10-LC-015	NAD83 / UTM zone 14N	519109.0	6940123.0	211.00	26.0	-45.0	107.00	25-Jun-10	26-Jun-10

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2010	10-LC-016	NAD83 / UTM zone 14N	519109.0	6940123.0	211.00	26.0	-71.0	149.00	27-Jun-10	29-Jun-10
Diamond	2010	10-LC-017	NAD83 / UTM zone 14N	518627.0	6940384.0	215.29	26.0	-45.0	74.00	29-Jun-10	02-Jul-10
Diamond	2010	10-LC-018	NAD83 / UTM zone 14N	518626.0	6940383.0	215.43	26.0	-75.0	92.00	02-Jul-10	04-Jul-10
Diamond	2010	10-LC-019	NAD83 / UTM zone 14N	518626.0	6940383.0	215.37	26.0	-82.0	149.00	04-Jul-10	05-Jul-10
Diamond	2010	10-LC-020	NAD83 / UTM zone 14N	518818.0	6940204.0	209.47	26.0	-45.0	164.00	05-Jul-10	06-Jul-10
Diamond	2010	10-LC-021	NAD83 / UTM zone 14N	518817.0	6940203.0	209.32	26.0	-65.0	197.00	07-Jul-10	08-Jul-10
Diamond	2010	10-LC-022	NAD83 / UTM zone 14N	518817.0	6940203.0	209.21	26.0	-76.0	235.00	08-Jul-10	10-Jul-10
Diamond	2010	10-LC-023	NAD83 / UTM zone 14N	518624.0	6940267.0	217.09	26.0	-45.0	173.00	11-Jul-10	12-Jul-10
Diamond	2010	10-LC-024	NAD83 / UTM zone 14N	518624.0	6940266.0	217.18	26.0	-58.0	203.00	12-Jul-10	14-Jul-10
Diamond	2010	10-LC-025	NAD83 / UTM zone 14N	518624.0	6940266.0	217.21	26.0	-73.0	284.00	15-Jul-10	17-Jul-10
Diamond	2010	10-LC-026	NAD83 / UTM zone 14N	518464.0	6940396.0	225.93	26.0	-45.0	53.00	18-Jul-10	18-Jul-10
Diamond	2010	10-LC-027	NAD83 / UTM zone 14N	518464.0	6940395.0	226.42	26.0	-65.0	128.00	18-Jul-10	19-Jul-10
Diamond	2010	10-LC-028	NAD83 / UTM zone 14N	518464.0	6940394.0	226.46	26.0	-73.0	167.00	19-Jul-10	20-Jul-10
Diamond	2010	10-LC-029	NAD83 / UTM zone 14N	518465.0	6940396.0	225.83	36.0	-45.0	161.00	20-Jul-10	22-Jul-10
Diamond	2010	10-LC-030	NAD83 / UTM zone 14N	519075.0	6940164.0	208.26	26.0	-45.0	95.00	22-Jul-10	24-Jul-10
Diamond	2010	10-LC-031	NAD83 / UTM zone 14N	519074.0	6940163.0	208.20	26.0	-72.0	116.00	24-Jul-10	26-Jul-10
Diamond	2010	10-LC-032	NAD83 / UTM zone 14N	519074.0	6940163.0	208.10	26.0	-82.0	152.00	26-Jul-10	27-Jul-10
Diamond	2010	10-LC-033	NAD83 / UTM zone 14N	519074.0	6940163.0	208.13	26.0	-87.0	194.00	27-Jul-10	30-Jul-10
Diamond	2010	10-LC-034	NAD83 / UTM zone 14N	519161.0	6940112.0	208.35	26.0	-45.0	98.00	30-Jul-10	01-Aug-10
Diamond	2010	10-LC-035	NAD83 / UTM zone 14N	519160.0	6940111.0	208.29	26.0	-69.0	128.00	01-Aug-10	03-Aug-10
Diamond	2010	10-LC-036	NAD83 / UTM zone 14N	519160.0	6940110.0	208.30	26.0	-81.0	155.00	03-Aug-10	07-Aug-10
Diamond	2010	10-LC-037	NAD83 / UTM zone 14N	519215.0	6940110.0	209.80	26.0	-45.0	101.00	07-Aug-10	08-Aug-10
Diamond	2010	10-LC-038	NAD83 / UTM zone 14N	519215.0	6940109.0	209.75	26.0	-75.0	126.00	09-Aug-09	10-Aug-10
Diamond	2010	10-LC-039	NAD83 / UTM zone 14N	519215.0	6940109.0	209.68	26.0	-83.0	168.00	10-Aug-10	13-Aug-10
Diamond	2010	10-LC-040	NAD83 / UTM zone 14N	519066.0	6940027.0	206.88	26.0	-60.0	256.00	16-Aug-10	19-Aug-10
Diamond	2010	10-LC-041	NAD83 / UTM zone 14N	519031.0	6940190.0	209.63	26.0	-45.0	71.00	20-Aug-10	21-Aug-10
Diamond	2010	10-LC-042	NAD83 / UTM zone 14N	519030.0	6940188.0	209.63	26.0	-87.0	164.00	21-Aug-10	22-Aug-10
Diamond	2010	10-LC-043	NAD83 / UTM zone 14N	519031.0	6940189.0	209.56	26.0	-60.0	74.00	22-Aug-10	23-Aug-10
Diamond	2010	10-LC-044	NAD83 / UTM zone 14N	519031.0	6940188.0	209.53	26.0	-80.0	134.00	23-Aug-10	24-Aug-10
Diamond	2010	10-LC-045	NAD83 / UTM zone 14N	519376.0	6940095.0	209.78	26.0	-45.0	68.00	25-Aug-10	26-Aug-10
Diamond	2010	10-LC-046	NAD83 / UTM zone 14N	519376.0	6940094.0	209.58	26.0	-74.0	102.00	26-Aug-10	27-Aug-10

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2010	10-LC-047	NAD83 / UTM zone 14N	519376.0	6940094.0	209.57	26.0	-86.0	149.00	28-Aug-10	29-Aug-10
Diamond	2010	10-LC-048	NAD83 / UTM zone 14N	519524.0	6940056.0	209.17	26.0	-45.0	62.00	30-Aug-10	31-Aug-10
Diamond	2010	10-LC-049	NAD83 / UTM zone 14N	519523.0	6940054.0	208.87	26.0	-87.0	101.00	31-Aug-10	01-Sep-10
Diamond	2010	10-LC-050	NAD83 / UTM zone 14N	519277.0	6940121.0	210.63	26.0	-45.0	65.00	30-Jul-10	01-Aug-10
Diamond	2010	10-LC-051	NAD83 / UTM zone 14N	519277.0	6940120.0	210.43	26.0	-78.0	101.00	01-Aug-10	02-Aug-10
Diamond	2010	10-LC-052	NAD83 / UTM zone 14N	518995.0	6940230.0	210.78	26.0	-45.0	65.00	03-Aug-10	04-Aug-10
Diamond	2010	10-LC-053	NAD83 / UTM zone 14N	518995.0	6940229.0	210.62	26.0	-80.0	92.00	04-Aug-10	06-Aug-10
Diamond	2010	10-LC-054	NAD83 / UTM zone 14N	518947.0	6940243.0	210.68	26.0	-45.0	71.00	06-Aug-10	07-Aug-10
Diamond	2010	10-LC-055	NAD83 / UTM zone 14N	518947.0	6940241.0	210.86	26.0	-73.0	86.00	07-Aug-10	08-Aug-10
Diamond	2010	10-LC-056	NAD83 / UTM zone 14N	518896.0	6940136.0	209.63	26.0	-56.0	176.00	08-Aug-10	10-Aug-10
Diamond	2010	10-LC-057	NAD83 / UTM zone 14N	518896.0	6940136.0	209.50	26.0	-70.0	218.00	10-Aug-10	13-Aug-10
Diamond	2010	10-LC-058	NAD83 / UTM zone 14N	518881.0	6940222.0	209.61	26.0	-68.0	185.00	13-Aug-10	16-Aug-10
Diamond	2010	10-LC-059	NAD83 / UTM zone 14N	518764.0	6940206.0	210.86	26.0	-45.0	176.00	16-Aug-10	20-Aug-10
Diamond	2010	10-LC-060	NAD83 / UTM zone 14N	518763.0	6940205.0	210.95	26.0	-71.0	210.00	20-Aug-10	22-Aug-10
Diamond	2010	10-LC-061	NAD83 / UTM zone 14N	518763.0	6940204.0	210.92	26.0	-86.0	356.00	22-Aug-10	25-Aug-10
Diamond	2010	10-LC-062	NAD83 / UTM zone 14N	518676.0	6940371.0	211.36	26.0	-52.0	69.00	26-Aug-10	27-Aug-10
Diamond	2010	10-LC-063	NAD83 / UTM zone 14N	518676.0	6940370.0	211.15	26.0	-74.0	89.00	27-Aug-10	28-Aug-10
Diamond	2010	10-LC-064	NAD83 / UTM zone 14N	518676.0	6940370.0	211.42	26.0	-82.0	110.00	28-Aug-10	29-Aug-10
Diamond	2010	10-LC-065	NAD83 / UTM zone 14N	518710.0	6940329.0	209.85	26.0	-48.0	93.00	29-Aug-10	30-Aug-10
Diamond	2010	10-LC-066	NAD83 / UTM zone 14N	518710.0	6940328.0	210.04	26.0	-86.0	158.00	30-Aug-10	01-Sep-10
Diamond	2010	10-LC-067	NAD83 / UTM zone 14N	518710.0	6940328.0	209.86	26.0	-78.0	146.00	01-Sep-10	02-Sep-10
Diamond	2010	10-LC-068	NAD83 / UTM zone 14N	518580.0	6940289.0	219.17	26.0	-60.0	224.00	03-Sep-10	05-Sep-10
Diamond	2010	10-LC-069	NAD83 / UTM zone 14N	518580.0	6940289.0	219.11	26.0	-70.0	272.00	05-Sep-10	08-Sep-10
Diamond	2010	10-LC-070	NAD83 / UTM zone 14N	518423.0	6940425.0	227.00	26.0	-45.0	110.00	08-Sep-10	09-Sep-10
Diamond	2010	10-LC-071	NAD83 / UTM zone 14N	518423.0	6940425.0	227.00	26.0	-75.0	143.00	09-Sep-10	10-Sep-10
Diamond	2010	10-LC-072	NAD83 / UTM zone 14N	518423.0	6940425.0	227.00	26.0	-83.0	179.00	11-Sep-10	13-Sep-10
Diamond	2010	10-LC-073	NAD83 / UTM zone 14N	518423.0	6940419.0	228.64	26.0	-90.0	302.00	14-Sep-10	16-Sep-10
Diamond	2010	10-LC-074	NAD83 / UTM zone 14N	518536.0	6940428.0	220.00	26.0	-45.0	68.00	17-Sep-10	18-Sep-10
Diamond	2010	10-LC-075	NAD83 / UTM zone 14N	518536.0	6940428.0	220.00	26.0	-82.0	86.00	18-Sep-10	19-Sep-10
Diamond	2010	10-LC-076	NAD83 / UTM zone 14N	518536.0	6940428.0	220.00	26.0	-66.0	56.00	19-Sep-10	19-Sep-10
Diamond	2010	10-LC-077	NAD83 / UTM zone 14N	518513.0	6940268.0	223.00	26.0	-45.0	227.00	20-Sep-10	23-Sep-10

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2010	10-LC-078	NAD83 / UTM zone 14N	518513.0	6940268.0	223.00	26.0	-60.0	260.00	23-Sep-10	25-Sep-10
Diamond	2010	10-LC-079	NAD83 / UTM zone 14N	518513.0	6940268.0	223.00	26.0	-70.0	281.00	25-Sep-10	29-Sep-10
Diamond	2010	10-LC-080	NAD83 / UTM zone 14N	518513.0	6940268.0	223.00	26.0	-76.0	321.00	30-Sep-10	03-Oct-10
Diamond	2010	10-LC-081	NAD83 / UTM zone 14N	519601.0	6939989.0	208.74	26.0	-45.0	82.00	02-Sep-10	03-Sep-10
Diamond	2010	10-LC-082	NAD83 / UTM zone 14N	519600.0	6939987.0	208.68	26.0	-77.0	89.00	03-Sep-10	04-Sep-10
Diamond	2010	10-LC-083	NAD83 / UTM zone 14N	519641.0	6939952.0	207.12	26.0	-45.0	82.00	05-Sep-10	06-Sep-10
Diamond	2010	10-LC-084	NAD83 / UTM zone 14N	519641.0	6939951.0	207.04	26.0	-80.0	107.00	06-Sep-10	07-Sep-10
Diamond	2010	10-LC-085	NAD83 / UTM zone 14N	519666.0	6939889.0	203.59	26.0	-45.0	98.00	07-Sep-10	08-Sep-10
Diamond	2010	10-LC-086	NAD83 / UTM zone 14N	519665.0	6939888.0	203.63	26.0	-77.0	131.00	08-Sep-10	09-Sep-10
Diamond	2010	10-LC-087	NAD83 / UTM zone 14N	519550.0	6939995.0	209.37	26.0	-45.0	93.00	10-Sep-10	11-Sep-10
Diamond	2010	10-LC-088	NAD83 / UTM zone 14N	519549.0	6939994.0	209.21	26.0	-73.0	119.00	11-Sep-10	12-Sep-10
Diamond	2010	10-LC-089	NAD83 / UTM zone 14N	519548.0	6939995.0	208.00	26.0	-88.0	167.00	12-Sep-10	14-Sep-10
Diamond	2010	10-LC-090	NAD83 / UTM zone 14N	519473.0	6940069.0	210.00	26.0	-45.0	68.00	14-Sep-10	15-Sep-10
Diamond	2010	10-LC-091	NAD83 / UTM zone 14N	519473.0	6940066.0	209.11	26.0	-81.0	110.00	15-Sep-10	16-Sep-10
Diamond	2010	10-LC-092	NAD83 / UTM zone 14N	519395.0	6940024.0	208.00	26.0	-56.0	152.00	17-Sep-10	19-Sep-10
Diamond	2010	10-LC-093	NAD83 / UTM zone 14N	519395.0	6940024.0	208.00	26.0	-72.0	194.00	19-Sep-10	20-Sep-10
Diamond	2010	10-LC-094	NAD83 / UTM zone 14N	519395.0	6940024.0	208.00	26.0	-79.0	145.00	21-Sep-10	22-Sep-10
Diamond	2010	10-LC-095	NAD83 / UTM zone 14N	519395.0	6940024.0	208.00	26.0	-81.0	252.00	22-Sep-10	25-Sep-10
Diamond	2010	10-LC-096	NAD83 / UTM zone 14N	519250.0	6939954.0	208.00	26.0	-50.0	245.00	25-Sep-10	27-Sep-10
Diamond	2010	10-LC-097	NAD83 / UTM zone 14N	519250.0	6939954.0	208.00	26.0	-62.0	47.00	27-Sep-10	28-Sep-10
Diamond	2010	10-LC-098	NAD83 / UTM zone 14N	519250.0	6939954.0	208.00	26.0	-65.0	302.00	29-Sep-10	02-Oct-10
Diamond	2010	10-LC-099	NAD83 / UTM zone 14N	519469.0	6939948.0	207.00	26.0	-65.0	185.00	03-Oct-10	04-Oct-10
Diamond	2010	10-LC-100	NAD83 / UTM zone 14N	519469.0	6939948.0	207.00	26.0	-80.0	251.00	05-Oct-10	07-Oct-10
Diamond	2010	10-LC-101	NAD83 / UTM zone 14N	519551.0	6939887.0	204.00	26.0	-51.0	182.00	08-Oct-10	10-Oct-10
Diamond	2010	10-LC-102	NAD83 / UTM zone 14N	519551.0	6939887.0	204.00	26.0	-65.0	221.00	10-Oct-10	12-Oct-10
Diamond	2010	10-LC-110	NAD83 / UTM zone 14N	518420.0	6940306.0	227.00	26.0	-55.0	248.00	04-Oct-10	06-Oct-10
Diamond	2010	10-LC-111	NAD83 / UTM zone 14N	518420.0	6940306.0	227.00	26.0	-66.0	290.00	06-Oct-10	09-Oct-10
Diamond	2010	10-LC-112	NAD83 / UTM zone 14N	518958.0	6940154.0	213.00	26.0	-60.0	197.00	09-Oct-10	12-Oct-10
Diamond	2010	10-NE-001	NAD83 / UTM zone 14N	516599.0	6940564.0	210.00	35.0	-45.0	95.00	13-Oct-10	15-Oct-10
Diamond	2010	10-L52-001	NAD83 / UTM zone 14N	520670.0	6940105.0	205.00	40.0	-45.0	95.00	13-Oct-10	16-Oct-10
Diamond	2011	11-523-001	NAD83 / UTM zone 14N	519378.0	6940817.0	222.07	26.0	-45.0	89.00	2011-09-10	2011-09-11

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2011	11-523-002	NAD83 / UTM zone 14N	519378.0	6940817.0	222.07	26.0	-64.0	110.00	2011-09-12	2011-09-13
Diamond	2011	11-AG-001	NAD83 / UTM zone 14N	526800.0	6939840.0	187.52	30.0	-45.0	80.00	2011-07-07	2011-07-08
Diamond	2011	11-BZ-001	NAD83 / UTM zone 14N	516599.4	6940561.5	221.00	35.0	-55.0	125.00	2011-04-11	2011-04-13
Diamond	2011	11-BZ-002	NAD83 / UTM zone 14N	516599.4	6940561.5	221.00	35.0	-62.0	152.00	2011-04-14	2011-04-15
Diamond	2011	11-BZ-003	NAD83 / UTM zone 14N	516599.4	6940561.1	221.00	35.0	-75.0	161.00	2011-04-16	2011-04-17
Diamond	2011	11-BZ-004	NAD83 / UTM zone 14N	516576.4	6940573.2	221.35	35.0	-45.0	119.00	2011-04-18	2011-04-20
Diamond	2011	11-BZ-005	NAD83 / UTM zone 14N	516576.2	6940572.7	221.37	35.0	-65.0	160.00	2011-04-21	2011-04-22
Diamond	2011	11-BZ-006	NAD83 / UTM zone 14N	516576.0	6940572.1	221.39	35.0	-80.0	91.00	2011-04-22	2011-04-24
Diamond	2011	11-BZ-007	NAD83 / UTM zone 14N	516620.8	6940545.2	221.00	35.0	-45.0	117.29	2011-04-25	2011-04-27
Diamond	2011	11-BZ-008	NAD83 / UTM zone 14N	516620.4	6940544.7	221.00	35.0	-65.0	147.21	2011-04-27	2011-04-28
Diamond	2011	11-BZ-009	NAD83 / UTM zone 14N	516620.2	6940544.3	221.00	35.0	-75.0	167.86	2011-04-28	2011-04-30
Diamond	2011	11-BZ-010	NAD83 / UTM zone 14N	516555.5	6940585.0	221.32	35.0	-45.0	161.00	2011-05-01	2011-05-04
Diamond	2011	11-BZ-011	NAD83 / UTM zone 14N	516554.4	6940583.2	221.43	35.0	-63.0	149.00	2011-05-05	2011-05-07
Diamond	2011	11-BZ-012	NAD83 / UTM zone 14N	516553.4	6940582.0	221.51	35.0	-80.0	137.00	2011-05-07	2011-05-09
Diamond	2011	11-BZ-013	NAD83 / UTM zone 14N	516532.9	6940599.5	221.06	35.0	-45.0	134.00	2011-05-09	2011-05-11
Diamond	2011	11-BZ-014	NAD83 / UTM zone 14N	516532.4	6940598.9	221.06	35.0	-60.0	140.00	2011-05-11	2011-07-13
Diamond	2011	11-BZ-015	NAD83 / UTM zone 14N	516531.3	6940597.7	221.07	35.0	-75.0	146.00	2011-05-13	2011-05-14
Diamond	2011	11-BZ-016	NAD83 / UTM zone 14N	516574.3	6940643.2	220.74	180.0	-45.0	119.00	2011-05-15	2011-07-16
Diamond	2011	11-BZ-017	NAD83 / UTM zone 14N	516574.2	6940643.9	220.72	180.0	-63.0	155.00	2011-05-16	2011-07-18
Diamond	2011	11-BZ-018	NAD83 / UTM zone 14N	516526.1	6940634.2	220.95	130.0	-45.0	170.00	2011-05-18	2011-05-20
Diamond	2011	11-BZ-019	NAD83 / UTM zone 14N	516525.0	6940635.0	220.93	130.0	-60.0	143.00	2011-05-21	2011-05-23
Diamond	2011	11-BZ-020	NAD83 / UTM zone 14N	516559.0	6940637.0	220.89	125.0	-60.0	170.00	2011-09-13	2011-09-15
Diamond	2011	11-BZ-021	NAD83 / UTM zone 14N	516559.0	6940637.0	220.89	125.0	-45.0	188.00	2011-09-15	2011-09-17
Diamond	2011	11-BZ-022	NAD83 / UTM zone 14N	516559.0	6940637.0	220.89	125.0	-74.0	113.00	2011-09-17	2011-09-19
Diamond	2011	11-BZ-023	NAD83 / UTM zone 14N	516559.0	6940637.0	220.89	125.0	-82.0	206.00	2011-09-19	2011-09-23
Diamond	2011	11-BZ3-001	NAD83 / UTM zone 14N	516074.0	6940581.0	218.44	35.0	-61.0	101.00	2011-09-11	2011-09-13
Diamond	2011	11-J9-001	NAD83 / UTM zone 14N	522837.5	6938195.8	203.08	25.0	-45.0	101.00	2011-07-01	2011-07-04
Diamond	2011	11-J9-002	NAD83 / UTM zone 14N	522836.8	6938194.8	203.11	25.0	-65.0	110.00	2011-07-06	2011-07-06
Diamond	2011	11-JML-001	NAD83 / UTM zone 14N	525482.0	6938430.0	181.91	0.0	-45.0	131.00	2011-06-23	2011-06-24
Diamond	2011	11-JML-002	NAD83 / UTM zone 14N	525482.0	6938429.1	181.91	0.0	-60.0	128.00	2011-06-25	2011-06-26
Diamond	2011	11-JML-003	NAD83 / UTM zone 14N	525482.0	6938428.3	181.91	0.0	-80.0	101.00	2011-06-26	2011-06-27

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2011	11-JML-004	NAD83 / UTM zone 14N	525455.1	6938417.1	183.09	0.0	-45.0	104.00	2011-06-28	2011-07-01
Diamond	2011	11-LC-001	NAD83 / UTM zone 14N	518370.4	6940426.2	231.02	26.0	-76.0	209.37	2011-04-23	2011-04-27
Diamond	2011	11-LC-002	NAD83 / UTM zone 14N	518369.8	6940426.3	231.03	26.0	-83.0	239.00	2011-04-27	2011-05-01
Diamond	2011	11-LC-003	NAD83 / UTM zone 14N	518369.6	6940426.4	231.03	26.0	-90.0	329.00	2011-05-02	2011-05-07
Diamond	2011	11-LC-004	NAD83 / UTM zone 14N	518371.0	6940426.0	231.01	26.0	-65.0	3.00	2011-05-07	2011-07-07
Diamond	2011	11-LC-005	NAD83 / UTM zone 14N	518371.0	6940426.0	231.01	26.0	-65.0	173.00	2011-05-08	2011-05-09
Diamond	2011	11-LC-006	NAD83 / UTM zone 14N	519426.2	6939969.3	203.75	26.0	-60.0	215.00	2011-05-10	2011-05-14
Diamond	2011	11-LC-007	NAD83 / UTM zone 14N	519426.1	6939969.1	203.74	26.0	-72.0	212.00	2011-05-14	2011-05-16
Diamond	2011	11-LC-008	NAD83 / UTM zone 14N	519425.8	6939968.5	203.73	26.0	-82.0	254.00	2011-05-16	2011-05-20
Diamond	2011	11-LC-009	NAD83 / UTM zone 14N	519425.8	6939968.5	203.73	26.0	-86.0	299.00	2011-05-20	2011-05-25
Diamond	2011	11-LC-010	NAD83 / UTM zone 14N	519589.6	6939846.5	201.07	26.0	-50.0	194.00	2011-05-26	2011-07-29
Diamond	2011	11-LC-011	NAD83 / UTM zone 14N	519589.3	6939845.8	201.03	26.0	-67.0	182.00	2011-05-30	2011-05-31
Diamond	2011	11-LC-012	NAD83 / UTM zone 14N	519589.1	6939845.4	200.99	26.0	-85.0	212.00	2011-06-01	2011-06-02
Diamond	2011	11-LC-013	NAD83 / UTM zone 14N	519613.4	6939782.0	199.01	26.0	-45.0	143.00	2011-06-03	2011-06-09
Diamond	2011	11-LC-014	NAD83 / UTM zone 14N	519612.9	6939780.7	199.01	26.0	-55.0	215.00	2011-06-09	2011-06-13
Diamond	2011	11-LC-015	NAD83 / UTM zone 14N	519753.9	6939763.9	198.48	358.0	-45.0	121.00	2011-06-12	2011-06-14
Diamond	2011	11-LC-016	NAD83 / UTM zone 14N	519753.9	6939761.9	198.37	358.0	-70.0	122.00	2011-06-14	2011-06-16
Diamond	2011	11-LC-017	NAD83 / UTM zone 14N	519653.4	6939633.0	198.00	26.0	-45.0	149.00	2011-06-16	2011-06-18
Diamond	2011	11-LC-018	NAD83 / UTM zone 14N	519652.9	6939632.1	198.00	26.0	-60.0	122.00	2011-06-18	2011-06-19
Diamond	2011	11-LC-019	NAD83 / UTM zone 14N	519807.9	6939726.0	196.41	26.0	-45.0	101.00	2011-06-19	2011-06-22
Diamond	2011	11-LC-020	NAD83 / UTM zone 14N	517936.5	6940560.0	233.78	26.0	-45.0	95.00	2011-05-23	2011-05-24
Diamond	2011	11-LC-021	NAD83 / UTM zone 14N	517936.0	6940559.0	233.85	26.0	-66.0	113.00	2011-05-25	2011-05-25
Diamond	2011	11-LC-022	NAD83 / UTM zone 14N	517935.7	6940558.3	233.90	26.0	-87.0	137.00	2011-05-25	2011-05-27
Diamond	2011	11-LC-023	NAD83 / UTM zone 14N	517972.1	6940528.0	231.99	26.0	-45.0	104.00	2011-05-27	2011-05-28
Diamond	2011	11-LC-024	NAD83 / UTM zone 14N	517971.2	6940526.2	232.02	26.0	-78.0	134.00	2011-05-28	2011-05-30
Diamond	2011	11-LC-025	NAD83 / UTM zone 14N	518134.5	6940398.3	229.95	26.0	-45.0	160.00	2011-06-01	2011-06-02
Diamond	2011	11-LC-026	NAD83 / UTM zone 14N	518134.2	6940397.9	229.95	26.0	-60.0	173.00	2011-06-02	2011-07-07
Diamond	2011	11-LC-027	NAD83 / UTM zone 14N	518299.6	6940514.4	231.87	26.0	-45.0	119.00	2011-06-08	2011-06-10
Diamond	2011	11-LC-028	NAD83 / UTM zone 14N	518299.0	6940513.4	231.88	26.0	-70.0	95.00	2011-06-10	2011-06-11
Diamond	2011	11-LC-029	NAD83 / UTM zone 14N	517600.9	6940786.8	232.41	26.0	-45.0	143.00	2011-06-12	2011-06-14
Diamond	2011	11-LC-030	NAD83 / UTM zone 14N	517600.6	6940786.4	232.41	26.0	-60.0	164.00	2011-06-14	2011-06-16

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2011	11-LC-031	NAD83 / UTM zone 14N	517600.1	6940786.1	232.40	26.0	-75.0	212.00	2011-06-16	2011-06-19
Diamond	2011	11-LC-032	NAD83 / UTM zone 14N	517599.9	6940785.9	232.39	26.0	-86.0	185.00	2011-06-19	2011-06-21
Diamond	2011	11-LC-033	NAD83 / UTM zone 14N	517644.1	6940763.4	234.52	26.0	-45.0	104.00	2011-06-21	2011-06-22
Diamond	2011	11-LC-034	NAD83 / UTM zone 14N	517643.8	6940762.9	234.51	26.0	-64.0	137.00	2011-06-23	2011-06-25
Diamond	2011	11-LC-035	NAD83 / UTM zone 14N	517643.7	6940762.4	234.51	26.0	-79.0	137.00	2011-06-25	2011-06-26
Diamond	2011	11-LC-036	NAD83 / UTM zone 14N	517643.5	6940762.0	234.49	26.0	-88.0	209.00	2011-06-26	2011-07-01
Diamond	2011	11-LC-037	NAD83 / UTM zone 14N	517697.8	6940759.6	237.53	26.0	-45.0	59.00	2011-07-01	2011-07-02
Diamond	2011	11-LC-038	NAD83 / UTM zone 14N	517697.7	6940759.1	237.53	26.0	-79.0	95.00	2011-07-03	2011-07-04
Diamond	2011	11-LC-039	NAD83 / UTM zone 14N	517643.1	6940646.6	235.44	26.0	-50.0	194.00	2011-07-05	2011-07-07
Diamond	2011	11-LC-040	NAD83 / UTM zone 14N	517642.8	6940646.3	235.42	26.0	-68.0	233.00	2011-07-07	2011-07-10
Diamond	2011	11-LC-041	NAD83 / UTM zone 14N	517642.4	6940645.4	235.39	26.0	-78.0	284.00	2011-07-10	2011-07-13
Diamond	2011	11-LC-042	NAD83 / UTM zone 14N	517723.4	6940696.8	239.52	26.0	-45.0	101.00	2011-07-14	2011-08-15
Diamond	2011	11-LC-043	NAD83 / UTM zone 14N	517723.0	6940696.0	239.51	26.0	-71.0	131.00	2011-07-16	2011-07-17
Diamond	2011	11-LC-044	NAD83 / UTM zone 14N	517722.7	6940695.3	239.50	26.0	-87.0	185.00	2011-07-17	2011-07-19
Diamond	2011	11-LC-045	NAD83 / UTM zone 14N	517764.3	6940668.5	240.00	26.0	-45.0	110.00	2011-07-19	2011-07-20
Diamond	2011	11-LC-046	NAD83 / UTM zone 14N	517763.9	6940668.0	240.00	26.0	-75.0	140.00	2011-07-20	2011-07-23
Diamond	2011	11-LC-047	NAD83 / UTM zone 14N	517763.0	6940666.1	240.00	26.0	-86.0	182.00	2011-07-23	2011-07-24
Diamond	2011	11-LC-048	NAD83 / UTM zone 14N	517800.1	6940627.3	239.86	26.0	-45.0	119.00	2011-07-25	2011-07-26
Diamond	2011	11-LC-049	NAD83 / UTM zone 14N	517798.8	6940625.4	239.98	26.0	-65.0	140.00	2011-07-26	2011-07-27
Diamond	2011	11-LC-050	NAD83 / UTM zone 14N	519807.3	6939724.8	196.38	26.0	-65.0	104.00	2011-06-21	2011-06-23
Diamond	2011	11-LC-051	NAD83 / UTM zone 14N	517559.0	6940814.8	230.36	26.0	-45.0	116.00	2011-07-09	2011-07-11
Diamond	2011	11-LC-052	NAD83 / UTM zone 14N	517558.7	6940814.4	230.35	26.0	-64.0	146.00	2011-07-11	2011-07-12
Diamond	2011	11-LC-053	NAD83 / UTM zone 14N	517558.6	6940813.9	230.34	26.0	-79.0	179.00	2011-07-12	2011-07-14
Diamond	2011	11-LC-054	NAD83 / UTM zone 14N	517558.3	6940813.3	230.33	26.0	-88.0	221.00	2011-07-17	2011-07-18
Diamond	2011	11-LC-055	NAD83 / UTM zone 14N	517530.4	6940870.5	227.23	26.0	-45.0	95.00	2011-07-17	2011-07-19
Diamond	2011	11-LC-056	NAD83 / UTM zone 14N	517530.3	6940870.3	227.24	26.0	-73.0	122.00	2011-07-18	2011-07-20
Diamond	2011	11-LC-057	NAD83 / UTM zone 14N	517530.1	6940870.0	227.24	26.0	-89.0	158.00	2011-07-20	2011-07-21
Diamond	2011	11-LC-058	NAD83 / UTM zone 14N	517477.0	6940877.3	223.93	26.0	-45.0	119.00	2011-07-22	2011-07-24
Diamond	2011	11-LC-059	NAD83 / UTM zone 14N	517476.8	6940876.7	223.92	26.0	-71.0	167.00	2011-07-24	2011-07-26
Diamond	2011	11-LC-060	NAD83 / UTM zone 14N	517476.6	6940876.2	223.92	26.0	-84.0	194.00	2011-07-26	2011-07-28
Diamond	2011	11-LC-061	NAD83 / UTM zone 14N	517438.6	6940911.6	220.40	26.0	-45.0	140.00	2011-07-28	2011-07-30

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2011	11-LC-062	NAD83 / UTM zone 14N	517437.6	6940909.0	220.47	26.0	-87.0	239.00	2011-07-31	2011-08-02
Diamond	2011	11-LC-063	NAD83 / UTM zone 14N	517396.7	6941016.2	213.57	26.0	-65.0	83.00	2011-08-02	2011-08-03
Diamond	2011	11-LC-064	NAD83 / UTM zone 14N	517396.2	6941015.3	213.57	26.0	-90.0	134.00	2011-08-04	2011-08-06
Diamond	2011	11-LC-065	NAD83 / UTM zone 14N	517331.0	6941033.4	209.79	26.0	-45.0	95.00	2011-08-06	2011-08-09
Diamond	2011	11-LC-066	NAD83 / UTM zone 14N	520093.3	6939851.4	195.72	26.0	-45.0	110.00	2011-08-09	2011-08-13
Diamond	2011	11-LC-067	NAD83 / UTM zone 14N	520093.4	6939851.4	195.72	26.0	-55.0	152.00	2011-08-11	2011-08-12
Diamond	2011	11-LC-068	NAD83 / UTM zone 14N	520093.0	6939850.7	195.70	26.0	-69.0	161.00	2011-08-12	2011-08-13
Diamond	2011	11-LC-069	NAD83 / UTM zone 14N	520092.9	6939850.4	195.69	26.0	-80.0	149.00	2011-08-14	2011-08-15
Diamond	2011	11-LC-070	NAD83 / UTM zone 14N	520092.8	6939850.3	195.68	26.0	-87.0	242.00	2011-08-15	2011-08-17
Diamond	2011	11-LC-071	NAD83 / UTM zone 14N	520133.8	6939820.9	195.00	26.0	-45.0	122.00	2011-08-17	2011-08-19
Diamond	2011	11-LC-072	NAD83 / UTM zone 14N	520133.1	6939819.5	195.00	26.0	-69.0	125.00	2011-08-19	2011-08-21
Diamond	2011	11-LC-073	NAD83 / UTM zone 14N	520133.0	6939819.4	195.00	26.0	-80.0	200.00	2011-08-21	2011-08-23
Diamond	2011	11-LC-074	NAD83 / UTM zone 14N	520048.1	6939872.0	196.00	26.0	-45.0	116.00	2011-08-23	2011-08-26
Diamond	2011	11-LC-075	NAD83 / UTM zone 14N	520467.7	6939594.2	193.46	26.0	-45.0	113.00	2011-08-25	2011-08-26
Diamond	2011	11-LC-076	NAD83 / UTM zone 14N	520468.0	6939594.6	193.48	26.0	-71.0	176.00	2011-08-27	2011-08-28
Diamond	2011	11-LC-077	NAD83 / UTM zone 14N	520468.3	6939595.7	193.52	26.0	-83.0	221.00	2011-08-28	2011-08-29
Diamond	2011	11-LC-078	NAD83 / UTM zone 14N	520518.4	6939581.6	193.40	26.0	-45.0	107.00	2011-08-30	2011-08-31
Diamond	2011	11-LC-079	NAD83 / UTM zone 14N	520518.0	6939580.4	193.39	26.0	-69.0	131.00	2011-09-01	2011-09-02
Diamond	2011	11-LC-080	NAD83 / UTM zone 14N	517798.2	6940623.8	240.00	26.0	-81.0	209.00	2011-07-27	2011-07-29
Diamond	2011	11-LC-081	NAD83 / UTM zone 14N	517843.9	6940607.9	239.57	26.0	-45.0	141.28	2011-07-29	2011-07-31
Diamond	2011	11-LC-082	NAD83 / UTM zone 14N	517843.6	6940607.3	239.61	26.0	-63.0	152.00	2011-07-31	2011-08-02
Diamond	2011	11-LC-083	NAD83 / UTM zone 14N	517843.5	6940607.0	239.64	26.0	-77.0	164.00	2011-08-02	2011-08-04
Diamond	2011	11-LC-084	NAD83 / UTM zone 14N	520445.7	6939636.7	194.00	26.0	-45.0	65.00	2011-08-04	2011-08-05
Diamond	2011	11-LC-085	NAD83 / UTM zone 14N	520444.6	6939635.1	194.00	26.0	-71.0	110.00	2011-08-05	2011-08-07
Diamond	2011	11-LC-086	NAD83 / UTM zone 14N	520444.3	6939634.6	194.00	26.0	-84.0	158.00	2011-08-07	2011-08-08
Diamond	2011	11-LC-087	NAD83 / UTM zone 14N	520392.4	6939660.6	194.01	26.0	-45.0	77.00	2011-08-09	2011-08-09
Diamond	2011	11-LC-088	NAD83 / UTM zone 14N	520392.2	6939660.3	194.01	26.0	-70.0	116.00	2011-08-09	2011-08-11
Diamond	2011	11-LC-089	NAD83 / UTM zone 14N	520392.1	6939660.1	194.01	26.0	-81.0	158.00	2011-08-11	2011-08-12
Diamond	2011	11-LC-090	NAD83 / UTM zone 14N	520391.9	6939659.8	194.01	26.0	-89.0	274.00	2011-08-12	2011-08-15
Diamond	2011	11-LC-091	NAD83 / UTM zone 14N	520345.7	6939687.4	194.47	26.0	-45.0	104.00	2011-08-14	2011-08-16
Diamond	2011	11-LC-092	NAD83 / UTM zone 14N	520346.5	6939688.2	194.55	26.0	-66.0	119.00	2011-08-16	2011-08-17

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2011	11-LC-093	NAD83 / UTM zone 14N	520346.0	6939687.8	194.50	26.0	-79.0	155.00	2011-08-17	2011-08-18
Diamond	2011	11-LC-094	NAD83 / UTM zone 14N	520345.5	6939687.2	194.45	26.0	-88.0	206.00	2011-08-18	2011-08-20
Diamond	2011	11-LC-095	NAD83 / UTM zone 14N	520308.2	6939723.8	194.75	26.0	-45.0	80.00	2011-08-20	2011-08-21
Diamond	2011	11-LC-096	NAD83 / UTM zone 14N	520307.9	6939723.3	194.73	26.0	-72.0	107.00	2011-08-21	2011-08-22
Diamond	2011	11-LC-097	NAD83 / UTM zone 14N	520307.9	6939723.0	194.72	26.0	-84.0	161.00	2011-08-22	2011-08-23
Diamond	2011	11-LC-098	NAD83 / UTM zone 14N	520307.9	6939722.8	194.72	26.0	-90.0	200.00	2011-08-23	2011-08-25
Diamond	2011	11-LC-099	NAD83 / UTM zone 14N	520047.8	6939871.1	195.99	26.0	-75.0	134.00	2011-08-25	2011-08-26
Diamond	2011	11-LC-100	NAD83 / UTM zone 14N	520047.5	6939870.5	195.98	26.0	-89.0	182.00	2011-08-27	2011-08-28
Diamond	2011	11-LC-101	NAD83 / UTM zone 14N	520266.1	6939749.9	194.98	26.0	-45.0	83.00	2011-08-28	2011-08-29
Diamond	2011	11-LC-102	NAD83 / UTM zone 14N	520265.8	6939749.1	194.97	26.0	-73.0	113.00	2011-08-30	2011-08-30
Diamond	2011	11-LC-103	NAD83 / UTM zone 14N	520265.7	6939748.6	194.97	26.0	-83.0	164.00	2011-08-31	2011-09-02
Diamond	2011	11-LC-104	NAD83 / UTM zone 14N	520223.8	6939777.4	195.00	26.0	-45.0	77.00	2011-09-03	2011-09-04
Diamond	2011	11-LC-105	NAD83 / UTM zone 14N	520223.3	6939776.4	195.00	26.0	-73.0	131.00	2011-09-04	2011-09-05
Diamond	2011	11-LC-106	NAD83 / UTM zone 14N	520223.0	6939775.9	195.00	26.0	-88.0	225.60	2011-09-05	2011-09-07
Diamond	2011	11-LC-107	NAD83 / UTM zone 14N	520711.0	6939529.0	191.00	26.0	-45.0	80.00	2011-09-08	2011-09-09
Diamond	2011	11-LC-108	NAD83 / UTM zone 14N	520711.0	6939529.0	191.00	26.0	-77.0	131.00	2011-09-09	2011-09-11
Diamond	2011	11-LC-109	NAD83 / UTM zone 14N	517532.0	6940650.0	230.01	26.0	-60.0	305.00	2011-09-21	2011-09-24
Diamond	2011	11-LC-110	NAD83 / UTM zone 14N	520517.8	6939580.0	193.39	26.0	-83.0	152.00	2011-09-02	2011-09-03
Diamond	2011	11-LC-111	NAD83 / UTM zone 14N	520563.0	6939567.0	192.69	26.0	-45.0	77.00	2011-09-04	2011-09-05
Diamond	2011	11-LC-112	NAD83 / UTM zone 14N	520563.0	6939567.0	192.69	26.0	-74.0	131.00	2011-09-05	2011-09-06
Diamond	2011	11-LC-113	NAD83 / UTM zone 14N	520563.0	6939567.0	192.69	26.0	-87.0	47.00	2011-09-06	2011-09-07
Diamond	2011	11-LC-114	NAD83 / UTM zone 14N	520563.0	6939567.0	192.69	26.0	-87.0	200.00	2011-09-08	2011-09-09
Diamond	2011	11-LC-115	NAD83 / UTM zone 14N	517453.0	6940720.0	227.17	26.0	-55.0	299.00	2011-09-11	2011-09-17
Diamond	2011	11-LC-116	NAD83 / UTM zone 14N	517453.0	6940720.0	227.17	26.0	-64.0	314.00	2011-09-17	2011-09-20
Diamond	2011	11-LC-117	NAD83 / UTM zone 14N	517453.0	6940720.0	227.17	26.0	-71.0	410.00	2011-09-21	2011-09-25
Diamond	2011	11-LC-118	NAD83 / UTM zone 14N	517453.0	6940720.0	227.17	26.0	-75.0	407.00	2011-09-25	2011-09-29
Diamond	2011	11-LC-119	NAD83 / UTM zone 14N	517532.0	6940650.0	230.01	26.0	-67.0	329.00	2011-09-24	2011-09-27
Diamond	2011	11-LC-120	NAD83 / UTM zone 14N	517532.0	6940650.0	230.01	26.0	-73.0	350.00	2011-09-27	2011-10-02
Diamond	2012	12-774-001	NAD83 / UTM zone 14N	522286.7	6938779.6	215.00	26.0	-45.0	110.00	2012-06-17	2012-06-17
Diamond	2012	12-774-002	NAD83 / UTM zone 14N	522286.4	6938779.1	215.00	26.0	-64.0	125.00	2012-06-18	2012-06-18
Diamond	2012	12-774-003	NAD83 / UTM zone 14N	522286.1	6938778.6	215.00	26.0	-78.0	128.00	2012-06-19	2012-06-20

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-774-004	NAD83 / UTM zone 14N	522285.8	6938778.2	215.00	26.0	-88.0	194.00	2012-06-20	2012-06-22
Diamond	2012	12-774-005	NAD83 / UTM zone 14N	522430.5	6938712.3	211.94	26.0	-45.0	29.00	2012-06-23	2012-06-23
Diamond	2012	12-774-006	NAD83 / UTM zone 14N	522429.6	6938710.9	211.99	26.0	-47.0	95.00	2012-06-24	2012-06-24
Diamond	2012	12-774-007	NAD83 / UTM zone 14N	522429.2	6938710.3	212.01	26.0	-71.0	119.00	2012-06-25	2012-06-26
Diamond	2012	12-774-008	NAD83 / UTM zone 14N	522428.6	6938709.5	212.05	26.0	-88.0	147.00	2012-06-26	2012-06-28
Diamond	2012	12-774-009	NAD83 / UTM zone 14N	522563.6	6938648.0	206.78	26.0	-45.0	122.00	2012-06-28	2012-06-30
Diamond	2012	12-774-010	NAD83 / UTM zone 14N	522329.0	6938740.0	215.00	26.0	-45.0	110.00	2012-08-20	2012-08-22
Diamond	2012	12-774-011	NAD83 / UTM zone 14N	522329.0	6938740.0	215.00	26.0	-60.0	425.00	2012-08-22	2012-08-26
Diamond	2012	12-774-012	NAD83 / UTM zone 14N	522329.0	6938740.0	215.00	26.0	-77.0	176.00	2012-08-26	2012-08-28
Diamond	2012	12-774-013	NAD83 / UTM zone 14N	522329.0	6938740.0	215.00	26.0	-90.0	221.00	2012-08-29	2012-08-31
Diamond	2012	12-774-014	NAD83 / UTM zone 14N	522510.0	6938655.0	208.40	26.0	-45.0	155.00	2012-09-01	2012-09-03
Diamond	2012	12-774-015	NAD83 / UTM zone 14N	522510.0	6938655.0	208.40	26.0	-60.0	491.00	2012-09-02	2012-09-08
Diamond	2012	12-774-016	NAD83 / UTM zone 14N	522510.0	6938655.0	208.40	26.0	-82.0	149.00	2012-09-08	2012-09-10
Diamond	2012	12-BIF-001	NAD83 / UTM zone 14N	526881.3	6933835.0	174.90	330.0	-45.0	137.00	2012-07-18	2012-07-19
Diamond	2012	12-BIF-002	NAD83 / UTM zone 14N	526881.5	6933834.6	174.90	330.0	-55.0	131.00	2012-07-19	2012-07-21
Diamond	2012	12-BIF-003	NAD83 / UTM zone 14N	526841.1	6933807.2	175.41	330.0	-45.0	230.00	2012-07-21	2012-07-24
Diamond	2012	12-BIF-004	NAD83 / UTM zone 14N	526936.3	6933838.4	173.76	330.0	-45.0	143.00	2012-07-25	2012-07-28
Diamond	2012	12-BIF-005	NAD83 / UTM zone 14N	526935.9	6933839.4	173.78	330.0	-65.0	200.00	2012-07-28	2012-07-30
Diamond	2012	12-BZ3-001	NAD83 / UTM zone 14N	515958.2	6940607.0	211.10	35.0	-45.0	68.21	2012-06-21	2012-06-22
Diamond	2012	12-BZ4-001	NAD83 / UTM zone 14N	515654.2	6940522.9	210.01	35.0	-45.0	104.00	2012-06-10	2012-06-10
Diamond	2012	12-BZ4-002	NAD83 / UTM zone 14N	515653.6	6940522.0	210.03	35.0	-66.0	149.00	2012-06-11	2012-06-13
Diamond	2012	12-BZ4-003	NAD83 / UTM zone 14N	515562.4	6940564.1	200.34	35.0	-45.0	125.00	2012-06-13	2012-06-14
Diamond	2012	12-BZ4-004	NAD83 / UTM zone 14N	515562.0	6940563.3	200.33	35.0	-57.0	128.00	2012-06-15	2012-06-15
Diamond	2012	12-BZ4-005	NAD83 / UTM zone 14N	515627.3	6940570.8	205.76	35.0	-45.0	86.00	2012-06-16	2012-06-17
Diamond	2012	12-BZ4-006	NAD83 / UTM zone 14N	515626.7	6940569.7	205.77	35.0	-67.0	110.00	2012-06-17	2012-06-18
Diamond	2012	12-BZ4-007	NAD83 / UTM zone 14N	515702.7	6940506.2	213.09	35.0	-45.0	104.00	2012-06-19	2012-06-20
Diamond	2012	12-HOT-001	NAD83 / UTM zone 14N	522067.8	6940593.1	183.00	26.0	-45.0	152.00	2012-07-21	2012-07-23
Diamond	2012	12-HOT-002	NAD83 / UTM zone 14N	522067.3	6940592.0	183.00	26.0	-70.0	212.00	2012-07-23	2012-07-25
Diamond	2012	12-HOT-003	NAD83 / UTM zone 14N	522161.0	6940556.5	184.02	26.0	-45.0	179.00	2012-07-25	2012-07-27
Diamond	2012	12-HOT-004	NAD83 / UTM zone 14N	522160.6	6940555.6	184.03	26.0	-70.0	215.00	2012-07-28	2012-07-29
Diamond	2012	12-HOT-005	NAD83 / UTM zone 14N	522160.2	6940554.6	184.03	26.0	-88.0	203.00	2012-07-29	2012-07-30

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-HOT-006	NAD83 / UTM zone 14N	522199.5	6940521.5	184.03	26.0	-45.0	131.00	2012-07-31	2012-08-01
Diamond	2012	12-HOT-007	NAD83 / UTM zone 14N	522199.0	6940520.5	184.04	26.0	-67.0	200.00	2012-08-01	2012-08-02
Diamond	2012	12-J2E-001	NAD83 / UTM zone 14N	522384.9	6938534.4	214.67	26.0	-45.0	152.00	2012-06-10	2012-06-12
Diamond	2012	12-J2E-002	NAD83 / UTM zone 14N	522384.4	6938533.4	214.69	26.0	-67.0	129.00	2012-06-12	2012-06-13
Diamond	2012	12-J2E-003	NAD83 / UTM zone 14N	522222.2	6938620.8	216.53	26.0	-45.0	131.00	2012-06-12	2012-06-16
Diamond	2012	12-J4-001	NAD83 / UTM zone 14N	522786.2	6938873.6	210.00	26.0	-45.0	131.00	2012-06-30	2012-07-01
Diamond	2012	12-J4-002	NAD83 / UTM zone 14N	522786.0	6938873.2	210.00	26.0	-55.0	152.00	2012-07-01	2012-07-02
Diamond	2012	12-J4-003	NAD83 / UTM zone 14N	522785.6	6938872.6	210.00	26.0	-72.0	155.00	2012-07-02	2012-07-03
Diamond	2012	12-J4-004	NAD83 / UTM zone 14N	522785.3	6938871.8	210.00	26.0	-89.0	192.35	2012-07-04	2012-07-05
Diamond	2012	12-J4-005	NAD83 / UTM zone 14N	522820.9	6938831.4	210.00	26.0	-45.0	158.00	2012-07-06	2012-07-07
Diamond	2012	12-J4-006	NAD83 / UTM zone 14N	522820.4	6938830.2	210.00	26.0	-72.0	215.00	2012-07-07	2012-07-09
Diamond	2012	12-J4-007	NAD83 / UTM zone 14N	522820.1	6938829.4	210.00	26.0	-90.0	257.00	2012-07-09	2012-07-12
Diamond	2012	12-J4-008	NAD83 / UTM zone 14N	522740.0	6938893.0	210.00	26.0	-45.0	131.00	2012-07-11	2012-07-13
Diamond	2012	12-J4-009	NAD83 / UTM zone 14N	522739.6	6938892.3	210.00	26.0	-73.0	149.00	2012-07-13	2012-07-15
Diamond	2012	12-J4-010	NAD83 / UTM zone 14N	522739.6	6938892.0	210.00	26.0	-88.0	191.00	2012-07-14	2012-07-15
Diamond	2012	12-J4-011	NAD83 / UTM zone 14N	522698.6	6938918.8	210.00	26.0	-45.0	131.00	2012-07-16	2012-07-18
Diamond	2012	12-J4-012	NAD83 / UTM zone 14N	522697.3	6938916.6	210.00	26.0	-76.0	161.00	2012-07-18	2012-07-19
Diamond	2012	12-J4-013	NAD83 / UTM zone 14N	522697.0	6938916.1	210.00	26.0	-88.0	200.00	2012-07-19	2012-07-21
Diamond	2012	12-J4-014	NAD83 / UTM zone 14N	522639.3	6938913.0	210.78	26.0	-45.0	146.00	2012-07-21	2012-07-22
Diamond	2012	12-J4-015	NAD83 / UTM zone 14N	522638.3	6938911.3	210.79	26.0	-72.0	170.00	2012-07-23	2012-07-24
Diamond	2012	12-J4-016	NAD83 / UTM zone 14N	522638.0	6938910.5	210.78	26.0	-85.0	194.00	2012-07-24	2012-07-26
Diamond	2012	12-J4-017	NAD83 / UTM zone 14N	522589.3	6938925.4	211.00	26.0	-45.0	155.00	2012-07-26	2012-07-28
Diamond	2012	12-J4-018	NAD83 / UTM zone 14N	522587.9	6938923.8	211.00	26.0	-68.0	170.00	2012-07-28	2012-07-30
Diamond	2012	12-J4-019	NAD83 / UTM zone 14N	522587.6	6938922.8	211.00	26.0	-85.0	200.01	2012-07-30	2012-07-31
Diamond	2012	12-J4-020	NAD83 / UTM zone 14N	522864.0	6938803.2	208.55	26.0	-45.0	65.80	2012-07-31	2012-08-01
Diamond	2012	12-J4-020B	NAD83 / UTM zone 14N	522864.6	6938804.6	208.53	26.0	-45.0	14.00	2012-08-01	2012-08-01
Diamond	2012	12-J4-021	NAD83 / UTM zone 14N	522863.4	6938802.1	208.57	26.0	-50.0	149.01	2012-08-02	2012-08-03
Diamond	2012	12-J4-022	NAD83 / UTM zone 14N	522863.3	6938801.7	208.57	26.0	-72.0	170.00	2012-08-04	2012-08-05
Diamond	2012	12-J4-023	NAD83 / UTM zone 14N	522863.2	6938801.1	208.58	26.0	-84.0	230.00	2012-08-05	2012-08-08
Diamond	2012	12-J4-024	NAD83 / UTM zone 14N	522904.7	6938773.8	204.44	26.0	-45.0	146.00	2012-08-08	2012-08-09
Diamond	2012	12-J4-025	NAD83 / UTM zone 14N	522904.2	6938772.8	204.49	26.0	-67.0	170.01	2012-08-10	2012-08-11

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-J4-026	NAD83 / UTM zone 14N	522903.9	6938772.3	204.53	26.0	-85.0	215.00	2012-08-12	2012-08-14
Diamond	2012	12-J4-027	NAD83 / UTM zone 14N	522946.0	6938753.0	201.44	26.0	-45.0	165.10	2012-08-14	2012-08-16
Diamond	2012	12-J4-028	NAD83 / UTM zone 14N	522946.0	6938753.0	201.44	26.0	-68.0	158.01	2012-08-17	2012-08-19
Diamond	2012	12-J4-029	NAD83 / UTM zone 14N	522946.0	6938753.0	201.44	26.0	-88.0	170.00	2012-08-19	2012-08-20
Diamond	2012	12-J4-030	NAD83 / UTM zone 14N	522560.5	6938756.3	209.06	26.0	-60.0	341.00	2012-08-01	2012-08-05
Diamond	2012	12-J4-031	NAD83 / UTM zone 14N	522662.1	6938739.0	209.86	26.0	-60.0	329.00	2012-08-05	2012-08-08
Diamond	2012	12-J4-032	NAD83 / UTM zone 14N	522738.9	6938662.1	202.47	26.0	-70.0	374.00	2012-08-09	2012-08-14
Diamond	2012	12-J4-033	NAD83 / UTM zone 14N	522458.0	6938776.0	211.47	26.0	-60.0	329.00	2012-08-14	2012-08-16
Diamond	2012	12-J4-034	NAD83 / UTM zone 14N	522458.0	6938776.0	211.47	26.0	-74.0	431.00	2012-08-17	2012-08-24
Diamond	2012	12-J4-035	NAD83 / UTM zone 14N	523009.0	6938649.0	200.00	26.0	-45.0	185.00	2012-08-24	2012-08-26
Diamond	2012	12-J4-036	NAD83 / UTM zone 14N	523009.0	6938649.0	200.00	26.0	-90.0	272.00	2012-08-27	2012-08-29
Diamond	2012	12-J4-037	NAD83 / UTM zone 14N	523009.0	6938649.0	200.00	26.0	-73.0	221.00	2012-08-29	2012-08-31
Diamond	2012	12-J4-038	NAD83 / UTM zone 14N	523056.0	6938642.0	199.49	26.0	-45.0	230.01	2012-09-01	2012-09-04
Diamond	2012	12-J4-039	NAD83 / UTM zone 14N	523056.0	6938642.0	199.49	26.0	-70.0	176.00	2012-09-05	2012-09-08
Diamond	2012	12-J4-040	NAD83 / UTM zone 14N	522549.4	6938953.7	211.07	26.0	-45.0	128.00	2012-08-03	2012-08-05
Diamond	2012	12-J4-041	NAD83 / UTM zone 14N	522548.4	6938952.4	211.02	26.0	-70.0	158.00	2012-08-05	2012-08-07
Diamond	2012	12-J4-042	NAD83 / UTM zone 14N	522547.8	6938951.2	211.01	26.0	-86.0	224.00	2012-08-06	2012-08-09
Diamond	2012	12-J4-043	NAD83 / UTM zone 14N	522503.7	6938979.9	212.00	26.0	-45.0	161.00	2012-08-09	2012-08-11
Diamond	2012	12-J4-044	NAD83 / UTM zone 14N	522503.4	6938979.0	212.00	26.0	-65.0	170.01	2012-08-11	2012-08-12
Diamond	2012	12-J4-045	NAD83 / UTM zone 14N	522502.5	6938977.9	212.00	26.0	-85.0	188.00	2012-08-11	2012-08-14
Diamond	2012	12-J4-046	NAD83 / UTM zone 14N	522420.0	6938926.0	212.57	26.0	-60.0	227.00	2012-08-14	2012-08-16
Diamond	2012	12-J4-047	NAD83 / UTM zone 14N	522420.0	6938926.0	212.57	26.0	-80.0	236.00	2012-08-17	2012-08-20
Diamond	2012	12-J4-048	NAD83 / UTM zone 14N	522976.0	6938693.0	200.50	26.0	-45.0	188.00	2012-08-21	2012-08-23
Diamond	2012	12-J4-049	NAD83 / UTM zone 14N	522976.0	6938693.0	200.50	26.0	-67.0	176.00	2012-08-23	2012-08-25
Diamond	2012	12-J4-050	NAD83 / UTM zone 14N	522976.0	6938693.0	200.50	26.0	-90.0	212.00	2012-08-25	2012-08-27
Diamond	2012	12-J4-051	NAD83 / UTM zone 14N	523178.0	6938541.0	188.88	26.0	-45.0	173.00	2012-08-26	2012-08-29
Diamond	2012	12-J4-052	NAD83 / UTM zone 14N	523178.0	6938541.0	188.88	26.0	-71.0	203.00	2012-08-29	2012-08-31
Diamond	2012	12-J4-053	NAD83 / UTM zone 14N	523178.0	6938541.0	188.88	26.0	-88.0	230.00	2012-08-31	2012-09-02
Diamond	2012	12-J4-054	NAD83 / UTM zone 14N	522820.0	6938606.0	197.41	26.0	-60.0	329.00	2012-09-03	2012-09-06
Diamond	2012	12-J4-055	NAD83 / UTM zone 14N	522891.0	6938523.0	193.00	26.0	-60.0	377.00	2012-09-07	2012-09-10
Diamond	2012	12-J4-056	NAD83 / UTM zone 14N	523093.0	6938594.0	195.81	26.0	-45.0	212.00	2012-09-11	2012-09-13

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-J4-057	NAD83 / UTM zone 14N	523093.0	6938594.0	195.81	26.0	-72.0	230.00	2012-09-13	2012-09-15
Diamond	2012	12-J4-060	NAD83 / UTM zone 14N	523056.0	6938642.0	199.49	26.0	-88.0	179.00	2012-09-07	2012-09-08
Diamond	2012	12-J4-061	NAD83 / UTM zone 14N	522344.0	6938998.0	212.00	26.0	-45.0	170.00	2012-09-09	2012-09-10
Diamond	2012	12-J4-062	NAD83 / UTM zone 14N	522344.0	6938998.0	212.00	26.0	-72.0	203.00	2012-09-10	2012-09-12
Diamond	2012	12-J4-063	NAD83 / UTM zone 14N	522344.0	6938998.0	212.00	26.0	-85.0	275.00	2012-09-12	2012-09-15
Diamond	2012	12-J4-070	NAD83 / UTM zone 14N	522880.0	6938602.0	197.16	26.0	-65.0	308.00	2012-09-11	2012-09-13
Diamond	2012	12-LC-001	NAD83 / UTM zone 14N	517533.0	6940647.5	230.02	26.0	-81.0	479.48	2012-03-28	2012-04-06
Diamond	2012	12-LC-002	NAD83 / UTM zone 14N	517532.9	6940647.4	230.02	26.0	-88.0	215.00	2012-04-05	2012-04-09
Diamond	2012	12-LC-003	NAD83 / UTM zone 14N	517532.9	6940647.1	230.02	26.0	-89.0	36.00	2012-04-10	2012-04-11
Diamond	2012	12-LC-004	NAD83 / UTM zone 14N	517584.6	6940526.4	230.94	26.0	-65.0	416.00	2012-04-12	2012-04-18
Diamond	2012	12-LC-005	NAD83 / UTM zone 14N	517413.9	6940750.9	224.36	26.0	-60.0	296.64	2012-04-18	2012-04-23
Diamond	2012	12-LC-006	NAD83 / UTM zone 14N	517413.5	6940750.5	224.33	26.0	-69.0	308.00	2012-04-23	2012-04-27
Diamond	2012	12-LC-007	NAD83 / UTM zone 14N	517413.3	6940750.3	224.32	26.0	-80.0	409.88	2012-04-28	2012-05-02
Diamond	2012	12-LC-008	NAD83 / UTM zone 14N	517413.0	6940750.0	224.31	26.0	-90.0	482.00	2012-05-04	2012-05-10
Diamond	2012	12-LC-009	NAD83 / UTM zone 14N	517790.7	6940493.0	238.94	26.0	-60.0	263.00	2012-05-09	2012-05-12
Diamond	2012	12-LC-010	NAD83 / UTM zone 14N	517695.5	6940519.1	235.99	26.0	-62.0	314.00	2012-04-13	2012-04-17
Diamond	2012	12-LC-011	NAD83 / UTM zone 14N	517695.4	6940518.7	235.97	26.0	-75.0	353.00	2012-04-17	2012-04-21
Diamond	2012	12-LC-012	NAD83 / UTM zone 14N	517851.1	6940502.4	240.39	26.0	-60.0	212.00	2012-05-14	2012-05-15
Diamond	2012	12-LC-013	NAD83 / UTM zone 14N	517290.8	6940719.4	219.33	26.0	-65.0	428.00	2012-05-16	2012-05-21
Diamond	2012	12-LCE-001	NAD83 / UTM zone 14N	519999.1	6939890.0	196.91	26.0	-45.0	140.00	2012-07-10	2012-07-11
Diamond	2012	12-LCE-002	NAD83 / UTM zone 14N	519998.9	6939889.4	196.89	26.0	-64.0	182.00	2012-07-12	2012-07-14
Diamond	2012	12-LCE-003	NAD83 / UTM zone 14N	519998.8	6939889.0	196.87	26.0	-77.0	230.00	2012-07-14	2012-07-17
Diamond	2012	12-LCE-004	NAD83 / UTM zone 14N	519971.0	6939939.5	198.00	26.0	-45.0	116.00	2012-07-18	2012-07-19
Diamond	2012	12-LCM-001	NAD83 / UTM zone 14N	519186.6	6939929.2	207.22	26.0	-60.0	226.30	2012-06-13	2012-06-16
Diamond	2012	12-LCM-002	NAD83 / UTM zone 14N	519186.3	6939928.8	207.23	26.0	-59.0	341.00	2012-06-16	2012-06-19
Diamond	2012	12-LCM-003	NAD83 / UTM zone 14N	519186.2	6939928.5	207.24	26.0	-70.0	371.00	2012-06-20	2012-06-24
Diamond	2012	12-LCM-004	NAD83 / UTM zone 14N	519306.9	6939931.6	207.87	26.0	-64.0	293.00	2012-06-23	2012-06-27
Diamond	2012	12-LCM-005	NAD83 / UTM zone 14N	519306.9	6939931.3	207.86	26.0	-79.0	407.00	2012-06-27	2012-07-01
Diamond	2012	12-LCM-006	NAD83 / UTM zone 14N	519337.7	6939788.3	205.00	26.0	-60.0	398.00	2012-07-02	2012-07-06
Diamond	2012	12-LCM-007	NAD83 / UTM zone 14N	519456.2	6939801.8	201.22	26.0	-60.0	320.00	2012-07-07	2012-07-09
Diamond	2012	12-LCM-010	NAD83 / UTM zone 14N	518962.1	6940036.5	209.24	26.0	-60.0	374.00	2012-06-23	2012-06-27

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-LCM-011	NAD83 / UTM zone 14N	518961.9	6940035.7	209.24	26.0	-75.0	284.00	2012-06-27	2012-06-30
Diamond	2012	12-LCM-012A	NAD83 / UTM zone 14N	519097.3	6939975.8	208.00	26.0	-55.0	25.90	2012-07-02	2012-07-03
Diamond	2012	12-LCM-012B	NAD83 / UTM zone 14N	519096.9	6939976.1	208.00	26.0	-55.0	93.73	2012-07-02	2012-07-03
Diamond	2012	12-LCM-013	NAD83 / UTM zone 14N	519096.2	6939976.1	208.00	26.0	-60.0	278.00	2012-07-03	2012-07-07
Diamond	2012	12-LCM-014	NAD83 / UTM zone 14N	518673.2	6940135.2	218.07	26.0	-60.0	341.00	2012-07-07	2012-07-12
Diamond	2012	12-LCM-015	NAD83 / UTM zone 14N	518224.5	6940358.7	230.25	26.0	-60.0	300.00	2012-07-13	2012-07-17
Diamond	2012	12-LCSW-001	NAD83 / UTM zone 14N	517290.6	6940719.1	219.33	26.0	-78.0	128.00	2012-05-21	2012-05-22
Diamond	2012	12-LCSW-002	NAD83 / UTM zone 14N	517290.3	6940718.7	219.32	26.0	-90.0	170.00	2012-04-22	2012-05-23
Diamond	2012	12-LCSW-003	NAD83 / UTM zone 14N	517333.6	6940695.5	221.11	26.0	-45.0	104.00	2012-05-24	2012-05-25
Diamond	2012	12-LCSW-004	NAD83 / UTM zone 14N	517333.5	6940695.2	221.11	26.0	-60.0	137.00	2012-05-25	2012-05-27
Diamond	2012	12-LCSW-005	NAD83 / UTM zone 14N	517333.4	6940695.0	221.10	26.0	-75.0	182.00	2012-05-27	2012-05-29
Diamond	2012	12-LCSW-006	NAD83 / UTM zone 14N	517333.3	6940694.8	221.10	26.0	-83.0	167.00	2012-05-29	2012-05-31
Diamond	2012	12-LCSW-007	NAD83 / UTM zone 14N	517333.1	6940694.6	221.08	26.0	-90.0	173.00	2012-05-31	2012-06-02
Diamond	2012	12-LCSW-008	NAD83 / UTM zone 14N	517372.8	6940659.4	223.31	26.0	-45.0	125.00	2012-06-02	2012-06-03
Diamond	2012	12-LCSW-009	NAD83 / UTM zone 14N	517372.5	6940659.1	223.29	26.0	-67.0	155.00	2012-06-04	2012-06-05
Diamond	2012	12-LCSW-010	NAD83 / UTM zone 14N	517422.3	6940537.1	223.35	26.0	-45.0	137.00	2012-06-06	2012-06-07
Diamond	2012	12-LCSW-011	NAD83 / UTM zone 14N	517422.1	6940536.6	223.33	26.0	-65.0	137.00	2012-06-08	2012-06-09
Diamond	2012	12-OHM-001	NAD83 / UTM zone 14N	520299.1	6939470.7	192.00	26.0	-45.0	110.00	2012-06-10	2012-06-11
Diamond	2012	12-OHM-002	NAD83 / UTM zone 14N	520298.4	6939469.6	192.00	26.0	-75.0	122.00	2012-06-12	2012-06-13
Diamond	2012	12-PL-001	NAD83 / UTM zone 14N	518980.5	6940910.0	230.11	26.0	-45.0	110.00	2012-04-22	2012-04-23
Diamond	2012	12-PL-002	NAD83 / UTM zone 14N	518980.2	6940909.7	230.13	26.0	-63.0	125.00	2012-04-23	2012-04-24
Diamond	2012	12-PL-003	NAD83 / UTM zone 14N	518980.2	6940909.4	230.15	26.0	-79.0	170.00	2012-04-26	2012-04-28
Diamond	2012	12-PL-004	NAD83 / UTM zone 14N	518954.3	6940940.1	229.41	26.0	-45.0	104.00	2012-04-28	2012-04-30

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2012	12-PL-005	NAD83 / UTM zone 14N	518954.1	6940939.5	229.42	26.0	-68.0	143.00	2012-04-30	2012-05-02
Diamond	2012	12-PL-006	NAD83 / UTM zone 14N	518953.9	6940939.0	229.42	26.0	-85.0	209.00	2012-05-02	2012-05-04
Diamond	2012	12-PL-007	NAD83 / UTM zone 14N	519023.1	6940909.6	230.14	26.0	-45.0	116.00	2012-05-04	2012-05-05
Diamond	2012	12-PL-008	NAD83 / UTM zone 14N	519023.0	6940909.2	230.16	26.0	-73.0	110.00	2012-05-05	2012-05-09
Diamond	2012	12-PL-009	NAD83 / UTM zone 14N	519022.7	6940908.8	230.18	26.0	-86.0	176.00	2012-05-06	2012-05-09
Diamond	2012	12-PL-010	NAD83 / UTM zone 14N	519407.4	6940785.2	220.30	26.0	-45.0	140.00	2012-05-11	2012-05-14
Diamond	2012	12-PL-011	NAD83 / UTM zone 14N	519407.4	6940784.8	220.31	26.0	-69.0	140.00	2012-05-14	2012-05-15
Diamond	2012	12-PL-012	NAD83 / UTM zone 14N	519407.1	6940784.3	220.33	26.0	-80.0	56.00	2012-05-16	2012-05-17
Diamond	2012	12-PL-013	NAD83 / UTM zone 14N	519406.9	6940784.1	220.33	26.0	-80.0	185.00	2012-05-17	2012-05-18
Diamond	2012	12-PL-014	NAD83 / UTM zone 14N	519449.6	6940758.9	219.08	26.0	-45.0	110.00	2012-05-19	2012-05-20
Diamond	2012	12-PL-015	NAD83 / UTM zone 14N	519449.4	6940758.4	219.09	26.0	-70.0	167.00	2012-05-21	2012-05-22
Diamond	2012	12-PL-016	NAD83 / UTM zone 14N	519449.4	6940758.0	219.09	26.0	-76.0	194.00	2012-05-23	2012-05-24
Diamond	2012	12-PL-017	NAD83 / UTM zone 14N	519489.9	6940727.6	218.35	26.0	-45.0	176.79	2012-05-24	2012-05-26
Diamond	2012	12-PL-018	NAD83 / UTM zone 14N	519489.2	6940726.5	218.37	26.0	-68.0	170.00	2012-05-26	2012-05-27
Diamond	2012	12-PL-019	NAD83 / UTM zone 14N	519489.0	6940726.2	218.38	26.0	-77.0	176.00	2012-05-27	2012-05-29
Diamond	2012	12-PL-020	NAD83 / UTM zone 14N	519799.1	6940564.4	209.00	26.0	-45.0	81.50	2012-05-30	2012-05-30
Diamond	2012	12-PL-021	NAD83 / UTM zone 14N	519798.9	6940564.0	209.00	26.0	-65.0	122.00	2012-05-30	2012-05-31
Diamond	2012	12-PL-022	NAD83 / UTM zone 14N	519798.1	6940562.8	209.00	26.0	-80.0	131.00	2012-05-31	2012-06-02
Diamond	2012	12-PL-023	NAD83 / UTM zone 14N	519797.9	6940562.5	209.00	26.0	-90.0	200.00	2012-06-02	2012-06-03
Diamond	2012	12-PL-024	NAD83 / UTM zone 14N	519899.2	6940473.3	208.00	26.0	-45.0	134.00	2012-06-04	2012-06-06
Diamond	2012	12-PL-025	NAD83 / UTM zone 14N	519898.5	6940472.4	208.00	26.0	-67.0	131.00	2012-06-06	2012-06-07
Diamond	2012	12-PL-026	NAD83 / UTM zone 14N	520127.0	6940320.2	206.31	26.0	-45.0	125.00	2012-06-07	2012-06-09
Diamond	2012	12-PL-027	NAD83 / UTM zone 14N	520126.6	6940319.8	206.32	26.0	-65.0	128.00	2012-06-09	2012-06-10
Diamond	2013	13-J1-001	NAD83 / UTM zone 14N	521610.0	6939202.0	192.00	26.0	-45.0	131.80	2013-04-09	2013-04-10
Diamond	2013	13-J1-002	NAD83 / UTM zone 14N	521610.0	6939202.0	192.00	26.0	-78.0	140.00	2013-04-10	2013-04-11
Diamond	2013	13-J1-003	NAD83 / UTM zone 14N	521610.0	6939202.0	192.00	26.0	-90.0	98.00	2013-04-12	2013-04-12
Diamond	2013	13-J1-004	NAD83 / UTM zone 14N	521452.0	6939222.0	190.68	26.0	-45.0	92.00	2013-04-13	2013-04-14
Diamond	2013	13-J1-005	NAD83 / UTM zone 14N	521452.0	6939222.0	190.68	26.0	-70.0	122.00	2013-04-15	2013-04-15
Diamond	2013	13-J1-006	NAD83 / UTM zone 14N	521452.0	6939222.0	190.68	26.0	-88.0	206.00	2013-04-16	2013-04-17
Diamond	2013	13-J1-007	NAD83 / UTM zone 14N	520994.0	6939426.0	187.00	26.0	-45.0	86.00	2013-04-22	2013-04-26
Diamond	2013	13-J4W-001	NAD83 / UTM zone 14N	521123.0	6939573.0	187.00	26.0	-45.0	143.00	2013-04-19	2013-04-21

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2013	13-ML-001	NAD83 / UTM zone 14N	523683.0	6938892.0	209.73	26.0	-45.0	185.00	2013-05-18	2013-05-20
Diamond	2013	13-ML-002	NAD83 / UTM zone 14N	523683.0	6938892.0	209.73	26.0	-67.0	173.00	2013-05-20	2013-05-22
Diamond	2013	13-ML-003	NAD83 / UTM zone 14N	524695.0	6938458.0	185.00	26.0	-45.0	131.00	2013-05-22	2013-05-24
Diamond	2013	13-ML-004	NAD83 / UTM zone 14N	524695.0	6938458.0	185.00	26.0	-66.0	143.00	2013-05-24	2013-05-25
Diamond	2013	13-VGR1-001	NAD83 / UTM zone 14N	495658.0	6943912.0	286.00	315.0	-90.0	412.00	2013-05-03	2013-05-17
Diamond	2013	13-VGR1-001A	NAD83 / UTM zone 14N	495658.0	6943912.0	286.00			38.00	2013-05-03	2013-05-07
Diamond	2015	15-DP-001	NAD83 / UTM zone 14N	492900.0	6932165.0	283.00	135.0	-45.0	23.53	2015-07-05	2015-07-08
Diamond	2015	15-DP-002	NAD83 / UTM zone 14N	492900.0	6932165.0	283.00	135.0	-60.0	119.00	2015-07-08	2015-07-10
Diamond	2015	15-DP-003	NAD83 / UTM zone 14N	492900.0	6932165.0	283.00	135.0	-50.0	80.00	2015-07-10	2015-07-11
Diamond	2015	15-DP-004	NAD83 / UTM zone 14N	492900.0	6932165.0	283.00	135.0	-70.0	116.00	2015-07-12	2015-07-14
Diamond	2015	15-DP-005	NAD83 / UTM zone 14N	492859.0	6932136.0	282.60	135.0	-65.0	107.00	2015-07-15	2015-07-17
Diamond	2015	15-DP-006	NAD83 / UTM zone 14N	492859.0	6932136.0	282.60	135.0	-75.0	122.00	2015-07-17	2015-07-18
Diamond	2015	15-DP-007	NAD83 / UTM zone 14N	492859.0	6932136.0	282.60	135.0	-90.0	127.60	2015-07-19	2015-07-20
Diamond	2015	15-DP-008	NAD83 / UTM zone 14N	492786.0	6932138.0	282.00	135.0	-45.0	152.76	2015-07-22	2015-07-24
Diamond	2015	15-DP-009	NAD83 / UTM zone 14N	492920.0	6932216.0	283.00	135.0	-45.0	110.00	2015-07-25	2015-07-26
Diamond	2022	22-DP-001	NAD83 / UTM zone 14N	6932246.0	492959.0	283.00	135.0	-70.0	78.00	2022-07-22	2022-07-25
Diamond	2022	22-DP-002	NAD83 / UTM zone 14N	6932246.0	492959.0	283.00	135.0	-75.0	184.00	2022-07-25	2022-07-28
Diamond	2022	22-DP-003	NAD83 / UTM zone 14N	6932281.0	492992.0	283.00	135.0	-45.0	16.00	2022-07-28	2022-07-30
Diamond	2022	22-DP-003a	NAD83 / UTM zone 14N	6932281.0	492992.0	283.00	135.0	-45.0	114.00	2022-07-30	2022-08-05
Diamond	2022	22-DP-004	NAD83 / UTM zone 14N	6932410.0	493142.0	282.00	135.0	-70.0	169.00	2022-08-01	2022-08-05
Diamond	2022	22-DP-005	NAD83 / UTM zone 14N	6932410.0	493142.0	284.00	135.0	-82.0	211.00	2022-08-05	2022-08-08
Diamond	2022	22-DP-006	NAD83 / UTM zone 14N	6932281.0	492992.0	283.00	135.0	-65.0	148.00	2022-08-05	2022-08-07
Diamond	2022	22-DP-007	NAD83 / UTM zone 14N	6932281.0	492992.0	283.00	135.0	-82.0	202.07	2022-08-07	2022-08-09
Diamond	2022	22-DP-008	NAD83 / UTM zone 14N	6932381.0	493104.0	283.81	135.0	-45.0	166.47	2022-09-08	2022-10-08
Diamond	2022	22-DP-009	NAD83 / UTM zone 14N	6932304.0	492869.0	282.00	135.0	-55.0	250.00	2022-08-10	2022-08-14
Diamond	2022	22-DP-010	NAD83 / UTM zone 14N	6932381.0	493104.0	283.81	135.0	-70.0	178.00	2022-08-10	2022-08-12
Diamond	2022	22-DP-011A	NAD83 / UTM zone 14N	6932304.0	492869.0	282.00	135.0	-70.0	322.00	2022-08-15	2022-08-19
Diamond	2022	22-DP-012	NAD83 / UTM zone 14N	6932351.0	492908.0	282.00	135.0	-62.0	334.00	2022-08-23	2022-08-27
Diamond	2022	22-DP-013	NAD83 / UTM zone 14N	6932327.0	493155.0	284.00	135.0	-45.0	130.00	2022-08-28	2022-08-31
Diamond	2022	22-DP-014	NAD83 / UTM zone 14N	6932348.0	493067.0	283.30	135.0	-45.0	160.00	2022-09-01	2022-09-03

Type	Year	Drillhole ID	UTM	Easting (m)	Northing (m)	Elevation (m)	Azimuth (°)	Dip (°)	Final depth (m)	Start Date	End Date
Diamond	2022	22-J4W-001	NAD83 / UTM zone 14N	6939113.0	521740.0	193.85	26.0	-45.0	79.00	2022-08-14	2022-08-15
Diamond	2022	22-J4W-002	NAD83 / UTM zone 14N	6939113.0	521740.0	193.85	26.0	-75.0	109.00	2022-08-15	2022-08-17
Diamond	2022	22-J4W-003	NAD83 / UTM zone 14N	6939144.0	521708.0	193.00	26.0	-90.0	149.00	2022-08-17	2022-08-19
Diamond	2022	22-J4W-004	NAD83 / UTM zone 14N	6939090.0	521783.0	194.84	26.0	-45.0	79.00	2022-08-20	2022-08-21
Diamond	2022	22-J4W-005	NAD83 / UTM zone 14N	6939090.0	521783.0	194.84	26.0	-75.0	101.50	2022-08-22	2022-08-23
Diamond	2022	22-J4W-006	NAD83 / UTM zone 14N	6939069.0	521827.0	196.31	26.0	-45.0	79.00	2022-08-23	2022-08-24
Diamond	2022	22-J4W-007	NAD83 / UTM zone 14N	6939069.0	521827.0	196.31	26.0	-75.0	35.12	2022-08-24	2022-08-24
Diamond	2022	22-J4W-007A	NAD83 / UTM zone 14N	6939069.0	521827.0	196.31	26.0	-75.0	109.00	2022-08-24	2022-08-27
Diamond	2022	22-J4W-008	NAD83 / UTM zone 14N	6939231.0	521564.0	192.00	26.0	-45.0	79.00	2022-08-28	2022-08-31
Diamond	2022	22-J4W-009	NAD83 / UTM zone 14N	6939231.0	521564.0	192.00	26.0	-75.0	110.00	2022-08-31	2022-09-03
Diamond	2023	23-LC-001	NAD83 / UTM zone 14N	519794.7	6939863.1	205.50	25.0	-59	234.00		
Diamond	2023	23-LC-002	NAD83 / UTM zone 14N	519322.0	6939769.0	206.00	26.0	-55	260.00		
Diamond	2023	23-LC-003	NAD83 / UTM zone 14N	519668.0	6939864.0	203.00	26.0	-55	197.00		
Diamond	2023	23-LC-004	NAD83 / UTM zone 14N	519002.3	6939999.0	210.80	24.0	-55	317.00		
Diamond	2023	23-LC-005	NAD83 / UTM zone 14N	519488.0	6939864.0	201.00	26.0	-58	269.00		
Diamond	2023	23-LC-006	NAD83 / UTM zone 14N	519002.0	6940001.0	210.00	24.0	-70	362.00		
Diamond	2023	23-LC-007	NAD83 / UTM zone 14N	519306.0	6939838.0	206.00	25.0	-60	380.00		
Diamond	2023	23-LC-008	NAD83 / UTM zone 14N	518957.0	6939966.0	209.00	23.0	-74	402.80		
Diamond	2023	23-LC-009	NAD83 / UTM zone 14N	519131.0	6940067.0	210.00	2.0	-72	242.00		
Diamond	2023	23-LC-010	NAD83 / UTM zone 14N	519131.0	6940067.0	210.00	12.0	-85	326.00		
Diamond	2023	23-LC-011	NAD83 / UTM zone 14N	518717.0	6940232.0	203.00	25.0	-54	215.00		
Diamond	2023	23-LC-012	NAD83 / UTM zone 14N	518458.1	6940273.5	233.50	26.0	-60	347.00		
Diamond	2023	23-LC-013	NAD83 / UTM zone 14N	519074.0	6939978.0	213.20	25.0	-85	17.00		
Diamond	2023	23-LC-013A	NAD83 / UTM zone 14N	519064.3	6939973.8	206.40	12.0	-80.5	413.00		
Diamond	2023	23-LC-014	NAD83 / UTM zone 14N	518458.3	6940273.9	233.30	25.0	-70	362.00		
Diamond	2023	23-LC-015	NAD83 / UTM zone 14N	518202.0	6940390.5	230.50	28.0	-60	359.00		
Diamond	2023	23-LC-016	NAD83 / UTM zone 14N	518394.1	6940471.3	235.20	24.0	-60	26.00		
Diamond	2023	23-LC-016A	NAD83 / UTM zone 14N	518394.1	6940471.3	234.90	24.0	-58	44.00		
Diamond	2023	23-LC-016B	NAD83 / UTM zone 14N	518401.6	6940485.8	230.10	24.0	-60	143.00		
Diamond	2023	23-LC-017	NAD83 / UTM zone 14N	518024.4	6940421.3	234.90	35.0	-45	302.00		
Diamond	2023	23-LC-018	NAD83 / UTM zone 14N	518205.8	6940320.3	229.80	26.0	-60	444.00		

APPENDIX B:
SIGNIFICANT HIGHLIGHT DRILLHOLE SAMPLE INTERVALS AND RESULTS

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%	Cu%
2009	09-LC-001	102.87	105.54	2.67	0.77	0.03
2009	Including	102.87	104.23	1.36	1.49	0.03
2009	09-LC-002	126.54	129.42	2.88	2.06	0.04
2009	09-LC-003	47.40	49.60	2.20	0.65	0.02
2009	09-LC-004*	46.38	50.92	4.54	0.23	0.04
2009	Including	46.38	47.32	0.94	0.26	0.06
2009	Including	50.54	50.92	0.38	2.11	0.02
2009	09-LC-005	78.96	79.58	0.62	1.35	0.07
2009	09-LC-006	139.22	141.67	2.45	1.2	0.24
2009	Including	139.22	140.00	0.78	3.66	0.66
2009	09-LC-007	122.88	124.75	1.87	0.25	0.06
2009	09-LC-008	126.26	127.75	1.49	0.07	0.02
2009	Including	126.64	127.35	0.71	0.11	0.01
2009	09-LC-010	58.68	63.42	4.74	0.25	0.14
2009	Including	59.63	60.67	1.04	0.84	0.19
2009	09-LC-012	92.30	94.08	1.78	1.87	0.91
2009	09-LC-013	75.73	76.36	0.63	0.6	2.73
2009	09-LC-014*	56.93	61.49	4.56	0.4	0.08
2009	Including	56.93	58.50	1.57	0.56	0.15
2009	Including	60.00	61.49	1.49	0.63	0.06
2009	09-LC-014	68.29	69.20	0.91	1.08	0.06
2009	09-LC-015	49.70	51.17	1.47	2.13	0.1

All samples subject to ICP 1 Analysis by SRC in Saskatoon, SK. ICP1 results >1000 ppm Uranium subject to SRC U3O8 Assay
ICP1 results for Cu are reported by SRC in parts per million (ppm). 1 ppm = 1gm/t, 10000 ppm = 1%

* Full intervals include ICP U analysis in ppm converted to U3O8%. Conversion to U3O8% = ppm x 0.01179%

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%
2010	10-LC-003	148.06	148.50	13.98	0.70
2010	10-LC-005	182.08	182.38	2.41	0.87
2010	10-LC-006	207.72	208.22	0.50	2.90
2010	10-LC-007	170.38	170.68	1.02	2.68
2010	10-LC-011	172.85	173.23	2.78	0.84
2010	10-LC-012	153.00	153.37	0.37	0.90
2010	10-LC-015	96.00	96.40	1.05	2.87
2010	10-LC-016	138.70	139.14	0.84	2.59
2010	10-LC-018	81.54	82.66	3.58	1.61
2010	10-LC-023*	116.70	117.10	1.10	0.75
2010	10-LC-024	184.00	184.50	2.63	0.83
2010	10-LC-027	116.10	116.40	0.30	0.57
2010	10-LC-028	129.48	129.78	0.30	1.72
2010	10-LC-031	100.80	101.35	0.55	1.89

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%
2010	10-LC-034	85.80	86.20	0.40	0.53
2010	10-LC-035*	45.97	46.54	1.74	2.11
2010	10-LC-039	145.60	146.42	6.84	1.05
2010	10-LC-043	65.00	65.32	1.70	0.59
2010	10-LC-044	120.23	120.60	5.65	0.33
2010	10-LC-045	57.10	57.55	1.65	0.20
2010	10-LC-046	89.00	89.65	1.35	2.92
2010	10-LC-048	46.20	46.60	0.90	4.26
2010	10-LC-050	52.24	52.95	1.85	1.10
2010	10-LC-051	87.80	88.70	2.04	2.23
2010	10-LC-053	67.47	67.77	1.52	1.23
2010	10-LC-054	56.48	57.10	1.20	1.16
2010	10-LC-055	76.40	76.80	2.70	1.55
2010	10-LC-063	62.00	62.58	0.92	2.35
2010	10-LC-078	229.40	230.35	1.35	0.87
2010	10-LC-081	52.70	53.00	0.80	3.79
2010	10-LC-082	77.80	78.20	1.00	5.70
2010	10-LC-087	80.80	81.25	0.45	1.22
2010	10-LC-088	102.50	102.87	0.37	1.56
2010	10-LC-089	144.20	144.80	1.13	6.86
2010	10-LC-090	54.40	54.90	1.12	2.25
2010	10-LC-091	94.00	94.55	3.75	1.02
2010	10-LC-092	135.90	136.56	0.66	1.49
2010	10-LC-095*	40.10	40.90	2.45	1.03
2010	10-LC-096	238.54	239.00	0.88	2.17
2010	10-LC-099	170.86	171.16	1.04	3.56
2010	10-LC-101	155.00	155.30	1.69	2.77
2010	10-LC-110*	111.40	111.80	0.40	1.06
2010	10-NE-001	34.92	35.29	1.35	0.83

*intervals are strongly mineralized gash veins as opposed to Main Zone

** large weakly mineralized breccias

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%	Cu ppm
2011	11-523-002	68.00	68.39	0.39	2.51	108
2011	11-BZ-005	52.55	52.95	0.40	2.19	3600
2011	11-BZ-005	95.53	95.80	0.27	2.35	459
2011	11-BZ-010	44.00	44.50	0.50	3.41	18900
2011	11-BZ-010	44.50	45.00	0.50	1.02	18300
2011	11-BZ-010	47.45	47.85	0.40	5.65	4390
2011	11-BZ-010	47.85	48.40	0.55	0.64	1460
2011	11-BZ-010	48.40	49.05	0.65	0.86	25600
2011	11-BZ-010	49.05	50.15	1.10	2.25	3680
2011	11-BZ-010	62.00	62.30	0.30	0.83	16300

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%	Cu ppm
2011	11-BZ-010	62.80	63.80	1.00	3.28	18300
2011	11-BZ-010	63.80	64.75	0.95	1.45	6470
2011	11-BZ-010	64.75	65.75	1.00	3.21	7680
2011	11-BZ-010	65.75	66.05	0.30	0.52	8020
2011	11-BZ-010	66.05	66.80	0.75	0.61	33700
2011	11-BZ-014	58.25	58.55	0.30	3.28	1390
2011	11-BZ-016	46.40	46.70	0.30	0.62	2850
2011	11-BZ-016	47.40	47.76	0.36	6.83	23400
2011	11-BZ-016	48.36	48.74	0.38	0.54	3610
2011	11-BZ-017	68.40	68.76	0.36	3.36	4710
2011	11-BZ-019	97.55	97.85	0.30	1.79	587
2011	11-BZ-019	97.85	98.16	0.31	1.42	111
2011	11-BZ-019	98.16	98.46	0.30	0.6	257
2011	11-BZ-019	98.46	98.76	0.30	1.2	227
2011	11-BZ-019	99.54	99.84	0.30	5.8	1710
2011	11-BZ-019	122.90	123.20	0.30	2.45	481
2011	11-BZ-020	46.60	47.10	0.50	1.46	5480
2011	11-BZ-020	47.10	47.90	0.80	2.25	7660
2011	11-BZ-021	35.15	35.55	0.40	3.38	1980
2011	11-BZ-021	35.55	36.10	0.55	0.65	2630
2011	11-BZ-022	18.07	18.46	0.39	2.77	2440
2011	11-LC-008	241.58	242.00	0.42	5.94	151
2011	11-LC-009	248.85	249.25	0.40	3.01	890
2011	11-LC-009	261.25	261.65	0.40	1.18	281
2011	11-LC-032	156.72	157.06	0.34	2.16	2030
2011	11-LC-035	123.75	124.05	0.30	2.61	4260
2011	11-LC-036	185.00	185.48	0.48	1.05	455
2011	11-LC-036	185.48	185.80	0.32	10.35	1140
2011	11-LC-036	185.80	186.17	0.37	3.63	1270
2011	11-LC-043	112.40	112.75	0.35	0.7	21900
2011	11-LC-043	112.75	113.15	0.40	7.39	10800
2011	11-LC-046	123.00	123.43	0.43	2.18	294
2011	11-LC-054	193.66	194.00	0.34	2.56	28600
2011	11-LC-054	194.00	194.46	0.46	1.54	13200
2011	11-LC-054	194.46	194.90	0.44	1.71	13300
2011	11-LC-056	100.42	100.78	0.36	4.78	4810
2011	11-LC-056	101.12	101.45	0.33	1.1	15700
2011	11-LC-057	119.15	119.47	0.32	5.18	3550
2011	11-LC-057	119.47	119.77	0.30	2.76	2120
2011	11-LC-057	142.24	142.58	0.34	4.13	328
2011	11-LC-057	142.58	143.00	0.42	9.12	6500
2011	11-LC-057	143.00	143.60	0.60	1.44	6560
2011	11-LC-060	163.25	163.56	0.31	3.1	5910
2011	11-LC-060	163.56	163.86	0.30	12.5	5770
2011	11-LC-060	163.86	164.16	0.30	1.04	5700

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%	Cu ppm
2011	11-LC-060	164.16	164.46	0.30	0.61	11200
2011	11-LC-060	165.24	165.54	0.30	2.89	3360
2011	11-LC-066	92.25	92.55	0.30	8.7	9120
2011	11-LC-066	92.55	92.85	0.30	0.78	126
2011	11-LC-075	104.70	105.00	0.30	4.25	1340
2011	11-LC-078	70.63	70.95	0.32	3.16	743
2011	11-LC-078	71.25	71.65	0.40	0.95	180
2011	11-LC-082	105.42	105.75	0.33	4.04	16100
2011	11-LC-082	105.75	106.05	0.30	0.55	13000
2011	11-LC-084	49.05	49.50	0.45	0.63	426
2011	11-LC-084	49.50	49.80	0.30	2.59	1720
2011	11-LC-090	256.55	257.05	0.50	1.29	257
2011	11-LC-090	257.75	258.25	0.50	1.34	1820
2011	11-LC-092	98.50	99.10	0.60	2.32	1570
2011	11-LC-093	127.15	127.65	0.50	3.37	1300
2011	11-LC-094	190.95	191.25	0.30	10.4	1100
2011	11-LC-094	191.65	192.00	0.35	2.76	534
2011	11-LC-094	192.00	192.40	0.40	0.84	439
2011	11-LC-096	88.80	89.15	0.35	2.64	878
2011	11-LC-096	89.15	89.52	0.37	5.94	19900
2011	11-LC-097	106.63	107.00	0.37	1.36	672
2011	11-LC-097	107.00	107.55	0.55	1.29	2640
2011	11-LC-097	110.55	110.85	0.30	1.16	979
2011	11-LC-097	112.75	113.05	0.30	2.17	607
2011	11-LC-097	113.05	113.65	0.60	0.69	282
2011	11-LC-099	114.55	114.95	0.40	6.07	2160
2011	11-LC-102	92.25	92.55	0.30	0.98	11800
2011	11-LC-102	92.55	92.85	0.30	6.32	27200
2011	11-LC-103	96.75	97.05	0.30	5.98	9310
2011	11-LC-105	90.80	91.10	0.30	2.97	240
2011	11-LC-105	91.40	91.75	0.35	0.62	237
2011	11-LC-109	280.00	280.40	0.40	0.59	4270
2011	11-LC-109	280.40	280.70	0.30	4.15	18800
2011	11-LC-109	280.70	281.00	0.30	2.05	22800
2011	11-LC-115	228.90	229.25	0.35	1.47	951
2011	11-LC-116	296.95	297.50	0.55	0.53	775
2011	11-LC-116	297.50	297.80	0.30	3.9	3420
2011	11-LC-119	304.05	304.45	0.40	2.04	2940
2011	11-LC-119	304.45	304.85	0.40	1.03	9170
2011	11-LC-120	341.10	341.50	0.40	1.26	14200
2011	11-LC-120	341.50	341.80	0.30	2.62	10200

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U308%
2012	12-LC-007	369.70	374.70	5.00	0.33
2012	12-LC-010	263.10	269.40	6.20	0.12
2012	12-LC-013	73.90	76.20	2.30	0.13
2012	12-LCE-001	109.90	110.70	0.80	0.44
2012	12-LCE-002	131.20	131.90	0.70	0.15
2012	12-LCM-003	335.50	335.80	0.30	0.32
2012	12-LCM-006	354.80	355.10	0.30	0.43
2012	12-LCM-014	276.90	277.50	0.60	0.23
2012	12-LCSW-001	93.70	94.40	0.70	0.35
2012	12-LCSW-002	146.10	149.10	3.00	0.80

Estimated true width was calculated using the orientation of the drill hole, the apparent thickness of the mineralization and the dip of the mineralized zone on that section.

From, To, and core length measurements are rounded to nearest 0.1 decimal place

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)	Cu (%)
2012	12-J4-001	95.60	95.90	0.30	1.33	0.10
2012	12-J4-001	99.80	100.10	0.30	1.85	0.07
2012	12-J4-002	99.50	99.80	0.30	1.66	0.05
2012	12-J4-003	87.70	89.10	1.40	1.91	0.30
2012	12-J4-003	118.80	119.20	0.40	3.76	0.06
2012	12-J4-004	182.70	183.00	0.30	1.50	0.28
2012		91.90	92.70	0.80	1.32	0.31
2012	12-J4-005	113.20	113.50	0.30	0.53	0.08
2012		122.20	122.50	0.30	1.75	0.03
2012		68.40	69.30	0.90	0.64	0.29
2012	12-J4-008	85.10	85.40	0.30	0.87	0.10
2012	12-J4-009	111.60	112.10	0.50	1.14	0.22
2012	12-J4-012	79.80	84.60	4.80	0.34	1.24
2012	includes	79.80	81.50	1.70	0.92	2.89
2012	12-J4-015	134.10	135.00	0.90	2.56	0.20
2012	12-J4-016	189.60	190.80	1.20	0.83	0.45
2012	12-J4-19	155.90	157.00	1.10	0.99	0.32
2012	12-J4-021	96.30	97.40	1.10	1.71	0.36
2012	12-J4-025	149.40	149.70	0.30	0.97	0.28
2012	12-J4-028	87.00	87.90	0.90	3.91	0.01
2012	12-J4-029	153.50	159.60	6.10	2.42	0.25
2012	12-J4-030	299.80	304.40	4.60	0.85	0.15
2012	12-J4-031	279.20	280.90	1.70	1.16	0.17
2012	12-J4-033	281.60	281.90	0.30	3.60	0.15
2012	12-J4-034	390.60	392.10	1.50	1.24	0.35
2012		99.00	100.50	1.50	2.86	0.19
2012	12-J4-042	167.70	169.70	2.00	1.17	0.54
2012	12-J4-045	156.50	157.90	1.40	0.67	0.66

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)	Cu (%)
2012	(12-J4-050)	114.90	117.50	2.60	1.77	0.07
2012		122.60	123.50	0.90	1.24	0.20
2012		139.60	142.40	2.80	0.58	0.28
2012	12-J4-062	119.10	120.70	1.60	2.85	0.66
2012	includes	119.60	119.90	0.30	15.40	1.64
2012	12-774-001	74.30	74.70	0.40	0.51	0.07
2012	12-774-010	96.10	97.40	1.30	0.66	0.03
2012		113.80	114.90	1.10	0.72	0.10
2012	12-774-011	373.90	374.20	0.30	0.57	0.08
2012	12-774-016	51.90	52.60	0.70	0.97	0.03
2012	12-PL-001	52.20	53.00	0.80	0.63	0.03
2012	12-J2E-002	10.00	11.00	1.00	0.53	0.07
2012	12-HOT-004	84.10	87.10	3.00	0.85	0.07
2012	12-HOT-005	132.80	133.40	0.60	0.53	0.18

All samples were subject to ICP analysis at the SRC. Results >1000 ppm U were re-analysed by SRC's U₃O₈ assay; 1 ppm = 1g/t; 10000 ppm = 1%; Conversion to U₃O₈% = ppm x 0.01179%.

11 exploratory diamond drill holes did not intersect mineralization.

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)	Cu (%)
2013	13-ML-001	90.20	94.50	4.30	0.46	0.48
2013	Includes	90.20	91.40	1.20	1.42	0.64
2013	13-J1-002	60.10	60.50	0.40	1.06	0.28
2013	13-J1-003	77.20	78.30	1.10	0.56	0.05
2013	Includes	77.20	77.70	0.50	1.31	0.09
2013	13-J1-005	16.60	17.00	0.40	0.17	0.02
2013	13-J1-005	114.80	115.10	0.30	0.15	0.05

All samples are subjected to ICP1 Analysis by Saskatchewan Research Council Geoanalytical Laboratories SRC, Canada.

ICP1 results >1,000 ppm U are subjected to SRC U₃O₈ assay

ICP1 results for Cu are reported by SRC in parts per million (ppm). 1 ppm= 1gm/t, 10,000 ppm= 1%

Intervals include ICP U analysis in ppm converted to U₃O₈%. Conversion to U₃O₈% = ppm x 0.01179.

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)
2015	15-DP-001*	21.60	22.60	1.00	0.05
2015	15-DP-002	38.10	39.90	1.80	0.02
2015	15-DP-002	73.10	78.00	4.90	0.07
2015	Includes	75.50	77.60	2.10	0.14
2015	15-DP-005	27.90	35.90	8.00	0.17
2015	Includes	27.90	29.80	1.90	0.37
2015	and Includes	34.00	35.00	1.00	0.42
2015	15-DP-006	35.50	42.20	6.70	0.18
2015	Includes	35.50	36.70	1.20	0.35

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)
2015	and Includes	39.10	41.10	2.00	0.34
2015	15-DP-009	27.80	31.30	3.50	0.88
2015	Includes	28.30	29.60	1.30	2.34
2015	Includes	49.30	50.40	1.10	0.77

Hole 15-DP-001 lost at 23.5m due to drilling conditions.

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

All samples are subjected to ICP1 Analysis by SRC. ICP1 results >1,000 ppm U are subjected to SRC U3O8 assay

ICP1 results for Cu are reported by SRC in parts per million (ppm). 1 ppm= 1gm/t, 10,000 ppm= 1%

Intervals include ICP U analysis in ppm converted to U3O8%. Conversion to U3O8% = ppm x 0.01179.

Year	Drillhole ID	From (m)	To (m)	Interval Core length (m)	U ₃ O ₈ (%)	Cu (%)
2022	22-DP-002	57.83	58.47	0.64	1.10	0.03
2022	and	101.71	102.42	0.71	0.41	0.02
2022	22-DP-006	75.00	76.00	1.00	0.34	0.02
2022	22-DP-008	141.73	142.36	0.63	0.61	0.05
2022	22-DP-009	166.93	167.43	5.00	0.24	0.01
2022	and	171.57	172.08	0.51	0.32	0.03
2022	and	175.34	175.87	0.53	0.29	0.02
2022	22-DP-010	152.46	155.15	1.54	1.40	0.34
2022	includes	153.40	154.00	0.60	3.40	0.56
2022	22-DP-012	288.95	289.53	0.58	0.54	0.02
2022	22-J4W-001	55.65	56.30	0.65	0.40	0.07
2022	22-J4W-003	79.87	80.59	0.72	0.60	0.02

Core samples submitted to SRC for assay via ICP1, ICP2, and U3O8

ICP1 results >1,000 ppm U are subjected to SRC % U3Os assay

ICP1 results for Cu are reported by SRC in parts per million (ppm). 1 ppm= 1gm/t, 10,000 ppm= 1%

All From, To, and core length interval measurements are metres (m) down-hole. True widths are yet to be determined.

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U ₃ O ₈ (%)	Cu (%)
2023	23-LC-002	125.60	126.00	0.40	0.42	0.01
2023	23-LC-005	165.30	165.60	0.30	0.79	0.02
2023	23-LC-005	218.00	219.60	1.60	7.54	0.06
2023	23-LC-006	72.80	76.70	3.90	0.19	0.65
2023	Includes	73.90	74.80	0.90	0.54	1.31
2023	23-LC-008	125.30	127.10	1.80	0.25	0.09
2023	23-LC-008	128.20	130.60	2.40	0.23	0.03
2023	23-LC-009	76.90	77.50	0.60	0.30	0.06
2023	23-LC-009	175.00	175.70	0.70	0.28	0.01
2023	23-LC-010	100.20	101.00	0.80	0.39	0.01
2023	23-LC-010	106.00	106.70	0.70	0.40	0.01
2023	23-LC-011	81.30	81.70	0.40	0.33	0.10
2023	23-LC-011	170.50	172.30	1.80	0.29	0.23
2023	Includes	171.20	171.70	0.50	0.73	0.07

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U ₃ O ₈ (%)	Cu (%)
2023	23-LC-012	234.60	237.00	2.40	0.57	0.22
2023	Includes	234.60	235.70	1.10	1.21	0.40
2023	23-LC-013A	282.90	283.80	0.90	0.68	0.00
2023	23-LC-014	139.90	140.40	0.50	0.30	0.02
2023	23-LC-015	229.80	230.30	0.50	2.05	0.07
2023	23-LC-015	299.10	300.00	0.90	1.93	0.03
2023	Includes	299.10	299.60	0.50	2.88	0.02
2023	Includes	34.50	35.00	0.50	0.29	2.12
2023	23-LC-018	213.00	213.60	0.60	0.29	0.01

All From, To and Interval measurements are metres (m) down-hole. True widths are yet to be determined.

Samples were sent to the Saskatchewan Research Council (SRC) lab facilities for U3O8 geochemical analysis.

The SRC is an ISO/IEC 17025/2005 and Standards Council of Canada certified analytical independent laboratory.

Samples returning 100 ppm U or greater were analyzed for wt% U3O8.

Where consecutive samples are above 0.02% U3O8, intersections are weighted averages of composited samples.

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U ₃ O ₈ (%)	Cu (ppm)
2024	BLZ-DD-034	94.00	94.50	0.50	0.21	884
2024	BLZ-DD-034	94.50	95.00	0.50	0.99	802
2024	BLZ-DD-034	95.00	95.50	0.50	0.26	220
2024	BLZ-DD-034	164.00	164.50	0.50	0.17	744
2024	EEX-DD-052	345.50	346.00	0.50	0.25	214
2024	EEX-DD-052	374.50	375.00	0.50	0.18	56
2024	EEX-DD-053	111.20	111.70	0.50	0.36	3400
2024	EEX-DD-053	306.40	306.90	0.50	0.82	1260
2024	HOT-DD-008	123.20	123.70	0.50	0.40	2410
2024	HOT-DD-010	290.20	290.60	0.40	0.20	1900
2024	J4R-DD-085	111.80	112.30	0.50	5.85	1250
2024	J4R-DD-085	428.10	428.60	0.50	0.65	125
2024	J4R-DD-085	429.10	429.60	0.50	1.52	537
2024	J4R-DD-085	457.40	457.90	0.50	0.63	75
2024	J4R-DD-086	35.50	36.00	0.50	0.62	58
2024	J4R-DD-086	393.50	394.00	0.50	3.92	591
2024	J4R-DD-086	394.00	394.50	0.50	0.41	2320
2024	J4R-DD-089	404.10	404.40	0.30	3.19	3970
2024	J4R-DD-089	404.40	404.50	0.10	2.89	4540
2024	J4R-DD-090	132.30	132.70	0.40	0.45	232
2024	ML-DD-009	102.50	103.00	0.50	0.77	6050
2024	MZ-DD-175	74.60	75.10	0.50	0.43	698
2024	MZ-DD-175	78.60	79.10	0.50	0.33	385
2024	MZ-DD-175	370.60	371.10	0.50	0.44	126
2024	MZ-DD-175	372.70	373.20	0.50	0.46	251
2024	MZ-DD-176	331.00	331.50	0.50	0.11	144
2024	PL-DD-030	262.50	263.00	0.50	1.62	1330

Year	Drillhole ID	From (m)	To (m)	Interval Core Length (m)	U₃O₈ (%)	Cu (ppm)
2024	PL-DD-030	263.00	263.50	0.50	0.97	386
2024	WEX-DD-079	40.30	40.80	0.50	0.11	21100

All From, To and Interval measurements are metres (m) down-hole. True widths are yet to be determined.