Future Mineral Resource Potential of the Revell Site

NWMO-TR-2024-03

January 2024

Alba Geosolutions and Consulting



Nuclear Waste Management Organization 22 St. Clair Avenue East, 6th Floor

22 St. Clair Avenue East, 6th Floor Toronto, Ontario M4T 2S3 Canada

Tel: 416-934-9814 Web: www.nwmo.ca

Future Mineral Resource Potential of the Revell Site

NWMO-TR-2024-03 January 2024

Dr Nicholas WilsonAlba Geosolutions and Consulting

This report has been prepared under contract to NWMO. The report has been reviewed by NWMO, but the views and conclusions are those of the authors and do not necessarily represent those of the NWMO.

All copyright and intellectual property rights belong to NWMO.

Document History

Title:	Future Mineral Resou	ırce Potential of the Re	vell Site								
Report Number:	NWMO-TR-2024-03										
Revision:	R000	Date:	January, 2024								
ALBA GEOSOLUTIONS AND CONSULTING											
Authored by:	Nicholas Wilson (ALBA GEOSOLUTIONS AND CONSULTING)										
Reviewed by:	Stefan Markovic (NW	MO)									
Accepted by:	Andy Parmenter (NW	MO)									

	Revi	sion Summary
Revision Number	Date	Description of Changes/Improvements
R0	2024-01	INITIAL ISSUE

ABSTRACT

Title: Future Mineral Resource Potential of the Revell Site

Report No.: NWMO-TR-2024-03 **Author(s):** Dr. Nicholas Wilson

Company: Alba Geosolutions and Consulting

Date: January 2024

The Initial Borehole Drilling and Testing project in the Wabigoon Lake Ojibway Nation (WLON) – Ignace Area, Ontario is part of Phase 2 Geoscientific Preliminary Field Investigations of the Nuclear Waste Management Organization's (NWMO) Adaptive Phased Management Site Selection Phase. This project involves the drilling and testing of several deep boreholes at the Revell Site, located within the northern portion of the Revell batholith. This report presents a review and interpretation of the lithogeochemical data set for 93 samples that were collected from three of these boreholes (IG_BH01, IG_BH02, and IG_BH03) in order to:

- 1) Evaluate the mineral resource potential of the Revell Site,
- 2) Evaluate the appropriateness of lithogeochemical analyses to assess mineral potential of the Revell Site, and
- 3) Interpret the lithogeochemistry data, to evaluate the potential prospectivity of the Revell Site for mineral resources at any point in the future based on current knowledge of minerals of economic importance.

Samples were selected for lithogeochemical analyses by NWMO from all 3 boreholes, across all rock types and alteration facies, and were submitted to Activation Laboratories Ltd. (ActLabs) for analysis using industry standard INAA, FUS-ICP/MS, and TD-ICP methods to determine element concentrations. Duplicate analyses, using different techniques, were consistent for elements over a range of concentrations.

The lithogeochemistry data set was screened by removing analyses at detection levels, averaged, and compared to published average upper crust concentrations which provided a background trend for the data. The data set was averaged for all samples, felsic samples, and for 'amphibolite' mafic samples. Only 2 out of 46 elements (Li and Cr) for the all samples data set were present at levels (only 2 - 3 times) above average upper crust values; all other elements were inconsistently detected, or at or below average upper crust background values. Only 1 element (Li) was present at levels (<3 times) above average upper crust values from the felsic data set but at concentrations orders of magnitude below economic enrichment levels. The 'amphibolite' mafic samples occur as xenoliths or dikes in the pluton, and only 3 elements (Cr, Ni, and Li) were above (up to 5 - 12 times; average 5 - 7 times) average upper crust background values. As the amphibolite units are not volumetrically significant (occur as approximately 2 % of the cored lithology), do not show significant enrichment (are orders of magnitude below economic enrichment levels), they do not have future mineral potential.

Direct comparison with adjacent plutons, of similar age and composition, indicates that the northern portion of the Revell batholith, at the Revell Site, has lower overall concentrations for almost all elements analyzed. The lack of any anomalous element concentrations, above a few times average upper crust levels, and the fact that most samples have background-level upper crust values, indicates that the Revell Site has extremely low mineral potential and that is unlikely to change even if we look tens to hundreds of thousands of years into the future.

TABLE OF CONTENTS

		Page
ABSTRACT	Гііі	
1.	INTRODUCTION AND OBJECTIVES	1
2.	BACKGROUND INFORMATION	3
2.1 2.2 2.3 2.4	Geological SettingRock Types and Alteration	6
3.	DATA PROVIDED AND REVIEWED	13
3.1 3.2 3.3	Methods and Approach Duplicate Analyses Mineral Potential	15
4.	DATA INTERPRETATION	17
4.1 4.2 4.3 4.4 4.5	All Samples Felsic Samples 'Amphibolite' Mafic Samples Alteration Comparison to other Plutons	17 20 22
5.	DISCUSSION AND CONCLUSIONS	24
REFERENC	CES	27
Appendix '	1 Lithogeochemistry Data for IG_BH01, IG_BH02, and IG_BH03	
Appendix 2 Logged Lith	2 Screened Lithogeochemistry Data used in this Report by Sample	Group and

LIST OF TABLES

Page
Table 1. Typical ore grade and tonnages for VMS deposits (Barrie and Hannington, 1997)9 Table 2. Suite of elements analyzed and detection limits. Lithium was only analyzed in IG_BH02 and IG_BH03 samples. There is some variation in the element suite analyzed between boreholes and the elements analyzed for each borehole are summarized in
Appendix 1
Table 4. Summary of the background average upper crust elemental concentrations, detection limits, and minimum, average, and maximum values for the single, felsic and amphibolite data set. Filled green cells (average columns) indicate elements that are above average upper crust averages; red filled cells indicate elements that have only a few analyses; orange filled cells indicate elements that are equivalent to average upper crust values, and unfilled cells are below average upper crust values
LIST OF FIGURES Page
Figure 1. Location map of the Revell Site (blue oval) within the northern portion of the Revell batholith. The map also shows the surface collar locations of all six deep boreholes drilled at the site. Rocks from IG_BH01, IG_BH02, and IG_BH03 were sampled for this lithogeochemistry study (modified from Parmenter et al., 2020)
modified from Parmenter et al., (2020)
visually identified as an amphibolite sample, but its major oxide geochemical composition is consistent with a felsic lithology and was reclassified as such in this report
Figure 4. Current mineral claims in the area are located on the western, northern, and northeastern edges of the Revell batholith in the surrounding greenstone belts, or to the

southeast, over 20 km from the Revell Site (from Parmenter et al., 2020). Red circles	
identify locations discussed in the text.	11
Figure 5. Geochemical samples, location of thesis studies, geochronology samples, gravity, and	d
magnetic susceptibility measurements over the Revell batholith and surrounding area (from	m
Parmenter et al., 2020).	12
Figure 6. Cross plot of element concentrations (PPM) with duplicate analyses from INAA and	
FUS-MS methods. There is excellent correlation between most of the elements and data	
points. Elements have been scaled to fit on a single figure (Cr/100, Co/5, and Ce/30)	15

1. INTRODUCTION AND OBJECTIVES

The Initial Borehole Drilling and Testing project in the Wabigoon Lake Ojibway Nation (WLON) – Ignace Area, Ontario is part of Phase 2 Geoscientific Preliminary Field Investigations of the Nuclear Waste Management Organization's (NWMO) Adaptive Phased Management Site Selection Phase. This project includes the drilling and testing of six deep boreholes at the Revell Site, as well as additional on-going studies, located within the northern portion of the Revell batholith (Figure 1). Information from the first three boreholes (IG_BH01, IG_BH02, and IG_BH03) was available at the time of writing this report.

The first three boreholes were drilled in order to characterize the subsurface geology of the northern portion of the Revell batholith. All three were drilled to ~1000 m along their length. IG_BH01 was oriented vertically, whereas IG_BH02 and IG_BH03 were drilled at an inclination of 70° towards the southwest and south, respectively. Summary descriptions of the geological findings from each borehole are included in geological integration reports for IG_BH01 (NWMO and Golder, 2022), IG_BH02 (NWMO, 2022a), and IG_BH03 (NWMO, 2022b). An updated bedrock map and accompanying report provides supplemental geological information for the regional area surrounding the Revell Site (Parmenter et al., 2020; Figure 1).

This report focuses specifically on the future mineral potential of the Revell Site, with a forward projection of up to a million years into the future based on a current evaluation of mineral potential within the rocks. As part of the site characterization process, 93 lithogeochemical samples were collected from the borehole samples to:

- 1) Evaluate the mineral resource potential of the Revell Site,
- Evaluate the appropriateness of lithogeochemical analyses to assess mineral potential of the Revell Site, and
- 3) Interpret the lithogeochemistry data and relevant supporting documents, to evaluate the potential prospectivity of the Revell Site for mineral resources at any point in the future based on current knowledge of minerals of economic importance.

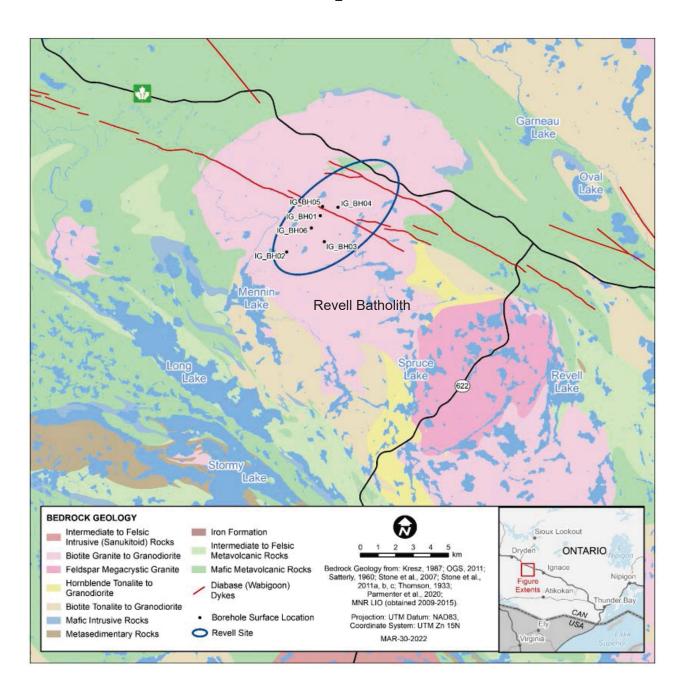


Figure 1. Location map of the Revell Site (blue oval) within the northern portion of the Revell batholith. The map also shows the surface collar locations of all six deep boreholes drilled at the site. Rocks from IG_BH01, IG_BH02, and IG_BH03 were sampled for this lithogeochemistry study (modified from Parmenter et al., 2020).

2. BACKGROUND INFORMATION

2.1 Geological Setting

The regional geological setting has been described in Parmenter et al. (2020) which covers the local and regional geology in significant detail. The summary below has been compiled from that report.

The topography of the regional area around the Revell batholith (Figure 1) is characterized by a broadly rolling surface of Canadian Shield bedrock (Thurston, 1991). Several areas along the northern and northwestern margin of the Revell batholith are covered by extensive overburden. In proximity to the boreholes at the Revell Site the bedrock is generally either exposed or covered by only a thin veneer of Quaternary cover.

The Revell batholith is situated in the northwestern part of the Superior Province of the Canadian Shield – a stable craton created from a collage of ancient plates and accreted juvenile arc terranes that were progressively amalgamated over a period of more than 2 billion years (Figure 2), and now form the core of the North American continent. The Superior Province has been divided historically into various regionally extensive east-northeast- trending subprovinces based on rock type, age, and metamorphism (Figure 2; Thurston, 1991). More recently, the Superior Province has been subdivided into lithotectonic terranes, defined as tectonically bounded regions with characteristics distinct from adjacent regions prior to their accretion into the Superior Province (Percival and Easton, 2007; Stott et al., 2010). The Revell Regional Area is situated in the southcentral part of the Western Wabigoon terrane, adjacent to the boundary with the Marmion terrane (Figure 2).

A summary of the Archean and Proterozoic geological events that have shaped the bedrock in the Revell Regional Area is shown in Figure 2. The western Wabigoon terrane, interpreted to represent a volcanic island arc, is predominantly composed of two main groups of rock. This includes ca. 2.745 to 2.711 Ga supracrustal rocks, comprising Archean mafic to intermediate to felsic metavolcanic rocks and subordinate sedimentary rocks distributed in greenstone belts, and ca. 2.70 to 2.67 Ga rocks of granitoid affinity predominantly consisting of felsic plutonic rocks. These two major rock groups are a common characteristic of granite-greenstone belts and subprovinces, across the entire western Superior Province.

The supracrustal rocks distributed to the southwest of the batholith belong to the Bending Lake greenstone belt and those distributed to the northeast of the batholith belong to the Raleigh Lake greenstone belt. These greenstone belts, as well as the additional supracrustal rocks wrapping around the northern boundary of the batholith (Figure 2), represent contiguous parts of the Kakagi Lake-Savant Lake greenstone belt that underlies the entire western Wabigoon terrane.

Four main rock units are identified in the supracrustal rock group: mafic metavolcanic rocks, intermediate to felsic metavolcanic rocks, metasedimentary rocks, and mafic intrusive rocks (Figure 1). Sedimentation within the supracrustal rock assemblage was largely synvolcanic, although sediment deposition in the Bending Lake area may have continued past the volcanic period (Stone, 2010a; Stone, 2010b; Stone et., 2011). All supracrustal rocks are affected, to varying degrees, by penetrative brittle-ductile to ductile deformation under greenschist- to amphibolite-facies metamorphic conditions (Blackburn and Hinz, 1996; Stone et al., 1998). In some locations, primary features, such as pillow basalt or bedding in sedimentary rock units are preserved, in other locations, primary relationships are completely masked by penetrative

deformation. Uranium-lead (U-Pb) geochronological analysis of the supracrustal rocks produced zircon ages that range between 2734.6±1.1 Ma and 2725±5 Ma (Stone et al. 2010b).

The Revell batholith is roughly rectangular in shape, trends northwest, is approximately 40 km in length, 15 km in width, and covers an area of approximately 455 km² (Figure 2). Szewczyk and West (1976) interpreted this batholith to be a sheet-like intrusion that is approximately 1.6 km thick. Based on recent geophysical modelling, the batholith has a relatively flat base that extends to depths of nearly 4 km in some regions (Mushayandebvu et al., 2020).

Three main suites of plutonic rock are recognized in the Revell batholith (Figure 1), including, from oldest to youngest: a Biotite Tonalite to Granodiorite suite, a Hornblende Tonalite to Granodiorite suite, and a Biotite Granite to Granodiorite suite. Plutonic rocks of the Biotite Tonalite to Granodiorite suite occur along the southwestern and northeastern margins of the Revell batholith. The principal type of rock within this suite is a white to grey, medium-grained, variably massive to foliated or weakly gneissic, biotite tonalite to granodiorite. One sample of foliated and medium- grained biotite tonalite produced a U-Pb zircon age of 2734.2±0.8 Ma (Stone et al., 2010b). The Hornblende Tonalite to Granodiorite suite occurs in two irregularly shaped zones surrounding the central core of the Revell batholith. Rocks of the Hornblende Tonalite to Granodiorite suite range compositionally from tonalite through granodiorite to granite and include significant proportions of quartz diorite and quartz monzodiorite. One sample of coarse-grained grey mesocratic hornblende tonalite produced a U-Pb zircon age of 2732.3±0.8 Ma (Stone et al., 2010b). Rocks of the Biotite Granite to Granodiorite suite underlie most of the northern, central, and southern portions of the Revell batholith. Rocks of this suite are typically coarse-grained, massive to weakly foliated, and white to pink in colour. The Biotite Granite to Granodiorite suite ranges compositionally from granite through granodiorite to tonalite. This suite includes the ovalshaped potassium-feldspar megacrystic granite body in the central portion of the Revell batholith (Figure 1). One sample of coarse-grained, pink, massive potassium feldspar megacrystic biotite granite produced a U-Pb zircon age of 2694.0±0.9 Ma (Stone et al., 2010b).

A large west-northwest trending mafic dyke interpreted from aeromagnetic data and observed during detailed mapping to be approximately 15-20 m wide, extends across the northern portion of the Revell batholith (Figure 1; Golder and PGW, 2017). This dyke is associated with several similarly orientated mafic dykes that stretch across the northern portion of the Revell batholith and into the surrounding greenstone belts. These mafic dykes are interpreted to be part of the Wabigoon dyke swarm and produced a U-Pb age of 1887±13 Ma (Stone et al., 2010b), indicating that these mafic dykes are Proterozoic in age, and based on surface measurements, are assumed to be subvertical (Golder and PGW, 2017).

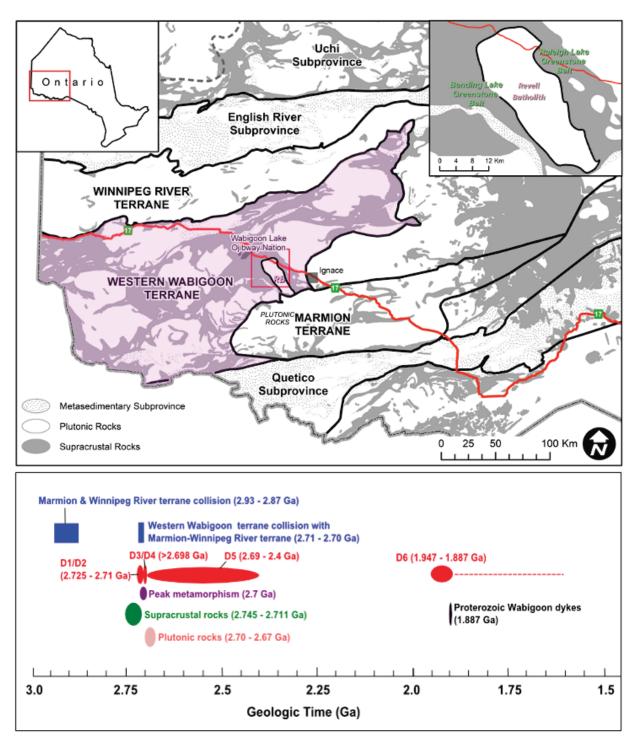


Figure 2. Upper. Geological setting of the Superior Province in northwestern Ontario (after Thurston, 1991) around the Revell Regional Area (red outline on main map and insert map) and the Revell batholith (RB). The Winnipeg River, Marmion and Western Wabigoon terranes (pink) are part of the Wabigoon subprovince. Lower. Chart summarizing the major Archean and Proterozoic geological events for the Revell Regional Area. Both figures are modified from Parmenter et al., (2020).

2.2 Rock Types and Alteration

The location of boreholes IG_BH01, IG_BH02, and IG_BH03 are shown in Figure 1. Summary descriptions of the geological findings from each borehole are included in geological integration reports for boreholes IG_BH01 (NWMO and Golder, 2022), IG_BH02 (NWMO, 2022a), and IG_BH03 (NWMO, 2022b). These reports present lithological, geochemical, alteration, and structural descriptions of the boreholes based on the integration of information from visual core logging and borehole geophysical logging information, petrography (NWMO, 2022c; NWMO, 2022d) and lithogeochemical analyses.

The range of rock types present in the studied core samples are shown in Figure 3. The dominant rock type encountered is a biotite granodiorite-tonalite to biotite tonalite, which represents more than 97 % of the recovered core by length. Several subordinate felsic phases, and a distinct 'amphibolite' mafic rock discussed in more detail below, comprise the remainder of the recovered core. Figure 3 also illustrates the range of alteration types identified during core logging. Alteration types identified include potassic, hematization, silicification, chloritization, bleaching, sericitization, carbonatization, and argillization (i.e., formation of clay minerals).

The intensity of alteration is variable between the boreholes; most of borehole IG_BH01 is not altered, borehole IG_BH02 has up to 74.8% of the cored interval with some alteration, and borehole IG_BH03 is altered over 62.3% of the cored interval. The order of prevalence of alteration between each of the borehole is also varied but most cores show dominantly potassic, silicification, hematization and chloritization.

Alteration levels in the logging reports (NWMO and Golder, 2022; NWMO, 2022a; and NWMO, 2022b) were classified as slightly altered (A2; confined to fractures), moderately altered (A3; penetrates the wall rock and creates an alteration halo), and highly altered (A4; pervasive). In borehole IG_BH01 only 8.45% of the cored interval was considered slightly altered, 1.4% was considered moderately altered and no occurrences of highly altered samples were documented; in borehole IG_BH02, 74.8% of the cored interval was slightly altered, 32.6% was moderately altered, and 1.8% was assigned as highly altered; in borehole IG_BH03, 47.8% of the cored interval was slightly altered, 13.4% was moderately altered and 1.1% was assigned as highly altered. Overall, alteration is present at low levels across most of the cored intervals and no clearly defined 'patterns' of alteration were documented during the core logging.

This alteration is most likely related to low-level hydrothermal fluid flow during granitoid emplacement and regional deformation. In contrast, the abundance of hematization less than 50 m below the surface (e.g., NWMO and Golder, 2022) could be related to recent surficial processes; hematization is also present at higher levels in the subsurface around mafic rock samples (likely due to the elevated Fe in these samples; 9.2wt% Fe₂O₃ in mafic samples vs 2.2wt% Fe₂O₃ in felsic samples; see Appendix 2).

2.3 'Amphibolite' Mafic Rocks

The field term 'amphibolite' was used in the field mapping and during the initial visual core logging process for the small amount of mafic and related rocks found in the cores from boreholes IG_BH-01, IG_BH-02 and IG_BH-03 (Parmenter et al., 2020). For consistency with earlier NWMO reports, this term 'amphibolite' is used in this report for these mafic rocks. These amphibolite rocks are present in approximately 2 % of the recovered core, by length.

Regardless of this prior usage, these rocks are better described as mafic to ultramafic rocks and based on sample petrography (NWMO 2022c, 2022d), these amphibolites have only been affected by greenschist facies metamorphism (actinolite-albite-chlorite assemblages), like their host felsic rocks, and are not metamorphic amphibolites. Petrography indicates (NWMO, 2022d) the amphibolites are composed of dominantly amphibole (86-88%) with lesser biotite (5-10%), plagioclase (4-5%), epidote (2-4%), and minor chlorite (0.5-1%). This is different to metamorphic amphibolite that contains dominantly plagioclase and hornblende.

The amphibolite sample set can be divided (see Appendix 2) into 2 groups based on total rare earth element content, as well as P_2O_5 contents, with one group (n=7) characterized by total rare earth contents of 230 to 372 ppm and P_2O_5 contents of 0.32 to 0.49 wt.%. As well, this group typically has higher Cr, Ni and V compared to the other amphibolites, but not always.

It is unclear whether the amphibolite units are dykes or xenoliths from the surrounding greenstone belts, or a combination, because primary contact relationships such as chilled margins are obscured by penetrative deformation localized along the amphibolite contacts (NWMO and Golder, 2022).

In terms of rock classification, the amphibolites classify as tholeiitic to alkaline basalts to basaltic trachyandesites, are alkaline, and plot as high-magnesium tholeiites or komatiitic basalts on a Jensen (1976) discrimination diagram. This suggests that the amphibolite may represent mafic dykes that intruded into the granodiorite- tonalite synchronous with either granitoid emplacement or regional deformation (NWMO and Golder, 2022), but they have undergone different degrees of fractionation. A more thorough discussions of the range of encountered rock types, including classification based on major oxide data and summaries of the REE data are given in the summary reports for the three boreholes (NWMO and Golder, 2022; NWMO, 2022a; and NWMO, 2022b).

Therefore, the field, petrographic, major and REE data suggests that the amphibolites may represent mafic dykes that intruded into the granodiorite- tonalite synchronous with either granitoid emplacement or regional deformation (NWMO and Golder, 2022).



Figure 3. Examples of core sample rock and alteration types. Sample IG_BH03_LG013_B was visually identified as an amphibolite sample, but its major oxide geochemical composition is consistent with a felsic lithology and was reclassified as such in this report.

2.4 Mineral Deposits and Leases

Volcanogenic massive sulphide (VMS) deposits are common in this part of northern Ontario and mineralization is typically hosted by bimodal mafic volcanic-dominated oceanic rift arc and bimodal felsic-dominated siliciclastic rocks in continental back-arc settings (Galley et al., 2007). VMS deposits are major sources of Zn, Cu, Pb, Ag, and Au, and significant sources of Co, Sn, Se, Mn, Cd, In, Bi, Te, Ga, and Ge (Galley et al., 2007).

VMS mineralization is hosted within the greenstone belts themselves and not within intrusive rocks associated with these belts. Nonetheless, these intrusions might supply heat source and/or

metals to the VMS systems, but are not mineralized (Galley et al., 2007). Due to the large-scale hydrothermal systems associated with VMS mineralization, VMS districts are commonly characterized by extensive semi-conformable zones of hydrothermal alteration that have well defined 'mappable' alteration patterns (Galley et al., 2007). Average grades for VMS deposits are shown in Table 1 with ore grades typical being at the percent level for most elements (Barrie and Hannington, 1997).

Table 1. Typical ore grade and tonnages for VMS deposits (Barrie and Hannington, 1997).

TABLE 2. Grade and Tonnage for VMS Types by Time Periods

	n	Total tonnes in MT	Average Tonnes in MT	Average Cu grade in wt %	Average Pb grade in wt %	Average Zn grade in wt %	Average Au grade in g/t	Average Ag grade in g/t
MAFIC								
Archean	1	1.5	1.5	(1.5)1		(4.15)		
Early Proterozoic	3	1.9	0.6	(4.83)		(0.34)	(1.72)	(5.23)
Middle and Late Proterozoic	0	0.0	0.0	(1.00)		(0.51)	(1.72)	(3.23)
Early Phanerozoic	23	60.0	2.6	1.77	(0.05)	2.86	(3.02)	(18.0)
Late Phanerozoic	35	115.9	3.3	2.00	(0.10)	(1.13)	(1.74)	(25.2)
BIMODAL-MAFIC								
Archean	121^{2}	606.7	0.5	1.66	0.42	5.04	1.32	38.6
Early Proterozoic	73	410.2	5.6	2.20	0.98	4.32	1.47	28.7
Middle and Late Proterozoic	17	24.5	1.4	2.06	(0.97)	2.64	(1.42)	(37.9)
Early Phanerozoic	54	278.8	5.2	1.93	(0.35)	3.02	2.40	44.4
Late Phanerozoic	19	130.6	6.9	1.74	(0.43)	2.54	(1.60)	28.4
MAFIC-SILICICLASTIC								
Archean	2	1.4	(0.7)	(1.37)		(1.46)		(42.5)
Early Proterozoic	7	159.8	(22.8)	(2.38)	(0.01)	(1.27)	(0.49)	(25.7)
Middle and Late Proterozoic	16	307.4	19.2	1.68	(2.91)	(2.44)	(0.51)	(17.4)
Early Phanerozoic	25	256.3	10.3	1.46	(1.73)	4.21	0.80	(33.2)
Late Phanerozoic	63	519.4	8.2	1.81	(0.02)	0.80	1.00	(12.4)
BIMODAL-FELSIC								
Archean	24	170.2	7.1	1.09	1.23	6.23	0.83	125.2
Early Proterozoic	42	222.9	5.3	1.05	0.72	4.45	1.65	49.3
Middle and Late Proterozoic	14	68.0	4.9	1.53	0.85	4.07	1.47	109.2
Early Phanerozoic	82	375.0	4.6	1.53	2.50	6.69	2.63	85.8
Late Phanerozoic	93	472.6	5.1	1.64	1.52	5.29	2.04	115.7
BIMODAL-SILICICLASTIC								
Archean	2	0.6	0.3	(1.23)	(1.67)	(4.60)	(1.36)	(37.7)
Early Proterozoic	9	24.6	2.7	(1.60)	(1.82)	(5.45)	(1.09)	(63.2)
Middle and Late Proterozoic	4	13.3	3.3	(1.15)	(1.61)	(5.28)	0.97	(57.1)
Early Phanerozoic	75	2451.1	32.7	0.93	1.74	3.83	0.76	54.8
Late Phanerozoic	7	14.9	2.1	(2.06)	(2.13)	(4.48)	(2.85)	(238.3)

¹Grades in parentheses for averages based on less than 10 values

Of more relevance to this study are mineral deposit types that are associated with intrusive rocks. In the Archean, these consist mainly of orogenic gold deposits and critical minerals (Li, Cs, Rb, Ta) associated with large pegmatite dikes associated with so-called "fertile granites" (Breaks et al., 2003).

Critical minerals may also be found in association with alkaline complexes and carbonatite complexes; however, no rocks of this type occur in the Revell Regional area. Similarly, porphyry copper deposits can be associated with intermediate to felsic composition intrusions and these large hydrothermal systems are accompanied by large well-defined alteration zones which are used in exploration for these deposits. Porphyry copper deposits, however, are rare in Archean terranes, and there is no field or geochemical evidence for any potential deposits of this sort in the Revell region.

Orogenic gold deposits are found in association with regional deformation zones and are associated with abundant quartz veining and/or host-rock alteration (e.g., Dubé et al., 2020). Such

 $^{^2\}mbox{Values}$ in bold highlight data appreciably higher than other grade-tonnage data

features are not present in the Revell region. Although gold occurrences are present locally in the surrounding supracrustal rocks in the Revell region, their locations have been previously documented (Ontario Geological Survey 2023a), and none are within the Revell Site.

Exploration for critical minerals associated with pegmatite dikes has increased recently because of the use of lithium in electric vehicle batteries. The rocks that form most of the Revell batholith are mostly tonalites, granodiorites and biotite granites rock types that generally do not produce large pegmatite bodies containing critical minerals, consistent with the lack of pegmatites present in the cored intervals (logged less than 0.001%; ~92cm in IG BH02).

Pegmatite dikes containing varied quantities of Li, Cs, Rb and Ta have been documented from the Raleigh Lake area in the extreme southeastern corner of the Revell Regional Area (Breaks 1993). These pegmatite bodies are hosted in deformed mafic to intermediate composition metavolcanic rocks near the Crocker Bay and Raleigh Lake stocks (Breaks, 1993). Six occurrences and one prospect are listed in the Ontario Mineral Inventory (Ontario Geological Survey, 2023a, 2023b). There is no indication that any of these pegmatites are related to the Revell batholith.

Currently there are active mineral claims occur around the periphery of the Revell batholith, mainly to the west, north and northeast, and are associated with abandoned mines, gold \pm copper \pm nickel \pm silver occurrences, and historical drill holes (Figure 4; Ontario Geological Survey, 2020). A molybdenum \pm silica sand \pm fluorite occurrence to the north of the Revell pluton was initially examined in the 1960s as a molybdenum occurrence; work in the 1990s indicates that it was predominantly a fluorite occurrence with associated quartz and sericite (MDI52F08NW00040; Ontario Geological Survey, 2023a). The veins are thin and discontinuous and are not sizable enough to meet the criteria of a Prospect (Ontario Geological Survey, 2023b). To the northeast of the Revell batholith a gold \pm copper \pm nickel \pm silver occurrence straddles the margin of the pluton (Figure 4).

The few active claims on the batholith are close to roads or are occurrences of building stone (Figure 4). Building stone operations are typically near surface operations, and do not require deep exploration drilling or other significant site disturbances. Consequently, they are not considered an issue with respect to the Revell Site. One active claim to the northwest of the Revell Site, has no obvious association with any previously identified mineral occurrence. There is a large contiguous operational mining claim to the southeast part of the pluton, over 20 km away from the Revell Site. This claim is in part related to mineral exploration in the Raleigh Lake pegmatite field (cf. Breaks 1993).

The PETROCH lithogeochemical database (Haus and Pauk, 2010) contains geochemical information on rock samples collected by Ontario Geological Survey (OGS) staff geoscientists during field projects undertaken since the mid-1970s. Additional geochemical data can also be found in Stone (2010c). The majority of lithogeochemistry samples have been taken from the greenstone belt rocks surrounding the Revell batholith and only a very few samples have been taken from within the pluton itself (Figure 5).

Sampling for site evaluation studies was also undertaken for surface measurements for magnetic susceptibility (SRK and Golder, 2015; Golder and PGW, 2017; Biswas 2019), specific gravity (Rainsford et al. 2018) to support gravity modelling and geophysics (SGL, 2020), and for geochronology (Ontario Geological Survey, 2019, K-Ar, Rb-Sr and U-Pb ages).

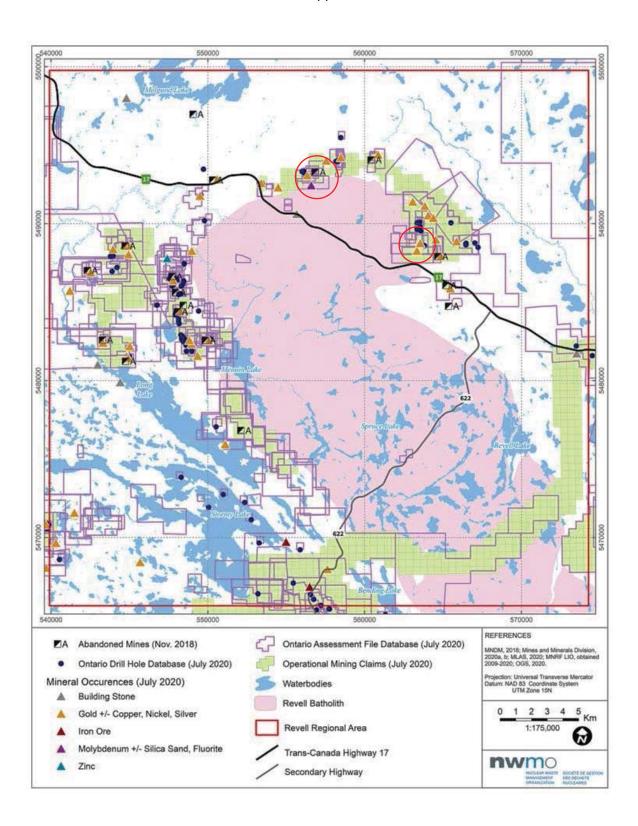


Figure 4. Current mineral claims in the area are located on the western, northern, and northeastern edges of the Revell batholith in the surrounding greenstone belts, or to the southeast, over 20 km from the Revell Site (from Parmenter et al., 2020). Red circles identify locations discussed in the text.

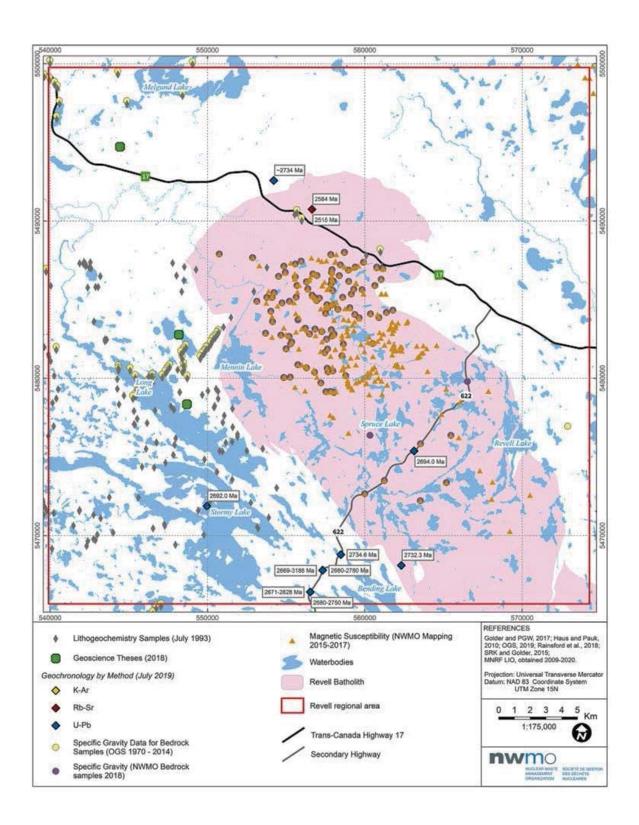


Figure 5. Geochemical samples, location of thesis studies, geochronology samples, gravity, and magnetic susceptibility measurements over the Revell batholith and surrounding area (from Parmenter et al., 2020).

3. DATA PROVIDED AND REVIEWED

To facilitate this work NWMO provided:

- A suite of lithogeochemical data for 93 samples, as obtained by appropriate methods, including results from instrumental neutron activation analysis (INAA), fusion inductively coupled optical emission mass spectrometry (FUS-ICP/FUS-MS), and total digestion inductively coupled plasma (TD-ICP) methods from Activation Laboratories Ltd., documented in NWMO and Golder, 2022; NWMO, 2022a; NWMO, 2022b.
- 2. Petrographic information, including photographs and mineralogical data obtained from thin sections that were used for lithogeochemical analyses (Actlabs A19-14907 and A20-05149, 2020), documented in NWMO (2022c) and NWMO (2022d).
- 3. Reports for regional geology (Parmenter et al., 2020) and boreholes IG_BH01, IG_BH02, and IG_BH03 including logging and interpretation (NWMO and Golder, 2022; NWMO, 2022a; NWMO, 2022b).

Table 2 shows the suite of elements analyzed with detection limits for each method. Data interpretation was performed on the samples with the lowest detections limits and using the analytical method presented below. Lithium was included in the analyses of boreholes IG_BH02 and IG BH03 (n = 48) but was not measured in the earlier analyses of the IG BH01 borehole.

The data set was received in Microsoft Excel file format and data screening and interpretation was undertaken using Microsoft Excel software. The raw data set (Appendix 1) contained duplicate analyses, using different methods, for 21 elements, in addition to those shown in Table 1, which allowed comparison of different analytical techniques over a range of elements and concentrations. Duplicated samples were compared for consistency and are discussed in Section 3.2 and individual elements and screened data used in Section 4 are presented in Appendix 2.

Table 2. Suite of elements analyzed and detection limits. Lithium was only analyzed in IG_BH02 and IG_BH03 samples. There is some variation in the element suite analyzed between boreholes and the elements analyzed for each borehole are summarized in Appendix 1.

	: A20-XX	10.000																								
Report Date: dd	/mm/yyy	ry																								
Analyte Symbol	Au	As	Br	Cr	Ir	Sc	Se	Sb	Mass	SiO2	A120	Fe20	03(T)	MnO	MgO	CaO	Na2O	K2O	TiO2	P2O5	LOI	Total	Sc	Be	v	Cr
Unit Symbol	ppb	ppm	ppm	ppm	ppb	ppm	ppm	ppm	g	%	%	%		%	%	%	%	%	%	%	%	%	ppm	ppm	ppm	ppn
Detection Limit	2	0.5	0.5	5	5	0.1	3	0.2		0.01	0.01	0.01		0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01	1	1	5	20
Analysis Method	INA	INA A	INA A	INA A	INA A	INA A	INA A	INA	INA A	FUS-	FUS- ICP	FUS	-ICP	FUS- ICP	GRA V	FUS- ICP	FUS- ICP	FUS- ICP	FUS- ICP	FUS MS						
Analyte Symbol Unit Symbol	Co	Ni	Cu	Z		Cd	S %	Ga	Ge	As	Rb	Sr	Y	Zr	Nb	Мо	Ag	In	Sn	Sb	Cs	Ba	La	Ce	Pr	Nd
			(1888)								1000	-	1000	1		100000	1.0	10.000				1.000	21178001		-	
Unit Symbol	ppm	ppm	ppm	Pi	pm	ppm	200	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm
Detection Limit	1	1	1	1		0.5	0.001	1	0.5	5	1	2	0.5	1	0.2	2	0.3	0.1	1	0.2	0.1	2	0.05	0.05	0.01	0.05
Analysis Method	FUS- MS	TD- ICP	TD-IC	P T	D- CP	TD- ICP	TD- ICP	FUS- MS	FUS- MS	FUS- MS	FUS- MS	FUS- ICP	FUS- MS	FUS- ICP	FUS- MS	FUS- MS	TD- ICP	FUS- MS	FUS- MS	FUS- MS	FUS- MS	FUS- ICP	FUS- MS	FUS- MS	FUS- MS	FUS- MS
Analyte Symbol	Sm	Eu	Gd	T	ь	Dy	Но	Er	Tm	,	rb 1	Ju	Hf	Ta	W	TI	Pb	Li	В	T	ı t	1				

0.005

3.1 Methods and Approach

The raw lithogeochemistry data set received from Activation Laboratories Ltd. is presented in Appendix 1.

The lithogeochemistry data set was initially screened to remove analyses for elements that were below the detection limit of the analytical techniques and averaged for the sample groups. Some elements were inconsistently detected (e.g., Au) and in those samples, analyses below the detection limits were removed, and averages were based on the number of analyses above the detection limit.

The average screened lithogeochemistry data were then compared to background data for average upper crust concentrations to generate a background trend for the data set. This would then identify any elements that were at expected upper crustal levels and elements that were above levels expected within the average upper crust. We used the average upper crust values proposed by Rudnick and Gao (2003), as they provide a review of all current estimates of the composition of the upper, lower and total crust and discuss the various methods used to obtain these estimates. The values of Rudnick and Gao (2003) are very similar to those proposed by McLennan (2001), providing additional confidence in the choice of the values used in this report.

This data screening was initially performed on averages of the complete data set (including both felsic and amphibolite samples), and the data set was also averaged for all felsic, tonalite, and amphibolite samples. This was necessary as certain elements are geochemically compatible with the mafic amphibolite samples (e.g., Ni, Cr, V, and platinum group elements; Cawthorn et al., 2005) and these elements would have higher concentrations than in the felsic samples. This separation by lithology produced more representative averages and ranges of data and is summarized in Appendix 2. The data set were also averaged for each lithological division based on core logging and are summarized in Appendix 2. Average element concentrations for the tonalite samples allowed direct comparison with other plutons in the region.

The major oxide data were not considered in detail in this report as potential economic minerals and are discussed in the borehole summaries (NWMO and Golder, 2022, NWMO, 2022a, and NWMO, 2022b). Major oxide data was used to correct sample rock types (Figure 3), to ensure consistent data averaging, and to confirm compatible element data to explain elevated concentrations related to specific rock types.

Prior to evaluating the lithogeochemistry data the regional geological data were reviewed as well as core study reports and petrographic studies (NWMO, 2022c; NWMO, 2022d) to ensure that the lithogeochemical data were interpreted within the correct context.

3.2 Duplicate Analyses

The received lithogeochemistry data set included duplicate analyses for 21 elements using FUS-MS and INAA analyses (As, Ce, Co, Cr, Cs, Eu, Hf, La, Lu, Mo, Nd, Rb, Sb, Sm, Ta, Tb, Th, U, W, Yb and Zn). These data were cross plotted to evaluate the consistency between the different techniques in detecting the elements over a range of concentrations (Figure 6). Overall, there is excellent correlation for the range of elements, across a range of concentrations, and gives support that the measured concentrations are consistent between analytical techniques and representative of the rocks.

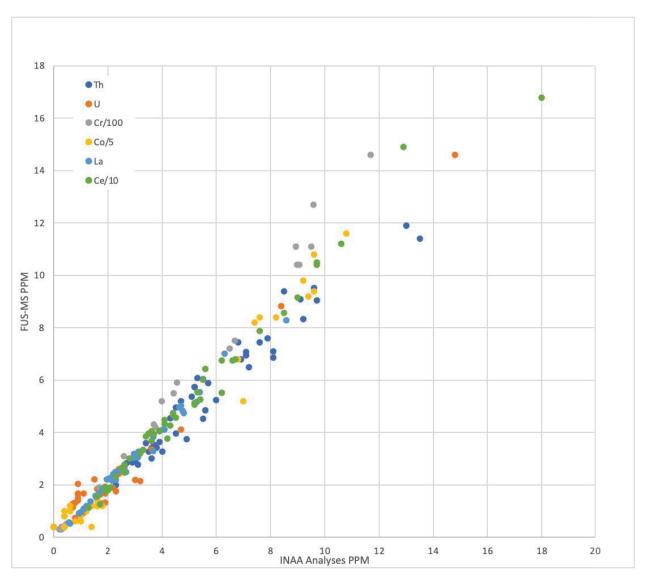


Figure 6. Cross plot of element concentrations (PPM) with duplicate analyses from INAA and FUS-MS methods. There is excellent correlation between most of the elements and data points. Elements have been scaled to fit on a single figure (Cr/100, Co/5, and Ce/30).

3.3 Mineral Potential

The determination of something having 'mineral potential' or something being 'mineralized' typically involves multiple methods from field-scale mapping to bedrock sampling and coring, integration with geophysical and surface methods, to precise laboratory analyses and interpretation. The element(s) in question would show significant enrichment levels (grade) above a background value, over a consistent area, in the required volumes (tonnage), in the right geological environment, and at the right metals price, that can be efficiently processed, to make an economic deposit. This of course depends on many other variables such as size, geographic location and political jurisdiction, whether surface of undergrounded mining, and on government and environmental regulations.

By using average upper crust concentrations to determine if something has 'mineral potential', if an element is at, or below, the average upper crust value, then it shows no enrichment or 'mineral potential'. It is at background. This method would highlight anomalous data and avoid potential arguments of 'potential' based on the unrealistic high prices, cut-off grades, or tonnages, as any other location in the upper crust would have equally as much 'potential' if at similar concentrations.

If something has 'mineral potential' then element concentrations need to be consistently highly enriched above background levels. Table 3 summarizes typical ore grades for a range of elements; as can be seen for Ni, grades are typically around 20,000 ppm (with an upper crust background of 47 ppm) and Zn grades are 50,000 ppm (verses 67 ppm average upper crust values) so enrichment levels of over x1000 times background values are typically required to get close to an ore grade. Ontario Geological Survey cut off values for mineral occurrences are included (Ontario Geological Survey, 2023b) and are typically 1/10th of base economic grades. So significant enrichment is required to get close to economic levels.

Table 3. Typical ore grades for a range of metallic elements (Dostal, 2017; De Los Hoyos, 2022; Ontario Geological Survey, 2023b) with background values taken from Rudnick and Gao (2003).

Element	Typical Economic Grade	ON Cut Off ³	Typical Background	Concentration Factor
Ag	1,000 ppm	35 ppm	0.05 ppm	20000 times
Au	6 ppm	0.5 ppm	3 ppb	2000 times
Cr	10% (100000 ppm)	10000 ppm	92 ppm	> 1000 times
Cu	10,000 ppm	2500 ppm	28 ppm	357 times
Li 1	0.58 - 1.18% wt% Li	1000 ppm	21 ppm	276 - 561 times
Mo	1,000 ppm	800 ppm	1.1 ppm	909 times
Ni	20,000 ppm	800 ppm	47 ppm	426 times
Pb	50,000 ppm	10000 ppm	17 ppm	2941 times
REE 2	>0.8-1.43 wt% TREO	1000 ppm		
U	10,000 ppm	300	2.7 ppm	3704 times
V	5,000 ppm	1000 ppm	97 ppm	52 times
w	3,000 - 5,000 ppm	1000 ppm	1.9 ppm	1578 - 2631 times
Zn	50,000 ppm	5000	67 ppm	746 times

¹ Hard rock source; ² All REE from this data set are below background upper crust values.

³ Cut off requirement to be a Mineral Occurence in Ontario, OGS

4. DATA INTERPRETATION

The screened data set is shown in Table 4 and in Figure 7. Table 4 includes the analytical technique used, detection limit, upper crust average concentration, and the elemental data set as an average for all samples (n=93), for felsic samples (n=78), and for amphibolite samples (n=15). The initial spreadsheet (Appendix 1) had 16 amphibolite samples but based on the major element geochemistry data one sample was reclassified as a felsic sample (see Figure 3). During data screening some elements were at or below detection levels (Cd, In, Mo, Se, and Tb from IG_BH01; Cd, In, Ir, and Se from IG_BH02; As, Cd, Ir, Sb, Se, and Sn from IG_BH03) and were removed from the screened data set.

4.1 All Samples

Most elements are at concentrations below the average upper crust values when compared as a single data set (38 out of 46 elements). The cells in Table 4 are color filled depending on the average concentration of the Revell data set compared to the average upper crust value; green filled cells are elements that are above the average upper crust values and unfilled cells are at (e.g. Be, Ga, and Ni; n = 3), or below, the average upper crust values (n = 35).

Elements that are above average upper crust values (green filled cells), and were not consistently detected, are red filled in Table 4. As can be seen in Table 4, 6 elements have average concentrations above average upper crust values (Ag, Au, Bi, Br, Sb and W) are inconsistently analysed within the sample suite (detected in only 5 to 25 of the 93 samples); if they were averaged over all samples (n = 93) average values would be well below average upper crust concentrations. Therefore, these samples are not considered to have mineral potential as they are inconsistently measured, across a range of rock types, at concentrations only a few times above background average crust levels.

Based on screening of the 'all data' sample set 44 out of the 46 elements analysed are at, or below, average upper crust values, or inconsistently detected. Only Li and Cr are present at average concentrations above average upper crust values. As Cr is at higher concentrations in the amphibolite samples, averaging over the all samples data set has skewed the values; when samples are averaged for felsic or tonalite samples the Cr concentrations are below average upper crust averages (Table 4).

4.2 Felsic Samples

The average data set for felsic samples (Table 4; n = 78) is similar to the 'all samples' average data set with 39 of 46 elements at concentrations below the average upper crust values. Of the remaining 7 elements, only Li is detected consistently above upper crust values. The remaining elements (Ag, Au, Bi, Br, Sb and W) are not consistently detected (detected between 3 to 22 times out of 93 samples). From this sample set, only Li is present at average concentrations less than 3 times average upper crust values.

maximum values for the single, felsic and amphibolite data set. Filled green cells (average columns) indicate elements that are above average upper crust averages; red filled cells indicate elements that have only a few analyses; orange filled cells indicate elements Table 4. Summary of the background average upper crust elemental concentrations, detection limits, and minimum, average, and that are equivalent to average upper crust values, and unfilled cells are below average upper crust values.

			7		All samples Average	S Aveida				dillipe olelle i	cisic samples Average		1		Amphibolite samples Average	ampies Aver	añe
Element Analyses	Level	Detection	Gao (2003) Upper Crust					Upper Crust average					Upper Crust average	<i>x</i>			
			afignav	Min	Average	Max	n of 93	s s	Min	Average	Max	n of 78	8	Min	Average	Max	n of 15
TD-ICP	PPM	0.3	0.053	4.0	0.4	9.0	23		9.0	0.4	9.0	22		4.0	4.0	0.4	-
INAA	PPM	0.5	4.8	9.0	1.3	3.1	27	×	9.0	1.2	3.1	21	×	0.7	1.4	2.5	9
INAA	PPB	2	1.5	2.00	3.80	00.9	2		2.0	3.8	0.9	4		4.0	4.0	4.0	-
FUS-ICP	PPM	2	624	7	495	1606	93	×	7.0	485.5	1606.0	78	×	213	541.9	951.0	15
FUS-ICP	PPM	-	2.1	2.0	2.1	3.0	17	×	2.0	2.0	2.0	1	×	2.0	2.3	3.0	9
FUS-MS	PPM	0.1	0.16	0.2	0.27	0.7	13		0.2	0.3	0.7	9		0.2	0.2	0.2	က
INAA	PPM	0.5	1.6	1.	8.9	18.5	23		1.	9.5	18.5	21		1.4	2.8	4.1	2
FUS-MS	PPM	0.05	63	6.25	42.9	168	93	×	6.25	34.8	91.5	78	×	37.6	84.8	168.0	15
FUS-MS	PPM	-	17.3	2.0	11.0	62.0	79	×	2.0	3.5	7.0	64	×	26.0	43.1	62.0	15
INAA	PPM	2	92	7	154	1170	69		7.0	20.3	46.0	54		168.0	637.3	1170.0	15
FUS-MS	PPM	0.1	4.9	0.3	2.37	13.7	93	×	9.0	1.8	4.5	78	×	0.3	5.2	13.7	15
TD-ICP	PPM	-	28	2	15	74	93	×	2.0	13.1	48.0	78	×	2.0	24.7	74.0	15
FUS-MS	PPM	0.01	3.9	0.45	1.3	4.36	93	×	0.5	6.0	3.1	78	×	1.7	3.1	4.4	15
FUS-MS	PPM	0.01	2.3	0.23	9.0	2.02	93	×	0.2	4.0	1.6	78	×	1.1	1.5	2.0	15
FUS-MS	PPM	0.005		0.048	0.7	3.22	93	×	0.05	0.49	0.94	78	×	0.7	1.8	3.2	15
FUS-MS	PPM	-	17.5	12	17.9	25	93	×	13.0	18.5	25.0	78	×	12.0	15.0	19.0	15
FUS-MS	PPM	0.01	4	0.79	2.0	8.03	93	×	8.0	1.4	2.8	78	×	2.3	5.1	8.0	15
FUS-MS	Mdd	0.5	1.4	0.7	1.1	2	85	×	0.7	1.0	2.0	71	×	1.2	1.4	1.8	17
FUS-MS	PPM	0.1	5.3	1.0	2.9	4.9	93	×	1.0	3.0	4.9	78	×	1.8	2.8	4.6	15
FUS-MS	PPM	0.01	0.83	80.0	0.2	8.0	93	×	0.1	0.2	9.0	78	×	4.0	9.0	0.8	15
FUS-MS	PPM	0.05	31	2.83	23.3	82.9	93	×	2.8	20.0	55.5	78	×	18.4	40.4	82.9	15
TD-ICP	PPM	-	21	7.0	65.3	242.0	48		7.0	55.1	93.0	43		55.0	153.2	242	S
FUS-MS	PPM	0.002	0.31	0.031	0.1	0.276	93	×	0.03	90.0	0.19	78	×	0.2	0.2	0.3	15
FUS-MS	PPM	0.2	12	2.0	1.4	12.5	93	×	2.0	3.8	7.3	78	×	2.2	5.3	12.5	15
FUS-MS	PPM	0.05	27	2.53	15.9	6.69	93	×	2.5	11.3	25.6	78	×	16.4	39.7	6.69	15
TD-ICP	PPM	-	47	2.0	47.4	535	80	×	2.0	6.1	175.0	65	×	75.0	226.4	535.0	15
TD-ICP	PPM	2	17	0.9	7.6	20	29	×	0.9	9.3	20.0	29	×	7.0	13.0	20.0	∞
FUS-MS	PPM	0.01	7.1	89.0	4.4	18.9	93	×	0.7	3.4	8.4	78	×	4.3	6.6	18.9	15
FUS-MS	PPM	-	84	24	77	200	93	×	24.0	0.77	200.0	78	×	28.0	76.0	145.0	15
TD-ICP	%	0.001	90.0	0.002	0.014	0.312	06	×	0.002	0.013	0.312	75	×	0.003	0.020	0.095	12
FUS-MS	W d	0.2	0.4	0.3	1.3	4 6	4 2	,	0.3	9.1	0.4	en 5	,	0.4	0.4	0.4	
FUS-ICE	2 :	- 10	4 .	2.00	1.7	88	18	< :	2.0	2.0	0.0	00	< :	0.71	0.02	38.0	2 :
SM-SD-	2 2	10.0	4.7	0.87	2.7	1.17	500	< >	9.0	8.1	3.0	70	< >	4.00	647.0	7.11	2 4
FUS-MS	Mdd	0.01	0.9	0.14	0.60	1 97	63	×	0.0	0.202	200	78	×	0.1	0.3	0.5	5 5
FUS-MS	PPM	0.01	0.7	60.0	0.3	0.93	93	×	0.1	0.2	0.5	78	×	0.3	9.0	6.0	15
FUS-MS	PPM	0.05	10.5	1.72	5.2	11.9	93	×	1.7	5.3	11.9	78	×	2.8	5.2	11.4	15
FUS-MS	PPM	0.05	6.0	0.11	0.38	96.0	92	×	0.11	0.38	96.0	77	×	0.15	0.43	0.94	15
FUS-MS	PPM	0.005	0.3	0.03	0.1	0.278	93	×	0.030	0.061	0.217	78	×	0.2	0.2	0.3	15
FUS-MS	PPM	0.01	2.7	0.61	2.1	14.6	93	×	0.61	2.21	14.60	78	×	0.75	1.4	2.6	15
FUS-ICP	PPM	2	26	6.0	48.4	233	84	×	00.9	19.12	41.00	69	×	139.00	183.3	233.0	15
FUS-MS	PPM	0.5	1.9	9.0	2.8	26.1	25		9.0	2.7	26.1	21		9.0	3.1	10.1	4
FUS-MS	PPM	0.5	21	2.5	9.9	20.8	93	×	2.5	4.9	18.8	78	×	9.4	15.4	20.8	15
FUS-MS	PPM	0.01	2	0.22	0.5	1.83	93	×	0.2	0.4	1.5	78	×	1.0	1.3	1.8	15
TD-ICP	PPM	-	29	1	53.2	121	93	×	11.0	47.1	118.0	78	×	68.0	85.0	121.0	15
FUS-ICP	Mdd	,	193	14	103.8	204	65	×	440	1034	0000	78	>	67.0	107 5	0 400	4

Elements with few analyses

Analysis close to upper crust averages

Analysis above upper crust average values

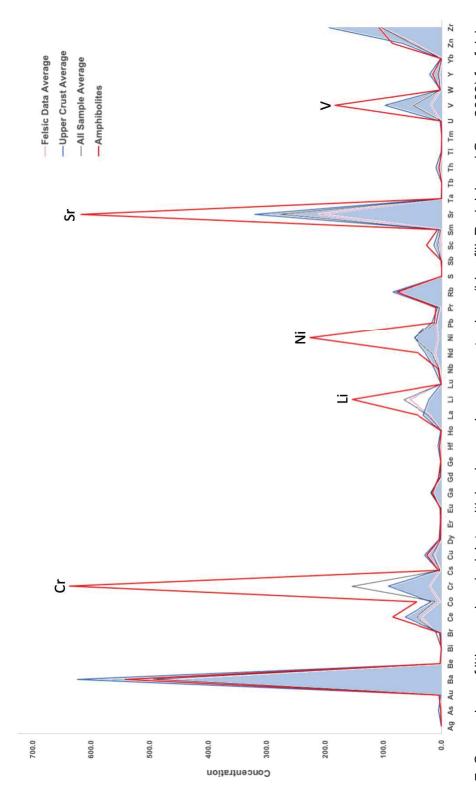


Figure 7. Comparison of lithogeochemical data with background upper crust values (blue fill; Rudnick and Gao, 2003) for felsic samples (pink), amphibolite (red), and average concentrations for the complete data set (black). Chromium, Li, Ni and Sr and V are the elements present at higher concentrations in amphibolite samples and skew the single data set averages.

The similarity of the felsic sample data set is also evident in chondrite-normalized rare earth element plots of the data (Figure 8, and shown in Figure 21 from NWMO and Golder, 2022 for all rock types). As expected, 71 of the 78 samples have nearly identical rare earth patterns that are typical of Archean tonalite-granodiorite intrusions. Seven samples are characterized by flat rare earth patterns with large negative Eu anomalies and low total rare earth element contents (17.5 ppm to 40.0 ppm), and all have SiO₂ contents greater than 74.5 wt.%. Six of these samples are aplite dikes, and one sample is logged as a silicified tonalite. The rare earth patterns in these samples can be explained by removal of a considerable amount of feldspar from the melt before emplacement. In the case of all the felsic samples, there is no suggestion of any sort of elemental enrichment that would hint at mineral potential within the Revell batholith.

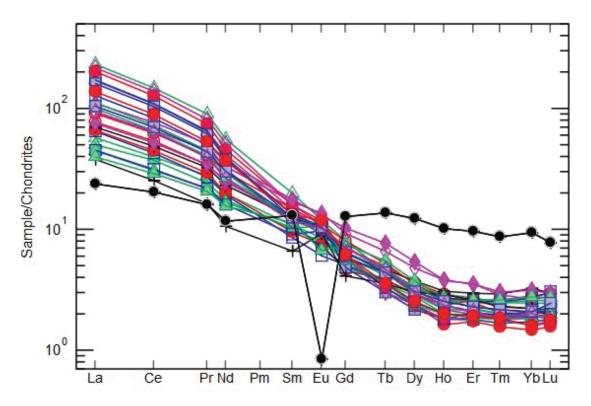


Figure 8. Chondrite normalized rare earth element plot for felsic samples from BH-01. The tonalites and granodiorites have sloping patterns and a sample (aplite dike) has low total rare earth element content and an intense Eu anomaly, likely reflecting feldspar extraction from the melt. The plots for samples from BH-02 and BH-03 (NWMO, 2022a and 2022b) show the similar overall pattern but include all rock types.

4.3 'Amphibolite' Mafic Samples

In the amphibolite sample set (n = 15) 30 out of 46 elements are at, or below, average upper crust values (Table 4). Four of the elements, Ag, Au, Br and W were only detected in a single sample (Au and Ag) or a few samples (Br and W; n = 2-4), and after removing samples with only a few analyses there are 12 elements that are above average upper crust concentrations (Ce, Co, Cr, La, Li, Nd, Ni, Sc, Sm, Sr, V, and Zn). The higher number of samples with concentrations above average upper crust values is somewhat expected as the upper crust averages would be from

rocks with a bulk geochemistry closer to a felsic sample rather than the mafic amphibolite samples. Many of these elements (e.g. Ce, Cr, Ni, Sr, V, and Zn) are geochemically compatible elements with the mafic compositions of the amphibolite (Deer et al., 1992), and therefore would be expected at high concentrations in the amphibolite samples. These elevated concentrations in the amphibolites would skew the 'all samples' data set averages (e.g. Cr).

As has been showing in Table 3, economic ore grades are typically concentrated 1000s of times greater than background values; of these 12 elements, only 3 elements are at concentrations above 3 times the upper crust backgrounds (Cr, Li and Ni; Figure 9), and the other elements (Ce, Co, La, Nd, Sc, Sm, Sr, V, and Zn) are only at levels of 1.5 – 2 times upper crust averages.

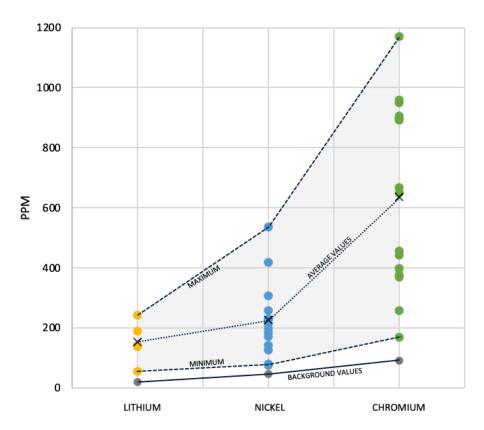


Figure 9. Cross plot of elements detected above background values (Li, Ni, and Cr) verses ppm concentrations. Minimum and maximum values are marked by the dashed lines and average of the amphibolite samples is shown by the dotted line. The range of data points that makes up the average analyses is shown by the points.

Chromium occurs at concentrations from 1.8 to 12.7 times background (168 - 1170 ppm; background of 92 ppm) with an average value of 637 ppm (6.9 times background average upper crust values). For reference average concentrations for ultramafic rocks, which are major sources of Cr, are 2700 ppm (dunite and peridotite; Whittaker, 1986), and the average concentration of Cr in the amphibolite samples of 637 ppm are significantly lower than the values for ultramafic rocks, below OGS mineral occurrences levels (10000 ppm cut off; Ontario Geological Survey, 2023b) and orders of magnitude below levels that would be considered ore grades.

Nickel occurs at concentrations from 1.6 to 11.4 times background (75 - 535ppm; background of 47 ppm) with average values of 226 ppm, around 4.8 times background upper crust values. Typical ore grades for Ni would be around 20000 ppm (Table 3), therefore an average concentration of 226 ppm would be 100 times below ore grade.

In the case of both Cr and Ni, the higher values are correlated with each other, suggesting that they might be related to the presence of typical magmatic mineral phases such as olivine, and is consistent with the mafic to ultramafic character of the amphibolites.

Average lithium concentrations are above average upper crust values in all data sets and range from around 3 times average upper crust background values in the all data and felsic sample data sets, to around 7 times upper crust background values in the amphibolite samples. The maximum concentration is around 12 times average upper crust values in a single sample (242 ppm; see Figure 9). Typical hard rock ore grades for Lithium are around 10000 - 15000 ppm (Ontario Geological Survey, 2023b; De Los Hoyos, 2022). Therefore, the values are not significant and do not indicate any Li mineral potential at the Revell Site.

As these amphibolite samples make up a very small proportion of the rock volume (around 1% of logged rock types; see Section 2.2), these above background elevated concentrations in Cr, Ni or Li do not show significantly high enough enrichment above background values, and would be volumetrically insignificant within the larger pluton, and therefore have no economic mineral potential.

4.4 Alteration

In reviewing the core logged data for alteration in bore holes IG_BH01, IG_BH02 and IG_BH03 there is significant variation between each borehole and nothing was noted that was consistent and could be related to intrusion-related mineralization. This is consistent with almost all elements at average upper crust background values and alteration being related to hydrothermal fluid flow during granite emplacement and related immediate post-emplacement alteration, or later surficial alteration (e.g., hematization). The generally low levels of alteration will have slightly altered the original compositions of the bedrock, but we do not see any patterns or concentrations of any elements with specific alteration facies.

4.5 Comparison to other Plutons

The lithogeochemistry data set was also averaged from the biotite granodiorite-tonalite and biotite tonalite samples (n = 52) to allow comparison to plutonic samples from the Marmion Terrane just to the south of the Revell Site (Figure 2; data from Stone, 2010c). A comparison between the plutons is shown in Table 5.

The data indicates that elemental concentrations of the Revell Site lithogeochemistry data set are at similar levels (orange filled cells in Table 5) or below levels (green filled cells in Table 5) of the adjacent equivalent age plutons. Out of the 36 elements that were analysed in Stone (2010c), 15 elements are at similar levels to the Revell batholith, 20 elements are below concentrations in the adjacent plutons, and only Li is at slightly elevated values. The similarity in the lithogeochemical data sets between the Revell pluton and the adjacent plutons supports the lack of any elevated element concentrations and mineral potential within the Revell batholith.

				Rudnick	Revell I	Biotite Tonalit	e Samples A	verage	_	subprovince
Element	Analyses	Level	Detection	and Gao					Biotite	Biotite
	7	2010.	Limit	(2003)					Tonalite Al-	Tonalite Al
				Upper	Min	Average	Max	n	rich	poor
Ag	TD-ICP	PPM	0.3	0.053	0.4	0.4	0.6	12		
As	INAA	PPM	0.5	4.8	0.6	1.4	3.1	15		
Au	INAA	PPB	2	1.5	3.0	4.3	6.0	3		
Ba	FUS-ICP	PPM	2	624	7.0	497.3	988.0	52	407	677
Be	FUS-ICP	PPM	1	2.1	2.0	2.0	2.0	8	1	
Bi	FUS-MS	PPM	0.1	0.16	0.2	0.2	0.4	7		
Br	INAA	PPM	0.5	1.6	1.2	10.7	18.5	16		
Ce	FUS-MS	PPM	0.05	63	9.9	37.6	91.5	52	42.4	48.6
Co	FUS-MS	PPM	1	17.3	2.0	3.0	7.0	45	10	
Cr	INAA	PPM	5	92	7.0	19.8	35.0	33	23	22
Cs	FUS-MS	PPM	0.1	4.9	0.6	1.7	3.8	52	1.2	2.2
Cu	TD-ICP	PPM	1	28	3.0	13.1	48.0	52	17	10
Dy	FUS-MS	PPM	0.01	3.9	0.5	0.8	2.6	52	1.3	1.6
Er	FUS-MS	PPM	0.01	2.3	0.2	0.4	1.4	52	0.7	0.8
Eu	FUS-MS	PPM	0.005	1	0.1	0.5	0.9	52	0.8	0.6
Ga	FUS-MS	PPM	1	17.5	13.0	18.3	23.0	52	19	16
Gd	FUS-MS	PPM	0.01	4	0.8	1.3	2.8	52	1.9	2.2
Ge	FUS-MS	PPM	0.5	1.4	0.7	0.9	1.8	47	1.9	2.2
Hf	FUS-MS	PPM	0.5	5.3	1.6	3.0	4.9	52	3.4	3.5
Но	FUS-MS	PPM	0.01	0.83	0.08	0.14	0.49	52	0.2	0.3
La	FUS-MS	PPM	0.05	31	4.4	21.9	55.5	52	22.1	25.2
Li	TD-ICP	PPM	1	21	7.0	51.6	86.0	27	34	41
Lu	FUS-MS	PPM	0.002	0.31	0.03	0.06	0.18	52	0.1	0.1
Nb	FUS-MS	PPM	0.2	12	2.0	3.7	7.3	52	5.6	6.5
Nd	FUS-MS	PPM	0.05	27	4.0	11.9	25.6	52	15.4	17.1
Ni	TD-ICP	PPM	1	47	2.0	6.8	175.0	45	11	7
Pb	TD-ICP	PPM	5	17	6.0	8.2	16.0	39	9	15
Pr	FUS-MS	PPM	0.01	7.1	1.1	3.6	8.4	52	4.5	5.2
Rb	FUS-MS	PPM	1	84	24.0	74.0	200.0	52	45	93
S	TD-ICP	%	0.001	0.06	0.002	0.017	0.312	50		
Sc	FUS-ICP	PPM	1	14	2.0	2.4	5.0	45	4	2
Sm	FUS-MS	PPM	0.01	4.7	1.0	1.8	3.0	52	2.5	2.7
Sr	FUS-ICP	PPM	2	320	8.0	220.2	411.0	52	371	241
Ta	FUS-MS	PPM	0.01	0.9	0.2	0.7	2.0	52	0.5	1
Tb	FUS-MS	PPM	0.01	0.7	0.1	0.2	0.4	52	0.3	0.3
Th	FUS-MS	PPM	0.05	10.5	1.8	5.7	11.9	52	4.1	10.1
TI	FUS-MS	PPM	0.05	0.9	0.11	0.40	0.80	51		
Tm	FUS-MS	PPM	0.005	0.3	0.03	0.06	0.20	52	0.1	0.1
U	FUS-MS	PPM	0.01	2.7	0.6	2.0	12.9	52	0.7	2.4
٧	FUS-ICP	PPM	5	97	6.0	16.7	41.0	48	40	14
W	FUS-MS	PPM	0.5	1.9	0.6	3.3	26.1	14		
Υ	FUS-MS	PPM	0.5	21	2.5	4.5	16.2	52	7	8
Yb	FUS-MS	PPM	0.01	2	0.2	0.4	1.3	52	0.6	0.8
Zn	TD-ICP	PPM	1	67	12.0	46.7	118.0	52	64	45
Zr	FUS-ICP	PPM	1	193	23.0	102.8	202.0	52	147	113
				.50				J-	1	

Table 5. Average chemical analyses of biotite tonalite (including biotite granodiorite-tonalite and biotite-tonalite endmembers) samples (n=52) for the Revell Site compared with average chemical analyses of plutonic suites from the central Wabigoon subprovince areas (two right-most columns; from Stone, 2010c). The Revell Site biotite tonalite contains an average of $13.9 \% \text{ Al}_2\text{O}_3$ (up to 18.7%; Appendix 2) and would range between the Al-rich (15.9 weight percent) and the Al-poor (13.97 weight percent) samples from Stone (2010c). The cell shading indicates whether the Revell Site average data is below the plutonic suite averages (green), within the range of the Al-rich to Al-poor plutonic suites (orange), or above the plutonic suite averages (red). Samples with a low number of analyses are marked in red but were not analyzed by Stone (2010c).

5. DISCUSSION AND CONCLUSIONS

Evaluation of the Revell Site lithogeochemistry data indicates that almost all elements, from an average of the single data set (n = 93), have concentrations below average upper crust levels. Of those elements that are above background upper crust values (Cr and Li) they are at values 2 - 3 times greater than background upper crust values. This is consistent for the felsic sample data set where Li is the only element consistently detected at up to 3 times average upper crust values, but still orders of magnitude below potential economic concentrations. For the amphibolite samples only Cr, Ni, and Li are present at greater than average upper crust background values. Importantly, amphibolite occurrences represent only approximately 2 % of the logged rock types by length of recovered core, with average values at levels 5 - 7 times greater than background upper crust values, and therefore this rock type is present at levels far below the enrichment necessary to have any mineral potential.

Direct comparison of the Revell Site data set to adjacent plutons, of similar age and composition, indicates that the granitoid rock of the Revell Site has lower overall concentrations for all elements that were measured and does not show evidence of mineral potential based on our current understanding.

Reviews of the core logging and petrographic data supports this conclusion:

- The petrographic reports supplied (NWMO, 2022c; NWMO, 2022d) are consistent with the lithogeochemical analyses with only sulfide and opaque minerals observed at trace levels in some samples and absent in the majority of the samples. Sulfide minerals are dominantly pyrite, with minor pyrrhotite, chalcopyrite, and sphalerite as sub-mm scale grains as inclusions in quartz and biotite and are consistent with primary igneous processes and do not show any concentration. Only trace levels of sulfur were measured in the lithogeochemical data set and consistent with the petrographic data suggest a lack of sulfide mineral potential at the Revell Site.
- The level of alteration identified was generally quite low during core logging, was sporadic and varied, was not pervasive and no discernable alteration zoning patterns were identified in the recovered core. Alteration types did not coincide with enrichment in any elements nor in any clearly 'mappable' alteration zones, that would be expected from mineralizing systems if they were present. These alteration textures are consistent with low level hydrothermal fluid flow during granite emplacement and related immediate postemplacement alteration, or later surficial alteration (e.g., hematization).
- In the core log photos that were reviewed there is no evidence of any additional phases or episodes of mineralization. Cross-cutting fractures, veins, and textures (sample and petrographic) are consistent with barren primary igneous textures and alteration types that would be expected during emplacement of the intrusion.

When the Revell Site lithogeochemistry data is compared to background upper crust levels, it is at or below average upper crust concentrations, and indicates that the Revell Site is mineralogically incredibly homogeneous with only minor variations due to the presence of amphibolite. It also suggests that many other areas of Archean crust in northwestern Ontario would have significantly more mineral potential than the Revell Site. This is important as based on current understanding, it is very unlikely upon projecting a million years into the future, that the Revell Site would be a focus for mineral exploration.

Some key questions were included in the work scope for this project and answers are summarized below:

1. Are there any quality issues with the collected data? Are collected data sufficient to answer key questions, or are further analyses recommended?

As the data were collect by qualified laboratories using accepted world-wide standards (ISO 9001:2015 and ISO 9002), quality assurance and quality control procedures, and calibration, it is highly unlikely that there are any quality issues with the collected data. Blank and duplicate samples were run during the analyses and these data were included within the results. These data were reviewed, and the analysis were repeatable and precise. The duplicate data (Figure 6) showed excellent consistency between the elements analysed at a range of concentrations, using different techniques.

The data set for the Revell Site is also very similar to the average elemental values from Marmion Terrane plutons (Stone, 2010a, 2010c) and independently supports the validity of the collected data. The lithogeochemistry data are also consistent with the findings of prior geological and core studies that did not document evidence of mineralization within the core samples, and which indirectly supports the quality of the lithogeochemistry data.

The consistency of the lithogeochemical data set suggests that the current data set is adequate for assessing the mineral resource potential of the Revell Site. It does not seem necessary to collect additional lithogeochemical data as it is expected to be almost identical to the current data set, and the analyses already completed are likely representative of the rock types encountered at the site.

If additional boreholes are drilled, there may be some value in additional lithogeochemistry sampling; not to evaluate mineral potential, but to ensure that there is lithological consistency across the Revell Site for complimentary and engineering studies.

The lithogeochemistry study, integrated with the previous coring and geological studies, are consistent with the understanding that there is limited existing or future mineral potential at the Revell Site and therefore further analyses to evaluate the mineral potential of the Revell Site are not considered to be necessary.

2. Is the mineral resource potential of the study area different from nearby regions, or from geologically-similar regions worldwide?

The lithogeochemistry data set indicates that there is limited future mineral potential at the Revell Site. Current exploration licenses are located at the contact of the Revell batholith and the greenstone belts to the west, north and northeast and mineralization is within the mafic metavolcanic rocks and other, older, host rocks. Genetic models for VMS deposits have focused on the surrounding greenstone belts and are constantly evolving, but the plutons have not been an exploration target in the past, and based on the lithogeochemical data, are unlikely to hold future potential.

In terms of intrusion-hosted mineralization, as outlined in Section 2.3, the tonalite and granodiorite rocks of the Revell batholith contain typical elemental values for these rock types, either below or at background upper crust levels. These includes elements such as Cs, Nb, Rb, and Ta, which are of current exploration interest, but are all at or below average upper crust concentrations; Li would also be a target but only shows enrichment to less than 3 times average upper crust values at levels far below economic concentrations. Therefore, the mineral potential of the Revell batholith is very low.

Pegmatites containing Cs, Li, Nb, Rb, and Ta do occur approximately 22 km southeast of the Revell site but are hosted in supracrustal rocks away from the margins of the Revell batholith. The lithogeochemistry data for the Revell Site pegmatite samples are shown in Appendix 2 and show no enrichment in these elements (only 2 pegmatite samples were logged in the IG_BH01 borehole and absent in the other boreholes, therefore, Li concentrations were not analysed.)

Therefore, the mineral resource potential of the Revell Site is very low and is supported by the background values for almost every element in the lithogeochemical data set and lack of evidence of mineralization or enrichment within the cored intervals. These background values as consistent with and almost identical to other plutons in the area (Stone, 2010a, 2010c).

3. How robust is any answer to the question of potential future mineral resource prospectivity on the timescales in question? Can uncertainties be defined?

In terms of answering the question based on the lithogeochemistry data set alone, it is very clear that there is little to no mineral potential at the Revell Site. The data set indicates almost every element is at a background level or lower, show no elemental concentration to ore grade levels, and elements at concentrations above background (Cr, Ni, and Li) occur in volumetrically insignificant amphibolite occurrences. So, this is a very robust assessment. This low to no mineral potential is clearly supported by the lithogeochemical data set, and there are few uncertainties regarding this interpretation.

Exploration models and potential resources do evolve through time, but as there is no evidence of any significant element concentrations in the lithogeochemistry data set, the potential mineral prospectivity of the Revell Site is unlikely to change tens to hundreds of thousands of years into the future.

REFERENCES

- Barrie, C.T. and Hannington, M.D. 1997. Classification of volcanic-associated massive sulfide deposits based on host-rock composition. In: Volcanic Associated Massive Sulfide Deposits: Processes and examples in modern and ancient settings. Society of Economic Geologists, Volume 8, Chapter 1, p.1-11.
- Biswas, S. 2019. Ontario Precambrian bedrock magnetic susceptibility geodatabase for 2001 to 2017; Ontario Geological Survey, Miscellaneous Release—Data 273–Revision 2.
- Blackburn, C.E. and Hinz, P. 1996. Gold and base metal potential of the northwest part of the Raleigh Lake greenstone belt, northwestern Ontario-Kenora Resident Geologist's District; in Summary of Field Work and Other Activities 1996, Ontario Geological Survey, Miscellaneous Paper 166, p.113-115.
- Breaks, F.W. 1993. Granite-related mineralization in northwestern Ontario: I. Raleigh Lake and Separation Rapids (English River) rare-element pegmatite fields; *in* Summary of Field Work and Other Activities, 1993, Ontario Geological Survey, Miscellaneous Paper 162, p.104-110.
- Breaks, F.W., Selway, J.B. and Tindle, A.G. 2003. Fertile peraluminous granites and related rareelement mineralization in pegmatites, Superior Province, northwest and northeast Ontario: Operation Treasure Hunt; Ontario Geological Survey, Open File Report 6099, 179p.
- Cawthorn, R.G., Barnes, S.J., Malitch, K.N., Ballhaus, C. 2005. Platinum group element, chromium, and vanadium deposits in mafic and ultramafic Rocks, Economic Geology 100th Anniversary Volume, p.215-249.
- Deer, W.A., Howie, R.A., and Zussman, J. 1992. An introduction to the rock-forming minerals (2nd Edition), 696p.
- De Los Hoyos, C.R. 2022. Geology of Economic Natural Lithium Deposits, April, 2022, AAPG's ICE Conference, Cartagena, Colombia, oral presentation, https://cartagena2022.iceevent.org/program/technical-program/on-demand-only.
- Dostal, J. 2017. Rare earth element deposits of alkaline igneous rocks, Resources v.6, article 34, 12p. https://doi.org/10.3390/resources6030034
- Dubé, B., Mercier-Langevin, P., Ayer, J., Pilote, J., Monecke, T. 2020. Chapter 3: Gold Deposits of the World-Class Timmins-Porcupine Camp, Abitibi Greenstone Belt, Canada in Sillitoe, R.H., Goldfarb, R.J., Robert, F. and Simmons, S. (eds.), Special Publication of the Society of Economic Geologists, Geology of the World's Major Gold Deposits and Provinces, Society of Economic Geologists, Volume 23. 850p.
- Galley, A.G., Hannington, M.D., and Jonasson, I.R. 2007. Volcanogenic massive sulphide deposits, *in* Goodfellow, W.D., ed., Mineral Deposits of Canada: A Synthesis of Major Deposit-Types, District Metallogeny, the Evolution of Geological Provinces, and Exploration Methods: Geological Association of Canada, Mineral Deposits Division, Special Publication No. 5, p.141-161.

- Golder (Golder Associates Ltd.) and PGW (Paterson, Grant, and Watson), 2017. Phase 2 Geoscientific Preliminary Assessment, Geological Mapping, Township of Ignace and Area, Ontario. Prepared for Nuclear Waste Management Organization (NWMO). NWMO Report APM- REP-01332-0225. Toronto, Canada, 205p.
- Haus, M. and Pauk, T. 2010. Data from the PETROCH Lithogeochemical database, Ontario Geological Survey, Miscellaneous Release Data 250, ISBN 978-1-4435-3732-2 [CD] ISBN 978-1-4435-3731-5 [zip file].
- Jensen, L.S. 1976. A new cation plot for classifying subalkalic volcanic rocks, Ontario Division of Mines, Miscellaneous Paper 66, 22p.
- McLellan, S.M. 2001. Relationship between the trace element content of sedimentary rocks and the composition of upper continental crust; Geochemistry, Geophysics, Geosystems, v.2, paper 2000GC000109, 24p.
- Mushayandebvu, M., DesRoches, A., Bates, M., Parmenter, A. and Kouhi, B. 2023. Subsurface geometry of the Revell Batholith by constrained geophysical modelling, NW Ontario, Canada, Applied Computing and Geosciences, Volume 19, September 2023.
- NWMO and Golder. 2022. Phase 2 Initial Borehole Drilling and 1 Testing, Ignace Area WP10 Geological Integration Report for Borehole IG_BH01. APM-REP-01332-0260, Toronto, Canada.
- NWMO. 2022a. Phase 2 Initial Borehole Drilling and 1 Testing, Ignace Area WP10 Geological Integration Report for Borehole IG BH02. APM-REP-01332-0265, Toronto, Canada.
- NWMO. 2022b. Phase 2 Initial Borehole Drilling and Testing, Ignace Area WP10 Geological Integration Report for Borehole IG_BH03. APM-REP-01332-0266, Toronto, Canada.
- NWMO. 2022c. Phase 2 Initial Borehole Drilling and Testing, Ignace Area. Thin Section Petrography and Lithogeochemical Analysis of Core Samples from IG_BH01. APM-REP-01332-0375, Toronto, Canada.
- NWMO. 2022d. Phase 2 Initial Borehole Drilling and Testing, Ignace Area. Thin Section Petrography and Lithogeochemical Analysis of Core Samples from IG_BH02 and IG_BH03. APM-REP-01332-0376, Toronto, Canada.
- Ontario Geological Survey 2019. Geochronology Inventory of Ontario; Ontario Geological Survey, Online Database, July 2019 version.

 https://www.geologyontario.mines.gov.on.ca/persistent-linking?publication=GeochrON
- Ontario Geological Survey. 2020. Mineral Deposit Inventory for Ontario; Ontario Geological survey. online database.
- Ontario Geological Survey. 2023a. Ontario Mineral Inventory; Ontario Geological Survey, online database, April 2023 update.
- Ontario Geological Survey. 2023b. Ontario Mineral Inventory; Ontario Geological Survey, online database, April 2023 update, OMI definitions and cut-off values.pdf.

- Parmenter, A., Waffle, L. and DesRoches, A. 2020. Bedrock Geology of the Revell Batholith and Surrounding Greenstone Belts. Nuclear Waste Management Organization, NWMO-TR-2020-08, Toronto, Ontario, 61p.
- Percival, J.A. and Easton, R.M. 2007. Geology of the Canadian Shield in Ontario: an update. Ontario Power Generation, Report No. 06819-REP-01200-10158-R00 and Ontario Geological Survey, Open File Report 6196, 65p.
- Rainsford, D.R.B., Carter-McAuslan, A. and Ashick-Stinson, L.C. 2018. Ontario specific gravity data for bedrock samples acquired from 1970 to 2014, Ontario Geological Survey, Miscellaneous Release—Data 371.
- Rudnick, R.L. and Gao, S. 2003. Composition of the continental crust, Chapter 3; in Holland, H.D. and Turekian, K.K., Eds., Treatise on Geochemistry, Vol. 3, The Crust, Elsevier-Pergamon, Oxford, 64p.
- SGL (Sander Geophysics Ltd.). 2020. 3d Geophysical Forward and Inversion Modelling of the Revell Batholith and Surrounding Greenstone Belt. Prepared for Nuclear Waste Management Organization (NWMO). NWMO Report APM-REP-01332-0270. Toronto, Canada, 119p.
- SRK (SRK Consulting Inc.) and Golder (Golder Associates Ltd.), 2015. Phase 2 Geoscientific Preliminary Assessment, Observation of General Geological Features, Township of Ignace, Ontario. Prepared for Nuclear Waste Management Organization (NWMO). NWMO Report APM-REP-06145-0004. Toronto, Canada, 88p.
- Stone, D., Halle, J. and Chaloux, E. 1998. Geology of the Ignace and Pekagoning Lake Areas, Central Wabigoon Subprovince, in Summary of Field Work and Other Activities 1998, Ontario Geological Survey, Miscellaneous Paper 169, p.127-135.
- Stone, D. 2010a. Precambrian geology of the central Wabigoon Subprovince area, northwestern Ontario, Ontario Geological Survey, Open File Report 5422, 130p.
- Stone, D. 2010b. Geology of the Stormy Lake area, northwestern Ontario; in Summary of Field Work and Other Activities 2010, Ontario Geological Survey, Open File Report 6260, p.13-1 to 13-12.
- Stone, D. 2010c. Geochemical analyses of rocks, minerals and soil in the central Wabigoon Subprovince area, northwestern Ontario; Ontario Geological Survey, Miscellaneous Release—Data 242.
- Stone, D., Hellebrandt, B. and Lange, M. 2011. Precambrian geology of the Bending Lake area (north sheet); Ontario Geological Survey, Preliminary Map P.3623, scale 1:20 000.
- Stott, G.M., Corkery, M. T., Percival, J.A., Simard, M. and Goutier, J. 2010. A revised terrane subdivision of the Superior Province; in Summary of Field Work and Other Activities 2010, Ontario Geological Survey, Open File Report 6260, p.20-1 to 20-10.
- Szewczyk, Z.J. and West, G.F. 1976. Gravity study of an Archean granitic area northwest of Ignace, Ontario, Canadian Journal of Earth Sciences, v.13, p.1119-1130.

Thurston, P. 1991. Geology of Ontario: Introduction; in Geology of Ontario, Ontario Geological Survey, Special Volume 4, Part 1, p. 3-25.

Whittaker, P., 1986, Chrome deposits in Ontario, Ontario Geological Survey, Study 55.

Appendix 1

Lithogeochemistry Data for IG_BH01, IG_BH02, and IG_BH03

All analyses performed by Activation Laboratories Ltd. in Thunder Bay or Ancaster, Ontario. In addition to the results included below, 28 were analysed by Actlabs to ensure overall quality control in the lithogeochemical analyses. Certified reference material measurements averaged within 10% of certified values (>90% accuracy). Duplicate sample measurements averaged within 7% of original measured certified reference materials (to measure accuracy), 7 duplicates (to measure precision), and 8 method blanks (to monitor contamination) values (>93% precision), and the method blanks all returned measurements below detection limit (no discernable contamination). Samples BH-01_LG-038 and BH-01_LG-39 have been flipped in this raw data table. BH-01_LG-038 is an amphibolite (low silica) and 039 is a tonalite (high silica), but table shows the opposite. The data are correct in NWMO (2022c).

Lithogeochemical results of various tests carried out by Activation Laboratories (NWMO, 2022c).

Report Number: A19-14907	A19-14907																					
Report Date: 16/03/2020	03/2020																					
Analyte Symbol	ı		γn	As	Br	Cr	S	Co	Ce	Eu	Hf	Ir	La	Lu	Mo	PN	Rb	Sb	Se	Sm	Ta	Th
Unit Symbol	From (position	To (position along	qdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd	qdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd	mdd
Detection Limit	borehole;	borehole; m)	2	0.5	6.5	3	1	1	3	0.2	1	2	0.5	0.05	2	2	20	0.2	3	0.1	0.5	0.2
Analysis Method	ì		INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA
IG_BH01_LG001	1.6	1.82	< 2	< 0.5	< 0.5	< × 5	^	3	26	0.2	4	< × 5	13.5	< 0.05	< × S	< × 5	90	< 0.2	> 3	1.4	< 0.5	3.6
IG_BH01_LG002	24.71	24.83	< 2	< 0.5	< 0.5	< × 5	^	5	23	0.3	3	< × 5	12.1	< 0.05	< × S	< × 5	< 20	0.4	< 3	1.3	< 0.5	3.5
IG_BH01_LG003	67.1	67.33	< 2	< 0.5	< 0.5	< × ×	^	1	25	< 0.2	3	< × 5	15.6	< 0.05	< × ×	∞	100	< 0.2	<3	1.2	< 0.5	4.5
IG_BH01_LG004	80.77	80.9	9	6.0	< 0.5	< × 5	3	^	35	< 0.2	3	\$ >	21.4	< 0.05	< × × 5	6	08	< 0.2	> 3	1.5	< 0.5	9.2
IG_BH01_LG005	131.88	131.97	< 2	9.0	< 0.5	< × 5	^	^	52	< 0.2	2	\$ >	29.5	< 0.05	< × × 5	10	90	< 0.2	< 3	1.5	< 0.5	6.9
IG_BH01_LG006	170.1	170.29	< 2	3.1	< 0.5	< × 5	1	2	62	< 0.2	3	\$ >	39	< 0.05	< × ×	14	90	< 0.2	> 3	2	< 0.5	6.7
IG_BH01_LG007	201.6	201.73	4	2	< 0.5	∞	^	7	33	< 0.2	3	\ \ \	17.9	< 0.05	< > 5	9	90	< 0.2	> 3	1.7	< 0.5	3.7
IG_BH01_LG008	214.1	214.23	< 2	-1	< 0.5	< × 5	2	2	26	< 0.2	4	< × 5	15.4	< 0.05	< × 5	9	70	< 0.2	> 3	1.1	< 0.5	3.8
IG_BH01_LG009	72.7.27	227.35	< 2	2.3	< 0.5	< × 5	4	7	28	< 0.2	4	\ \ \	16.6	< 0.05	< > 5	\ S	< 20	< 0.2	> 3	1.5	< 0.5	3.6
IG_BH01_LG010	234.07	234.22	< 2	8.0	< 0.5	> 5	-	^	06	9:0	4	> 5	53.9	< 0.05	< × ×	20	100	< 0.2	< 3	2.7	< 0.5	13
IG_BH01_LG011	287.97	288.27	< 2	6.0	< 0.5	< × 5	^	1	43	< 0.2	2	< × 5	24.5	< 0.05	< × S	14	20	< 0.2	> 3	1.4	< 0.5	7.2
IG_BH01_LG012	298.11	298.32	< 2	6.0	< 0.5	7	3	2	52	< 0.2	3	< × 5	31	< 0.05	< × 5	11	100	< 0.2	> 3	1.5	< 0.5	7.6
IG_BH01_LG013	299.95	300.13	< 2	< 0.5	< 0.5	< × 5	2	2	36	< 0.2	2	< × 5	23	< 0.05	< × S	10	08	< 0.2	< 3	1.2	< 0.5	9
IG_BH01_LG014	315.04	315.12	< 2	< 0.5	< 0.5	258	> 1	35	180	2.6	3	> >	85.8	< 0.05	< 5	62	< 20	< 0.2	< 3	11.1	< 0.5	13.5
IG_BH01_LG015	335.11	335.3	< 2	< 0.5	< 0.5	> 5	2	~	53	< 0.2	3	< × 5	31.3	< 0.05	\ \ \ \ \ \	12	70	< 0.2	< 3	1.6	< 0.5	7.9

400.35 <	IG_BH01_LG016	394.98	395.13	< 2	< 0.5	< 0.5	> 5	3	× ×	19	< 0.2	3	< ×	9.4	< 0.05	< ×	∞	120	< 0.2	< 3	1.3	< 0.5	5.6
Marie Mari	_BH01_LG017	400.37	400.53	< 2	< 0.5	< 0.5	< > 5	\ \ \	~	13	< 0.2	2	< > 5	6.1	0.07	9	< ×	160	< 0.2	< 3	1.3	< 0.5	4.9
Heart	_BH01_LG018	402.82	402.93	< 2	< 0.5	< 0.5	> 5	4	~	17	< 0.2	4	< ×	5.7	0.15	∞	< > 5	200	< 0.2	× 3	1.7	1.1	5.5
48216 46228 <	BH01_LG019	443.22	443.3	< 2	1.4	< 0.5	> 5	3	^	29	0.3	3	\ \$	40.3	< 0.05		10	100	0.3	> 3	1.7	< 0.5	7.1
470.11 470.21 470.11	BH01_LG020	452.16	452.28	< 2	< 0.5	< 0.5	< 5	^	6	21	< 0.2	4	> 5	11	< 0.05		< > 5 <	< 20	< 0.2	> 3	1.4	< 0.5	1.8
589.18 589.18 6.0 6	BH01_LG021	470.11	470.22	< 2	< 0.5	< 0.5	< > 5	1	-	19	< 0.2	4	< × ×	11.1	< 0.05	> 5	∞	< 20	< 0.2	<3	1.3	< 0.5	2.3
58014 5801 6201 6301 <t< td=""><td>BH01_LG022</td><td>529.93</td><td>530.02</td><td>< 2</td><td>0.7</td><td>4.4</td><td>< 5</td><td>^ 1</td><td>5</td><td>99</td><td>0.3</td><td>5</td><td>> 5</td><td>41</td><td>< 0.05</td><td>< × 5</td><td>19</td><td>40</td><td>< 0.2</td><td>< 3</td><td>1.7</td><td>< 0.5</td><td>8.1</td></t<>	BH01_LG022	529.93	530.02	< 2	0.7	4.4	< 5	^ 1	5	99	0.3	5	> 5	41	< 0.05	< × 5	19	40	< 0.2	< 3	1.7	< 0.5	8.1
89014 58078 COR COS COS <td>BH01_LG023</td> <td>549.15</td> <td>549.33</td> <td>< 2</td> <td>0.8</td> <td>< 0.5</td> <td>10</td> <td>3</td> <td>∞</td> <td>39</td> <td>0.4</td> <td>4</td> <td>< ×</td> <td>22.5</td> <td>< 0.05</td> <td>< × 5</td> <td>12</td> <td>09</td> <td>< 0.2</td> <td>× 3</td> <td>2.3</td> <td>< 0.5</td> <td>4.5</td>	BH01_LG023	549.15	549.33	< 2	0.8	< 0.5	10	3	∞	39	0.4	4	< ×	22.5	< 0.05	< × 5	12	09	< 0.2	× 3	2.3	< 0.5	4.5
88704 8872 <t< td=""><td>BH01_LG024</td><td>560.64</td><td>560.78</td><td>< 2</td><td>< 0.5</td><td>< 0.5</td><td>7</td><td>4</td><td>2</td><td>32</td><td>0.5</td><td>4</td><td>> 5</td><td>18.6</td><td>< 0.05</td><td>> 5</td><td>12</td><td>< 20</td><td>< 0.2</td><td>× 3</td><td>1.7</td><td>< 0.5</td><td>3.9</td></t<>	BH01_LG024	560.64	560.78	< 2	< 0.5	< 0.5	7	4	2	32	0.5	4	> 5	18.6	< 0.05	> 5	12	< 20	< 0.2	× 3	1.7	< 0.5	3.9
589.12 589.21	BH01_LG025	587.04	587.2	< 2	< 0.5	< 0.5	15	^ \	∞	34	0.5	3	> 5	21.4	< 0.05	< × 5	12	< 20	< 0.2	< 3	1.9	< 0.5	4.3
666.04 636.2 62 63 62 61 61 61 62 63 62 63 62 61 61 62 63 62 63 62 63 62 63 62 63 64 63 64 63 60 63 63 64 63 60 63	BH01_LG026	593.12	593.27	< 2	< 0.5	< 0.5	> 5	-	3	92	0.5	3	\ \$	47.5	< 0.05	< ×	17	30	< 0.2	× 3	2.1	< 0.5	9.6
656.32 656.48 54.6	BH01_LG027	636.09	636.2	< 2	< 0.5	< 0.5	< 5	^ \	^	16	< 0.2	3	> 5	9.4	< 0.05	< 5	< > 5 <	< 20	0.2	> 3	6.0	< 0.5	3.1
793.04 699.25	BH01_LG028	656.32	656.48	< 2	0.7	< 0.5	168	3	34	76	2.4	4	> 5	48	< 0.05	< × 5	35	40	< 0.2	< 3	9.8	< 0.5	7.1
791.13 701.25 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2 <2	BH01_LG029	659.05	659.25	< 2	< 0.5	< 0.5	< 5	^ \	^	20	< 0.2	2		10	< 0.05	< 5	< > 5 <	40	0.3	<3	1.4	< 0.5	4.7
735.09 735.33 < 2 2.5 < 60.5 < 88 15 48 42 0.7 < 6.5 6.9 6.0 6.5 6.5 6.0 6.5 6.0 <t< td=""><td>BH01_LG030</td><td>701.13</td><td>701.25</td><td>< 2</td><td>2.1</td><td>< 0.5</td><td>< 5</td><td>2</td><td>2</td><td>99</td><td>0.4</td><td>4</td><td>< × S</td><td>36.8</td><td>< 0.05</td><td>< × S</td><td>12</td><td>< 20</td><td>< 0.2</td><td>< 3</td><td>1.7</td><td>< 0.5</td><td>8.9</td></t<>	BH01_LG030	701.13	701.25	< 2	2.1	< 0.5	< 5	2	2	99	0.4	4	< × S	36.8	< 0.05	< × S	12	< 20	< 0.2	< 3	1.7	< 0.5	8.9
775.3 752.36 <2 1	BH01_LG031	735.09	735.33	< 2	2.5	< 0.5		15	48	42	0.7	2	< × ×	19.5	< 0.05	< × ×	16	190	< 0.2	<3	3	< 0.5	2.9
776.2 776.28 44 0.5 44	BH01_LG032	752.3	752.35	< 2	-	< 0.5	11	-	3	85	9.0	S	< × S	52.9	< 0.05	< × S	17	100	< 0.2	< 3	2.3	< 0.5	9.1
795.26 793.03 4 6.05 <t< td=""><td>BH01_LG033</td><td>776.2</td><td>776.28</td><td>< 2</td><td>0.7</td><td>< 0.5</td><td>13</td><td>3</td><td>3</td><td>4</td><td>0.5</td><td>4</td><td>\ S</td><td>25.2</td><td>< 0.05</td><td>\ S</td><td>6</td><td>< 20</td><td>< 0.2</td><td>> 3</td><td>2.5</td><td>< 0.5</td><td>5.2</td></t<>	BH01_LG033	776.2	776.28	< 2	0.7	< 0.5	13	3	3	4	0.5	4	\ S	25.2	< 0.05	\ S	6	< 20	< 0.2	> 3	2.5	< 0.5	5.2
795.65 795.75 45 3 38 55 1.1 2 <5 50.6 <0.05 <5 24 48 44 0.8 1.1 <5 1.9 <5 20 <5 24 60 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.48 797.49 797.48 797.49	BH01_LG034	792.92	793.03	4	< 0.5	< 0.5		S	46	45	0.7	2		21.2	< 0.05		18	130	< 0.2	< 3	4.1	< 0.5	2.7
841.99 842.11 <2 <0.5 <0.5 <0.5 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <0.6 <	BH01_LG035	795.65	795.75	< 2	< 0.5	< 0.5		3	38	55	1.1	2	< × ×	29.6	< 0.05	< 5	24	09	0.3	< 3	5.1	< 0.5	3.4
841.99 842111 <2 <0.5 184 13 <1 3 41 0.3 4 <5 26.1 <0.05 <5 9 20 895.67 895.97 <2	BH01_LG036	797.48	797.62	< 2	< 0.5	< 0.5		5	48	4	8.0	-	< > 5 ×	19.5	< 0.05	< ×	15	170	< 0.2	>3	4	< 0.5	3.6
895.67 895.97 <2 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <	BH01_LG037	841.99	842.11	< 2	< 0.5	18.4	13	^	3	41	0.3	4	> 5	26.1	< 0.05	< ×	6	20	< 0.2	< 3	1.7	< 0.5	5.1
950.02 950.15 <2 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5	BH01_LG038	895.67	895.97	< 2	< 0.5	< 0.5	∞	^ \	9	36	0.4	4	< 5	21.8	< 0.05		∞	30	< 0.2	< 3	1.5	< 0.5	5.3
950.02 950.13 <2 <0.5 <0.5 <0.5 <0.1 <0.4 <0.4 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <	BH01_LG039	901.98	902.16	< 2	< 0.5	< 0.5		2	37	129	2	3	< 5	63	< 0.05	< 5	53	50	< 0.2	< 3	9.01	< 0.5	8.5
953.41 953.62 <2 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5	BH01_LG040	950.02	950.13	< 2	< 0.5	< 0.5	6	^ \	7	41	0.4	4	< 5	22.7	< 0.05		7	30	< 0.2	< 3	1.8	< 0.5	5.2
968.28 968.44 <2 1.5 <0.5 1170 13 54 106 1.9 2 <5 46.3 <0.05 <5 49 110 978.81 978.81 978.95 <2 1.9 4.1 443 <1 41 97 1 <1 <5 46.9 <0.05 <5 34 30 995.04 995.15 <2 <0.5 <0.5 <0.5 <0.5 <0.5 <0.7 <0.9 <2 <5 <6 <0.05 <5 <10 <0.05 <5 <0.5 <0.05 <5 <0.5 <0.05 <5 <0.5 <0.05 <5 <0.05 <5 <0.05 <5 <0.05 <5 <0.05 <5 <0.05 <5 <0.05 <5 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05 <0.05	BH01_LG041	953.41	953.62	< 2	< 0.5	< 0.5	∞	^ \	^	37	0.3	3	< 5	20.7	< 0.05	> 5	6	50	< 0.2	<3	1.8	< 0.5	5.7
978.81 978.9 <2 1.9 4.1 443 <1 41 97 1 <1 <1 <5 46.9 <0.05 <5 34 30 995.04 995.05 <2 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5 <0.5	BH01_LG042	968.28	968.44	< 2	1.5	< 0.5		13	54	106	1.9	2		46.3	< 0.05	> 5	49	110	< 0.2	× 3	9.3	< 0.5	5.5
995.04 995.15 <2 <0.5 <0.5 <0.5	3H01_LG043	978.81	6.876	< 2	1.9	4.1		^ 1	41	76	-	^ \	< 5	46.9	< 0.05	< >	34	30	< 0.2	< 3	6.9	< 0.5	3
999.1 999.29 <2 <0.5 <0.5 <0.5 905 6 47 54 0.9 2 <5 26.8 <0.05 <5 16 <20	BH01_LG044	995.04	995.15	< 2	< 0.5	< 0.5	18	~	4	62	0.7	4	< 5	36.8	< 0.05	> 5	15	110	< 0.2	< 3	2.4	< 0.5	8.1
	BH01_LG045	999.1	999.29	< 2	< 0.5	< 0.5	902	9	47	54	6.0	2	< × 5	26.8	< 0.05	< × 5	16	< 20	< 0.2	< 3	5.4	< 0.5	4

																	3	3	•	5	3	j	
Unit Symbol	mdd	mdd	udd	mdd	0.0	%	%	%	%	%	%	%	%	%	%	%	mdd	mdd	mdd	mdd	mdd	mdd	mdd
Detection Limit	0.5	0.5	-	0.2		0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01	0.01	-	-	5	20	-	-	0.5
Analysis Method	INAA	INAA	INAA	INAA	INAA	FUS- ICP	FUS- ICP	FUS-ICP	FUS- ICP	FUS-	FUS-	FUS-	FUS- MS	FUS- MS	FUS- MS								
IG_BH01_LG001	< 0.5	< 0.5	< 1	< 0.2	30.9	70.18	15.35	2.69	0.04	0.62	2.6	4.72	2.37	0.268	90.0	99.39	3	1	27	< 20	5	20	8.0
IG_BH01_LG002	< 0.5	0.7	< 1	0.4	30	71	15.84	2.69	0.037	0.59	2.6	4.75	2.26	0.269	0.07	100.8	4	-	56	< 20	4	21	8.0
IG_BH01_LG003	< 0.5	6.0	< 1	< 0.2	29.4	68.7	15.07	2.34	0.028	0.5	1.25	3.89	5:55	0.204	0.05	98.64	2		17	< 20	4	18	6.0
IG_BH01_LG004	< 0.5	6.0	< 1	< 0.2	28.7	75.67	12.53	1.61	0.027	0.23	1.62	4.11	2.29	860.0	0.04	98.61	2	2	< × 5	< 20	-	17	1:1
IG_BH01_LG005	< 0.5	2.6	< 1	< 0.2	31.9	75.89	13.46	2.04	0.032	0.29	1.85	4.24	2.13	0.142	0.02	100.6	-	2	7	< 20	<u>^</u>	16	< 0.5
IG_BH01_LG006	< 0.5	< 0.5	< 1	< 0.2	28.7	74.16	13.3	2.12	0.031	0.37	2.06	4.23	1.95	0.158	0.03	68.86	2	-	7	< 20	2	18	0.8
IG_BH01_LG007	< 0.5	< 0.5	< 1	< 0.2	29.4	66.39	15.2	3.29	0.046	0.78	3.07	4.44	1.94	0.336	0.1	66	2	2	32	< 20	2	22	< 0.5
IG_BH01_LG008	< 0.5	1.9	< 1	0.4	29.7	67.04	17.28	2.85	0.04	9.0	2.56	5.43	3.15	0.269	0.04	100.7	3	2	25	< 20	4	23	0.8
IG_BH01_LG009	< 0.5	< 0.5	< 1	< 0.2	27.8	68.81	16.03	3.31	0.042	0.7	3.25	5.04	1.49	0.328	0.1	99.64	4	2	59	< 20	9	23	0.8
IG_BH01_LG010	< 0.5	11	< 1	< 0.2	28.6	69.37	15.63	2.33	0.03	0.38	2.09	4.42	3.74	0.175	90.0	98.71	2	^ \	12	< 20	2	61	_
IG_BH01_LG011	< 0.5	6.0	< 1	< 0.2	29.6	78.11	11.05	2.19	0.025	0.22	1.52	3.5	1.9	0.11	0.02	85.66	-	1	< × 5	< 20	2	14	6.0
IG_BH01_LG012	< 0.5	3	< 1	< 0.2	29.3	74.13	12.63	1.84	0.027	0.26	1.84	4.05	2.09	0.124	0.03	98.93	-	_	9	< 20	2	18	0.0
IG_BH01_LG013	< 0.5	2.2	< 1	< 0.2	30.2	81.66	8.74	1.56	0.02	0.18	1.26	2.71	1.44	0.092	0.03	61.66	-	^ \	7	< 20	2	13	_
IG_BH01_LG014	< 0.5	2.4	< 1	1.3	29.3	52.38	14.85	7.75	0.14	8.51	8.04	4.07	68.0	0.672	0.49	100.5	17	3	140	310	56	17	< 0.5
IG_BH01_LG015	< 0.5	2.4	\ 	< 0.2	30.8	75.04	13.03	2.13	0.03	0.33	1.99	4.15	2.07	0.14	0.03	99.35	2	-	7	< 20	2	18	6.0
IG_BH01_LG016	< 0.5	4.7	< 1	< 0.2	30.8	75.5	13.49	0.95	0.014	80.0	69:0	3.36	5.71	0.036	0.01	100		-	> >	< 20	-	61	1.2
IG_BH01_LG017	< 0.5	8.4	< 1	9.0	30.2	76.51	13.16	1.04	0.02	90.0	0.75	3.97	4.81	0.039	< 0.01	100.7	^ \	2	9	< 20	_ 	20	< 0.5
IG_BH01_LG018	< 0.5	14.8	< 1	1.4	31.6	73.94	13.51	0.94	0.087	0.05	0.43	4.12	5.2	0.03	0.02	98.52	3	-	> 5	< 20	<u>^</u>	25	2
IG_BH01_LG019	< 0.5	3.6	< 1	< 0.2	31.5	73.92	13.37	2.05	0.027	0.33	2.12	4.24	2.03	0.165	0.04	29.86	2	-	6	< 20	2	18	-
IG_BH01_LG020	< 0.5	< 0.5	\ \	0.2	33.1	68.39	15.97	3.41	0.038	0.84	3.72	4.8	1.25	0.384	60:0	99.41	5	-	37	< 20	9	21	0.7
IG_BH01_LG021	< 0.5	< 0.5	< 1	0.4	34.6	69.44	15.57	3.41	0.038	0.88	3.69	4.84	1.3	0.372	0.1	100.1	5	-	41	< 20	7	22	0.7
IG_BH01_LG022	< 0.5	1.6	< 1	< 0.2	33.9	73.89	13.05	2.34	0.028	0.45	2.45	4.1	1.67	0.223	90:0	9.86	2	-	14	< 20	3	18	0.8
IG_BH01_LG023	< 0.5	0.7	< 1	0.5	29.9	62:69	15.07	3.3	0.039	0.81	3.14	4.32	2.06	0.447	0.12	100.7	4	1	31	< 20	9	20	0.8
IG_BH01_LG024	< 0.5	< 0.5	< 1	< 0.2	31.4	71.63	15.47	2.57	0.031	0.61	3.22	4.42	1.6	0.293	0.08	100.2	3	-	23	< 20	4	20	0.8
IG_BH01_LG025	< 0.5	0.7	< 1	< 0.2	31.2	70.2	15.15	3.1	0.038	68.0	3.34	4.49	1.76	0.33	60:0	99.75	4	-	33	20	9	21	0.0
IG_BH01_LG026	< 0.5	1.7	\ \	< 0.2	30.7	75.11	13.35	2.59	0.034	0.48	2.36	4.01	1.92	0.24	80.0	9.001	2	2	16	< 20		16	< 0.5
	> 0 >	1.0	- 1	0.0	20.4	7,70	13.60		2000				, 0 ,		-00		-		;				

1.2	-	0.7	1.6	6.0	6.0	1.6	1.6	1.5	8.0	-	1.2	0.7	< 0.5	1.4	1.2	< 0.5	1.4
61	61	18	12	19	21	14	17	17	19	19	17	18	17	13	12	19	13
34	-	S	47	5	9	49	42	54	9	S	41	9	~	28	42	3	46
061	< 20	< 20	1040	< 20	< 20	1110	290	1270	< 20	< 20	520	< 20	< 20	1460	550	< 20	1040
891	6	24	193	24	34	203	227	206	29	23	139	29	56	139	179	29	223
7	-	-	-	-	2	-	-	-	-	-	2	-	-	2	3	2	2
61	7	2	30	7	4	33	27	35	3	2	21	3	7	26	26	3	33
66.66	99.2	100.2	18.66	99.24	100.2	99.74	100.4	98.12	98.49	68.86	100.3	98.52	100.2	99.86	6.66	99.38	78.66
0.38	0.03	60.0	0.17	80.0	0.12	0.16	0.22	0.17	0.1	0.07	0.48	0.1	90.0	0.37	0.32	0.12	0.19
0.978	0.088	0.317	0.557	0.32	0.443	0.577	0.732	0.582	0.337	0.299	0.649	0.335	0.201	0.636	0.627	0.377	209.0
2.68	2.28	1.72	5.38	1.76	1.65	4	1.87	3.58	1.67	1.8	2.22	1.68	2.32	3.49	2.41	1.66	1.53
2.87	4.71	3.96	0.17	3.91	4.59	68.0	2.91	1.13	4.17	4.1	2.87	4.04	4.1	0.64	1.69	3.63	2.71
7.64	1.91	2.76	9.14	2.65	3.06	9.53	96.6	8.62	3.07	2.68	9.37	2.93	2.18	9.47	14.39	2.8	10.47
9.9	0.22	29.0	10.98	0.71	0.82	12.04	8.53	13.09	98.0	69.0	10.93	98.0	0.45	16.82	9.12	0.95	11.03
0.117	0.023	0.038	0.156	0.039	0.039	0.168	0.154	0.175	0.043	0.042	0.127	0.044	0.039	0.169	0.152	0.05	0.164
69:8	1.29	3.09	9.22	3.33	3.31	9.64	9.93	10.94	3.49	3.15	8.05	3.56	2.49	9.43	9.45	3.67	9.76
14.02	14.19	13.73	10.22	13.38	14.61	10.63	13.68	11.07	14.09	13.47	13.3	13.81	13.46	9.2	69.6	13.07	11.13
54.05	74	73.29	47.18	72.67	70.34	47.66	49.87	46.08	70.24	72.22	49.84	70.83	74.68	43.91	41.51	72.51	49.1
31.3	30.5	28.9	22.8	27.1	29.8	28.6	29.3	26	27.2	29.9	25.2	30.7	29.8	25.5	24.6	25.1	29
1.5	< 0.2	< 0.2	1.1	< 0.2	0.5	1.1	1.3	6:0	< 0.2	0.3	1.2	< 0.2	< 0.2	-	-	0.3	1.2
-	-	~	~	-	~	~	~		~	~	~	~	~	~	~	~	< I >
6:0	6:0	< 0.5	< 0.5	1.7	< 0.5	8.0	< 0.5	< 0.5	< 0.5	3.2	1.5	< 0.5	8.0	2.3	< 0.5	< 0.5	1.1
< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
IG_BH01_LG028	IG_BH01_LG029	IG_BH01_LG030	IG_BH01_LG031	IG_BH01_LG032	IG_BH01_LG033	IG_BH01_LG034	IG_BH01_LG035	IG_BH01_LG036	IG_BH01_LG037	IG_BH01_LG038	IG_BH01_LG039	IG_BH01_LG040	IG_BH01_LG041	IG_BH01_LG042	IG_BH01_LG043	IG_BH01_LG044	IG_BH01_LG045

		_			ı																										
Tm	mdd	0.005	FUS- MS	0.062	0.062	0.047	0.05	0.05	0.045	0.057	0.065	0.073	0.051	0.048	0.045	0.03	0.197	0.047	0.053	0.112	0.217	0.05	0.054	0.063	0.042	0.072	0.045	0.051	0.039	0.058	0.278
垣	mdd	0.01	FUS- MS	0.44	0.4	0.34	0.35	0.3	0.32	0.42	0.42	0.48	0.35	0.31	0.32	0.23	1.65	0.31	0.33	8.0	1.56	0.32	0.37	0.42	0.29	0.57	0.33	0.4	0.28	0.42	2.02
Но	mdd	0.01	FUS- MS	0.16	0.15	0.13	0.11	0.11	0.12	0.16	0.13	0.17	0.12	0.11	0.11	0.08	0.64	0.11	0.13	0.26	0.56	0.11	0.14	0.15	0.1	0.21	0.13	0.15	0.09	0.16	8.0
Dy	mdd	0.01	FUS- MS	0.81	0.83	0.72	0.62	9.0	0.74	6.0	0.72	0.93	0.71	0.59	0.62	0.45	4.06	0.63	0.75	1.44	3.06	69:0	0.84	82.0	0.54	1.19	0.78	0.87	0.59	0.75	4.36
Tb	mdd	0.01	FUS- MS	0.16	0.16	0.14	0.13	0.12	0.16	0.18	0.14	0.17	0.16	0.12	0.13	0.09	0.88	0.13	0.17	0.24	0.5	0.13	0.15	0.16	0.11	0.24	0.17	0.19	0.12	0.13	0.93
PS	mdd	0.01	FUS- MS	1.18	1.12	1.02	1.09	66.0	1.31	1.4	0.92	1.27	1.62	66.0	1	0.79	7.96	1.02	1.19	1.49	2.57	66.0	1.22	1.17	76.0	1.86	1.45	1.61	1.22	0.83	96.9
Eu	mdd	0.005	FUS- MS	0.486	0.452	0.344	0.455	0.546	0.553	0.61	0.425	0.525	0.677	0.422	0.393	0.265	3.22	0.521	0.38	0.201	0.048	0.521	0.604	0.555	0.604	0.773	0.563	0.62	0.67	0.528	2.47
Sm	mdd	0.01	FUS- MS	1.61	1.48	1.28	1.74	1.7	2.17	1.93	1.42	1.87	2.97	1.54	1.7	1.26	11.7	1.85	1.43	1.28	1.95	1.97	1.54	1.36	1.83	2.53	1.88	2.21	2.37	66.0	9.85
PN	mdd	0.05	FUS- MS	9.1	8.1	8.88	12.1	13.3	19.1	11.5	9.19	10.7	25.6	12.5	14.5	10.7	6.69	15.4	6.75	4.1	5.4	18.2	7.75	7.4	18	15.2	11.4	13.8	21	4.9	52.6
Pr	mdd	0.01	FUS- MS	2.61	2.27	2.54	3.72	4.49	6.13	3.22	2.74	3.04	8.43	3.99	4.67	3.32	18.9	4.95	1.85	1.21	1.5	9	2.06	2.03	60.9	4.24	3.22	3.8	6.95	1.53	12.7
Ce	mdd	0.05	FUS- MS	24.6	23.1	26.4	39.7	50.7	9.79	33.3	27.7	30.1	91.5	42.7	51.4	36.9	168	55.4	17.5	11.3	12.6	6.79	19	19.4	67.5	40.7	32.1	38.6	78.7	15.7	105
La	mdd	0.05	FUS- MS	13.7	12	15.5	22.8	30.3	40.5	18.5	15.8	16.8	55.5	25.6	30.6	21.9	82.9	32.6	9.12	5.24	5.7	41.1	10.7	10.8	41.1	21.8	18	21.7	48.3	9.07	47.4
Ba	mdd	2	FUS- ICP	956	513	1606	405	424	403	583	582	441	886	390	434	339	216	392	1434	185	50	377	447	467	423	466	363	436	528	410	951
Cs	mdd	0.1	FUS- MS	1.6	1.4	1	2.3	1.6	1.2	1.2	2.5	2.9	1	1.5	2.2	1.4	1.3	1.8	2.3	1.1	2.5	1.9	8.0	6.0	1.2	2.3	4.5	1.2	1.4	1.2	2.8
SP	mdd	0.2	FUS- MS	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2
Sn	mdd	-	FUS- MS	1	_	_	_	_	_	_	1	_	1	1	1	1		1			1	1	< 1	1	_	_	_	_	1	< 1	_
П	mdd	0.1	FUS- MS	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.1
Мо	mdd	2	FUS- MS	< 2	< 2	< 2	> 2	< 2	> 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	2	< 2	< 2	< 2	> 2	> 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2
NP	mdd	0.2	FUS- MS	2	2.1	2.5	3.3	3.8	3.6	2	3.9	5.8	3.3	2.5	3.1	2.1	6.2	3.7	2.5	3	5.8	3	2.1	2.3	3.3	5.3	3.8	3.9	3.5	2.4	4.9
Zr	mdd	-	FUS-	95	100	09	58	84	83	126	114	127	82	56	71	49	204	77	25	36	32	83	132	138	143	150	127	129	158	81	157
Y	mdd	0.5	FUS- MS	4.3	4.3	3.6	3.6	3.3	3.5	4.4	4.1	4.9	3.5	3.3	3.4	2.5	18.9	3.5	3.9	7.9	18.8	3.6	4	4.2	3	5.6	3.6	4.4	3	4.5	20.8
Sr	mdd	2	FUS- ICP	297	312	206	161	187	197	356	253	357	243	160	223	114	1429	189	86	99	15	205	411	403	228	273	249	259	224	167	936
Rb	mdd	-	FUS- MS	09	55	93	71	64	62	52	102	63	82	99	70	47	28	70	135	121	179	99	44	47	54	09	40	48	55	53	89
As	mdd	5	FUS- MS	< 5	< 5	< 5	< > 5	< > 5	< > 5	< > 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< > 5	< 5	< 5	< 5	< 5	< 5	< 5	> 5	> 5	< 5	< 5	< 5	< 5
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH01_LG001	IG_BH01_LG002	IG_BH01_LG003	IG_BH01_LG004	IG_BH01_LG005	IG_BH01_LG006	IG_BH01_LG007	IG_BH01_LG008	IG_BH01_LG009	IG_BH01_LG010	IG_BH01_LG011	IG_BH01_LG012	IG_BH01_LG013	IG_BH01_LG014	IG_BH01_LG015	IG_BH01_LG016	IG_BH01_LG017	IG_BH01_LG018	IG_BH01_LG019	IG_BH01_LG020	IG_BH01_LG021	IG_BH01_LG022	IG_BH01_LG023	IG_BH01_LG024	IG_BH01_LG025	IG_BH01_LG026	IG_BH01_LG027	IG_BH01_LG028

0.05	0.049	0.166	0.046	0.076	0.188	0.239	0.183	0.045	0.065	0.226	0.054	0.046	0.171	0.187	0.046	0.212
0.36	0.33	1.07	0.3	0.57	1.32	1.7	1.32	0.3	0.42	1.68	0.37	0.34	1.32	1.36	0.31	1.41
0.15	0.1	0.35	0.1	0.21	0.47	0.59	0.47	0.11	0.14	0.65	0.14	0.13	0.53	0.52	0.11	0.5
0.93	0.58	1.7	0.59	1.34	2.54	3.06	2.5	0.59	0.83	4.02	92.0	0.73	3.28	3.09	0.63	2.61
0.2	0.12	0.31	0.11	0.28	0.47	0.59	0.45	0.13	0.17	68.0	0.16	0.17	0.74	99.0	0.13	0.53
1.51	1.15	2.29	1.15	2.06	3.49	4.2	3.47	1.11	1.3	8.03	1.3	1.41	69:9	4.85	1.24	4.01
0.38	0.599	0.748	0.623	0.741	1.14	1.55	1.23	0.649	0.592	3.01	0.592	0.577	2.53	1.67	0.673	1.38
1.7	1.99	3.05	2.38	2.64	4.48	5.5	4.47	1.71	1.87	11.7	1.87	1.73	9.76	7.63	1.99	5.29
7.19	17.5	16.4	23.6	16.6	23.6	29.8	23.6	13.6	12.3	68.7	13.9	12.1	56.5	43.5	17	26.9
1.93	5.79	4.31	7.61	4.77	5.67	7.33	5.65	4.22	3.82	17.4	4.1	3.68	13.7	10.8	4.98	6.46
17.9	64.2	37.6	85.7	47.4	45.7	60.1	46.6	8.44	40.4	149	43.2	39.4	112	104	55.2	52.5
9.52	38.4	18.4	51.8	26.4	22.1	31.9	22.1	26.8	24	70.1	25	22.6	49.6	50.3	33	25
446	403	648	447	414	959	406	591	431	387	870	403	710	622	254	434	344
1.1	1.8	13	1.8	2.6	4.8	2.8	5.4	1.3	1.3	3.2	1.5	1.3	13.7	0.5	1.2	4.9
< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2
1	1	1	1	1	1	1	1	1	1	1	1	1	1	<1	1	1
< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2
2.3	4	2.2	4.2	6.3	2.5	3.5	2.7	4.3	4.7	6.4	4.4	3.6	9.3	12.5	4.4	2.7
62	153	29	165	153	89	87	69	170	145	154	185	135	118	89	202	74
4.5	3.1	9.4	3.2	9	12.4	16.4	12	3.2	4.5	17.8	3.9	3.6	14.6	14.7	3.5	13.7
168	242	138	236	265	206	653	200	258	225	1137	250	194	266	241	242	529
49	99	145	65	49	113	51	101	51	65	52	90	65	94	71	50	43
< 5	< > 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< > 5	< 5	< > 5	< > 5	> 5	< > 5	< 5	< 5
IG_BH01_LG029	IG_BH01_LG030	IG_BH01_LG031	IG_BH01_LG032	IG_BH01_LG033	IG_BH01_LG034	IG_BH01_LG035	IG_BH01_LG036	IG_BH01_LG037	IG_BH01_LG038	IG_BH01_LG039	IG_BH01_LG040	IG_BH01_LG041	IG_BH01_LG042	IG_BH01_LG043	IG_BH01_LG044	IG_BH01_LG045

TOI	%		GRAV	0.48	0.71	1.06	0.39	0.46	0.47	9.4	1.46	0.54	0.48	0.93	1.92	1.48	2.72	0.41	0.17	0.31	0.19	0.38	0.5	0.4	0.34	1.61	0.24	0.34	0.42	0.43	1.97
Pb	mdd	3	TD- ICP	8	9	61	∞	8	9	4	8	5	11	7	9	9	12	∞	20	18	16	7	3	4	5	7	4	5	9	6	∞
Ag	udd	0.3	TD- ICP	< 0.3	0.4	0.4	< 0.3	0.4	< 0.3	0.3	0.5	0.4	< 0.3	0.4	< 0.3	< 0.3	0.4	< 0.3	< 0.3	< 0.3	< 0.3	0.4	< 0.3	< 0.3	0.4	0.4	< 0.3	0.4	0.3	< 0.3	0.3
S	%	0.001	TD- ICP	0.005	0.002	0.004	0.001	0.002	0.002	0.003	0.003	0.004	0.003	0.312	0.047	0.225	900.0	0.003	600.0	0.03	90000	0.003	0.003	0.003	0.003	0.002	0.003	0.004	0.004	0.003	0.095
Cd	mdd	0.5	TD- ICP	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
Zu	mdd	-	TD- ICP	54	53	47	38	4	47	61	99	72	49	33	40	85	98	44	11	14	15	42	09	61	47	61	54	56	49	25	98
Cu	mdd	-	TD- ICP	31	17	46	13	10	8	37	8	18	7	8	9	9	12	20	3	9	5	18	10	7	9	4	10	17	6	8	74
ž	mdd	-	TD- ICP	4	3	2	2	-	2	3	2	3	2	2	1	1	171	2	2	-	< 1	2	3	4	2	9	3	8	3	2	143
n	mdd	0.01	FUS- MS	0.61	1.15	1.65	1.66	2.63	86.0	89.0	1.67	0.93	1.66	1.42	2.18	1.89	2.59	2.43	4.12	8.83	14.6	3.35	89.0	1.25	1.85	1.28	1.05	1.15	1.3	1.32	2.04
Th	mdd	0.05	FUS- MS	3.41	3.28	3.96	8.34	8.9	9.04	3.53	3.43	3.4	11.9	6.5	7.44	5.23	11.4	7.6	4.86	3.75	4.53	6.94	1.81	2	7.1	4.95	3.63	4.55	9.53	2.77	7.07
Bi	mdd	0.1	FUS- MS	< 0.1	< 0.1	< 0.1	0.2	0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.4	< 0.1	0.2	0.2	0.2	< 0.1	< 0.1	0.1	0.2	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
I	udd	0.05	FUS- MS	0.36	0.34	0.51	0.42	0.28	0.39	< 0.05	0.55	0.44	0.42	0.35	0.42	0.26	0.15	0.43	0.72	0.54	96.0	0.39	0.28	0.29	0.34	0.38	0.26	0.29	0.37	0.28	0.46
M	mdd	0.5	FUS- MS	< 0.5	< 0.5	9.0	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	1.2	0.7	2.9	10.1	< 0.5	< 0.5	< 0.5	2.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
Ta	mdd	0.01	FUS- MS	0.3	0.25	0.46	88.0	0.52	0.64	0.22	0.72	1.16	0.43	0.44	0.72	0.5	0.28	0.62	0.75	0.4	1.21	0.55	0.2	0.21	0.58	9.0	0.45	0.48	0.36	0.44	0.34
Hf	mdd	0.1	FUS- MS	2.6	2.7	1.7	2	2.3	2.6	3.5	3.3	3.7	2.5	1.7	2.2	1.6	4.6	2.4	6.1	2.3	3.3	2.5	3.3	3.3	3.9	3.9	3.4	3.4	3.8	3.1	4.6
Lu	mdd	0.002	FUS- MS	0.064	0.061	0.047	0.049	0.052	0.049	0.054	0.073	0.074	0.046	0.048	0.055	0.038	0.216	0.052	0.046	0.12	0.193	0.054	0.055	0.076	0.045	690:0	0.042	0.051	0.039	0.048	0.276
Yb	mdd	0.01	FUS- MS	0.42	6.4	0.31	0.33	0.34	0.3	0.36	0.46	0.53	0.31	0.32	0.35	0.22	1.32	0.32	0.34	0.77	1.53	0.33	0.34	0.44	0.28	0.43	0.26	0.33	0.24	0.36	1.83
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH01_LG001	IG_BH01_LG002	IG_BH01_LG003	IG_BH01_LG004	IG_BH01_LG005	IG_BH01_LG006	IG_BH01_LG007	IG_BH01_LG008	IG_BH01_LG009	IG_BH01_LG010	IG_BH01_LG011	IG_BH01_LG012	IG_BH01_LG013	IG_BH01_LG014	IG_BH01_LG015	IG_BH01_LG016	IG_BH01_LG017	IG_BH01_LG018	IG_BH01_LG019	IG_BH01_LG020	IG_BH01_LG021	IG_BH01_LG022	IG_BH01_LG023	IG_BH01_LG024	IG_BH01_LG025	IG_BH01_LG026	IG_BH01_LG027	IG_BH01_LG028

	0.52	6.64	0.39	1.24	4.44	2.54	2.7	0.42	0.38	2.43	0.33	0.24	4.54	10.54	0.54	3.17
10	5	< 3	7	7	< 3	19	< 3	8	6	8	5	7	< 3	< 3	3	4
< 0.3	0.4	< 0.3	0.3	0.3	< 0.3	< 0.3	< 0.3	0.3	0.3	0.3	< 0.3	0.4	0.3	< 0.3	0.3	< 0.3
0.003	0.005	0.003	0.004	90000	0.003	0.005	0.003	90000	0.003	0.071	0.005	0.003	900.0	0.003	0.003	0.005
< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
23	89	84	79	\$9	84	27	121	118	85	89	69	46	28	16	89	69
7	12	2	12	∞	3	∞	4	20	11	09	18	9	16	22	7	∞
3	4	213	5	9	229	127	258	175	7	306	7	4	419	81	9	185
1.49	1.51	0.75	1.62	1.32	0.75	1.17	8:0	66:0	2.14	2.22	1.55	1.32	1.75	0.77	0.83	0.94
5.19	7.44	2.85	80.6	5.74	2.84	3.59	3.01	5.37	60.9	9.39	5.73	5.88	6.03	2.9	98.9	3.26
< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.1	0.2	< 0.1	< 0.1	< 0.1	0.1	0.1	< 0.1	0.2	< 0.1	< 0.1	0.1
0.26	0.34	0.94	0.37	0.3	7.00	0.32	7:0	0.32	0.35	0.34	0.31	0.29	0.56	0.32	0.13	0.26
< 0.5	< 0.5	< 0.5	< 0.5	6.0	< 0.5	0.5	< 0.5	< 0.5	< 0.5	9.0	< 0.5	< 0.5	9.0	1.2	< 0.5	0.5
0.36	2.0	0.14	0.57	89:0	0.14	61.0	0.14	0.41	0.82	0.3	0.52	0.38	68.0	0.51	0.32	0.15
2.3	3.9	6.1	4.3	4	6.1	2.4	2	4.2	3.8	3.8	4.7	3.5	3.2	1.8	4.9	2.1
0.049	0.052	0.191	0.05	0.074	0.195	0.238	0.179	0.049	0.068	0.191	0.061	0.043	0.156	0.182	0.044	0.203
0.3	0.34	1.14	0.31	0.51	1.26	1.54	1.16	0.3	0.45	1.33	0.34	0.33	1.02	1.2	0.26	1.34
IG_BH01_LG029	IG_BH01_LG030	IG_BH01_LG031	IG_BH01_LG032	IG_BH01_LG033	IG_BH01_LG034	IG_BH01_LG035	IG_BH01_LG036	IG_BH01_LG037	IG_BH01_LG038	IG_BH01_LG039	IG_BH01_LG040	IG_BH01_LG041	IG_BH01_LG042	IG_BH01_LG043	IG_BH01_LG044	IG_BH01_LG045

certified values in one element, but other elements in the material were within an acceptable range of > 90% accuracy so these precision), and the method blanks all returned measurements below, or within an acceptable range of, detection limit (no discernable discrepancies were deemed insignificant. Duplicate sample measurements averaged within 7 % of original measured values (>93% All analyses performed by Activation Laboratories Ltd. in Thunder Bay or Ancaster, Ontario. In addition to the results included below, 63 certified reference materials (to measure accuracy), 7 duplicates (to measure precision), and 18 method blanks (to monitor contamination) were analysed by Actlabs to ensure overall quality control in the lithogeochemical analyses. Certified reference material (CRM) measurements averaged within 0.1 % of certified values (>99 % accuracy). There were a few CRM samples that were >10% off their contamination)

Lithogeochemical results of various tests carried out by Activation Laboratories (NWMO, 2022d).

Report Number: A20-05149	er: A20-0514	61																					
Report Date: 18/6/2020	18/6/2020																						
Analyte Symbol	ţ	E	Au	As	Br	Cr	Ir	Sc	Se	Sb	Mass	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O	K20	TiO2	P205	IOI	Total
Unit Symbol	From (position	To (position	qdd	mdd	mdd	mdd	qdd	mdd	mdd	mdd	50	%	%	%	%	%	%	%	%	%	%	%	%
Detection Limit	borehole;	borehole;	2	0.5	0.5	5	5	0.1	3	0.2		0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01
Analysis Method	(III	(III	INA A	INA A	INA	NA A	INA A	NA A	INA	NA A	INA A	FUS-	FUS- ICP	FUS-ICP	FUS- ICP	FUS- ICP	FUS- ICP	FUS- ICP	FUS-	FUS- ICP	FUS-	GRA V	FUS- ICP
IG_BH02_LG001	15.29	15.41	< 2	< 0.5	< 0.5	17	< > 5	2.7	< 3	< 0.2	32.3	70.95	14.65	2.7	0.035	89.0	2.93	4.42	1.66	0.296	60.0	0.48	68.89
IG_BH02_LG002	15.75	15.9	< 2	< 0.5	< 0.5	28	< > 5	1.9	< 3	_	31.2	72.86	13.56	2.11	0.036	0.33	1.86	4.2	2.58	0.22	0.07	0.58	98.41
IG_BH02_LG003	60.05	60.2	< 2	< 0.5	< 0.5	17	< × 5	2	< 3	< 0.2	30.8	72.33	14.43	1.91	0.025	0.34	1.24	4.5	2.73	0.175	0.05	8.0	98.54
IG_BH02_LG004	73.47	73.68	< 2	< 0.5	< 0.5	19	\$ >	1.9	< 3	< 0.2	1.07	74.01	14.09	1.73	0.033	0.32	1.74	4.39	3.01	0.16	0.03	0.45	96.66
IG_BH02_LG005	113.1	113.28	< 2	< 0.5	< 0.5	22	< × ×	2.8	< 3	7.1	31.1	71.59	14.6	2.96	0.039	0.71	3.06	4.34	1.75	0.308	0.1	0.5	99.95
IG_BH02_LG006	132.25	132.4	< 2	0.5	< 0.5	22	> 5	2	< 3	< 0.2	1.03	74.22	14.18	1.86	0.036	0.37	2.02	4.53	2.47	0.174	0.04	89:0	100.6
IG_BH02_LG007	280.6	280.8	< 2	< 0.5	18.1	14	< 5	2.9	< 3	< 0.2	32.3	70.15	14.51	2.92	0.049	0.75	1.91	5.54	1.22	0.315	0.09	1.39	98.85
IG_BH02_LG008	284.5	284.7	< 2	< 0.5	1.2	20	< > 5	1.8	< 3	< 0.2	29.6	72.63	14.02	1.7	0.028	0.32	1.58	4.47	2.9	0.154	0.04	1.65	99.49
IG_BH02_LG009	346.75	347	< 2	< 0.5	11.7	35	< 5	2.1	< 3	< 0.2	28.1	72.64	13.75	1.81	0.034	0.38	1.42	4.52	2.91	0.164	0.05	0.93	98.61
IG_BH02_LG010	368.5	368.65	< 2	0.5	< 0.5	23	> 5	2.9	< 3	< 0.2	31.1	71.82	14.62	2.74	0.032	9.65	3.09	4.5	1.71	0.306	0.1	0.37	99.92
IG_BH02_LG011	372.24	372.44	< 2	< 0.5	< 0.5	29	> 5	1.8	< 3	0.4	30.7	73.51	13.5	1.55	0.031	0.27	1.53	4.38	3.24	0.131	0.04	0.48	99.86
IG_BH02_LG012	381.03	381.2	< 2	< 0.5	1.1	30	\$ >	2.2	< 3	< 0.2	30.4	73.34	13.55	1.88	0.037	0.42	1.56	4.4	2.81	0.16	0.05	0.87	90.66
IG_BH02_LG013	478.22	478.44	< 2	< 0.5	< 0.5	22	< 5	2.2	< 3	< 0.2	30.5	73.13	13.74	1.84	0.033	0.31	1.78	4.38	2.84	0.167	0.04	1.92	100.2
IG_BH02_LG014	531.25	531.47	< 2	< 0.5	< 0.5	22	< 5	1	< 3	< 0.2	32.5	75.78	13.04	0.78	0.015	90.0	0.59	4.3	4.59	0.038	0.01	0.19	99.41
IG_BH02_LG015	536.48	536.82	< 2	< 0.5	< 0.5	899	< 5	34.5	< 3	< 0.2	30.2	42.77	12	10.3	0.159	10.24	11.38	1.99	3.26	0.94	0.47	5.72	99.24

IG_BH02_LG016	567.26	567.44	< 2	< 0.5	11.2	13	< × ×	2.1	< 3	< 0.2	1.04	74.5	14.07	1.88	0.034	0.34	1.84	4.47	2.97	0.165	0.04	0.46	100.8
IG_BH02_LG017	573.51	573.68	< 2	< 0.5	18.5	21	< > 5	2.9	< 3	0.3	31.7	72.59	14.75	2.59	0.045	0.55	2.63	4.68	1.75	0.25	90.0	0.56	100.5
IG_BH02_LG018	600.43	600.62	< 2	< 0.5	4	24	< 5	1.6	< 3	< 0.2	30.5	74.38	13.74	1.43	0.026	0.32	1.06	4.11	4.51	0.13	0.04	89.0	100.4
IG_BH02_LG019	632.98	633.26	< 2	< 0.5	< 0.5	23	< 5	1.9	< 3	< 0.2	1.05	73.68	14.16	1.8	0.033	0.33	1.75	4.36	3.09	0.163	0.05	0.43	99.85
IG_BH02_LG020	668.29	668.44	2	< 0.5	1.7	23	< 5	3.8	< 3	< 0.2	30	72.9	13.25	2.65	0.037	89.0	2.7	3.93	1.75	0.26	0.07	1.13	99.37
IG_BH02_LG021	672.2	672.4	< 2	1	1.4	376	< 5	19	< 3	0.4	33.1	55.35	14.26	7.12	0.133	5.49	9.57	3.46	1.02	0.574	0.18	3.11	100.3
IG_BH02_LG022	680.74	680.91	3	1.7	8.9	22	> 5	2	< 3	< 0.2	30.3	74.15	13.99	1.82	0.033	0.36	1.86	4.46	2.63	0.167	0.04	99.0	100.2
IG_BH02_LG023	682.58	682.74	< 2	< 0.5	14.3	< 5	> 5	2.3	< 3	0.2	30.8	58.9	18.74	1.97	0.037	0.53	4.01	8.61	1.27	0.249	90.0	4.12	98.49
IG_BH02_LG024	702.01	702.19	< 2	< 0.5	5.1	21	< 5	1.7	< 3	< 0.2	31.2	75.04	13.86	1.62	0.029	0.25	1.62	4.64	2.93	0.13	0.03	0.47	100.6
IG_BH02_LG025	777.49	777.75	< 2	< 0.5	13.6	24	< 5	1.9	< 3	< 0.2	32	72.74	14.3	1.85	0.031	0.32	1.99	4.97	3.1	0.166	0.05	1.05	100.6
IG_BH02_LG026	904.5	904.74	< 2	< 0.5	13.7	21	< 5	1.9	< 3	< 0.2	29.2	73.11	14.25	1.97	0.027	0.42	1.44	4.7	2.69	0.188	0.05	1.66	100.5

		_	, ,		,										,	,													
Ba	mdd	2	FUS-	390	555	505	522	419	466	348	529	603	396	481	507	265	14	655	570	330	919	899	471	213	561	495	391	682	711
Cs	mdd	0.1	FUS- MS	2.4	1.9	2.6	3.3	2.3	3.8	1.3	1.8	1.5	1	2.2	1.7	1.9	1.3	6.4	2.6	2.9	0.7	2.1	2.3	0.3	1.6	9.0	1.9	9.0	2.8
Sb	udd	0.2	FUS- MS	0.3	9.0	< 0.2	< 0.2	4	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	9.4	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2
uS	uıdd	1	FUS- MS	1	1	1	1	1	1	1	1	1	1	1	1	1	< 1	-	1	1	1	1	1	1	1	1	1	1	-
In	uıdd	0.1	FUS- MS	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
gA	mdd	6.0	TD- ICP	< 0.3	6.0	< 0.3	< 0.3	< 0.3	< 0.3	0.4	0.3	< 0.3	0.5	< 0.3	6.0	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	6.0	< 0.3	< 0.3	6.0	£:0>	< 0.3	< 0.3
оМ	uıdd	2	FUS- MS	< 2	3	< 2	< 2	< 2	< 2	< 2	< 2	2	2	< 2	2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	2	< 2
9N	mdd	0.2	FUS- MS	4	3.6	3.8	3.8	4.2	4.3	4	4.1	4.1	3.6	3.4	4.1	4.3	3.5	6.4	3.9	4.7	2.3	4.1	3.3	5.2	4.1	2.7	3.3	4	3.7
Σr	mdd	1	FUS- ICP	145	112	88	58	134	86	152	62	98	143	74	88	88	15	125	91	501	99	68	611	601	86	125	02	86	108
Ā	mdd	6.0	FUS- MS	4.2	6.4	5.3	4.7	4.3	5.2	5.5	4.8	5	3.6	9	5.3	5.2	6.4	20.2	4.9	9.6	3.3	4.5	4.3	14.5	4.4	6	4	4.4	4.6
Sr	mdd	2	FUS- ICP	258	191	861	181	244	194	238	182	163	253	155	180	222	91	682	192	208	136	193	236	1226	907	310	153	192	203
Rb	ppm	1	FUS- MS	50	91	87	66	52	88	58	98	98	49	105	81	90	126	104	93	73	134	98	70	37	83	34	98	115	108
As	mdd	5	FUS- MS	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5
Ge	mdd	0.5	FUS- MS	1	1	1	1	1	1.1	1	1.1	1.1	6.0	1.1	1	1.1	1.8	1.4	1	8.0	8.0	1	8.0	1.8	6.0	1	1	1.1	6.0
Ga	mdd	1	FUS- MS	19	18	19	18	18	18	20	18	18	18	18	18	18	19	14	18	19	15	18	16	17	18	19	17	19	17
S	%	0.001	TD- ICP	0.003	0.003	0.003	0.003	0.004	0.002	0.003	0.003	0.002	0.004	0.003	0.002	0.014	0.001	0.067	0.003	0.003	0.003	0.003	0.002	0.004	0.002	0.003	0.003	0.001	0.022
Cd	mdd	0.5	TD- ICP	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
Zn	mdd	1	TD- ICP	99	57	46	4	29	48	39	39	38	64	37	39	43	14	62	42	58	21	42	48	81	48	48	35	17	21
Cu	mdd	1	TD-ICP	13	10	12	7	7	12	21	6	8	15	7	8	7	2	71	9	6	25	8	15	43	6	37	8	3	13
Ni	mdd	1	TD- ICP	4	4	3	2	4	1	5	2	2	4	2	2	4	× 1	197	2	2	2	2	9	75	2	4	1	2	3
Co	mdd	1	FUS- MS	4	2	2	2	5	2	5	2	2	5	2	2	2	^ \	45	2	4	2	2	5	26	2	2	2	2	2
Cr	mdd	20	FUS- MS	< 20	30	20	20	30	20	20	< 20	40	20	40	30	30	< 20	750	< 20	20	30	20	20	420	20	< 20	< 20	20	20
Λ	mdd	5	FUS-	25	14	12	10	26	12	29	10	11	24	6	12	12	< 5	233	13	23	10	12	29	140	14	21	6	14	14
Be	mdd	1	FUS-	1	1	1	1	1	2	-	-	1	\ \ \	1	1	1	1	-	-	1	^ 1	1	^ 1	1	1	1	-	^ \	-
Sc	mdd	1	FUS-	3	2	2	2	3	2	3	2	2	3	2	2	2	1	39	2	3	2	2	4	20	2	2	2	2	2
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH02_LG001	IG_BH02_LG002	IG_BH02_LG003	IG_BH02_LG004	IG_BH02_LG005	IG_BH02_LG006	IG_BH02_LG007	IG_BH02_LG008	IG_BH02_LG009	IG_BH02_LG010	IG_BH02_LG011	IG_BH02_LG012	IG_BH02_LG013	IG_BH02_LG014	IG_BH02_LG015	IG_BH02_LG016	IG_BH02_LG017	IG_BH02_LG018	IG_BH02_LG019	IG_BH02_LG020	IG_BH02_LG021	IG_BH02_LG022	IG_BH02_LG023	IG_BH02_LG024	IG_BH02_LG025	IG_BH02_LG026

							l	l			l							l	1										
n	mdd	0.01	FUS- MS	1.74	1.62	1.4	2.21	1.3	2.85	1.91	1.77	1.36	-	1.15	1.8	2.45	7.31	2.12	1.93	2.54	1.37	2.71	1.42	2.04	1.15	1.19	1.6	2.34	1.9
Th	udd	0.05	FUS- MS	4.41	5.06	6.77	6.22	4.73	6.7	4.4	5.59	6.46	4.12	6.49	6.19	7.02	1.97	5.95	6.21	4.34	3.05	5.78	4.71	5.41	5.48	6.52	4.47	4.6	5.12
Bi	udd	0.1	FUS- MS	0.7	< 0.1	< 0.1	< 0.1	0.1	< 0.1	0.4	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.2	< 0.1	0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
Li	uıdd	1	TD- ICP	93	69	92	71	77	08	68	62	44	98	99	42	08	11	242	61	71	25	\$9	28	55	41	54	64	28	58
Pb	mdd	5	TD- ICP	6	7	6	10	> 5	∞	12	6	7	> 5	∞	∞	∞	18	11	∞	∞	7	∞	7	20	∞	< > 5	11	< > 5	< 5
II	uıdd	90.0	FUS- MS	0.37	0.44	6.4	0.41	0.31	0.39	0.31	0.33	98.0	0.25	0.49	0.32	68.0	5:0	0.47	0.43	0.37	0.55	94.0	0.34	0.19	0.34	0.13	0.35	0.47	0.47
W	ppm	0.5	FUS- MS	1.2	< 0.5	< 0.5	3.9	< 0.5	0.7	9.0	2.6	1.8	4.4	< 0.5	< 0.5	2.1	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	0.7	< 0.5	0.5	26.1	9.0	< 0.5	0.7	0.9
Та	mdd	0.01	FUS- MS	0.54	0.82	0.91	62.0	0.61	6:0	0.53	1.07	1	0.47	0.82	6.0	0.83	1.14	0.38	98.0	1.28	0.39	6.0	0.55	0.33	0.94	1.41	69:0	0.67	0.78
Hf	mdd	0.1	FUS- MS	3.6	2.8	2.6	2.6	3.4	2.8	3.6	2.3	2.5	3.5	2.5	2.5	2.7	1.3	3.2	2.7	2.8	1.8	2.6	3	2.9	2.6	4	2.1	2.7	3
Lu	mdd	0.002	FUS-MS	0.045	0.053	90.0	0.051	0.045	0.056	0.061	0.051	0.055	0.036	0.074	0.058	90.0	0.126	0.248	0.061	0.067	0.031	0.048	90:0	0.218	0.059	0.082	0.042	0.054	0.048
Yb	mdd	0.01	FUS- MS	0.32	0.36	0.41	0.37	0.34	0.39	0.43	0.34	0.38	0.24	0.49	0.38	0.41	0.82	1.67	0.39	0.47	0.23	0.32	0.37	1.4	0.36	0.55	0.28	0.36	0.33
Tm	mdd	0.005	FUS-MS	0.05	0.056	0.064	0.057	0.055	0.059	0.062	0.053	0.058	0.039	0.071	0.059	0.061	0.121	0.275	0.057	0.073	0.034	0.054	0.055	0.201	0.051	0.088	0.048	0.054	0.05
Er.	mdd	0.01	FUS- MS	0.36	0.43	0.44	0.39	0.42	0.45	0.46	0.4	0.42	0.29	0.5	0.44	0.42	0.83	1.96	0.43	0.49	0.23	0.38	0.35	1.4	0.36	79.0	0.37	0.36	0.39
Но	mdd	0.01	FUS- MS	0.12	0.15	0.16	0.14	0.14	0.16	0.18	0.15	0.15	0.11	0.19	0.17	0.16	0.29	0.72	0.16	0.18	60.0	0.14	0.14	0.48	0.14	0.27	0.13	0.14	0.14
Dy	mdd	0.01	FUS- MS	0.84	0.88	1.01	0.88	0.92	1.01	1.1	96:0	0.91	0.78	1.05	1.08	0.93	1.54	4.11	68.0	1.03	0.58	0.83	0.81	2.57	0.85	1.7	92.0	0.92	0.87
Tb	mdd	0.01	FUS- MS	0.18	0.18	0.2	0.18	0.19	0.21	0.23	0.18	0.19	0.19	0.21	0.21	0.19	0.24	0.78	0.18	0.18	0.12	0.18	0.17	0.49	0.18	0.35	0.14	0.17	0.18
PD	mdd	0.01	FUS- MS	1.44	1.39	1.49	1.43	1.62	1.52	1.75	1.39	1.51	1.43	1.53	1.46	1.57	1.22	6.5	1.49	1.34	68.0	1.27	1.28	3.96	1.35	2.77	1.12	1.34	1.45
Eu	mdd	0.005	FUS-MS	0.614	0.494	0.412	0.431	0.675	0.463	0.746	0.409	0.444	0.619	0.401	0.467	0.408	0.139	2.42	0.476	0.478	0.468	0.478	0.509	1.35	0.464	0.937	0.381	0.509	0.399
Sm	mdd	0.01	FUS- F	1.9	1.91	2.08	1.82	2.17	1.98	2.26	1.76	1.91	2.07	1.97	1.94	1.99	86:0	8.86	1.94	1.81	1.09	1.86	1.83	5.21	1.77	2.95	1.4	1.84	1.79
PN	mdd	0.05	FUS- I	11.1	11.9	10.5	10.2	13.8	10.8	12.5	9.74	10.5	12.7	9.47	10.3	11.1	2.71	49.4	10.4	11.6	5.94	10.1	10.9	30.5	10.5	11.3	7.8	11.8	98.6
Pr	d udd	0.01	FUS- F MS	3.08	3.58	2.99	3	3.88	3.13	3.45	2.74	3.01	3.69	2.62	2.94	3.15	0.73	12.4	3.1	3.41	1.73	2.93	3.19	7.8	2.98	3.08	2.25	3.42	2.87
ce	d udd	0.05	FUS- F	31.1	36.2	29.3	28.9	38.3	30.5	37.5	26.8	29.5	36.7	25.4	29.2	31 3	6.48	101	30.6	34.5	16.9	29.4	32.2	2:69	29.9	30.1	22.3	34.4	28.5
La (ld udd	0.05 0.	FUS- FU MS N	16.2	20.7	16.9	16.4	21.7	17.2 3	18.4 3	15.1	16.5	20.4 3	14.1	16.2	18.1	3.31 6.	1 45.9	17.6 30	19.6 3-	10.4	16.8	18.2 3.	37.9 6	17.1	15.4 30	12.7 2:	21 3.	16.8
	Jd																												
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH02_LG001	IG_BH02_LG002	IG_BH02_LG003	IG_BH02_LG004	IG_BH02_LG005	IG_BH02_LG006	IG_BH02_LG007	IG_BH02_LG008	IG_BH02_LG009	IG_BH02_LG010	IG_BH02_LG011	IG_BH02_LG012	IG_BH02_LG013	IG_BH02_LG014	IG_BH02_LG015	IG_BH02_LG016	IG_BH02_LG017	IG_BH02_LG018	IG_BH02_LG019	IG_BH02_LG020	IG_BH02_LG021	IG_BH02_LG022	IG_BH02_LG023	IG_BH02_LG024	IG_BH02_LG025	IG_BH02_LG026

	0.52	6.64	0.39	1.24	4.44	2.54	2.7	0.42	0.38	2.43	0.33	0.24	4.54	10.54	0.54	3.17
10	5	< 3	7	7	< 3	19	< 3	8	6	8	5	7	< 3	< 3	3	4
< 0.3	0.4	< 0.3	0.3	0.3	< 0.3	< 0.3	< 0.3	0.3	0.3	0.3	< 0.3	0.4	0.3	< 0.3	0.3	< 0.3
0.003	0.005	0.003	0.004	90000	0.003	0.005	0.003	90000	0.003	0.071	0.005	0.003	900.0	0.003	0.003	0.005
< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
23	89	84	79	99	84	27	121	118	85	89	69	46	28	16	89	69
7	12	2	12	∞	3	∞	4	20	11	09	18	9	16	22	7	∞
3	4	213	5	9	229	127	258	175	7	306	7	4	419	81	9	185
1.49	1.51	0.75	1.62	1.32	0.75	1.17	8:0	66:0	2.14	2.22	1.55	1.32	1.75	0.77	0.83	0.94
5.19	7.44	2.85	80.6	5.74	2.84	3.59	3.01	5.37	60.9	9.39	5.73	5.88	6.03	2.9	98.9	3.26
< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.1	0.2	< 0.1	< 0.1	< 0.1	0.1	0.1	< 0.1	0.2	< 0.1	< 0.1	0.1
0.26	0.34	0.94	0.37	0.3	7.00	0.32	7:0	0.32	0.35	0.34	0.31	0.29	0.56	0.32	0.13	0.26
< 0.5	< 0.5	< 0.5	< 0.5	6.0	< 0.5	0.5	< 0.5	< 0.5	< 0.5	9:0	< 0.5	< 0.5	9.0	1.2	< 0.5	0.5
0.36	2.0	0.14	0.57	89:0	0.14	61.0	0.14	0.41	0.82	0.3	0.52	0.38	68.0	0.51	0.32	0.15
2.3	3.9	1.9	4.3	4	1.9	2.4	2	4.2	3.8	3.8	4.7	3.5	3.2	1.8	4.9	2.1
0.049	0.052	0.191	0.05	0.074	0.195	0.238	0.179	0.049	0.068	0.191	0.061	0.043	0.156	0.182	0.044	0.203
0.3	0.34	1.14	0.31	0.51	1.26	1.54	1.16	0.3	0.45	1.33	0.34	0.33	1.02	1.2	0.26	1.34
IG_BH01_LG029	IG_BH01_LG030	IG_BH01_LG031	IG_BH01_LG032	IG_BH01_LG033	IG_BH01_LG034	IG_BH01_LG035	IG_BH01_LG036	IG_BH01_LG037	IG_BH01_LG038	IG_BH01_LG039	IG_BH01_LG040	IG_BH01_LG041	IG_BH01_LG042	IG_BH01_LG043	IG_BH01_LG044	IG_BH01_LG045

certified values in one element, but other elements in the material were within an acceptable range of > 90% accuracy so these precision), and the method blanks all returned measurements below, or within an acceptable range of, detection limit (no discernable discrepancies were deemed insignificant. Duplicate sample measurements averaged within 7 % of original measured values (>93% All analyses performed by Activation Laboratories Ltd. in Thunder Bay or Ancaster, Ontario. In addition to the results included below, 63 certified reference materials (to measure accuracy), 7 duplicates (to measure precision), and 18 method blanks (to monitor contamination) were analysed by Actlabs to ensure overall quality control in the lithogeochemical analyses. Certified reference material (CRM) measurements averaged within 0.1 % of certified values (>99 % accuracy). There were a few CRM samples that were >10% off their contamination)

Lithogeochemical results of various tests carried out by Activation Laboratories (NWMO, 2022d).

Report Number: A20-05149	er: A20-0514	61																					
Report Date: 18/6/2020	18/6/2020																						
Analyte Symbol	ţ	E	Au	As	Br	Cr	Ir	Sc	Se	Sb	Mass	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O	K20	TiO2	P205	IOI	Total
Unit Symbol	From (position	To (position	qdd	mdd	mdd	mdd	qdd	mdd	mdd	mdd	50	%	%	%	%	%	%	%	%	%	%	%	%
Detection Limit	borehole;	borehole;	2	0.5	0.5	5	5	0.1	3	0.2		0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01
Analysis Method	(i	(III	INA A	INA A	INA	NA A	INA A	NA A	INA	NA A	INA A	FUS-	FUS- ICP	FUS-ICP	FUS- ICP	FUS- ICP	FUS- ICP	FUS- ICP	FUS-	FUS-	FUS-	GRA V	FUS- ICP
IG_BH02_LG001	15.29	15.41	< 2	< 0.5	< 0.5	17	< > 5	2.7	< 3	< 0.2	32.3	70.95	14.65	2.7	0.035	89.0	2.93	4.42	1.66	0.296	60.0	0.48	68.86
IG_BH02_LG002	15.75	15.9	< 2	< 0.5	< 0.5	28	< > 5	1.9	< 3	_	31.2	72.86	13.56	2.11	0.036	0.33	1.86	4.2	2.58	0.22	0.07	0.58	98.41
IG_BH02_LG003	60.05	60.2	< 2	< 0.5	< 0.5	17	< × 5	2	< 3	< 0.2	30.8	72.33	14.43	1.91	0.025	0.34	1.24	4.5	2.73	0.175	0.05	8.0	98.54
IG_BH02_LG004	73.47	73.68	< 2	< 0.5	< 0.5	19	\$ >	1.9	< 3	< 0.2	1.07	74.01	14.09	1.73	0.033	0.32	1.74	4.39	3.01	0.16	0.03	0.45	96.66
IG_BH02_LG005	113.1	113.28	< 2	< 0.5	< 0.5	22	\$ >	2.8	< 3	7.1	31.1	71.59	14.6	2.96	0.039	0.71	3.06	4.34	1.75	0.308	0.1	0.5	99.95
IG_BH02_LG006	132.25	132.4	< 2	0.5	< 0.5	22	< 5	2	< 3	< 0.2	1.03	74.22	14.18	1.86	0.036	0.37	2.02	4.53	2.47	0.174	0.04	89.0	100.6
IG_BH02_LG007	280.6	280.8	< 2	< 0.5	18.1	14	< > 5	2.9	< 3	< 0.2	32.3	70.15	14.51	2.92	0.049	0.75	1.91	5.54	1.22	0.315	60.0	1.39	98.85
IG_BH02_LG008	284.5	284.7	< 2	< 0.5	1.2	20	< > 5	1.8	< 3	< 0.2	29.6	72.63	14.02	1.7	0.028	0.32	1.58	4.47	2.9	0.154	0.04	1.65	99.49
IG_BH02_LG009	346.75	347	< 2	< 0.5	11.7	35	< 5	2.1	< 3	< 0.2	28.1	72.64	13.75	1.81	0.034	0.38	1.42	4.52	2.91	0.164	0.05	0.93	98.61
IG_BH02_LG010	368.5	368.65	< 2	0.5	< 0.5	23	< 5	2.9	< 3	< 0.2	31.1	71.82	14.62	2.74	0.032	0.65	3.09	4.5	1.71	0.306	0.1	0.37	99.92
IG_BH02_LG011	372.24	372.44	< 2	< 0.5	< 0.5	29	> 5	1.8	< 3	0.4	30.7	73.51	13.5	1.55	0.031	0.27	1.53	4.38	3.24	0.131	0.04	0.48	99.86
IG_BH02_LG012	381.03	381.2	< 2	< 0.5	1.1	30	\$ >	2.2	< 3	< 0.2	30.4	73.34	13.55	1.88	0.037	0.42	1.56	4.4	2.81	0.16	0.05	0.87	90.66
IG_BH02_LG013	478.22	478.44	< 2	< 0.5	< 0.5	22	< 5	2.2	< 3	< 0.2	30.5	73.13	13.74	1.84	0.033	0.31	1.78	4.38	2.84	0.167	0.04	1.92	100.2
IG_BH02_LG014	531.25	531.47	< 2	< 0.5	< 0.5	22	< 5	1	< 3	< 0.2	32.5	75.78	13.04	0.78	0.015	90.0	0.59	4.3	4.59	0.038	0.01	0.19	99.41
IG_BH02_LG015	536.48	536.82	< 2	< 0.5	< 0.5	899	< 5	34.5	< 3	< 0.2	30.2	42.77	12	10.3	0.159	10.24	11.38	1.99	3.26	0.94	0.47	5.72	99.24

IG_BH02_LG016	567.26	567.44	< 2	< 0.5	11.2	13	< × ×	2.1	< 3	< 0.2	1.04	74.5	14.07	1.88	0.034	0.34	1.84	4.47	2.97	0.165	0.04	0.46	100.8
IG_BH02_LG017	573.51	573.68	< 2	< 0.5	18.5	21	< > 5	2.9	< 3	0.3	31.7	72.59	14.75	2.59	0.045	0.55	2.63	4.68	1.75	0.25	90:0	0.56	100.5
IG_BH02_LG018	600.43	600.62	< 2	< 0.5	4	24	< 5	1.6	< 3	< 0.2	30.5	74.38	13.74	1.43	0.026	0.32	1.06	4.11	4.51	0.13	0.04	89.0	100.4
IG_BH02_LG019	632.98	633.26	< 2	< 0.5	< 0.5	23	< 5	1.9	< 3	< 0.2	1.05	73.68	14.16	1.8	0.033	0.33	1.75	4.36	3.09	0.163	0.05	0.43	99.85
IG_BH02_LG020	668.29	668.44	2	< 0.5	1.7	23	< 5	3.8	< 3	< 0.2	30	72.9	13.25	2.65	0.037	89.0	2.7	3.93	1.75	0.26	0.07	1.13	99.37
IG_BH02_LG021	672.2	672.4	< 2	1	1.4	376	< 5	19	< 3	0.4	33.1	55.35	14.26	7.12	0.133	5.49	9.57	3.46	1.02	0.574	0.18	3.11	100.3
IG_BH02_LG022	680.74	680.91	3	1.7	8.9	22	> 5	2	< 3	< 0.2	30.3	74.15	13.99	1.82	0.033	0.36	1.86	4.46	2.63	0.167	0.04	99.0	100.2
IG_BH02_LG023	682.58	682.74	< 2	< 0.5	14.3	< 5	< > 5	2.3	< 3	0.2	30.8	58.9	18.74	1.97	0.037	0.53	4.01	8.61	1.27	0.249	90:0	4.12	98.49
IG_BH02_LG024	702.01	702.19	< 2	< 0.5	5.1	21	< 5	1.7	< 3	< 0.2	31.2	75.04	13.86	1.62	0.029	0.25	1.62	4.64	2.93	0.13	0.03	0.47	100.6
IG_BH02_LG025	777.49	777.75	< 2	< 0.5	13.6	24	< 5	1.9	< 3	< 0.2	32	72.74	14.3	1.85	0.031	0.32	1.99	4.97	3.1	0.166	0.05	1.05	100.6
IG_BH02_LG026	904.5	904.74	< 2	< 0.5	13.7	21	< 5	1.9	< 3	< 0.2	29.2	73.11	14.25	1.97	0.027	0.42	1.44	4.7	2.69	0.188	0.05	1.66	100.5

					,															,									
Ba	mdd	2	FUS-	390	555	505	522	419	466	348	529	603	396	481	507	595	14	655	570	330	919	899	471	213	561	495	391	682	711
Cs	mdd	0.1	FUS- MS	2.4	1.9	2.6	3.3	2.3	3.8	1.3	1.8	1.5	1	2.2	1.7	1.9	1.3	6.4	2.6	2.9	0.7	2.1	2.3	0.3	1.6	9.0	1.9	9.0	2.8
Sb	mdd	0.2	FUS- MS	0.3	9.0	< 0.2	< 0.2	4	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	9.4	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2
uS	udd	1	FUS- MS	-	1	1	1	1	1	1	1	1	1	1	1	1	< 1	1	1	1	1	1	1	1	1	1	1	1	-
In	udd	0.1	FUS- MS	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
gA	mdd	6.0	TD- ICP	< 0.3	6.0	< 0.3	< 0.3	< 0.3	< 0.3	0.4	0.3	< 0.3	5:0	< 0.3	6.0	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	< 0.3	6.0	< 0.3	< 0.3	6.0	< 0.3	< 0.3	< 0.3
оМ	udd	2	FUS- MS	< ×	3	< 2	< 2	< 2	< 2	< 2	< 2	2	2	< 2	2	< 2	7>	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	2	< 2
9N	mdd	0.2	FUS- MS	4	3.6	3.8	3.8	4.2	4.3	4	4.1	4.1	3.6	3.4	4.1	4.3	3.5	6.4	3.9	4.7	2.3	4.1	3.3	5.2	4.1	2.7	3.3	4	3.7
Σr	mdd	1	FUS- ICP	145	112	88	85	134	86	152	62	98	143	74	88	88	15	125	91	501	99	68	611	601	86	125	02	86	108
Ā	mdd	5:0	FUS- MS	4.2	4.9	5.3	4.7	4.3	5.2	5.5	4.8	5	3.6	9	5.3	5.2	9.4	20.2	4.9	9.6	3.3	4.5	4.3	14.5	4.4	6	4	4.4	4.6
Sr	mdd	2	FUS- ICP	258	191	861	181	244	194	238	182	163	253	155	180	222	91	682	192	208	136	193	236	1226	907	310	153	192	203
Rb	ppm	1	FUS- MS	50	91	87	66	52	88	58	98	98	49	105	81	90	126	104	93	73	134	98	70	37	83	34	98	115	108
As	mdd	5	FUS- MS	> 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5
Ge	mdd	0.5	FUS- MS	1	1	1	1	1	1.1	1	1.1	1.1	6.0	1.1	1	1.1	1.8	1.4	1	8.0	8.0	1	8.0	1.8	6.0	1	1	1.1	6.0
Ga	mdd	1	FUS- MS	19	18	19	18	18	18	20	18	18	18	18	18	18	19	14	18	19	15	18	16	17	18	19	17	19	17
S	%	0.001	TD- ICP	0.003	0.003	0.003	0.003	0.004	0.002	0.003	0.003	0.002	0.004	0.003	0.002	0.014	0.001	0.067	0.003	0.003	0.003	0.003	0.002	0.004	0.002	0.003	0.003	0.001	0.022
Cd	mdd	0.5	TD- ICP	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
Zn	mdd	1	TD- ICP	99	57	46	4	29	48	39	39	38	64	37	39	43	14	62	42	58	21	42	48	81	48	48	35	17	21
Cu	mdd	1	TD-ICP	13	10	12	7	7	12	21	6	8	15	7	8	7	2	71	9	6	25	8	15	43	6	37	8	3	13
Ni	mdd	1	TD- ICP	4	4	3	2	4	1	5	2	2	4	2	2	4	< 1 ×	197	2	2	2	2	9	75	2	4	1	2	3
Co	mdd	1	FUS- MS	4	2	2	2	S	2	S	2	2	S	2	2	2	^ \	45	2	4	2	2	S	26	2	2	2	2	2
Cr	mdd	20	FUS- MS	< 20	30	20	20	30	20	20	< 20	40	20	40	30	30	< 20	750	< 20	20	30	20	20	420	20	< 20	< 20	20	20
Λ	mdd	5	FUS-	25	14	12	10	26	12	29	10	11	24	6	12	12	< 5	233	13	23	10	12	29	140	14	21	6	14	14
Ве	mdd	1	FUS-	-	1	1	1	1	2	1	1	1	^ _	1	1	1	1	1	1	1	^ _	1	^ _	1	1	1	1	^ \	-1
Sc	mdd	1	FUS-	8	2	2	2	3	2	3	2	2	3	2	2	2	1	39	2	3	2	2	4	20	2	2	2	2	2
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH02_LG001	IG_BH02_LG002	IG_BH02_LG003	IG_BH02_LG004	IG_BH02_LG005	IG_BH02_LG006	IG_BH02_LG007	IG_BH02_LG008	IG_BH02_LG009	IG_BH02_LG010	IG_BH02_LG011	IG_BH02_LG012	IG_BH02_LG013	IG_BH02_LG014	IG_BH02_LG015	IG_BH02_LG016	IG_BH02_LG017	IG_BH02_LG018	IG_BH02_LG019	IG_BH02_LG020	IG_BH02_LG021	IG_BH02_LG022	IG_BH02_LG023	IG_BH02_LG024	IG_BH02_LG025	IG_BH02_LG026

							l	l			l							l	1										
n	mdd	0.01	FUS- MS	1.74	1.62	1.4	2.21	1.3	2.85	1.91	1.77	1.36	-	1.15	1.8	2.45	7.31	2.12	1.93	2.54	1.37	2.71	1.42	2.04	1.15	1.19	1.6	2.34	1.9
Th	udd	0.05	FUS- MS	4.41	5.06	6.77	6.22	4.73	6.7	4.4	5.59	6.46	4.12	6.49	6.19	7.02	1.97	5.95	6.21	4.34	3.05	5.78	4.71	5.41	5.48	6.52	4.47	4.6	5.12
Bi	udd	0.1	FUS- MS	0.7	< 0.1	< 0.1	< 0.1	0.1	< 0.1	0.4	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.2	< 0.1	0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
Li	uıdd	1	TD- ICP	93	69	92	71	77	08	68	62	44	98	99	42	08	11	242	61	71	25	\$9	28	55	41	54	64	28	58
Pb	mdd	5	TD- ICP	6	7	6	10	> 5	∞	12	6	7	> 5	∞	∞	∞	18	11	∞	∞	7	∞	7	20	∞	< > 5	11	< > 5	< 5
II	uıdd	90.0	FUS- MS	0.37	0.44	6.4	0.41	0.31	0.39	0.31	0.33	98.0	0.25	0.49	0.32	68.0	5:0	0.47	0.43	0.37	0.55	94.0	0.34	0.19	0.34	0.13	98.0	0.47	0.47
W	ppm	0.5	FUS- MS	1.2	< 0.5	< 0.5	3.9	< 0.5	0.7	9.0	2.6	1.8	4.4	< 0.5	< 0.5	2.1	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	0.7	< 0.5	0.5	26.1	9.0	< 0.5	0.7	0.9
Та	mdd	0.01	FUS- MS	0.54	0.82	0.91	62.0	0.61	6:0	0.53	1.07	1	0.47	0.82	6.0	0.83	1.14	0.38	98.0	1.28	0.39	6.0	0.55	0.33	0.94	1.41	69.0	0.67	0.78
Hf	mdd	0.1	FUS- MS	3.6	2.8	2.6	2.6	3.4	2.8	3.6	2.3	2.5	3.5	2.5	2.5	2.7	1.3	3.2	2.7	2.8	1.8	2.6	3	2.9	2.6	4	2.1	2.7	3
Lu	mdd	0.002	FUS-MS	0.045	0.053	90.0	0.051	0.045	0.056	0.061	0.051	0.055	0.036	0.074	0.058	90:0	0.126	0.248	0.061	0.067	0.031	0.048	90:0	0.218	0.059	0.082	0.042	0.054	0.048
Yb	mdd	0.01	FUS- MS	0.32	0.36	0.41	0.37	0.34	0.39	0.43	0.34	0.38	0.24	0.49	0.38	0.41	0.82	1.67	0.39	0.47	0.23	0.32	0.37	1.4	0.36	0.55	0.28	0.36	0.33
Tm	mdd	0.005	FUS-MS	0.05	0.056	0.064	0.057	0.055	0.059	0.062	0.053	0.058	0.039	0.071	0.059	0.061	0.121	0.275	0.057	0.073	0.034	0.054	0.055	0.201	0.051	0.088	0.048	0.054	0.05
Er.	mdd	0.01	FUS- MS	0.36	0.43	0.44	0.39	0.42	0.45	0.46	0.4	0.42	0.29	0.5	0.44	0.42	0.83	1.96	0.43	0.49	0.23	0.38	0.35	1.4	0.36	79.0	0.37	0.36	0.39
Но	mdd	0.01	FUS- MS	0.12	0.15	0.16	0.14	0.14	0.16	0.18	0.15	0.15	0.11	0.19	0.17	0.16	0.29	0.72	0.16	0.18	60.0	0.14	0.14	0.48	0.14	0.27	0.13	0.14	0.14
Dy	mdd	0.01	FUS- MS	0.84	0.88	1.01	0.88	0.92	1.01	1.1	96:0	0.91	0.78	1.05	1.08	0.93	1.54	4.11	68.0	1.03	0.58	0.83	0.81	2.57	0.85	1.7	92.0	0.92	0.87
Tb	bpm	0.01	FUS- MS	0.18	0.18	0.2	0.18	0.19	0.21	0.23	0.18	0.19	0.19	0.21	0.21	0.19	0.24	0.78	0.18	0.18	0.12	0.18	0.17	0.49	0.18	0.35	0.14	0.17	0.18
PD	mdd	0.01	FUS- MS	1.44	1.39	1.49	1.43	1.62	1.52	1.75	1.39	1.51	1.43	1.53	1.46	1.57	1.22	6.5	1.49	1.34	68.0	1.27	1.28	3.96	1.35	2.77	1.12	1.34	1.45
Eu	mdd	0.005	FUS-MS	0.614	0.494	0.412	0.431	0.675	0.463	0.746	0.409	0.444	0.619	0.401	0.467	0.408	0.139	2.42	0.476	0.478	0.468	0.478	0.509	1.35	0.464	0.937	0.381	0.509	0.399
Sm	mdd	0.01	FUS- F	1.9	1.91	2.08	1.82	2.17	1.98	2.26	1.76	1.91	2.07	1.97	1.94	1.99	86:0	8.86	1.94	1.81	1.09	1.86	1.83	5.21	1.77	2.95	1.4	1.84	1.79
PN	mdd	0.05	FUS- I	11.1	11.9	10.5	10.2	13.8	10.8	12.5	9.74	10.5	12.7	9.47	10.3	11.1	2.71	49.4	10.4	11.6	5.94	10.1	10.9	30.5	10.5	11.3	7.8	11.8	98.6
Pr	d udd	0.01	FUS- F MS	3.08	3.58	2.99	3	3.88	3.13	3.45	2.74	3.01	3.69	2.62	2.94	3.15	0.73	12.4	3.1	3.41	1.73	2.93	3.19	7.8	2.98	3.08	2.25	3.42	2.87
2	d udd	0.05 0	FUS- F MS N	31.1 3	36.2 3	29.3	28.9	38.3 3	30.5	37.5 3	26.8 2	29.5	36.7 3	25.4 2	29.2	31 3	6.48 0	101	30.6	34.5	16.9	29.4	32.2	. 2.69	29.9	30.1	22.3	34.4 3	28.5 2
La (ld udd	0.05 0.	FUS- FU MS N	16.2	20.7	16.9	16.4	21.7	17.2 3	18.4 3	15.1	16.5	20.4 3	14.1	16.2	18.1	3.31 6.	1 45.9	17.6 30	19.6 3-	10.4	16.8	18.2 3.	37.9 6	17.1	15.4 3	12.7	21 3.	16.8
	Jd																												
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH02_LG001	IG_BH02_LG002	IG_BH02_LG003	IG_BH02_LG004	IG_BH02_LG005	IG_BH02_LG006	IG_BH02_LG007	IG_BH02_LG008	IG_BH02_LG009	IG_BH02_LG010	IG_BH02_LG011	IG_BH02_LG012	IG_BH02_LG013	IG_BH02_LG014	IG_BH02_LG015	IG_BH02_LG016	IG_BH02_LG017	IG_BH02_LG018	IG_BH02_LG019	IG_BH02_LG020	IG_BH02_LG021	IG_BH02_LG022	IG_BH02_LG023	IG_BH02_LG024	IG_BH02_LG025	IG_BH02_LG026

off their certified values in one element, but other elements in the material were within an acceptable range of > 90% accuracy so these discrepancies were deemed insignificant. Duplicate sample measurements averaged within 7 % of original measured values (>93% precision), and the method blanks all returned measurements below, or within an acceptable range of, detection limit (no discernable (CRM) measurements averaged within 0.1 % of certified values (>99 % accuracy). There were a few CRM samples that were >10% contamination) were analysed by Actlabs to ensure overall quality control in the lithogeochemical analyses. Certified reference material All analyses performed by Activation Laboratories Ltd. in Thunder Bay or Ancaster, Ontario. In addition to the results included below, 63 certified reference materials (to measure accuracy), 7 duplicates (to measure precision), and 18 method blanks (to monitor contamination)

Lithogeochemical results of various tests carried out by Activation Laboratories (NWMO, 2022d)

Report Number: A20-05149	20-05149																						
Report Date: 18/6/2020	2020																						
Analyte Symbol	ı		ηV	As	Br	Cr	Ir	Sc	Se	Sb	Mass	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O	K20	TiO2	P205	TOI	Total
Unit Symbol	From (position	To (position	qdd	mdd	mdd	mdd	qdd	mdd	mdd	mdd	50	%	%	%	%	%	%	%	%	%	%	%	%
Detection Limit	borehole;	along borehole;	2	0.5	0.5	5	5	0.1	3	0.2		0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01
Analysis Method	(m)	m)	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	FUS-	FUS-	FUS-ICP	FUS- ICP	FUS-	FUS-	FUS-	FUS-	FUS-	FUS-	GRAV	FUS-
IG_BH03_LG001	74.00	74.30	< 2	< 0.5	< 0.5	27	< 5	2	< 3	< 0.2	1.06	73.39	14.24	1.77	0.03	0.39	1.99	4.52	2.46	0.189	90.0	9.0	99.64
IG_BH03_LG002	131.01	131.25	< 2	< 0.5	< 0.5	28	< 5	2.1	< 3	< 0.2	1.04	72.92	13.84	1.77	0.032	0.38	1.99	4.54	2.21	0.183	0.05	0.59	98.51
IG_BH03_LG003	149.48	149.64	< 2	6:0	< 0.5	369	< 5	16.6	< 3	< 0.2	34	53.93	13.46	8.28	0.125	7.9	8.18	3.24	2.12	0.728	0.2	2.08	100.2
IG_BH03_LG004	184.72	184.83	< 2	6.0	< 0.5	18	< 5	1.2	< 3	< 0.2	32.7	77.01	12.83	6.0	0.022	0.17	0.5	4.14	4.61	0.077	0.01	0.27	100.5
IG_BH03_LG005	231.97	232.16	< 2	9:0	< 0.5	21	< 5	4	< 3	< 0.2	36.4	71.09	14.53	3.22	0.04	0.81	3.17	4.26	1.86	0.447	0.12	0.5	100.1
IG_BH03_LG006	363.51	363.73	< 2	< 0.5	6.5	17	< 5	2.2	< 3	< 0.2	34	72.26	14.69	1.67	0.023	0.37	1.31	4.27	3.33	0.164	0.05	1.27	99.4
IG_BH03_LG007	477.80	477.98	< 2	< 0.5	1.7	20	< 5	0.7	< 3	< 0.2	32.5	75.89	12.9	0.59	0.014	90.0	0.51	4.42	4.28	0.036	<0.01	0.33	99.04
IG_BH03_LG008	489.97	490.15	< 2	< 0.5	< 0.5	46	< 5	4.1	< 3	< 0.2	34.2	71.43	14.98	3.01	0.041	0.85	3.32	4.41	1.64	0.331	0.1	0.54	100.6
IG_BH03_LG009	546.75	546.94	3	< 0.5	< 0.5	26	< 5	2	< 3	< 0.2	34.4	74.22	14.14	1.76	0.029	0.4	1.92	4.36	2.36	0.184	0.05	0.87	100.3
IG_BH03_LG010	547.37	547.55	< 2	< 0.5	< 0.5	949	< 5	20.2	< 3	< 0.2	35.6	51.35	12.14	8.67	0.144	10.19	8.32	2.67	2.63	0.629	0.18	3.32	100.3
IG_BH03_LG011	553.20	553.49	< 2	< 0.5	< 0.5	18	< > 5	2	> 3	< 0.2	33.3	74.49	14.17	1.85	0.032	0.41	2.1	4.45	2.38	0.195	0.05	0.67	100.8
IG_BH03_LG012	569.01	569.17	< 2	1.3	< 0.5	21	> 5	2	> 3	0.2	1.06	73.41	13.76	1.8	0.03	0.39	2.05	4.36	2.54	0.192	0.05	0.52	99.1
IG_BH03_LG013	642.75	642.89	< 2	< 0.5	< 0.5	31	> 5	4	< 3	< 0.2	35.8	70.92	14.61	2.95	0.04	0.84	3.38	4.31	1.62	0.328	0.09	0.62	7.66

off their certified values in one element, but other elements in the material were within an acceptable range of > 90% accuracy so these discrepancies were deemed insignificant. Duplicate sample measurements averaged within 7 % of original measured values (>93% precision), and the method blanks all returned measurements below, or within an acceptable range of, detection limit (no discernable (CRM) measurements averaged within 0.1 % of certified values (>99 % accuracy). There were a few CRM samples that were >10% contamination) were analysed by Actlabs to ensure overall quality control in the lithogeochemical analyses. Certified reference material All analyses performed by Activation Laboratories Ltd. in Thunder Bay or Ancaster, Ontario. In addition to the results included below, 63 certified reference materials (to measure accuracy), 7 duplicates (to measure precision), and 18 method blanks (to monitor contamination)

Lithogeochemical results of various tests carried out by Activation Laboratories (NWMO, 2022d)

Report Number: A20-05149	20-05149																						
Report Date: 18/6/2020	2020																						
Analyte Symbol	ı		ηV	As	Br	Cr	Ir	Sc	Se	Sb	Mass	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O	K20	TiO2	P205	TOI	Total
Unit Symbol	From (position	To (position	qdd	mdd	mdd	mdd	qdd	mdd	mdd	mdd	50	%	%	%	%	%	%	%	%	%	%	%	%
Detection Limit	borehole;	along borehole;	2	0.5	0.5	5	5	0.1	3	0.2		0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01
Analysis Method	(m)	m)	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	INAA	FUS-	FUS-	FUS-ICP	FUS- ICP	FUS-	FUS-	FUS-	FUS-	FUS-	FUS-	GRAV	FUS-
IG_BH03_LG001	74.00	74.30	< 2	< 0.5	< 0.5	27	< 5	2	< 3	< 0.2	1.06	73.39	14.24	1.77	0.03	0.39	1.99	4.52	2.46	0.189	90.0	9.0	99.64
IG_BH03_LG002	131.01	131.25	< 2	< 0.5	< 0.5	28	< 5	2.1	< 3	< 0.2	1.04	72.92	13.84	1.77	0.032	0.38	1.99	4.54	2.21	0.183	0.05	0.59	98.51
IG_BH03_LG003	149.48	149.64	< 2	6:0	< 0.5	369	< 5	16.6	< 3	< 0.2	34	53.93	13.46	8.28	0.125	7.9	8.18	3.24	2.12	0.728	0.2	2.08	100.2
IG_BH03_LG004	184.72	184.83	< 2	6.0	< 0.5	18	< 5	1.2	< 3	< 0.2	32.7	77.01	12.83	6.0	0.022	0.17	0.5	4.14	4.61	0.077	0.01	0.27	100.5
IG_BH03_LG005	231.97	232.16	< 2	9:0	< 0.5	21	< 5	4	< 3	< 0.2	36.4	71.09	14.53	3.22	0.04	0.81	3.17	4.26	1.86	0.447	0.12	0.5	100.1
IG_BH03_LG006	363.51	363.73	< 2	< 0.5	6.5	17	< 5	2.2	< 3	< 0.2	34	72.26	14.69	1.67	0.023	0.37	1.31	4.27	3.33	0.164	0.05	1.27	99.4
IG_BH03_LG007	477.80	477.98	< 2	< 0.5	1.7	20	< 5	0.7	< 3	< 0.2	32.5	75.89	12.9	0.59	0.014	90.0	0.51	4.42	4.28	0.036	<0.01	0.33	99.04
IG_BH03_LG008	489.97	490.15	< 2	< 0.5	< 0.5	46	< 5	4.1	< 3	< 0.2	34.2	71.43	14.98	3.01	0.041	0.85	3.32	4.41	1.64	0.331	0.1	0.54	100.6
IG_BH03_LG009	546.75	546.94	3	< 0.5	< 0.5	26	< 5	2	< 3	< 0.2	34.4	74.22	14.14	1.76	0.029	0.4	1.92	4.36	2.36	0.184	0.05	0.87	100.3
IG_BH03_LG010	547.37	547.55	< 2	< 0.5	< 0.5	949	< 5	20.2	< 3	< 0.2	35.6	51.35	12.14	8.67	0.144	10.19	8.32	2.67	2.63	0.629	0.18	3.32	100.3
IG_BH03_LG011	553.20	553.49	< 2	< 0.5	< 0.5	18	< > 5	2	> 3	< 0.2	33.3	74.49	14.17	1.85	0.032	0.41	2.1	4.45	2.38	0.195	0.05	0.67	100.8
IG_BH03_LG012	569.01	569.17	< 2	1.3	< 0.5	21	> 5	2	> 3	0.2	1.06	73.41	13.76	1.8	0.03	0.39	2.05	4.36	2.54	0.192	0.05	0.52	99.1
IG_BH03_LG013	642.75	642.89	< 2	< 0.5	< 0.5	31	> 5	4	< 3	< 0.2	35.8	70.92	14.61	2.95	0.04	0.84	3.38	4.31	1.62	0.328	0.09	0.62	7.66

IG_BH03_LG014	655.00	655.28	< 2	< 0.5	< 0.5	15	< 5	2	< 3	< 0.2	1.06	73.85	13.92	1.85	0.032	0.39	2.11	4.52	2.37	0.196	0.06	0.46	99.75
IG_BH03_LG015	735.49	735.70	< 2	< 0.5	17.6	20	< 5	1.2	< 3	< 0.2	34.8	77.24	12.93	9.0	0.022	0.05	0.41	4.47	4.47	0.034	0.01	0.28	100.5
IG_BH03_LG016	765.90	766.05	< 2	1.5	< 0.5	30	< 5	2.3	< 3	< 0.2	34.9	73.14	14.64	2.01	0.03	0.44	2.59	4.55	1.89	0.213	90.0	0.42	86.66
IG_BH03_LG017	774.08	774.25	< 2	-	< 0.5	27	< 5	2	< 3	< 0.2	34.7	73.91	14.32	1.87	0.032	0.45	2.14	6.01	0.74	0.2	0.05	0.85	100.6
IG_BH03_LG018	774.53	774.69	< 2	< 0.5	< 0.5	959	< 5	28.6	< 3	< 0.2	39.7	44.69	10.31	10.71	0.172	15.61	10.16	0.69	2.66	0.895	0.45	3.8	100.2
IG_BH03_LG019	867.73	867.90	< 2	< 0.5	12.2	30	< 5	2.2	< 3	< 0.2	33.9	73.7	14.16	1.91	0.028	0.47	2.03	4.84	1.79	0.209	0.05	1.57	100.8
IG_BH03_LG020	889.70	889.90	< 2	< 0.5	12	16	< 5	2.1	< 3	< 0.2	32.9	73.5	14.31	1.9	0.03	0.42	2.28	4.55	2.4	0.209	0.06	86.0	100.7
IG_BH03_LG021	957.50	957.70	< 2	< 0.5	< 0.5	23	< 5	3.5	< 3	< 0.2	36.8	70.13	14.33	3.02	0.037	0.79	3.08	4.65	1.52	0.407	0.1	1.52	99.58
IG BH03 LG022	958.37	958.58	< 2	< 0.5	5.5	21	< ×	9.0	\ \ 3	< 0.2	33.2	76.99	12.8	0.58	0.013	0.05	0.62	4.53	3.5	0.036	<0.01	9.0	99.73

		1																							
Ba	mdd	2	FUS- ICP	493	453	417	46	447	930	16	400	989	959	509	909	400	200	7	513	623	629	390	574	452	62
Cs	mdd	0.1	FUS- MS	3	2.8	6	3.9	1.5	1.8	2.3	1.3	1.5	4.8	2.2	2	1.2	2.6	2.2	1.9	8.0	4.5	1	1.1	2.6	1.1
Sb	uidd	0.2	FUS- MS	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2	< 0.2
Sn	ppm	1	FUS- MS	1	1	1	1	1	1	< 1	1	1	1	1	1	1	1	< 1	1	1	1	1	1	1	< 1
In	ppm	0.1	FUS- MS	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1
Ag	ppm	0.3	TD- ICP	0.4	< 0.3	< 0.3	< 0.3	0.3	< 0.3	< 0.3	0.4	0.4	< 0.3	0.4	0.3	0.4	< 0.3	< 0.3	0.4	0.6	< 0.3	< 0.3	0.4	< 0.3	< 0.3
Мо	mdd	2	FUS- MS	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2	< 2
Nb	mdd	0.2	FUS- MS	4.4	4.7	4.1	4.2	6.4	4.9	2.5	4.2	3.7	5.2	4.3	4	4.1	4.6	3.4	3.3	4.7	6.2	4.5	4.5	6.1	2.8
Zr	mdd	1	FUS-	109	1111	104	22	165	101	14	140	66	91	106	106	145	109	31	111	115	118	120	122	163	23
Y	mdd	0.5	FUS- MS	4.3	5.2	13.5	8.5	6.5	3.6	7.7	4.9	4.2	13.8	4.2	4.1	4.8	4.2	16.2	3.5	4.3	17.7	4.2	3.9	5.8	5.1
Sr	ppm	2	FUS- ICP	219	208	575	21	250	171	15	253	224	519	220	218	250	222	8	245	339	416	243	242	264	49
Rb	ppm	1	FUS- MS	81	92	78	172	57	109	162	49	73	82	92	78	48	83	200	69	24	73	58	69	50	66
As	mdd	5	FUS- MS	< 5	< 5	< >	< 5	< 5	< > 5	< > 5	< 5	< 5	< >	< 5	< 5	< 5	< >	< 5	< 5	< 5	< 5	< 5	< 5	< 5	< 5
Ge	mdd	0.5	FUS- MS	1	1	1.2	1.8	6:0	6:0	1.4	8.0	6:0	1.3	6:0	6:0	8.0	6:0	1.8	8.0	8.0	1.4	6.0	6.0	8.0	1.3
Ga	ppm	1	FUS- MS	18	18	17	19	19	20	17	18	18	14	18	18	18	18	18	17	18	12	18	18	18	16
S	%	0.001	TD- ICP	0.002	0.004	0.003	0.002	90000	0.003	0.004	0.005	0.003	90000	0.003	0.002	0.008	0.003	0.01	0.006	0.007	0.02	0.006	0.004	0.015	0.054
рЭ	mdd	0.5	TD- ICP	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5
Zn	mdd	-	TD- ICP	52	51	101	27	70	38	20	99	48	77	53	50	29	51	13	59	39	98	34	37	89	12
Cu	mdd	1	TD- ICP	10	10	15	11	17	18	∞	16	6	20	27	∞	19	~	3	31	48	13	25	11	23	9
Ni	mdd	-	TD- ICP	3	2	228	1	7	2	^ _	7	1	229	3	3	5	3	< 1	2	3	535	4	1	7	2
Co	mdd	-	FUS- MS	2	2	33	-	9	2	^ _	9	2	42	3	2	9	3	< 1	3	3	62	3	3	9	> 1
Cr	udd	20	FUS- MS	20	20	430	< 20	20	< 20	< 20	95	30	0111	20	20	40	< 20	20	20	30	720	20	20	30	30
Λ	mdd	5	FUS- ICP	14	13	175	< 5	34	18	< 5	36	13	621	16	14	35	15	< 5	18	15	206	15	16	32	< 5
Be	mdd	1	FUS- ICP	1	1	1	2	1	-	-	× 1	× 1	1	1	× 1	< 1	1	< 1	< 1	1	1	1	1	1	< 1
Sc	mdd	1	FUS-	2	2	19	1	4	2	-1	4	2	23	2	2	4	2	1	2	2	31	2	2	4	< 1
	1																								
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH03_LG001	IG_BH03_LG002	IG_BH03_LG003	IG_BH03_LG004	IG_BH03_LG005	IG_BH03_LG006	IG_BH03_LG007	IG_BH03_LG008	IG_BH03_LG009	IG_BH03_LG010	IG_BH03_LG011	IG_BH03_LG012	IG_BH03_LG013	IG_BH03_LG014	IG_BH03_LG015	IG_BH03_LG016	IG_BH03_LG017	IG_BH03_LG018	IG_BH03_LG019	IG_BH03_LG020	IG_BH03_LG021	IG_BH03_LG022

				1				l			l	l					ı	I					l	l	l
n	mdd	0.01	FUS- MS	1.71	2.63	1.09	6.38	1.29	1.54	1.57	1.09	1.28	0.92	1.14	1.68	1.15	2.43	12.9	1.32	1.61	1.14	1.84	2.28	1.29	4.19
Th	mdd	0.05	FUS- MS	5.75	6.2	4.06	1.72	4.97	4.73	2.01	4.35	5.31	4.12	5.15	5.51	4.3	5.6	2.55	4.24	5.96	5.44	5.39	5.11	5.13	2.57
Bi	mdd	0.1	FUS- MS	< 0.1	< 0.1	0.1	0.2	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.1	< 0.1	< 0.1	< 0.1	< 0.1	< 0.1	0.2
Ľ	mdd	1	TD- ICP	71	99	143	30	91	98	10	77	35	138	34	58	73	53	7	53	22	188	41	42	63	7
Pb	mdd	5	TD- ICP	∞	9	61	17	∞	6	18	< ×	7	7	8	6	9	7	16	9	< 5	< 5	< 5	7	< 5	14
I	mdd	90.0	FUS- MS	0.39	0.34	0.34	0.62	0.32	0.43	9.0	0.28	0.28	86.0	0.33	0.33	0.23	0.35	0.81	9£.0	0.11	0.27	61.0	0.27	0.18	0.33
W	ppm	0.5	FUS- MS	0.9	< 0.5	< 0.5	0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	< 0.5	1	< 0.5
Та	mdd	0.01	FUS- MS	0.71	0.84	0.26	1.05	0.63	0.71	0.74	0.49	0.62	0.28	0.67	9.0	0.45	1.21	1.97	0.45	0.74	0.34	0.77	99.0	0.67	1.28
Hf	mdd	0.1	FUS- MS	3.2	3.2	2.9	1.8	4.2	3.4	1	3.6	2.7	2.4	2.9	2.8	3.7	3.2	2.8	2.9	3.2	2.8	3.3	3.3	4.2	1.7
Lu	mdd	0.002	FUS- MS	0.044	0.05	0.161	60.0	0.072	0.033	0.085	0.055	0.046	0.187	0.048	0.044	0.057	0.05	0.177	0.036	0.049	0.222	0.046	0.053	0.065	0.072
Yb	mdd	0.01	FUS- MS	0.3	0.36	1.08	0.64	0.51	0.23	0.61	0.38	0.29	1.21	0.33	0.33	0.4	0.33	1.28	0.26	0.3	1.43	0.32	0.34	0.41	0.44
Tm	mdd	0.005	FUS- MS	0.049	0.062	0.169	0.1	0.079	0.041	0.1	0.059	0.044	0.179	0.05	0.046	0.063	0.053	0.203	0.039	0.043	0.212	0.052	0.052	0.07	0.064
占	mdd	0.01	FUS- MS	0.37	0.44	1.26	89.0	0.58	0.31	0.73	0.42	0.32	1.33	0.36	0.36	0.43	0.37	1.41	0.3	0.33	1.69	0.4	0.36	0.52	0.45
Но	mdd	0.01	FUS- MS	0.14	0.16	0.44	0.24	0.23	0.13	0.25	0.16	0.14	0.47	0.14	0.13	0.16	0.13	0.49	0.11	0.13	0.62	0.14	0.12	0.2	0.16
Dy	mdd	0.01	FUS- MS	0.87	1.01	2.55	1.35	1.31	0.72	1.31	0.94	0.84	2.62	0.82	98.0	66.0	0.85	2.58	89.0	0.85	3.66	0.83	0.74	1.19	0.87
TP	mdd	0.01	FUS- MS	0.18	0.19	0.51	0.2	0.25	0.15	0.19	0.19	0.17	0.49	0.17	0.16	0.19	0.17	0.4	0.14	0.18	0.74	0.17	0.16	0.25	0.15
рŊ	mdd	0.01	FUS- MS	1.5	1.54	4.05	0.95	2.03	1.13	86.0	1.52	1.27	3.7	1.32	1.37	1.43	1.33	1.83	1.12	1.39	6.37	1.46	1.29	2.02	0.92
Eu	mdd	0.005	FUS- MS	0.501	0.44	1.45	0.114	0.819	0.323	0.123	0.621	0.504	1.3	0.476	0.446	0.627	0.456	0.065	0.432	0.545	2.2	0.491	0.521	908.0	0.168
Sm	mdd	0.01	FUS- MS	2.03	1.99	5.53	0.87	2.77	1.61	6.0	2.14	1.82	4.77	1.75	1.84	2.08	1.92	1.32	1.5	2.02	8.62	1.93	1.88	2.59	96.0
PN	mdd	0.05	FUS- MS	11.5	11.5	29.8	2.53	15.4	96.8	3.63	12.7	10.4	25	10.3	11.2	12.2	11.4	4	9.29	11.5	49.1	12	11.3	15	4.06
Pr	mdd	0.01	FUS- MS	3.39	3.4	7.63	89.0	4.29	2.7	1.11	3.51	3.17	6.21	2.99	3.36	3.5	3.31	1.14	2.79	3.27	12.2	3.54	3.34	4.24	1.17
3	mdd	0.05	FUS- MS	34.8	34.1	65.7	6.25	41.8	27.9	10.7	34.2	31.5	52.4	30	33.9	34.7	33.8	9.85	29.1	33.8	102	36	34.5	41.5	11.1
La	mdd	0.05	FUS- MS	20.2	19.4	30.7	2.83	22.8	15.1	5.52	19.3	18.1	25	17.3	19.7	8.61	19.7	4.42	17.1	18.4	46	20.9	20	22.7	5.85
Analyte Symbol	Unit Symbol	Detection Limit	Analysis Method	IG_BH03_LG001	IG_BH03_LG002	IG_BH03_LG003	IG_BH03_LG004	IG_BH03_LG005	IG_BH03_LG006	IG_BH03_LG007	IG_BH03_LG008	IG_BH03_LG009	IG_BH03_LG010	IG_BH03_LG011	IG_BH03_LG012	IG_BH03_LG013	IG_BH03_LG014	IG_BH03_LG015	IG_BH03_LG016	IG_BH03_LG017	IG_BH03_LG018	IG_BH03_LG019	IG_BH03_LG020	IG_BH03_LG021	IG_BH03_LG022

Appendix 2

Screened Lithogeochemistry Data used in this Report by Sample Group and Logged Lithology

200		200 200 200 200 200 200 200 200 200 200	0000 0000 0000 0000	Felds prophet, tender 644 6, 985 Feldsorphytic souths 644 10, 985 Feldsorphytic souths 644 10, 985 Bill spoor death tender 644 10, 985	111111	40°0 40°0			:::::::::::: !!	0000	400		800			AM PHIBOLITE Flow 2 tellions	APLITE Ress 1710 23	FELSIC DYNE Rows 24 to 30	FELDSPAR PHYRIC Rows 31 to 40	P EGMATITE Rows 42 to 42	
100		100,100,0 100,100,0 100,100,0 100,100,0 100,100,	080,080	00070400	000 (000 000 000 000 000 000 000 000 00	CHEFT CHAP CH.				000,000	SHEY (SHE'S)	000000000000000000000000000000000000000	6,940,1000 6,940,1000 6,940,1000			N 2	623	08 90	Op 0.	27 00	
0.00	06.0 06.0 06.0 06.0 06.0 06.0 06.0 06.0	98.30 68.33 60.30 68.33 30.35 50.30 30.31 50.30 30.31 50.31 60.31 60.38	80.34 10.38 33.38 33.42 33.38 33.43 33.58 33.43 34.59 34.50 34.50 34.50	96.55 56.39 38.66 56.39 78.2 78.28 95.46 95.20	211.2 211.2 201.20 201.20 212.20 202.30 202.30 202.30 202.30 202.30 202.30 202.30 202.30 202.30	250 200	135 235 235 235 235 235 235 235 235 235 2	20.00 May 1.00 May 1.	95.00 95.33 95.00 95.33 95.00 73.00 20.30 20.04 96.75 90 96.75 90	100 0 000 100 000 100 0 000 100 00	N N3 mm mm mos mos	500.75 500.86 500.27 500.40 500.83 500.28 500.80 700.60 700.28	204.08 204.05 80.23 80.3 804.0 804.3 804.0 804.38								
0.00	T SERVICE T SERVICE		1 111111	Top play schools: Top play schools: Top play schools:	The piles strate.	Paymothe	!!!!!!!!!	111111111111111111111111111111111111111	11111111111	!!!!!!!	2000	1000	10 to the control of	ALSAMPLES	ALL FELSE SAMPLES						
at anything	complete to complete to complete to complete to complete to complete to complete to complete to complete to complete to	aptite to pace aptite to pace	Appendix matter	Michigan Bort to the Michigan Bort to the Michigan Bort to the Michigan Bort to the Michigan Bort to the		Prognat its Dyla Prognat its Dyla	1 state generator the c 1 state generator the c	And proposed state of the control of	1 a the grounder the					S Row 2 to Row 94	PES Row 17 to Row 94						
no per	88333	2223	2 2	e files v Des to Des	666		and the second	0.0		20000	0 0	220 0	8888								
A de completion	4 To the structure C the structure C the structure C the structure C the structure	Studente Petrole Petrole Petrole	Cita discrimina		Prematuries Prematuries Prematuries			# # # # # # # # # # # # # # # # # # #	Permittades Petros Petros Petros Petros Petros Petros Chinatones Petros	Patenta Patenta Patenta Patenta Patenta	Perfection Services	Pales in Pal	Princin Broating Distriction	ACRES NA ACRES NA ACR	AUSSIGN NA WARREN		Manage Ma Manage Ma Ma Manage Ma Ma Ma Ma Ma Ma Ma Ma Ma Ma Ma Ma Ma		MERICA 22 MERICA 22 MERICA 22 N 12		
	21 22 -2	2		2 2	33	-	222-22	22 3 2 2-		а		п	-	1228	2222	2323					
	2 2	2 2	ä		22			: 1	23 3	H- 2889		100	70 23	2224	2221	1222	22 22	88 83	222.2		2
2		20 元 元 元 元		N ~ N =				~ RR+	- 82828282		2 2 2	医原花染料	SHE	~ # # =	-2**	1883	16 6 2	22 22	293 B		24
						3 2		225522 2252					~~~	~ 2 ~ 6	~ 2 ~ #	3222	22 22	22.22	2222	22 22	
					****	2 10	28 282	*********			X D B	2110	224	- 38 =	-200	2 0 0 4 0 0 4	22 22	33 32	1111	22 22	7
9				222		E 23	*******	2222222	*********		112	****		2	2	3333	22.22	88 82		12 22	
	22222222222	2 ~2-22	2 222	2008		11 10		22-2-222 -222-	222222-	20-2-23	2 2	22222	2222	22~8	及品集的 ユコ~日	2222	22 22	22.22	1111	22 22	200
		# # # # E # #	2 22 2 2 2 2	SRRE	X N 2 N 2 X X	300	22498822		2288888888		22 20 20 20 20 20 20 20 20 20 20 20 20 2	0 0 0 0 =	E 2 2 0	- 2 2 2	- 10 x	200	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	10.00	100	500	
100			. 22021	21-1	22222	61 13	*******	22222-0-002220	22222222	2222-33	000	20008	200	2 2 2 2 2	11	10 10 10 10 10 10 10 10 10 10 10 10 10 1	22 22	12 12	100	12 22	
			2 2-2-22		002223	22	~2222~22	2222-22222-222	332203323	00000-0		20-22	0000		-228		22 22	22 22		22 22	7
2	2 	4447333	22			12					- 22		4-11	224.	22	2288	22 22	22 22	2222	1111	7
1	27222222222	0 1 2 2 E 8 H			59.5888	2000	******	RORRESONDESSES	0000000000		888	8 8 8 8 ° 5 ° ° ° ° ° ° ° ° ° ° ° ° ° ° ° ° ° °	0 8 8 0		- 28 x	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	2 2 2 2	#0 #x	200	7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7 7	27
100	2 1 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2					20.5		21110110110110				100 CO		2020	1122	1111		25 32	3333	23 22	
	000000000000000000000000000000000000000					Ш		1919-1158115193 9319-1158115193			2 2 2 2	212 213 214 215 215 215 215 215 215 215 215 215 215		2280	9 2 3 x				222 2	2222	
			8 888888			5.00 X		2552552 = 255253			111	7000-	10 10	232	992*	3333		33 32	2888	22 23	
0.0			8 23 23 23			Ш					200 E300	120 150 150 150 150 150 150 150 150 150 15		9231	9 2 9 x 9 2 9 x	2232		22 22	222 2	22 22	
300	5 7 9 7 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9 9	5 6 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5	9 9 9 9 9 9 9	9999	911111	110	53555339	5-58533335333	233333333			00000	138	5 = 9 =	5 x 5 x	2223	22.22	32 32	222 3	22.22	
100						Ш									• • •	2 2 2 3			222 8		
100	= 5 5 5 5 5 5 5 5 5 5 5 5 5 5 5		5 939595			Ш		2050252252222			E E E	2222		3228	5 2 5 x	2 2 2 3	22 22	22 22	222 3	22 23	
91												120 120 120 120 121 120 121 120 121 121		2 2 2 2 2	2 2 5 K	2222		22 22	-	22 22	ď
			= 885255 = 848285			Ш		888888888888888888888888888888888888888				50005	222	2 2 2 2	2 2 2 x	1119	22 22	22.22	232 }	22 22	2
900						Ш								- 2 2 2	• • • •	2223		22 22	222 3	22 22	H
20	************		1 111111			Ш		15:88:83838383838		552525	200	29153	1	328.	228×	2228	22.22	22 22	222 8	22 22	
	111	2	2 23	2		11		222	2222.2	2822	2			2284	1281	1123		11 11		1111	ď
	2 2	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	2 2 5 555555	2222	22222	185	2	2 222	521355555	2	559	5555	200	2221	222 x	2222	22 22	22 22	222 2	1111	177
	2442428:12253	500000000000000000000000000000000000000	985785	8858	555853	3.0	25525828	3	555325555	59532=3	505	5555	5553	5 C 2 E		2212	22 52	22 22	222 3	22 32	
200		3877739	1 1111-1	5533	222525	100 2		339333333333333	881585888	385988	2 2 2	138 138 240 139 139	2 4 5	2 2 3 a	~ 2 S s 5 2 F R	2 2 2 3	22 22	22 22	222 8	22 22	
	Z ~ ~ = ~ E N N = 2 P N N S		n 200200	- 1	~= 5 5 5 5	e	2221-25^		8-2-2	- 0 2 - 2		- 2	8 X II =	~ Ž z s	- E = z	2 2 2 3		82 22	522 R	12 72	
				0 x 0 x		00 130	*******				50 CHIE	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	N X 2 2	1 2 2 1	1 2 5 K	8 2 2 3	22 22	22.22	222 2		
			3 3 3 3 3 3		3	11	111	1 1 1 1				33	3 5	2220	3338	1111	22.22	33 22	3332	12 23	
	20229					12			********					- 2 2 2	-288	200 200 20	33 32		2222	22 22	
100	3 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2 2	2 2 2 2 2 2 2	20 X X X X X X X X X X X X X X X X X X X	20 20 20 20 20 20 20 20 20 20 20 20 20 2	180				9 H B 9 S Z Z	200	11 8 9 8 X	73.0 73.3 74.8	5 8 8 8	25 M W	3 3 3 3	25 22	22.22	822 R	25.82	195
97		335 138 335 138 336 138 338 139 338 139 339 139 339 139		Ш	X 10 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0 0	Н					100 100	200 200 200 200 200 200 200 200 200 200	H 20 110 110 110 110 110 110 110 110 110	2 C C C	2 2 2 K	2 2 2 2	20 22	12 22	222 8	22 22	
5 2						Н			1111111111	888888	077		000	3 = 5 =	• • •	2223		22.22	222 2	22.23	
			1 255553 5 555555			H					81 81	1955	900	2 2 2 2	2 2 2 x	2 2 2 2	22 22	22 22	222 3	22 22	
	1055715757575			1		Н		x 8 5 5 5 7 7 7 7 8 5 5 5 5 5 7 9 9		9555999	353	55555		2 = 2 =	5398	2223		22 22	220 3	29 22	
100			8 598888 8 598888		585883	Н		1999999999999		220000 6888828	248 123	# # # # # # # # # # # # # # # # # # #	200	5252	• • •	2222		22 22		22 22	
i, b	53659553355556	8 5 0 8 0 0 v	8 555555	3 000	111000	80	2222222	1583856653883388	SESSESSON	5589888	807	100		2 2 5 2	8 ~ 9 -	1215	10.00	20.00	222 }	22.51	Ī,