The Best Place on Earth	a the
Ministry of Energy, Mines & Petroleum Resources Mining & Minerals Division BC Geological Survey	Assessment Report Title Page and Summary
TYPE OF REPORT [type of survey(s)]: Diamond Drilling	TOTAL COST : \$87,855
AUTHOR(S): Lesley Hunt	SIGNATURE(S):
NOTICE OF WORK PERMIT NUMBER(S)/DATE(S): MX-1-805, June 10	, 2011, Approval # 11-1650537 0610 YEAR OF WORK: 2014
STATEMENT OF WORK - CASH PAYMENTS EVENT NUMBER(S)/DATE(S):	5520086, September 2, 2014
PROPERTY NAME: Haskins Reed Property	
CLAIM NAME(S) (on which the work was done): 510709	
COMMODITIES SOUGHT: Zn, Au, Cu, Pb, Ag, Mo, W MINERAL INVENTORY MINFILE NUMBER(S), IF KNOWN:	
MINING DIVISION: Liard Mining Division	NTS/BCGS: 104P/6, M104P033, M104P023
LATITUDE: <u>59</u> ° <u>18</u> <u>36</u> LONGITUDE: <u>12</u>	29 ° 27 ' 40 " (at centre of work)
OWNER(S): 1) Pacific Bay Minerals Ltd.	_ 2)
MAILING ADDRESS: # 733 – 510 W. Hastings Street, Vancouver, BC, V6B 1L8	
OPERATOR(S) [who paid for the work]: 1) Pacific Bay Minerals Ltd.	2)
MAILING ADDRESS: # 733 – 510 W. Hastings Street, Vancouver, BC, V6B 1L8	
PROPERTY GEOLOGY KEYWORDS (lithology, age, stratigraphy, structure Mt. Haskins, Mt. Reed, Brett Zone, McDame Synclinorium, Auto	, alteration, mineralization, size and attitude): ochthonous Cassiar Platform, Skarn, magnetite sphalerite skarn,

Metasediments, argillites, thrust faulting, Cassiar Gold Camp, Della Mines, Canadian Superior

Rosella Boya Formation, McDame Group, Road River Group

REFERENCES TO PREVIOUS ASSESSMENT WORK AND ASSESSMENT REPORT NUMBERS: 48, 4481, 5721, 5121, 25254





TYPE OF WORK IN THIS REPORT	EXTENT OF WORK (IN METRIC UNITS)	ON WHICH CLAIMS	PROJECT COSTS APPORTIONED (incl. support)
GEOLOGICAL (scale, area)			
Ground, mapping			
Photo interpretation			
GEOPHYSICAL (line-kilometres) Ground			
Magnetic			
Induced Polarization			
Radiometric			
Seismic			
Other			
Airborne			
GEOCHEMICAL (number of samples analysed for)			
Soil			
Silt			
Rock			
Other			
DRILLING (total metres; number of holes, size)			
Core 408.6m, Six (6) NQ dri	ll holes	510709	\$79,455
Non-core		_	
RELATED TECHNICAL			
Sampling/assaying 115 drill co	ore samples		\$6,000
Petrographic			
Mineralographic			
Metallurgic			
PREPARATORY / PHYSICAL			
Line/grid (kilometres)		_	
Topographic/Photogrammetric (scale, area)			
Legal surveys (scale, area)			
Road, local access (kilometres)/t			
Trench (metres)			
Underground dev. (metres)			
Other Report Writing		510709	\$2,400
		TOTAL COST:	\$87.855

BC Geological Survey Assessment Report 35088

REPORT ON THE 2014 DIAMOND DRILLING PROGRAM

HASKINS REED PROPERTY

Cassiar District Liard Mining Division British Columbia, Canada

> UTM Zone 09 NAD 83 473839E, 6574697N NTS 104 P/06

Report Prepared For

Pacific Bay Minerals Ltd. Ltd. # 733 – 510 W. Hastings Street, Vancouver, BC, V6B 1L8

Prepared By

Lesley Hunt, B.Sc., Geol.

December 1, 2014

Table of Contents

SUMMA	ARY	1
1.0	INTRODUCTION	2
2.0	LOCATION, ACCESS AND INFRASTRUCTURE	4
3.0	CLIMATE VEGETATION & TOPOGRAPHY	5
4.0	CLAIM STATUS	6
5.0	MT. HASKINS REED & AREA PROPERTY HISTORY	9
JOE R B ZOM BRETT DAKO PLACI LODE	NT REED (DOME) EED VEIN NE (DELLA) T ZONE) 10 ER MINING GOLD MINING IAR ASBESTOS DEPOSIT	
6.0	GEOLOGICAL SETTING	
6.1 6.2 6.3	REGIONAL GEOLOGY PROPERTY GEOLOGY, MT. REED BRETT ZONE 6.2.1 Lithology: 6.2.2 Structure: 6.2.3 Alteration and Mineralization DEPOSIT TYPES	
7.0	2014 EXPLORATION	
7.1 Di	IAMOND DRILLING 7.1.1 Description of Work 7.1.2 Drilling Procedures 7.1.3 Sample Preparation, Analysis And Security 7.1.4 Drilling Results	20 20 21
8.0	CONCLUSIONS	
9.0	RECOMMENDATIONS	
10.0	REFERENCES	
11.0	STATEMENT OF QUALIFICATIONS	

List of Figures

FIGURE 1; PROPERTY LOCATION	3
FIGURE 2; REGIONAL TENURE LOCATION, HASKINS REED PROPERTY	
FIGURE 3; LOCAL TENURE LOCATION, HASKINS REED PROPERTY	8
FIGURE 4: CONCEPTUAL MODEL GEOLOGIC SETTING OF CARBONATE-HOSTED DEPOSITS	16
FIGURE 5; REGIONAL GEOLOGICAL MAP	17
FIGURE 5A; REGIONAL GEOLOGICAL MAP LEGEND	18
FIGURE 6; STRATIGRAPHIC COLUMN OF THE CASSIAR TERRANE	19

List of Tables

TABLE 1; PACIFIC BAY MINERALS LTD. MINERAL TENURES NOVEMBER 2014;	6
TABLE 2: DIAMOND DRILL HOLE COLLAR INFORMATION	22
TABLE 3; DESCRIPTIVE STATISTICS FOR ZINC (PPM): 94 ROCK SAMPLES	23
TABLE 4; 2014 BRETT ZONE DRILLING COMPOSITE SUMMARY	

List of Appendices

APPENDIX A:	SUMMARY OF DRILL CORE ANALYTICAL & ASSAY LAB RESULTS
APPENDIX B:	ORIGINAL DRILL CORE GEOCHEMISTRY LAB RESULTS
APPENDIX C:	DRILL CORE GEOCHEMISTRY ANALYTICAL PROCEDURE DESCRIPTION
APPENDIX D:	DRILL CORE GEOCHEMISTRY ASSAY PROCEDURE DESCRIPTION
APPENDIX E:	2014DRILL PROGRAM PLAN VIEW MAPS AND CROSS SECTIONS
APPENDIX F:	2014DRILL PROGRAM PLAN COST STATEMENT

SUMMARY

This report provides a summary of the results obtained during the 2014 Diamond Drilling Program conducted by Pacific Bay Minerals Ltd.. on the Brett Zone within the Haskins Reed Property situated in the Cassiar Gold District, Liard Mining Division of northwestern British Columbia. The work program consisted of a six (6) NQ drillholes.

The objective of the drill program was to further explore the potential of the Brett Zone, a zinc anomaly previously discovered during diamond drilling program in 1997.

The program design and execution was supervised by Lesley Hunt, a local consulting geologist with over 25 years experience in the Cassiar area. DJ Drilling of Watson Lake, YT was contracted to perform the drilling during period of August 18th to 29th, 2014.

Total applicable exploration expenses on the Haskins Reed Property during the 2014 exploration program amounted to **\$87,855**.

The Haskins-Reed property of Pacific Bay Minerals Ltd. Ltd comprises 11 mineral tenures totaling 3,354 hectares (Table 1). The property is located 22 km northeast of the old townsite of Cassiar, British Columbia, (See Figure 2, Regional Tenure Location). Road access to the property is provided from an access road departing the Stewart Cassiar Hwy at Kilometer 613 and following the well established gravel roads previously the Old Della Mines access roads, up to the lower of Mt. Reed.

The property covers several historic mineral showings and developed prospects including: the Joe Reed polymetallic vein with silver, lead and zinc; the Mount Reed skarn/stockwork with zinc copper molybdenum and tungsten; and the B Zone skarn containing, copper, lead, zinc, silver and bismuth.

Exploration work has been carried out on the Haskins-Reed property since 1937 with the bulk of exploration conducted since the 1960s.

The claims are underlain by mainly thrust imbricated strata belonging to the Boya and Rosella Formations of the Atan Group and the overlying Kechika Group. These strata are age dated at Cambrian to Ordovician. Eocene age quartz-feldspar porphyry plutons of the Mount Haskin and Mount Reed stock complex intrude these sedimentary units in the north and central portion of the property. Intrusion of these igneous bodies have created contact metamorphic aureole within the sedimentary strata in which massive sulphide mineralization has developed.

A total of 94 drill core samples were collected from the six diamond drill holes. In all, 115 samples including standards and blanks were shipped to ALS Global in Whitehorse for analysis and assay.

Further diamond drilling is recommended within the Brett zone.

1.0 INTRODUCTION

This report documents the results of the 2014 Exploration program on the Haskins Reed Property within the Cassiar Gold District, Liard Mining Division in northwestern British Columbia that was conducted by Pacific Bay Minerals Ltd. Ltd.. The program was conducted from August 18 to 29th, 2014, including a GPS survey of the a six (6) NQ diamond drill holes.

The 2012 work program was completed under Mineral Reclamation Permit number MX-1-805, Approval No. 11-1650537-0610. The drill program was completed on mineral tenure 510709 on the lower northwest slope of Mt. Reed.

Work conducted for this report was supervised by Lesley C. Hunt, B.Sc. Geol., who has worked throughout the Cassiar area specifically on Mt Haskins and Mt. Reed as a consulting geologist for over twenty years. The work conducted and reported on for this report was financed by Pacific Bay Minerals Ltd. whose head office is at #733 - 510 W. Hastings Street, Vancouver, BC, V6B 1L8.

Total applicable exploration expenses on the Haskins Reed Property during the 2014 exploration program that are documented in this report total **\$87,855**. A Cost Statement accompanies this report in Appendix F.

The objective of the drill program was to further explore the potential of the Brett Zone. It had been discovered as a zinc in skarn anomaly which had been briefly drilled but more recently outlined in 1997 by a soil geochemistry survey and subsequent drilling in the same year.

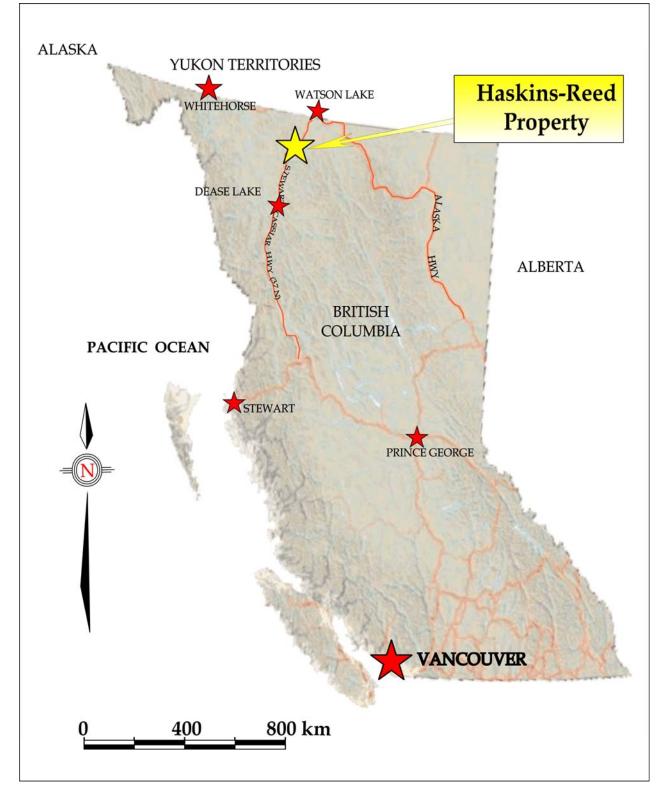
Some of the regional geology discussion contained in this report has been sourced from the report titled "Technical Report on the Haskins - Reed Property, Cassiar District, Liard Mining Division, British Columbia", dated August 8, 2008, authored by Thomas Carpenter, B.Sc., PGeo and William R. Gilmour, B.Sc., PGeo. This Technical Report is available on SEDAR.

Additional historical and geological information was obtained from digital reports, maps and knowledge obtained from the authors's reports and knowledge of the region. The author believes the data and the interpretations contained in this report to be a current and an accurate representation of the property's geology.

The author has relied upon verification by Pacific Bay Minerals Ltd. of the title to these claims and the underlying agreements. Claim locations are as indicated by the Mining Recorder and MT Online.

Drill core samples were shipped to the ALS Global Lab in Whitehorse, Yukon Territories for geochemical analysis and assay. A Summary of the Analytical and Assay results are presented in Appendix A and the Original Certificate of Analysis and Assay is presented in Appendix B. Analytical and Assay Procedures are described in Appendix C.

Figure 1; Property Location



2.0 LOCATION, ACCESS AND INFRASTRUCTURE

The Haskins-Reed Property located 59° 18' 36" north latitude and 129° 27' 40" west longitude within BCGS Map Sheets 104P.023, 033 and National Topographic System (NTS) Map Sheet 104P/ 06.

The Property is located in northwest British Columbia, 23 kilometres east-northeast of the old town site of Cassiar, BC, 120 km south of Watson Lake YT and 126 km north of Dease Lake, BC,(see Figure 1).

The Property comprises 11 mineral tenures containing 3.353.9 hectares (Table 1).

Access to the property is via four-wheel drive vehicle departing from Hwy 37N (Stewart Cassiar Hwy) approximately 13km northeast of the townsite of Jade City. The old Della Mines Access Road then splits off to the north from the access road to Hot Lake which continues on to the west.

Mineral exploration activities can be completed during the summer months, from late May through to October. Drilling operations can be conducted on a year-round basis subject only to adequate provision for snow removal from access roads and water for various work activities. The former Cassiar asbestos mine operated in the district from 1953 until 1992. The open pit mine was located at high elevation and seldom encountered unmanageable operating conditions.

Most general supplies and services are available in Watson Lake, Yukon Territories and limited supplies are available in Dease Lake, British Columbia. Scheduled commercial air service is available from Smithers to Dease Lake three days a week by Northern Thunderbird Air (NTAir). The Cassiar airstrip is available for use by charter aircraft. Alkan Air out of Whitehorse has in the past flown numerous charters into Cassiar in the last few years. NTAir has also flown charters from Vancouver to Cassiar returning the same day.

The nearest major centers are Whitehorse, Yukon with a population of 28,000, located approximately 560 kilometres via Hwy 37N and the Alaska Highway and Smithers, BC which services a population of 15,000, located 720 kilometres south via Hwy 37N and Hwy 16 east.

Only twenty two full time residents of which eight (8) are minor children in one family remain in the nearby townsite of Jade City. Power for the region has historically been provided by privately owned diesel generators, however BC Hydro has entered into an agreement to supply power to the local residents and businesses in the immediate Jade City area which should be in place in 2015.

There is a small but highly skilled population base in the area however most personnel needed for an exploration program would have to be hired from elsewhere. The former townsite of Cassiar was purchased in 1996 and a few buildings remain. In the past the buildings have been leased out to various local exploration companies to serve as base camp facilities.

3.0 CLIMATE VEGETATION & TOPOGRAPHY

The Property is mostly rugged terrain on both sides of McDame Creek. Elevations range from around 900 m along McDame Creek to in excess of 1,900 metres above sea level at the northwestern part of the property, at the summit of Mt. Haskins. The claims are drained westerly and southwesterly into Hot Lake and Hot Creek; southerly and northerly into McDame Creek; and northeasterly into Poorman Lake and Dennis Creek.

The climate is characterized by short, warm summers and long, cold winters. Underground mining can be conducted year round. Daily mean temperatures recorded at Jade City range from -20° C in January to $+15^{\circ}$ C in July. Snowfall between October and May has an average total accumulation of 227 cm. Highest summer temperatures are close to 25° C and winter temperatures may exceptionally reach -50° C. Precipitation, equally in the form of rain in summer and snow in winter, averages 750 mm annually (Environment Canada website). Cloud cover prevails in summer and low hanging fog frequently obscures the mountains.

Vegetation consists of forests of jack pine, lodge pole pine, black spruce, and poplar thinning to buck brush and alpine meadows above tree line at 1,400 to 1,500 metres above sea level. Valley bottoms comprise shallow lakes and swamps with thick, stunted growths of pine and spruce.

4.0 CLAIM STATUS

The Haskins Reed Property is covered by eleven contiguous mining claims and three crown grants all of which are 100% owned by Pacific Bay Minerals Ltd..

The 2014 exploration program was conducted on mineral tenure 510709.

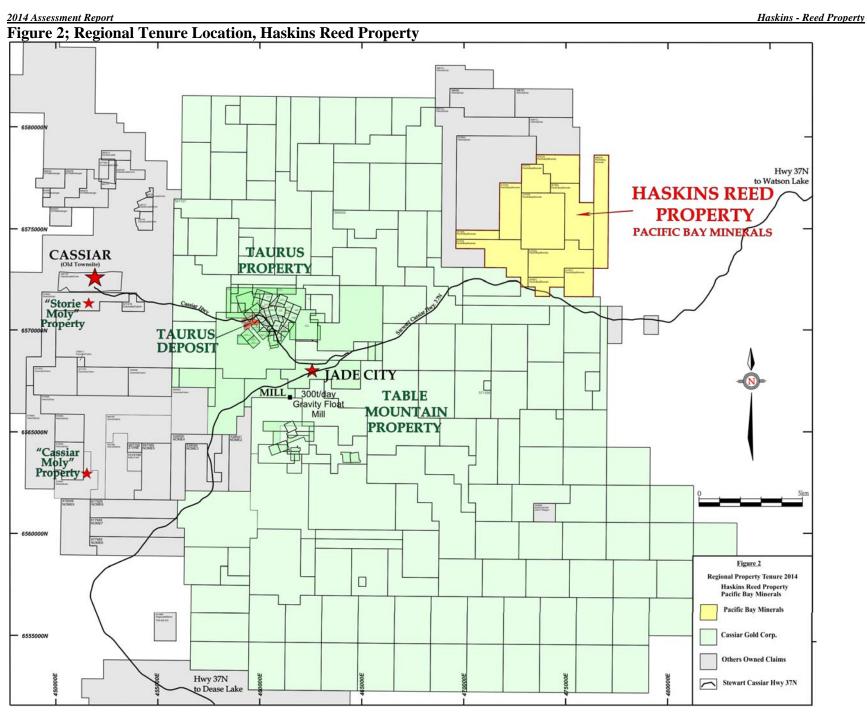
The author is not aware of any environmental issues specific to the Property. The Property is in the Statement of Interest area of the Kaska Dena Council. No LRMP has yet been planned for the area of the Property.

The author has checked the status of recorded ownership and expiry dates of the mineral tenures that cover the Haskins Reed Property, as listed on the BC MEMPR Mineral Titles Division website. The claim tenure numbers, names, expiry dates, and areas that comprise the property are all currently in good standing and are listed in Table 1 below.

Tenure Number	Claim Name Issue Date Good To Da		Good To Date	Area (Ha)
510709		2005/apr/13	2018/sep/05	594.7
510712		2005/apr/13	2018/sep/05	181.6
510720		2005/apr/13	2018/sep/05	297.3
510721		2005/apr/13	2018/sep/05	198.4
510722	JOE REED 1-25	2005/apr/13	2018/sep/05	413.2
510723	JOE REED 26-50	2005/apr/13	2018/sep/05	413.2
531855	NEW JR 1-6	2006/apr/12	2018/sep/05	99.1
552837	ZINC 1-7	2007/feb/26	2018/sep/05	115.7
561802	MORE JR	2007/jul/01	2018/sep/05	396.5
586219	FLANK 1	2008/jun/11	2018/sep/05	396.5
1026212	STEALTH	2014/feb/22	2018/feb/22	247.6

Table 1; Pacific Bay Minerals Ltd. Mineral Tenures November 2014;

Figures 2 and 3 below illustrate the Haskins Reed Property and its regional and local relationships to natural boundaries, adjacent mineral tenures, mining properties and infrastructure.



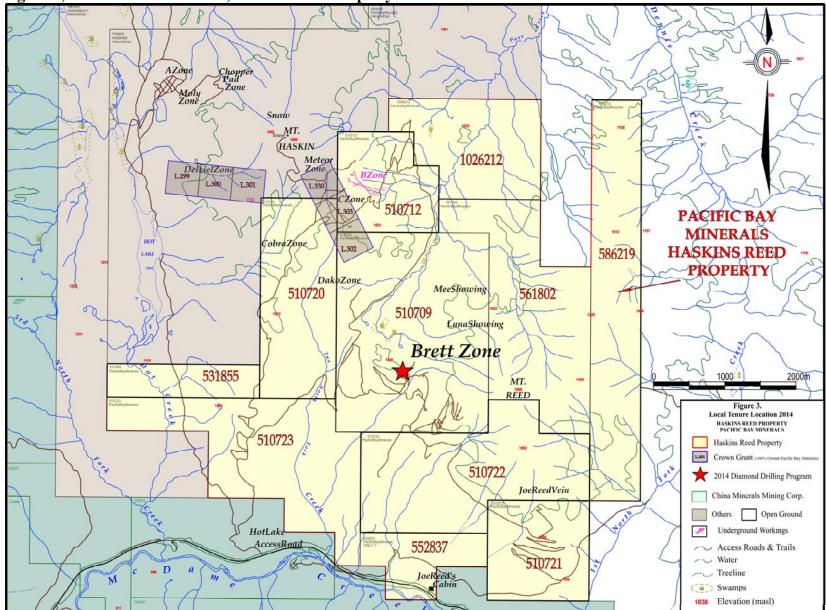


Figure 3; Local Tenure Location, Haskins Reed Property

5.0 MT. HASKINS REED & AREA PROPERTY HISTORY

The majority of the following Property and area History discussion contained in this report has been sourced from the report titled "Technical Report on the Haskins - Reed Property, Cassiar District, Liard Mining Division, British Columbia", dated August 8, 2008, authored by Thomas Carpenter, B.Sc., PGeo and William R. Gilmour, B.Sc., PGeo. This Technical Report is available on SEDAR.

In 1969, an aeromagnetic survey was flown for Brettland Mines Ltd. over an area that covered much of the same ground as does the present Property. A report by R. Crosby, P. Eng concluded that the survey "revealed magnetic features which are interpreted as indicating major faulting and possible skarn type mineralization" (Assessment Rpt No. 2228).

Demand Gold Ltd. in 1997 completed an extensive program of exploration in the Mount Haskins and Mount Reed areas. That company's work is detailed in AR 25253 and 25254 (available online from the BC MEMPR website) and included prospecting, geological mapping, ground magnetometer surveys and soil geochemical surveys.

The following are summaries of exploration specific to certain showings and prospects.

Mount Reed (Dome)

These showings are located at about 1,100 m elevation on the southwest side of Mount Reed. In 1969, the property was option from J. Reed by Glen Copper Mines Ltd ("Glen Copper") and Brettland Mines Ltd.. Work in 1969 included geological mapping, silt and soil geochemistry, induced polarization and magnetometer surveys. The geochemistry highlighted areas of molybdenum and tungsten mineralization. In late 1969, Pacific Petroleums Ltd optioned a onethird interest in the property. In 1970, geological mapping, magnetometer survey, trenching and a 14-hole 1250 m drilling program was completed. All the options were dropped by 1971.

In 1971, Reed Mines Ltd ("Reed Mines") optioned the property from J. Reed. The property was optioned from Reed Mines by Glen Copper and 4 holes totaling 375 m were drilled.

Canadian Superior Exploration Limited ("Canadian Superior") optioned the property from J. Ashton and A. Macdonald in 1978, and drilled seven holes, totaling 570 m, to determine the extent of skarn mineralization (Harivel, 1978). In 1979, 4 holes totaling 406 m were drilled. In 1980, 867 m were drilled in 7 holes. A report by R. Lasmanis, P. Eng stated that all holes "encountered significant intercepts of tungsten /molybdenum mineralization (AR 8421). The 1981 drilling indicated that the skarn zones lack continuity (Watkins, 1981).

Joe Reed Vein

The first mineral discovery in the immediate area was made in 1937 by Joe Reed who discovered a Pb-Zn-Ag vein on the southwestern flank of Reed Mountain and staked the first claims in the area on what is now known as the Joe Reed Vein (Minfile 104P 021).

In 1955 the Consolidated Mining and Smelting Company ("Cominco") optioned the property and, in 1956, drilled five diamond drill holes for a total of 457 m, testing the Joe Reed Vein to a depth of 60 m and over a strike length of 170 m.

In 1969, the soil geochemical surveys on the Mount Reed extended eastward to cover the area of the Joe Reed. In 1971, three holes totaling 153 metres were drilled by Glenn Copper.

B Zone (Della)

In 1965, United States Smelting, Refining & Mining Company discovered the "Main Zone" on Mount Haskins. This corresponds to the B Zone described in this Report.

In the early 1970s Della Mines Ltd. ("Della") made an attempt to mine this deposit. The company drove two tracked drifts to intersect and crosscut three main zones. Adverse metal prices and changing political climate at the time forced the company to discontinue exploration and development. The claims were held by cash in lieu from the mid-1970s until 1993.

Brett Zone

These showings adjoin the Mount Reed molybdenum-tungsten prospect, and are zinc-rich skarns. In 1980, Canadian Superior encountered zinc-bearing skarns in drilling adjacent to a molybdenum-tungsten zone.

In 1997, Demand Gold carried out a program of reconnaissance prospecting, geological mapping, a magnetometer survey and a soil geochemical survey. Nine drill holes totaling 1,019 m were drill to test magnetic highs thought to be related to zinc-lead-silver bearing skarns. In three of the holes, zinc zones were encountered.

Dako

This skarn outcrop showing is located on Tenure 510709. An old adit of unknown age and an old trench are shown by Hodgson (1977). In 1997, a magnetometer survey helped map the extent of the skarn and two holes were drilled by Demand Gold. No significant widths of skarn were intersected.

Placer Mining

Placer gold was first discovered in the McDame area in 1874. The town of Centerville was established during the ensuing gold rush from placer workings on McDame, Snow, Troutline, and Quartzrock Creeks. The district developed into one of British Columbia's major placer camps; most of its production occurred between 1874 and 1895. At that time and until the 1990's, it was the site of the largest nugget discovered in British Columbia, 73 ounces. It came from McDame Creek, just downstream from the Table Mountain Property. Smaller scale placer mining continues today. The total placer gold production from the area up to the early 2000's has been estimated at 108,000 ounces.

Lode Gold Mining

Although placer production in the district was significant, little was done prior to 1933 to locate lode gold deposits. In 1934, the first gold-bearing quartz veins were found in Quartzrock & Troutline Creeks and the first mineral claims were staked. A small exploration rush developed over the next few years as most of the near-surface, gold-bearing veins were discovered. The higher-grade portions of these veins were exploited by small-scale mining over the next forty years. At one point, half-a-dozen abandoned mill sites with capacities of less than 12 tons per day existed in the area.

The two largest gold operations in the area were the Cusac and Erickson mines known as the Table Mountain Mine located 5 km south east of the townsite of Jade City. Another gold mine, the Taurus Gold mine is located 4 km west of the abandoned townsite of Cassiar. Gold was first discovered in the area in 1934 and immediate staking of the Erickson and Taurus properties followed. Intermittent production of all three gold mines dates back to the sixties and continued with several mill expansions at the Table Mountain mine and new high-grade gold discoveries are recorded up until 1997. A more detailed description of the history of gold discovery, exploration and exploitation in the Cassiar vicinity can be found in the Update Of Technical Report on the Table Mountain Property, Liard Mining District, British Columbia, Prepared for Hawthorne Gold Corp., Prepared by Garth Kirkham, P.Geo., P.Geoph., W. Peter Stokes, P.Eng. and John Fox, P.Eng., Beacon Hill Consultants (1988) Ltd., dated June 1, 2008.

Cassiar Asbestos Deposit

In 1949, a GSC mapping crew first encountered the Cassiar asbestos deposit on McDame Mountain. A small 500 ton per day plant was built and in operation by 1952. The asbestos fibre produced was shipped from Whitehorse in the Yukon and all of the supplies for the mine were brought in along the Alaska Highway to Cassiar. Eventually, Highway 37N was constructed between Stewart and Cassiar, which gave access to supplies from Smithers or Terrace. Chrysotile fibre ore was trucked to Stewart with backhauls of diesel for power and heat. The chrysotile fibre was subsequently shipped from Stewart to markets around the world.

Between 1960 and 1992, Cassiar became the most notable infrastructure north of Stewart and west of Fort Nelson with the exception of Whitehorse. The town was sold off when government loan guarantees were not extended after the transition from open pit to underground operations and the mine was forced to close in February 1992.

6.0 GEOLOGICAL SETTING

6.1 Regional Geology

"The property lies within the northern extension of the Omineca lithotectonic domain. Mounts Reed and Haskin are underlain by a northwest trending belt of Cambrian-Ordovician Kechika Group and Lower Cambrian Atan Group sediments (Rosella and Boya Formations) which have been intruded by Eocene granitic stocks of the Cassiar Batholith. The oldest rocks exposed in the region are thick bedded limestone and dolomite with olive green to grey phyllitic partings, belonging to the Espee Formation. In the valley east of Mount Haskin, the Espee Formation outcrops as two limestone bands separated by a forested section which probably masks the phyllitic interval. Exposed on Mount Haskin are the Boya quartzites, Rosella limestones and the Kechika siltstones. The siltstones have been hornfelsed pervasively. The rocks exposed on Mount Reed are the Boya quartzites and Rosella limestones which have been locally skarnified above the quartzite contact. The Rosella Creek Fault structure runs northwesterly on the eastern flank of Mount Reed"(Nelson, J.L., 1993).

There are two components of the regional geology of the area. The autochthonous Cassiar Platform rocks, and the later intrusive stocks belonging to the Cassiar Batholith (Figure 5).

Shallow dipping thrust faults which imbricate the sequence are likely early and related to easterly directed, syn-accretionary thrust development during Mesozoic emplacement of the Sylvester Allochthon onto the siliclastic strata of the Cassiar Platform Terrane (Nelson and Bradford, 1993).

The Cassiar Platform autochthonous rocks underlie the Sylvester Allochthon, and comprise rocks ranging in age from Hadrynian to Early Mississippian. The autochthonous rocks are seen as north northwest striking, steeply east dipping sequences of metasediments and sediments, ranging in width from 100m to 1,000m with a few mostly sinistral offsetting faults.

The Cassiar Batholith granitic and granodioritic rocks of Middle to Late Cretaceous age intrude the Cassiar Platform rocks. The Cassiar batholith is dominated by muscovite - biotite granite and biotite \pm muscovite granodiorite along with subordinate biotite \pm hornblende granodiorite, quartz monzodiorite, and quartz monzonite (Driver et al, 2000). Megacrystic feldspar is seen throughout the intrusive rocks as well as local clusters and disseminations of magnetite.

A major dextral Kechika fault system lies to the east of Midway – Cassiar area. It is related to the prominent crustal structure - the Tintina Fault - that accounts for the major displacement of the continental margin. The Selwyn basin was dissected by it and the western part (Cassiar Terrane) was moved 450 km north from its original position. The dextral displacement took place during the Cretaceous-Tertiary.

6.2 Property Geology, Mt. Reed Brett Zone

Approximately 75% of the property is covered by forest and scrub vegetation. Outcrop accounts for approximately 20% of the claimed area and occurs primarily in isolated exposures on hillsides, ridges and along road cuts.

6.2.1 Lithology:

Geological mapping has identified the primary lithologies underlying the claims area as a package of Atan Group sediments composed of interbedded quartzite with phyllite and limestone with dolostone. The bedrock geology on the present tenures 510709 and 510712, consists of northwest to northerly striking, moderately to steeply southwest dipping Atan Group sediments.

"The Lower Cambrian Rosella Formation consists of thin to thick bedded limestone with recessive slatey/muddy interbeds. The limestone is partly replaced by orange-weathered, coarse secondary dolomite. The Rosella limestone ranges from 200m to 700m thick and rests conformably on the Boya Formation.

The Boya Formation occurs as rubbly subcrop of a siliclastic sequence consisting predominantly of quartzite with interbedded slate and siltstone. The Boya Formation probably represents a shallow marine fan-top facies evidenced by crossbedding in sandstones found in float" (Nelson, J.L. 1993).

Within present Tenure 510709 a blind or "cryptic" granitic intrusion was located by drilling done between 1968- 1972. "This granitic intrusion is dated from early to middle Eocene and is composed of 3 facies; a coarse granite, a fine granite and an aplitic facies. The granite stock is a coarse grained quartz and alkali-feldspar megacrystic monzogranite. It contains 30% quartz, 1-3% biotite and equal portions of K-feldspar and plagioclase. Xenoliths of aplite, cut by quartz veins, are found within the main granitic stock and, coarse granitic dykes cut the aplite facies which indicates that the aplitic rocks were emplaced prior to granitic intrusion. It has been determined, through past study, that the mineralization within the claim block is mostly related to the aplitic intrusive stage." (Nelson, J.L. 1993).

6.2.2 Structure:

"The general attitude of the stratigraphic sequences is to the northwest with moderate to steep dips to the southwest. The faults on the property follow two dominant trends. The first are northwest trending faults paralleling the stratigraphy and the long axis of the Mount Reed intrusion. Secondly, a series of faults trending northeast cut the northwesterly trending faults. The northeasterly trending faults are a controlling factor on the extent of skarn development and Ag-Pb-Zn bearing veins." (Nelson, J.L. 1993).

6.2.3 Alteration and Mineralization

There are four types of mineralization associated with the granitic intrusion:

- 1.) Massive sulphide (pyrrhotite, sphalerite, chalcopyrite and galena) skarn / replacement beds and lenses in carbonate rocks (example, B Zone).
- > 2.) Molybdenum tungsten and sphalerite in hornfels and skarns (Mount Reed).
- 3.) Molybdenum and pyrite with quartz stockwork in southeast portion of the Mount Reed intrusion (Mount Reed).
- 4.) Silver-rich, sphalerite-galena-pyrite vein mineralization occurs in 3 to 6 m wide northtrending shear (Joe Reed Vein).

"The patterns of mineralization and alteration at Mounts Reed and Haskin are related to the emplacement of two separate pulses of granitoid magmas that created complex adjacent hydrothermal systems. The metamorphic minerals formed in the Rosella carbonates include tremolite, actinolite and diopside. The intensity of metamorphism, especially around Mount Haskin, suggests the presence of a larger intrusion at depth. The exact age relationships between the two episodes of intrusion and development of the exoskarns are unclear. Evidence from previous drilling shows that the exoskarn system was cut by the granitic stock underlying the west flank of Mount Reed thus suggesting that the monzogranite emplacement post-dated the main stage skarn development. It is possible that the main stage exoskarns were associated with earlier aplitic intrusions.

The temporal relationship of the base metal skarn and the vein hosted sulphide mineralization is unknown; however, by analogy from other areas, they most likely both formed late and peripherally within the hydrothermal systems in the area." (Nelson, J.L., 1993).

The Mount Reed molybdenum and tungsten prospect, (Minfile 104P06 043), located 5 km southeast of Mount Haskins, has characteristics similar to those of the Joem prospect. Like the latter, Mount Reed has been explored by extensive drilling. A small porphyritic granitic intrusion of Eocene age emplaced in Atan Group sedimentary strata has a peripheral shell of mixed metacarbonate skarn that carries elevated values in molybdenum and tungsten. Skarn types include pyrrhotite-garnet, epidote-diopside-garnet, andradite-wollastonite, garnet-epidote - magnetite and magnetite-epidote. Although most of the molybdenum and tungsten mineralization is associated with skarn, there is some porphyry style mineralization within the intrusive rocks. On a Property scale, molybdenum and tungsten soil geochemistry delineates the area of the intrusive and surrounding hornfels and skarns. The cause of the IP chargeability anomaly to the west of most of the drilling is not known.

The Joe Reed Vein, located on Tenure 510721, has been traced by trenching and drilling over about 80 m. Soil geochemistry shows silver, lead and zinc anomalies up slope and down slope of the area of drilling. The soils also indicate a possible parallel zone about 500 m to the west.

The B Zone, located on Tenure 510712, has been traced on surface for 700 metres and comprises skarn with sulphides and magnetite. The underground development and drilling indicate the B Zone has strong vertical continuity and, by extrapolation, that somewhat similar configurations may apply to other skarn-type occurrences. Syenite dykes stemming from the intrusive, found within the main zones, were postulated by geologists in1969 to have been the source of mineralization.

Other base metal skarn showings are the Brett, Dako and Cobra, which follow a northwest trend from the Mount Reed zone. The C Zone is located between the Cobra and the B Zone. The skarn development appears to be strongly controlled by stratigraphy; this is, within carbonate units along the contact of other sedimentary units. The repetition of the stratigraphy is due to thrust faulting.

6.3 Deposit Types

On the Property there is evidence of four types of deposits, although their genesis may be related:

- 1) Carbonate Replacement
- 2) Skarn
- 3) Porphyry
- 4) Vein

Polymetallic carbonate-replacement deposits and associated skarn mineralization form by the reaction of high temperature hydrothermal fluids (>>250°C) generated in igneous (e.g., porphyry) environments with carbonate-bearing country rocks. These fluids can be of low to high salinity and may contain CO_2 and other gaseous components. In contrast, the MVT and Irish type platform-carbonate deposits form by the interaction of low to moderate temperature (generally <200°C), high salinity (10 to 30 equivalent weight percent NaCl), basinal brines or meteoric waters.

Polymetallic carbonate-replacement and skarn deposits are related genetically to magmas that intrude into sedimentary rocks (Figure 4). The deposits form when magmatic-hydrothermal fluids expelled from cooling magmas react chemically with carbonate-rich sedimentary rocks.

Skarn and polymetallic carbonate-replacement deposits are associated with many other types of magmatic-hydrothermal deposits in mineral districts. In fact, distinction between skarn and other deposit types is not always easy. In many districts, skarns form an intermediate "zone" of deposits between porphyry deposits in the centre of mining districts and outer zones of polymetallic vein, replacement and distal-disseminated deposits.

The Rosella Formation is known to host molybdenum-tungsten skarns and silver-lead-zinc mineralization adjacent to the Cassiar and Mount Haskin stocks. Molybdenum and tungsten mineralization is associated with metasomatic actinolite-garnet skarn, while silver-lead-zinc replacement bodies generally occur in un-skarned marble.

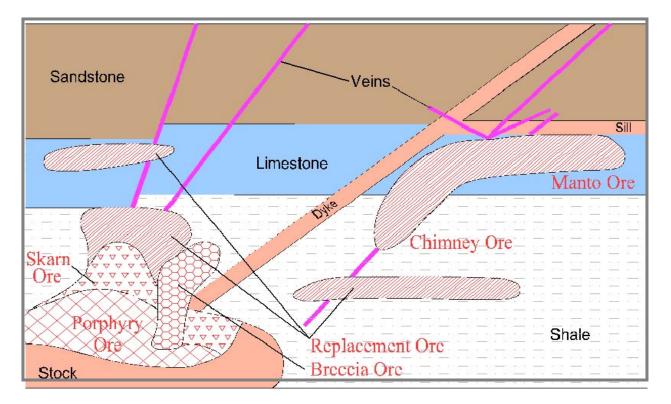


Figure 4: Generalized conceptual model for the geologic setting of high-temperature, carbonate-hosted and related deposits associated with igneous intrusions. (From Plumlee and others, 1999, figure 19.18)

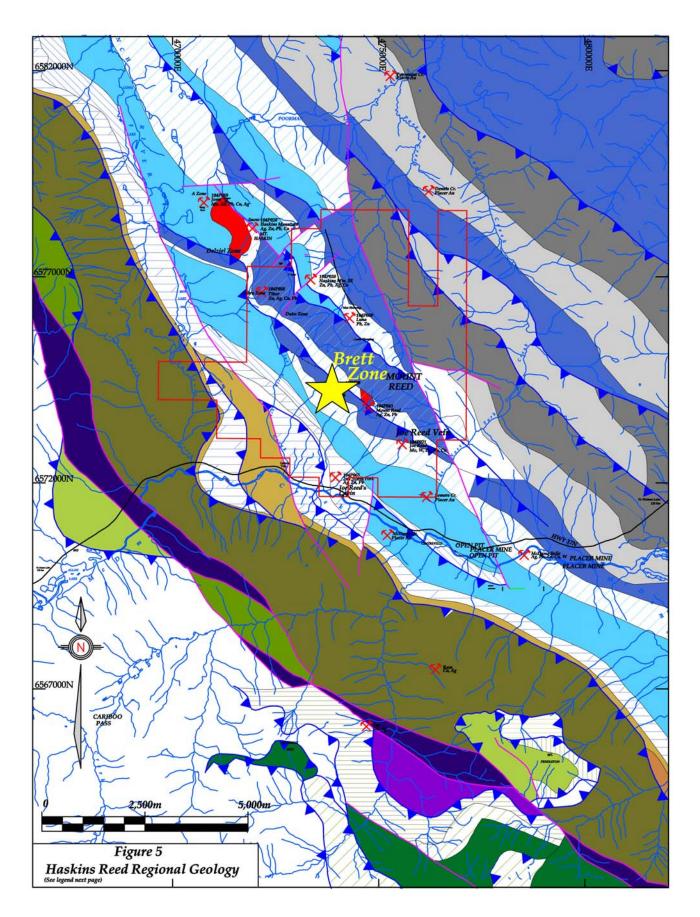


Figure 5a; Regional Geology Legend



Figure 6, below, illustrates the stratigraphic succession of the Cassiar Terrane

MPs uM-P		
un 1	0.0.0.0.0.	Chert, argillite
EARN GROUP	0 0 9 0 0 0 09 0 00 0 09 0 00 0	Conglomerate and interbedded sandstone (only in Toolsee Lake area)
uD-IM		Shale, siltstone and interbedded sandstone, exhalites
McDAME GROUP		Sandstone, shale, phyllilc sillstone
mD		Limestone and dolostone
TAPIOCA SANDSTONE S-1D		Doloslone, sandy doloslone, quartzite
	·· <u>··Z</u> ···· <u>·</u> Z····	Biolurbaled, fine dolomilic sillstone
ROAD RIVER GROUP OS		Dark grey sillstone Graphilic slate, argillaceous limesto
KECHIKA GROUP €0		Buff siltstone, limey siltstone phyllite
18		
Rosella Formation		Limestone, dolostone and shale
5 19		Haskins Reed Property
ATAN GROUP		
E		Quartzite
Boya Formation		Interbedded quartzite and phyllite
		Phyllite
		Quedella ekallita
		Quartzite, phyllite Variegated grey, red and
		green phyllite with limestone bands limestone
Stelkuz Formation		
INGENIKA		
GROUP		
(Late P€)		Cliff forming limestone
Fonas Formation	gr	rey phyllite Cliff forming
Espee Formation	gr	limestone rey phyllite
		Cliff forming limestone

7.0 2014 EXPLORATION

This report documents the 2014 exploration program consisting of a six (6) NQ diamond drill holes completed between August 18th and 29th, 2014 by Pacific Bay Minerals Ltd. on the mineral tenure 510709.

Drill site location was accomplished using Garmin eTrex GPS hand held units and pace and compass methods for backup and verification.

Professional surveying was unavailable as most the surveying companies were not available until after the snow was too deep to travel up the Mt. Haskins Reed access roads. This surveying will be completed in the early part of the 2015 exploration season.

7.1 Diamond Drilling

7.1.1 Description of Work

Pacific Bay Minerals Ltd. Ltd. completed a diamond drilling program on the Haskins Reed Property, Brett Zone, mineral tenure No. 510709, between August 18th and 29th 2014. The program consisted of 6 NQ diamond drill holes, designated 14-01 thru 14-06.

The program was supervised by the author Lesley Hunt, BSc Geol, a local geological consultant. DJ Drilling, of Watson Lake, Yukon Territories was the drilling contractor.

7.1.2 Drilling Procedures

DJ Drilling Ltd. of Surrey, B.C, contracted the 2014 diamond drilling. All drill core from the program was delivered from the drill site to the exploration camp where it was logged and sampled. The holes were drilled with Longyear LF 90 diamond drill rig using conventional NQ size equipment.

Access to the drill collar was obtained by 4X4 pick up and ATV quads using old exploration roads leading north and east from the Hot Lake road junction with the Stewart Cassiar Highway 37N, located 16 km north of the village of Jade City. There were no new access trails or drill pads constructed. The drill and rod sloop were skidded behind a D-6 Cat up the access roads to the drill collars. Drill pad set-ups, drill hole azimuths and dips were verified by the author using a hand held Garmin eTrex GPS.

Downhole surveys used a 'Flex-It' single-shot downhole survey instrument. Drill collar locations are marked with a 4"X4" wooden post and a metal Dymo tag marked with the hole number after the drill equipment had been was moved off each pad. Drill pad set-ups and access trails were re-contoured immediately after completion of the program.

The drill core was logged and sampled onsite by the author. Core recoveries and RQD measurements were recorded and entered into excel spreadsheets. Core photographs were taken

after logging and laying out core sample intervals but before splitting the core. The core is stored in core racks at the exploration base camp in Jade City, owned by the author.

7.1.3 Sample Preparation, Analysis And Security

Core samples from the 2014 Haskins Reed drill program were split with a conventional Longyear core splitter, bagged and delivered by management or a management designated employee directly to the ALS Global Laboratory in Whitehorse. Half the core was left in the core boxes at base camp as a permanent record.

Sampling consisted of marking the mineralized sections into sample intervals based on geological criteria, splitting the core in half along its length using a continuous line to prevent bias, and bagging one-half of the split core from each marked sample interval.

Standard samples were purchased from Canadian Resource Labs in Delta, BC and were inserted into the sample sequence as every tenth sample. One blank sample was inserted into the sample sequence every 20th sample.

Each sample was individually bagged in 6mm plastic sample bags and then several samples were sealed together in a large 6 mm poly bag. The large bags were in turn sealed in a woven rice bag to provide protection during shipping.

The drill core samples were analyzed for 33 elements by trace level methods, using conventional ICP-AES Analysis at ALS Global Labs in North Vancouver, BC. Gold was analyzed using Fire Assay Fusion and Atomic Absorption Spectrometry on 30g analytical pulp.

A Summary of Diamond Drill Core Analyses Results is located in Appendix A, and the original lab certificates are located in Appendix B.

In house check assays were conducted by ALS Global Labs on approximately 10% of the samples and all standards returned results within the upper and lower limits allowed.

At the exploration camp, drilling information was compiled on a master spreadsheet and relevant portions imported into Gemcom for geological modeling. Recorded data includes the following items:

- 1. Header Hole, X, Y, Z, Depth, Zone, Start, Finish, Logger, Purpose.
- 2. Surveys Hole, Depth, Azimuth, Dip.
- 3. Lithology Hole, From, To, Lithological Code, Structure Code, Notes.
- 4. Analyses Hole, From, To, Sample No., Width, ICP Analyses & Assays.
- 5. Drill Core Recovery and RQD data.
- 6. Copies of Original Assay Certificates

Drill Hole Collar information is summarized below.

	Diamond Drill Collar Information						
Hole ID	Easting (UTM Z09)	Northing (UTM Z09)	Elevation (meters)	Total Depth (m)	Azimuth	Dip	Project
14-01	473745.0	6574667.0	1372.0	32.0	225.0	-45.0	Brett Zone, Mt. Reed
14-02	473839.0	6574697.0	1359.0	96.0	130.0	-45.0	Brett Zone, Mt. Reed
14-03	473839.0	6574697.0	1359.0	83.8	220.0	-45.0	Brett Zone, Mt. Reed
14-04	473839.0	6574697.0	1359.0	67.1	220.0	-65.0	Brett Zone, Mt. Reed
14-05	473839.0	6574697.0	1359.0	61.9	180.0	-45.0	Brett Zone, Mt. Reed
14-06	473839.0	6574697.0	1359.0	67.8	155.0	-45.0	Brett Zone, Mt. Reed
			Totals	408.6			

 Table 2: Diamond Drill Hole Collar Information

7.1.4 Drilling Results

Five (5) of the six drill holes intersected the Brett Zone magnetite sphalerite skarn. Drill hole 14-01 was errantly planned which necessitated a new drill plan to be established on-site.

Drill holes 14-02 through 14-05 intersected the Brett Zone and delineated a strike length of approximately 60 metres and with a down dip extension of approximately 55 to 60 metres. The Brett Zone skarn appears to be striking at an azimuth of 100° to 110° and dipping steeply to the southward.

There appears to be two magnetite-sphalerite rich sections to the Brett Zone. Magnetite sphalerite skarn is at bedrock surface in drill holes 14-03, 14-05 and 14-06 with grades of zinc within the top 5 meters of 0.92%, 1.78% and 1.09% respectively. Molybdenum mineralization occurs in both skarn and hornfels lithologies, however the higher grades of molybdenum are seen in the skarn units.

Ninety-four (94) drill core samples were collected from a total of 408.6 metres of NQ sized drill core in six drill holes. Five (5) blank standards and 10 standard reference material samples were inserted into the sample sequence during the core logging procedure.

During the drilling program, the possibly mineralized skarn was explored for over 60 meters along strike length and 65 metres downdip. The zone occurs locally at the bedrock surface beneath 6 metres of overburden.

The results for zinc were significant and the descriptive statistics for Zn are summarized below in Table 3.

Table 4, Descriptive Statistics for Zinc (%): 94 Rock samples				
MIN	0.0038			
MAX	22.3			
RANGE 22.				
AVERAGE (MEAN)	1.61			
MODE	0.02			
MEDIAN	0.35			
STDEV	2.99			

 Table 3; Descriptive Statistics for Zinc (ppm)

Also some copper and tungsten and bismuth were significant over varying widths. Drill composites of the more significant results are summarized below in Table 4.

2014 I	2014 Brett Zone Drilling Composite Summary					
Hole ID	FROM	то	Total m	%		
14-02	15.80	29.50	13.70	2.33	%Zn	
14-02	54.70	64.40	9.70	7.27	%Zn	
14-02	15.80	64.40	48.60	2.31	%Zn	
14-03	9.50	20.10	10.60	2.35	%Zn	
14-03	9.50	39.90	30.40	1.47	%Zn	
14-03	29.70	39.90	10.20	1.92	%Zn	
14-03	29.70	39.90	10.20	0.21	% C u	
14-03	29.70	39.90	10.20	0.15	% W	
14-03	29.70	39.90	10.20	0.04	%Bi	
14-05	10.20	22.20	12.00	1.43	%Zn	
14-05	10.20	36.20	26.00	1.08	%Zn	
14-06	10.50	31.75	21.25	1.60	%Zn	
14-06	36.00	50.00	14.00	1.74	%Zn	
14-06	25.40	29.75	4.35	6.21	%Zn	
14-06	10.50	50.00	39.50	1.48	%Zn	

Table 4; 2014 Brett Zone Drilling Composite Summary

A complete set of lithogeochemical analytical and assay results for the 94 rock samples are located in Appendix A. The original certified copy of the results is located in Appendix B.

8.0 CONCLUSIONS

The 2014 diamond drilling survey on the Brett Zone was successful in verifying the moderate to high grades of zinc in magnetite sphalerite skarn that were intersected in the first two holes of the 1997 drilling campaign. The 1997 composite assays results returned longer intervals with significant zinc grades than the 2014 program results returned. This was a result of the drilling oriented down dip in the higher grade sections of the Brett Zone. There appears to be two magnetite-sphalerite rich sections to the Brett Zone, and it is likely that the 1997 drilling drilled the first two holes directly down dip of one of the zones.

The 2014 drilling delineated two high grade zinc sections in the Brett Zone, roughly 60 metres along strike at an azimuth of approximately 100° to 110°. The skarn with significant zinc values was intersected at bedrock surface and extended downdip for approximately 55 metres. It is open to downdip extension.

The skarn zones containing the highest of magnetite and sphalerite returned the highest zinc grades. However the grades of other base and precious metals known to exist elsewhere on the Haskins Reed property with higher grades, especially copper, silver, bismuth, molybdenum, and tungsten are relatively low

The Brett Zone remains untested along strike to the south east and north west and it is open along it's down dip extensions.

9.0 **RECOMMENDATIONS**

Further diamond drilling is warranted on the Brett Zone to determine its exploration potential.

As well, during winter months, the on-going geological compilation of the historic exploration on the Mt. Haskins Reed Property, may reveal a better understanding the potential for skarn multi-element mineralization peripheral to the intrusive stocks, molybdenum and tungsten mineralization in skarns and hornfels and molybdenum and tungsten mineralization within the intrusive stocks.

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11.0 STATEMENT OF QUALIFICATIONS

I, Lesley Catherine Hunt, do hereby certify that:

- 1) I am a consulting geoscientist with an office at Jade City, Km 603 Stewart Cassiar Hwy 37N, British Columbia, V0C 1E0.
- 2) This Statement of Qualifications applies to the "2014 Assessment Filing for the Haskins Reed Property by Pacific Bay Minerals Ltd.".
- 3) I am a graduate of Lakehead University in 1985 with a B.Sc. in Geology.
- 4) This report is based on exploration work on the Mt. Haskins Reed Property performed in the summer of 2014.
- 5) I was involved in the planning and execution of this program and supervised and performed the field work.

Dated at Jade City, B.C. this 1st day of December, 2014

Respectfully submitted:



Lesley Hunt, B.Sc.

APPENDIX A

Summary of Drill Core Geochemistry Analysis & Assay Results

Haskins Reed Property Analytical and Assay Result Summary

Drill Hole ID	FROM	то	SAMPLE No.	WIDTH	Au >=0.5	Ag >=5.0		>100 >=	Ba 300 Bepp om	m >=1	00 Ca %	Cd ppm Co		Cu >=1000	Fe % Ga			Mn ppm	Mo >=200	Na % Ni ppm P ppm	Pb >=50	S %	Sb ppm	Sc ppm S	Sr ppm T	'nppm Ti%	5 TI pp	pm U ppr	n V ppm	W >=1000	Zn >=0.5
14-02	10.80	11.95	282363	1.15	5 0.003	g/t 0.5	0.15	49		2.0	4 21.4	0 68.7	9 5.0) 35	5 17.15 3	0 0.01	10 0.5	5 6660	42	2 0.01 1 20	8	0.8	2.5	0.5	17	10 0.0	05	5	5 5	230	1.04
14-02	15.80	17.40		1.60			0.12	80			198 19.0		5 5.0				30 1.				31	2.43	6	0.5	19	10 0.0			20 3		4.11
14-02 14-02	17.40 18.40			1.00			0.15	34 29		7.2 9.5	6 24.2		5 7.0		0 14.60 2 9 17.60 2		20 1.1	1000		3 0.01 1 20 3 0.01 4 20	8	0.3	12	0.5	23 24	10 0.0 10 0.0		0	5 3	210	0.32
14-02	19.40			1.00			0.08	14		9.5 1.6	5 15.9		3 6.0		3 23.20 2		10 2.8				5	1.14	5	0.5	24	10 0.0			10 2	440	2.32
14-02	20.70			1.10			0.16	20		0.4	6 19.0		2 8.0				10 2.7				5	0.83	6	0.5	1	10 0.0		÷	5 4	590	0.05
14-02	21.80			1.10			0.73	17		6.9	25 15.0		9 11.0		2 19.65 2		20 3.5					1.52	2.5	1	11	10 0.	• •	÷	10 15		2.32
14-02	22.90			1.10			0.20	22			275 18.0		7 7.0				20 4.8			5 0.005 1 30	319	2.9	7	0.5	52	10 0.			5 4	60	4.49
14-02	24.00 25.40			1.40			0.05	2.5 2.5		2.5 0.5	7 31.0		6 3.0 2 2.0			5 0.01	10 2.9 10 3.0	1 2460 3 1710		8 0.005 0.5 20 9 0.005 0.5 20	5	0.52	2.5	0.5	275 331	10 0.0			5 1	30 20	0.89
14-02	26.80			1.40			0.03	34		7.1	5 23.		2 2.0				10 4.5			5 0.01 0.5 10	-2	1.55	2.5	0.5	129	10 0.0		0	5 4	80	2.47
14-02	28.10			1.40			0.14	24		8.6	69 20.0		0.0				10 5.8			4 0.005 0.5 20	19	0.00	2.5	0.5	95	10 0.0	05	0	5 4	30	5.44
14-02	34.10			1.00			0.13	11		5.4	9 18.		0 0.5				10 5.4			2 0.005 6 20	10	2.68	2.5	0.5	99	10 0.0		0	5 13		3.77
14-02	38.60 46.90			1.60			0.19	90 2.5		7.0 2.9	1 18.		2 1.0			0 0.005	5 5.7			3 0.005 0.5 20 6 0.005 0.5 10	10	2.57	2.5	0.5	19 234	10 0.0 10 0.0		-	5 10 5 12	20 860	3.73 0.04
14-02	54.70			1.05			0.08	2.5		2.0	1 31.4		4 0.5			5 0.005	10 3.0			8 0.005 0.5 10	9	1.24	2.5	0.5	234	10 0.0		5	5 2	10	2.15
14-02	55.75			1.40	0.003	2.1	0.22	30		6.3	5 9.3	4 669.0 7	9 0.5			0 0.01	5 4.0	1 23100	52	2 0.005 3 30	21	6.17	2.5	0.5	6	10 0.	01	5	5 9	70	9.62
14-02	57.15			1.60			0.28	105		4.7	1 18.9		6 1.0			0 0.01	5 2.				9	2.7	2.5	0.5	15	10 0.			10 12		2.64
14-02 14-02	58.75 60.50			1.75			0.15	84.5 209		6.4 0.7	6 15.0 2 34.0					0 0.01	5 1.76 5 2.2			0 0.005 7 30 1 0.005 0.5 20	16	6.255 0.93	2.5 23	0.5	37 257	10 0. 10 0.			10 7 5 2		9.57
14-02	61.60			1.10			0.09	2.5	-	0.7	10 7.3					0 0.005	5 2.2				16	>10.93	2.5	0.5	257	10 0.0		-	5 4	40	22.30
14-02	62.90	64.40	282387	1.50	0.005	0.8	0.22	26	5	8.2	2 18.2	0 172.0 3	4 2.0	22	1 16.25 2	0 0.03	10 3.8	1 8020	15	5 0.01 4 30	6	1.6	2.5	0.5	2	10 0.0	05	5	10 6	100	2.39
14-02	70.50			1.30			0.21	79		8.4	1 22.3		7 2.0				10 2.0				10	0.67	2.5	0.5	22	10 0.0			20 4		0.02
14-02 14-02	80.35 80.90			0.55			0.22	66 2.5		8.0 4.0	1 24.4		9 0.5 6 2.0		1 9.78 1 9 14.40 2	0 0.005	5 1.4 10 3.8				12	1.4	5 2.5	0.5	105 13	10 0.0 10 0.			5 8 5 33	540 1470	0.02
14-02	82.50			1.10			8.82	2.5		1.5	4 14.		2 70.0				40 1.7				8	0.18	5	17	286		75		5 104		0.01
14-02	83.60			1.00			8.07	2.5		2.5	1 9.0		9 67.0				20 1.5				7	0.13	2.5	12	378).7		5 78		0.01
14-02	84.60			1.70			8.08	2.5		7.9	1 2.		9 59.0				30 1.3				11	0.10	2.5	13	231	20 0.			5 63		0.01
14-03 14-03	7.60			1.90			0.29	21 21		4.3 7.6	6 24.4		4 2.0 5 1.0		3 19.75 4 1 23.40 4	• •••=	10 0.0				15	0.00	2.5 2.5	0.5	6	10 0.0		-	5 10 10 8		0.02
14-03	12.00			2.50			0.23	13		6.6	1 19.3		9 0.5		9 19.00 3		10 2.3			0.01 0.0 20	8	0.64	2.5	0.5	3	10 0.0			10 10		0.03
14-03	14.50		282399	1.80		3.1	0.11	6	5 1	4.3	1 11.4					0 0.01	5 4.0			0 0.01 0.5 10		4.31	2.5	0.5	1	10 0.0		-	5 2	100	5.27
14-03	16.30			1.80			0.12	8		3.8	3 12.4		8 1.0		0 22.30 2		10 3.8			0 0.01 0.0 10	14	1.1.1	6	0.5	2	10 0.0			5 2	200	6.22
14-03 14-03	18.10 20.10			2.00			0.24	25 14		6.6 4.0	3 17.		9 0.5		2 17.40 3 2 24.20 3	0 0.01	5 3.7			6 0.01 0.5 10 5 0.02 2 10	15	0.85	2.5 2.5	0.5	4	10 0.0 10 0.0		5	5 8 5 10	130	0.93
14-03	22.00			2.00			0.18	2.5		4.6	39 18.		0 0.5			0 0.01	5 5.				14		2.5	0.5	10	10 0.0		5	5 6		0.02
14-03	24.00			2.00			0.15	7			100 18.3		0 1.0		1 11.80 1	0 0.01	5 5.3			8 0.02 0.5 10	7	0.05	2.5	0.5	21	10 0.0			5 2		0.03
14-03 14-03	26.00 26.45			0.45			0.97	1655 106		2.6 8.3	93 17. 10 18.9		7 7.0		5 14.45 1 2 8.06 4	0 0.09	5 1.6 20 1.4				143	>10.0	34	1	57 81	10 0. 10 0.			5 10 5 55		0.02
14-03	28.30			1.65			2.11	403		7.9	5 20.9		6 12.0		5 9.02 2		10 1.1				7	3.25	2.5	3	67		2.5		5 23		0.02
14-03	29.70	31.70	68970	2.00		2.2	3.32	20	100	5.1	138 4.3	8 197.0 2	20.0	166	5 24.70 3		20 1.4			3 0.09 13 250	18	>10.0	2.5	4	16	10 0.		0	5 21	1450	3.01
14-03	31.70			2.00			3.27	21			410 4.0		3 21.0				20 1.8			7 0.06 12 270	18	>10.0	10	4	13	10 0.			5 20		4.05
14-03 14-03	33.70 35.70			2.00			3.24 2.71	17 19			426 2.5 382 1.0		7 24.0 4 22.0				30 2.1 10 1.5			5 0.05 17 200 2 0.04 19 180	1	>10.0	2.5 2.5	5	6 11		13).1		10 24 5 21	1580 1680	1.48 0.42
14-03	37.70			2.20			2.54	13			613 2.9						30 1.0			3 0.02 17 430	-2	>10.0	2.5	4	10	10 0.			10 17		0.75
14-03	39.90			2.60			5.47	45			155 7.3		1 41.0				40 1.:			2 0.2 10 630	2	6.41	2.5	9	113		26		5 44		0.18
14-03	42.50			2.60			5.60			8.2	78 5.4		7 42.0				40 1.3 70 0.6			4 0.18 7 730 4 0.11 6 2370	4	5.54	2.5	9	103	10 0.			5 46 5 40	560 50	0.18
14-03 14-03	45.10 55.20			2.20			3.78 4.20			2.3 2.1	4 1.4		6 40.0 9 32.0				70 0.6 50 0.4				3	0.84	2.5 2.5	/ 8	48 44	30 0. 20 0.	35 31		5 40		0.01
14-03	68.60		68981	2.40		0.3	4.68	2.5	445	2.9	5.5 1.4	1 0.3 13	5 51.0	11	1 3.66 1	0 1.935	70 0.78			0 0.24 17 925	4	1.44	2.5	8.5	59.5	30 0.3		5	5 49.5		0.00
14-04	57.20			2.00			0.28	43		4.8	51 22.9	0 150.0 2			7 11.15 1		10 3.0				17	1.44	2.5	0.5	78	10 0.			5 3	120	2.14
14-05 14-05	6.10 8.10			2.00			0.33	26 21		3.7 5.8	1 22.		3 3.0 1 2.0				10 0.0 10 0.5			0.02 0.0 10	4	0.01	2.5 2.5	0.5	1	10 0. 10 0.0			5 9 5 5		0.03
14-05	10.20			2.10			0.24	17		3.1	1 22.	i0 7.3	5 2.0		4 18.00 3		10 0.5				3	1.3	2.5	0.5	1	10 0.0			5 5		1.78
14-05	12.20	14.20	68986	2.00	0.003	0.5	0.25	14	5 2	9.5	1 16.	0 45.6 3	1.0) 52-	4 22.20 3	0 0.01	10 2.8	2 7660	99	9 0.02 0.5 20	-2	1.49	2.5	0.5	-1	10 0.0	05	5	5 8	570	0.67
14-05	14.20			2.00			0.28	7		4.4	1 13.3						10 3.3				-2	1.95	2.5	0.5	1	10 0.			10 13		0.22
14-05 14-05	16.20 18.20	18.20 20.20		2.00			0.20	51 10		6.5 5.4	1 16.		7 1.0		0 18.05 3 7 14.95 2		10 2.4 10 3.3			5 0.03 0.5 20 4 0.02 0.5 10	-2	3.13	2.5 2.5	0.5	22	10 0.0 10 0.0			10 9 5 4	620 220	1.54
14-05		22.20		2.00			0.15	277		4.6	2 22.0		2 0.5				10 1.3			3 0.01 0.5 20	-2	1.84	2.5	0.5	23	10 0.0			5 4		1.86
14-05	22.20	24.20	68991	2.00	0.003	0.3	0.35	35	5 4	6.4	1 18.9	0 9.0 2	4 0.5	5 50-	4 16.65 3	0 0.02	10 2.0			8 0.02 0.5 20	3	1.22	2.5	0.5	19	10 0.0	05		5 17		0.14
14-05 14-05	24.20	26.20 28.20	68993 68994	2.00		0.3	0.33	15 21		7.4 7.6	1 12.	5 <u>5.6</u> 3	8 0.5			0 0.02	5 4.4 10 4.1			1 0.02 0.5 20 8 0.02 2 20	4	1.58 0.89	2.5 2.5	0.5	9	10 0.0 10 0.0			5 14 5 18		0.12
14-05		30.20		2.00			0.39	16		7.7	2 12.0) 44			10 4.1				4	1.14	2.5	0.5	1	10 0.0			10 15		0.04
14-05	30.20	32.20	68997	2.00	0.003	0.3	0.40	25	5 3	5.7	1 14.1	5 4.8 2	5 1.0	32	4 24.90 3	0 0.02	10 3.1	2 8760	70	0 0.02 2 50	2	0.88	2.5	1	2	10 0.	01	5	10 18	1310	0.10
14-05		34.20		2.00			0.33	22		2.3	1 16.0						10 3.9				-2	1.87	2.5	0.5	1	10 0.			5 14		0.30
14-05 14-05	34.20 36.20			2.00			0.23	9		6.1 3.2	1 14.		0 2.0 6 2.0		5 16.60 2 1 16.95 2		10 4.5 10 4.8			0 0.02 10 20	-2	5.67 0.99	2.5 2.5	0.5	2	10 0. 10 0.		5	5 11	160 710	4.52 0.03
14-05	38.20			2.00			0.56	2.5		3.5	3 17.3	0 0.8 4	2 2.0		4 15.40 2		10 4.8				5	2.16	2.5	0.3	7	10 0.		10	5 18		0.03
14-05	40.50	42.50	282331	2.00	0.003	0.3	5.36	2.5	20 2	3.0	7 17.9	0 0.3	8 32.0) 1	0 7.55 2	0 0.24	60 2.6	9 11050	227	7 0.12 19 2010	-2	0.03	2.5	11	99	20 0.	47	5	5 61	300	0.02
14-05	42.50	44.50	282332	2.00	0.003	0.3	5.54	2.5	10 1	1.1	1 18.	0 2.4	6 33.0)	9 9.96 4	0 0.15	40 1.6	2 11550	219	9 0.15 15 1170	2	0.04	2.5	10	107	10 0.	34	5	5 45	280	0.04

Haskins Reed Property Analytical and Assay Result Summary

Drill			SAMPLE		Au	Ag		As	Ba		Bi		<u>.</u>	Co		Cu	E a Ga		La			Мо		Ni	Pb				-	-	-			w	Zn
Hole ID	FROM	то	No.	WIDTH	>=0.5	>=5.0	AI %	>100	>=300	Be ppm	>=100	Ca %	Cd ppm	ppm	Cr ppm	>=1000	Fe % ppr	n K%	ppm	Mg %	Mn ppr	>=200	Na %	ppm P ppm		S %	Sb ppm	Sc ppm Sr ppr	n Th ppm	Ti %	TI ppm	U ppm	V ppm	>=1000	>=0.5
44.00	0.5	40.50	000004	0.00	g/t	g/t	0.05	ppm	ppm		ppm	04.00	0.7		1.0	ppm	40.05	0 0.00			0 50	ppm		4 05 0	ppm	0.00	0.5	0.5	0 40	0.005	5	-	40	ppm	%
14-06	8.5		282334	2.00		0.3		24	5	4.4	3	21.80	8.7	1	1.0		3 19.35 4	0 0.02		0.1			9 0.0	0.5 3) /	0.06	2.5	0.5	2 10	0.005	5	5	10	260	0.12
14-06	10.5		282335	2.00		0.5		31	5	6.6	2	20.60	31.0	11	1.0	7	5 22.00 3	0.01		0.4			9 0.0	01 0.5 3	0 6	0.33	2.5	0.5	6 10	0.005	5	5	8	480	0.46
14-06	12.5		282336	0.90			0.14	329	5	2.6	1	25.40	1.3	-1	1.0		5 12.75 2	0.005		0 0			1 0.00	05 0.5 1) 3	3.27	2.5	0.5	52 10	0.005	5	5	5	90	0.02
14-06	13.4		282337	2.00			0.19	566	10	4.8	1	22.10	76.8	8	1.0	230	16.85	0.01		0.2			0.0	01 0.5 2) 2	2.27	2.5	0.5	29 10	0.005	5	5	4	180	
14-06	15.4		282338	2.00			0.16	8	5	27.3	1	17.60	102.0	27	1.0	350		0.01		3.0	-		7 0.0	02 0.5 2) 2	1.33	2.5	0.5	5 10	0.005	5	10	6	240	1.52
14-06	17.4		282339	2.00			0.32	34	5	22.8	1	18.80	3.2	10	1.0	53	3 19.00	0.01		2.2			5 0.0	02 0.5 3) 2	0.11	2.5	0.5	2 10	0.005	5	5	9	460	0.05
14-06	19.4		282340	2.00		0.3	0.37	17	5	30.3	2	16.20	1.8	22	1.0	68	3 22.40 3	0.01	1 10	3.2	1 72	0 102	2 0.0	01 0.5 4) 3	0.05	2.5	1	1 10	0.01	5	5	14	1840	0.05
14-06	21.4	23.40	282341	2.00	0.003	0.3	0.23	14	5	23.2	1	13.60	1.5	24	1.0	366	6 25.20 3	0.01	1 10	3.1	4 77	35	5 0.0	02 0.5 2) -2	0.53	2.5	0.5	1 10	0.005	5	10	10	410	0.05
14-06	23.4	25.40	282342	2.00	0.003	0.3	0.20	20	5	16.1	1	16.40	2.4	16	1.0	3	21.70	0.01	1 10	2.9	7 73	0 32	2 0.0	01 0.5 1) -2	0.02	2.5	0.5	1 10	0.005	5	10	6	230	0.06
14-06	25.4	27.40	282344	2.00	0.003	1.5	0.11	27	5	4.9	6	17.50	567.0	66	1.0	614	16.80	0.08	3 10	1.5	1 100	i0 30	0.0	01 0.5 2) 10	5.1	2.5	0.5	1 10	0.005	10	10	2	120	8.35
14-06	27.4	29.75	282346	2.35	0.003	1.1	0.12	20	5	5.7	5.5	18.80	305.0	37.5	2.0	51	5 13.23	0 0.2	2 10	3.0	7 87	95 98	B 0.0	01 0.5 2) 4	2.78	2.5	0.5	9 10	0.005	5	10	1	220	4.38
14-06	29.7	5 31.75	282347	2.00	0.003	0.3	0.04	7	10	0.6	8	33.70	13.8	2	1.0	4	0.68	5 0.005	5 5	i 2	4 12	0 29	9 0.00	05 0.5 2) -2	0.18	2.5	0.5 3	32 10	0.005	5	5	1	10	0.21
14-06	36.0	38.00	282348	2.00	0.006	0.6	0.07	39	10	2.8	96	34.50	78.3	8	1.0	4	5 1.73	5 0.005	5 10	1.6	7 19	25 48	8 0.00	05 0.5 2) 15	0.72	2.5	0.5 3	37 10	0.005	5	5	1	50	1.18
14-06	38.0	40.00	282349	2.00	0.003	0.3	0.05	44	10	-0.5	2	34.90	44.8	6	1.0	128	3 0.98	5 0.005	5 10	0.7	9 16	i5 37	7 0.00	05 0.5 1) 4	0.52	5	0.5 3	27 10	0.005	5	5	1	10	0.62
14-06	40.0	42.00	282350	2.00	0.003	0.3	0.05	28	5	0.7	2	33.20	120.0	15	1.0	13	5 1.91	5 0.005	5 10	1.0	5 28	0 26	6 0.00	05 0.5 2) -2	0.91	2.5	0.5 3	06 10	0.005	10	5	1	20	1.65
14-06	42.0	44.00	57401	2.00	0.003	1.1	0.17	39	5	5.6	1	17.70	273.0	33	1.0	402	2 19.00	0.005	5 10	1.8	1 113	0 78	B 0.00	05 1 2) 2	2.55	2.5	0.5	35 10	0.01	5	10	2	160	3.80
14-06	44.0	45.90	57402	1.90	0.003	0.3	0.25	45	5	9.5	1	19.60	23.7	9	2.0		12.20	0 0.01	1 10	4.0	9 57	0 47	7 0.0	01 0.5 2) 4	0.22	2.5	0.5	5 10	0.01	5	10	3	140	0.39
14-06	45.9	48.00	57404	2.10	0.003	1.6	0.20	17	5	11.0	1	16.50	243.0	26	2.0	403	3 16.15	0.02	2 10	4.0	6 72	0 38	B 0.0	01 1 3) 14	2.41	2.5	0.5	3 10	0.01	5	5	3	120	3.47
14-06	48.0	50.00	57405	2.00	0.003	1.6	0.52	74	5	11.3	3	19.70	62.5	28	3.0	1230	17.75	0.01	1 10	1.4	6 63	0 125	5 0.0	01 2 2) 10	0.86	2.5	0.5	2 10	0.01	5	5	12	200	0.91
14-06	50.0	52.00	57406	2.00	0.003	0.3	0.62	146	5	10.8	1	20.50	5.2	4	2.0	12	13.95	0.01	1 5	3.0			6 0.0	01 2 4) 7	0.17	2.5	0.5	2 10	0.01	5	5	11	50	0.08
14-06	52.0		57407	2.00		0.3	0.33	6	5	9.8	1	18.90	2.0	8	2.0	63	3 13.00	0.01	1 10	4.6	2 78	0 101	1 0.0	01 1 3) 8	0.05	2.5	0.5	2 10	0.01	5	5	5	350	0.04
14-06	54.0		57408	2.00		0.3	0.29	2.5	5	10.8	3	18.30	0.3	11	1.0	1	3 12.55	0 0.01		5.0	5 98		3 0.0	12 1 2) 3	0.07	2.5	0.5	3 10	0.005	5	10	5	350	0.02
14-06	56.0		57409	1.75		0.3		2.0	5	7.2	32		0.3		1.0		5 11.65	0 0.005		4.2	8 98		2 0.0	2 1 3) 3	0.06	2.5	0.5	4 10	0.01	5	5	6	560	0.02
14-06		67.80	57410	1.10		0.3		2.5	20	22.9	36		0.8	30	23.0	278	3 13.05	0 0.1) 1	6 136		B 0.2	29 18 469) 3	2.19	2.5	7	79 10	0.19	5	5	35	2900	0.05

APPENDIX B

Original Certificate of Drill Core Geochemistry

Analysis & Assay Results

Including Duplicates

WH14141	781 - Fin	alized												1																		1	1
CLIENT : F			c Bav	Minera	ls Ltd																												
# of SAMP			,																														
DATE REC	EIVED :	2014-1	0-08																														
PROJECT	: Mt.Has	kins/Re	ed																														
CERTIFIC	ATE COI	IMENT	'S :																														
PO NUMB	ER :																																
	ME-	ME-	ME-		ME-	ME-			ME-		ME-	ME-	ME-	ME-	ME-	ME-		ME-			ME-	ME-		ME-	ME-	ME-		ME-	ME-	ME- ME-	ME- ME-	7- 0000	A., A A O O
SAMPLE	ICP61 Aq	ICP61 AI	ICP61 As		ICP61 Be	ICP61 Bi			ICP61 Co	ICP61 Cr	ICP61 Cu	ICP61	ICP61 Ga	ICP61 K	ICP61	ICP61 Mg		ICP61 Mo			ICP61 P	ICP61 Pb		ICP61 Sb		ICP61 Sr		ICP61 Ti	ICP61 TI	ICP61 ICP61	ICP61 ICP61 W Zn	Zn-OG62 Zn	Au-AA23 Au
DESCRIP	U	%	ppm		ppm	ppm	%			-	ppm	Fe %		к %	La ppm	%		ppm			r ppm			ppm		ppm		%		ppm ppm	ppm ppm	211 %	ppm
282363		0.15	49		2 <u>2</u>	4 2	21.4		ppin ۵	5 ppin				0.01	10			42		2011 1	20		0.8	-5		17		-0	ppm -10		5 230 >100	1.04	-0.005
282364		0.12	80		4.1	-	19		35	5	697				30			57		2			2.43	6		19		-0	-10		150 >100	4.11	0.000
282365		0.15	34		7.2	6			5	7						1.13		53		- 1		-	0.3	12		23		-0	-10		210 3190		-0.005
282366		0.06	29		9.5	6			21	2						2.76		93		4				8	-1	24		-0	10		190 >100	1.78	-0.005
282367	1		14		11.6				33	6						2.87		165		1				5		1	-20	-0	-10		440 >100	2.32	
282368	0.7	0.16	20	-10	20.4	6	19.6	2.2	12	8	749	19.5	30	0.01	10	2.74	7200	112	0.01	-1	10) 5	0.83	6	-1	1	-20	-0	-10	-10 4	590 451		-0.005
282369	1.5	0.73	17	60	16.9	25	15	152	29	11	832	19.7	20	0.32	20	3.52	11250	184	0.04	1	90) 12	1.52	-5	1	11	-20	0.04	-10	10 15	5 720 >100	2.32	0.005
282370	22.9	0.2	22		5.7	1275	18	268	27		3850	12.8	10	-0	20	4.83	13150	15	-0				2.9	7	-1	52	-20	0.01	-10		60 >100	4.49	0.074
282371		8.02		1270	1.8				6	21				3.69				5				4010	0.62	70			-20	0.46	-10				1.435
282372	-0.5		-5		2.5		31		6	3				0.01		2.91	2460	38			-		0.52	-5		275	-20	-0	-10				-0.005
282373		5.12	-5		0.6				9	40	21		10			0.76		2					0.04	6			-20	0.26	-10				-0.005
282374		0.05	-5						2	2			-10			3.03		29		-1 -1			0.28	6			-20	-0	-10			0.47	-0.005
282375 282376		0.11	34 24		7.1 8.6	5 69			20 37	-1 -1	-					4.59 5.81	7580 10050	55 14			-		1.55 3.39	-5 -5		129 95	-20 -20	-0 -0	-10 -10		80 >100	2.47 5.44	-0.005 0.005
282370		0.14	11		5.4			268	40		1675					5.41		14						-5				-0	-10			3.77	-0.005
282378		0.13	90		4	-2			22	-1	360		10			5.71		13						-5		19		-0	-10			3.73	
282379		0.17	-5		2.9		30.6		2	1	57		10			2.72		496					0.11	7	-1	234	-20	-0	-10			0.70	-0.005
282380		0.08	7		2.0			151	14	-1	286							18					1.24	-5		232	-20	-0	-10			2.15	
282381	-	7.78	550	1300	1.8			31.4	7	19					-	0.31	1120	4			-	4230	0.65	71			-20	0.48	-10				1.4
282382	2.1	0.22	30		6.3	5	9.34	669	79	-1		25	20		-10	4.01	23100	52	-0	3	30	21	6.17	-5	-1	6	-20	0.01	-10			9.62	-0.005
282383	0.8	0.28	105	-10	4.7	-2	18.9	194	26	1	171	16.8	20	0.01	-10	2.7	7770	107	0.01	-1	30) 9	2.7	-5	-1	15	-20	0.01	-10	10 12	2 50 >100	2.64	-0.005
282384	2.5	0.15	85	-10	6.5	7	16.1	717	86	1	922	19.4	20	0.01	-10	1.81	17200	41	-0	6	30) 17	6.42	-5	-1	37	-20	0.01	-10	10 7	110 >100	9.57	-0.005
282385		0.11	209		0.7	2			11	1			-10	-0		2.25		41					0.93	23		257		0.01	-10	-10 2	2 10 >100	1.09	0.007
282386		0.09	-5		10.8			>1000	225		2520					3.45		21					>10.0	-5		6	-20	-0	-10		40 >100	22.3	-0.005
282387		0.22	26		8.2	2	18.2		34	2				0.03		3.81	8020	15		4			1.6	-5		2	-20	-0	-10		5 100 >100	2.39	0.005
282388		0.21	79					1.8	7	2				0.01		2.02		167		1	-			-5				-0	-10		290 242		-0.005
282389		0.22	66		8	-2		1	9	-1	1					1.42		373		4	-		1.4	5			-20	-0	-10		3 540 231 1 1 1 70 201		-0.005
282390 282391	-0.5 34.2		-5	-10 1310	24 1.8		19.2 0.33	0.9 31.4	16 5	2 19			20 20		30	3.81 0.31	12450 1115	495 6		15) 12) 4230	0.04	-5 72		13 402	-20 -20	0.02	-10 -10		1470 291 10 2730		0.005
282391		8.82	-5						12	70	35		40			1.74	-	675			1260		0.00	5			20	0.48	-10		1670 138		-0.005
282393		5.15	-5		0.6		-		9	38	20			0.20		0.72		2					0.04	-5			-20	0.26	-10				-0.005
282394		8.07	-5			-2		-0.5	9	67	17			0.73		1.56					1050		0.13	-5				0.7	-10		1050 110		-0.005
282395		8.08	-5		27.9				9	59	9					1.34		172						-5			20	0.46	-10		1		-0.005
282396		0.29	21		4.3	6		3.1	4	2	83					0.08		133		-1			0.03	-5		6	-20	-0	-10		1180 243		-0.005
282397	1.2	0.23	21	-10	17.6	4	18.1	62.3	25	1	511	23.4	40	0.01		1.66		101	0.01	-1			1.12	-5	-1	4	-20	-0	-10	10 8	1090 9200		-0.005
282398	-0.5	0.29	13	-10	26.6	-2	19.3	1.4	9	-1	379	19	30	0.01	10	2.34	7910	95	0.02	-1	20) 8	0.64	-5	-1	3	-20	-0	-10	10 10	960 286		-0.005
282399		0.11	6		14.3	-2		365	37		1450			0.01		4.02		20		-1			4.31	-5		1		-0	-10		2 100 >100	5.27	-0.005
282400		0.12	8	-	13.8	3	12.4	447	38		1070		20		10			53		-1			4.77	6	-1	2		-0	-10		200 >100	6.22	-0.005
68962	35		541		1.8	4	0.32	31.2	6	19					30		1090	5				4200	0.65	68				0.47	-10		1		1.41
68963		0.24	25			3	17.7		20	1					-10			16		-1			0.85	-5				-0	-10	-	130 9260		-0.005
68964		0.22		-10				1.6				24.2					9050		0.02		10		1.08				-20				130 602		-0.005
68965 68966		0.18			24.6			0.8 0.8	10 10			10.3		0.01		5.9	9650 10300		0.02				0.01 0.05								330 231 70 256		0.032
68966		0.15			2.6		18.2		7			14.5		0.01		5.3) 143						0.06					0.055
68968		4.5			28.3		18.9		5	31		8.06		0.09		1.43			0.03		560		0.84	6					-10		540 124		-0.005
68969		2.11			17.9		20.9		6			9.02		0.13		1.43			0.05		180		3.25						-10		1450 151		-0.005
68970		3.32						197	22			24.7		2.02		1.45			0.09		250		>10.0					0.12			1450 >100		
68971		3.27		130				259	23			25.3		2.07		1.88			0.06		270		>10.0					0.12			1110 >100		
68972		8.26	559	1340	1.9	-2	0.31	31.6	6			5.86		3.83		0.31			0.08			4170				409		0.48			-10 2540		1.39
68973		3.24		40					27			30		2.26			1740		0.05		200		>10.0					0.13			1580 >100		0.094
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SAMPLE	ICP61 Aq	ICP61 AI	ICP61 As		CP61 Be						ICP61 Cu	ICP61 Fe		ICP61 K	ICP61 La	ICP61 Mg		ICP61 Mo	ICP61 Na		ICP61 P			ICP61 Sb		ICP61 Sr		ICP61 Ti	ICP61 TI	ICP61		ICP61 IC W Z			Au-AA23 Au
DESCRIPT	U	Ai %	ppm		opm	-					ppm	ге %		%	ppm	%		ppm	1Na %		ppm	-				ppm		%		ppm p			om %		ppm
68974		5.29	-5	1. · · ·	0.7				10	39	26		1.1	0.84	10		460	2 2						-5	10			0.28	ppm -10		78		68		-0.005
68975		2.71	19		2.5		1.65		34		2390	35.9		1.61		1.59		2		19		-	>10.07	-5	4	11	-20	0.20	-10			1680 4			0.102
68976		2.54	13		2.5	613	2.9		38		2200			1.01		1.01	1430	3		17			>10.0	-5		10		0.08	-10			1820 7			0.192
68977	0.5		45	1			7.39		11	41	494			0.51	40			2		10		-	6.41	-5		113	-20	0.26	10		44				0.046
68978	-0.5	-		1	8.2		5.46		7	42	310			0.95		1.39			0.18	7		-		-5			-20	0.28	-10		46		795		0.016
68979	-0.5	3.78	12	430	2.3	4	1.4	0.5	6	40	54	3.13	10	1.92	70	0.63	735	4	0.11	6	2370) 3	0.84	-5	7	48	30	0.35	-10	-10	40	50	101		-0.005
68980	-0.5	4.2	128	200	2.1	-2	9.78	-0.5	9	32	55	3.35	10	1.77	50	0.41	1435	35	0.05	12	850) 4	2.56	-5	8	44	20	0.31	10	-10	48	50	38		-0.005
68981	-0.5	4.83			3	16	1.44	-0.5	14	53	114		10		70	0.81	700	10	0.25	18	950) 4	1.48	-5		62	30	0.33	-10	-10		1180	49		-0.005
68982		8.24		1330	1.9		0.31	30.9	6	19	134			3.83		0.31	1105	3				4130	0.65	54		407		0.47	-10		152				1.325
68983	-0.5	-	26		3.7		22.1	3.9	3	3	18			0.06		0.09		49		-1		-	0.01	-5		1	-20	0.01	-10		9		270		-0.005
68984		0.24	21		5.8		22.3	7.3	1	2	13		40			0.51	6920	61		-1		-	0.07	-5		2		-0	-10		5		925		-0.005
68985		0.19	17				20.3		15	2				0.01		1.59		143		-1			1.3	-5		1	-20	-0	-10		3			1.775	
68986 68987	0.5	0.25	14 7				16.1	45.6	31 37	1	524 855		30	0.01		2.82 3.34		99 79		-1 -1		-		-5 -5		-1	-20 -20	-0 0.01	-10 -10		8	570 6 1670 2	-		-0.005 -0.005
68988	1.3	-	51		26.5		13.3 16.5		27		1070			0.03		2.45		79		-1				-5 -5		22		-0	-10		9			1.535	-0.005
68989		0.14	10				17.8		30	-1	367	15		0.01		3.32		44		-1		-	2	-5		6		-0	-10		4			2.5	-0.005
68990		0.14					22.6		22	-1	271			0.01		1.33		33		-1				-5		23	-	-0	-10		4			1.86	-0.005
68991	-0.5	-			46.4		18.9	9	24	-1	504		30			2.03		108		-1			1.22	-5		19		-0	-10			1370 1		1.00	-0.005
68992		8.29		1340	2		0.35	31.1	6	20	138		20			0.31	1130	4				4170	0.66	57	15		-20	0.48	-10		154				1.435
68993	-0.5	0.33	15	-10	37.4	-2	12.9	5.6	38	-1	592	22	30			4.49	7810	61	0.02	-1	20) 4	1.58	-5	-1	9	-20	-0	-10	-10	14	880 1	175		-0.005
68994	-0.5	0.39	21	-10	37.6	-2	16.4	1	22	1	389	19.2	30	0.02	10	4.14	7550	38	0.02	2	20) 4	0.89	-5	-1	3	-20	-0	-10	-10	18	600	353		-0.005
68995	-0.5	7.19	7	730	1	-2	2.04	-0.5	10	52	35	3.75	10	1.3	10	1.09	731	2	2.35	29	690) 15	0.04	-5	14	299	-20	0.35	-10	-10	108	10	76		-0.005
68996	-0.5	0.29			37.7		12.7	17.2	29	1				0.01		4.87		32					1.14	-5		1	-20	-0	-10		15	390 3	030		-0.005
68997	-0.5	-	25				14.2		25	1	324			0.02		3.12		70		2				-5		2	-20	0.01	-10			1310 1			-0.005
68998		0.33	22			-2	16		28	2				0.03		3.93		32			30	-		-5		1	-20	0.01	10		14				-0.005
68999		0.23	9				14.2		80		1175			0.01		4.56		56		10		-		-5		2	-	0.01	-10		11			4.52	-0.005
69000 282330		0.34	8 -5		33.2	-2 3	16		26 42	2	411 674	17	20			4.88 4.26		138 256		1				-5 -5		2		0.01	-10 10		16		311 201		-0.005 -0.005
282330		5.36	-5 -5		23		17.3 17.9		42	32	10		20 20			2.69		236			2010	-		-5 -5		99		0.01	-10		61		165		-0.005
282332		5.54	-5		11.1		18.4	2.4	6	33	9			0.24		1.62		219			1170		0.03	-5			-20	0.47	-10		45		431		-0.005
282333		8.17		1320	1.9		0.33		6	19	133			3.77		0.31		4				4100		55				0.47	-10		149				1.435
282334		0.25			4.4		21.8		1	1	8		40			0.19		19		-1		-	0.06	-5		2	-20	-0	-10		10				-0.005
282335		0.22		1 1	6.6	-	20.6		11	1	75		30			0.43		59		-1			0.33	-5		6	-20	-0	-10		8				-0.005
282336		0.14	329	-10	2.6		25.4	1.3	-1	1	5		20		10	0.3	4860	41		-1	10) 3		-5		62	-20	-0	-10	-10	5		173		-0.005
282337	0.5	0.19	566		4.8	-2	22.1	76.8	8	1	230	16.9	30	0.01		0.23		110	0.01	-1) 2	2.27	-5	-1	29	-20	-0	-10	-10	4	180 >	1000	1.09	-0.005
282338	0.6	0.16	8	-10	27.3	-2	17.6	102	27	1	350	15.9	20	0.01		3.05	7830	47		-1			1.33	-5		5	-	-0	-10	10	6	240 >	1000	1.515	-0.005
282339		0.32	34				18.8		10	1	53			0.01		2.21	7610	25		-1		-	0.11	-5	-1	2		-0	-10		9		490		-0.005
282340	-0.5	-	17	1			16.2		22	1	68		30			3.21	7250	102		-1		-	0.05	-5	1	1	-20	0.01	-10				511		-0.005
282341	-0.5		14	1			13.6		24	1	366		30			3.14		35		-1		-	0.53	-5		1	-20	-0	-10		10		515		-0.005
282342	-0.5		20				16.4		16	1	31		30			2.97		32		-1			0.02	-5	-1	1	-20	-0	-10		6		630		-0.005
282343				1290				29.8	5			5.69		3.7			1035		0.07			4010					-20					10 2		0.05	1.34
282344 282345		0.11		-10 570			17.5	567	66 9	1 60	-	16.8 3.04		0.08			10050 587		0.01 2.17		20 580		5.1	-5			-	-0			2 91			8.35	
282345		0.12		-10				0.9 307	9 37			3.04				0.91 3.09			0.01				0.05 2.79	-5 -5		269 9		0.3 -0			91			4.38	-0.005 -0.005
282340		0.12						13.8	2	2 1		0.68				2.4							0.18	-5				-0			1			4 .30	-0.005
282347		0.04						78.3	2	1		1.73				1.67		48					0.72	-5				-0			1			1.175	
282349		0.05						44.8	6	1		0.98				0.79		37					0.52	5				-0			1				-0.005
282350		0.05						120	15	1		1.91				1.05		26					0.91	-5				-0			1			1.65	-0.005
57401		0.17						273	33	1							11300						2.55					0.01				160 >			-0.005
57402		0.25		-10				23.7	9			12.2		0.01			5710		0.01				0.22					0.01				140 3			-0.005
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	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	ICP61	Zn-OG62	Au-AA23
SAMPLE	Ag	AI	As	Ba	Be	Bi	Ca	Cd	Co	Cr	Cu	Fe	Ga	K	La	Mg	Mn	Мо	Na	Ni	Ρ	Pb	S	Sb	Sc	Sr	Th	Ti	ΤI	U	V	W	Zn	Zn	Au
DESCRIPT	ppm	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	%	ppm	%	ppm	%	ppm	ppm	%	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm
57403	37.6	8.03	548	1310	1.9	-2	2 0.36	30.6	5	19	134	5.75	20	3.74	30	0.31	1085	3	0.07	5	1810	4060	0.64	57	15	400	-20	0.49	-10	-10	151	-10	2510		1.37
57404	1.6	0.2	17	-10	11	-2	2 16.5	243	26	2	403	16.2	20	0.02	10	4.06	7200	38	0.01	1	30	14	2.41	-5	-1	3	-20	0.01	-10	-10	3	120	>1000	3.47	-0.005
57405	1.6	0.52	74	-10	11.3	3	3 19.7	62.5	28	3	1230	17.8	20	0.01	10	1.46	6310	125	0.01	2	20	10	0.86	-5	-1	2	-20	0.01	-10	-10	12	200	9110		-0.005
57406	-0.5	0.62	146	-10	10.8	-2	2 20.5	5.2	4	2	12	14	20	0.01	-10	3.05	5450	56	0.01	2	40	7	0.17	-5	-1	2	-20	0.01	-10	-10	11	50	823		-0.005
57407	-0.5	0.33	6	-10	9.8	-2	2 18.9	2	8	2	63	13	20	0.01	10	4.62	7840	101	0.01	1	30	8	0.05	-5	-1	2	-20	0.01	-10	-10	5	350	391		-0.005
57408	-0.5	0.29	-5	-10	10.8	3	3 18.3	-0.5	11	1	8	12.6	20	0.01	10	5.05	9820	493	0.02	1	20	3	0.07	-5	-1	3	-20	-0	-10	10	5	350	224		-0.005
57409	-0.5	0.42	8	-10	7.2	32	2 19.3	-0.5	9	1	5	11.7	10	-0	10	4.28	9810	642	0.02	1	30	3	0.06	-5	-1	4	-20	0.01	-10	-10	6	560	204		0.012
57410	-0.5	3.65	-5	20	22.9	36	6 13.4	0.8	30	23	278	13.1	30	0.1	50	1.6	13650	158	0.29	18	4690	3	2.19	-5	7	79	-20	0.19	-10	-10	35	2900	467		0.005
57411	2	0.28	43	-10	4.8	51	22.9	150	24	2	527	11.2	10	0.01	10	3.08	6060	46	0.01	1	40	17	1.44	-5	-1	78	-20	0.01	-10	-10	3	120	>1000	2.14	0.009
57412	-0.5	5.12	-5	1450	1.2	-2	2 3	-0.5	13	53	82	3.83	10	1.35	10	1.33	615		1.51		720	-2	0.34	-5	16	170	-20		-10	-10	153	30	147		-0.005
57413	-0.5	4.68	-5	550	0.6	-2	2 2.33	2.7	24	101	73	4.63	10	0.45	10	2.71		2	1.29	55	430	4	0.04	-5	21	93	-20	0.4	-10	-10	169	-10	465		-0.005
57414	-0.5	1.79	-5	910	0.6	-2	0.79	-0.5	5	29	35		-10	0.83	10	0.53		1	0.05	16	180	2	0.03	-5	5	23	-20	0.09			46	10	-		-0.005
282359	-0.5		-5	550	_	-2	2 0.89		11	55	76			2.89		-			0.55			-	1.16	-5	12	48	-20					-			-0.005
282360	-0.5		24	420	1.6	-2	2 2.1	1.6	10	70	62			2.14		2.32			0.06				-	-5	11	178		0.31			171	-10			0.009
282361	-0.5	4.85	-5	440	1.5	-2	0.88	-0.5	9	52	90	3.12	10	2.54	20	1.17	425	5	0.38	33	270	8	1.38	-5	11	57	-20	0.27	-10	-10	95	-10	116		-0.005

WH14141	781 - F	inalize	ed																																
CLIENT : I	PACBA	AY - Pa	acific I	Bay Mi	inerals	s Ltd																													
# of SAMF				Ĺ																															
DATE REC	CEIVE	D : 20	14-10-	-08																															
PROJECT	: Mt.H	laskin	s/Ree	b																															
CERTIFIC	ATE C	COMM	ENTS	:																															
PO NUMB	ER :																																		
		ME- ICP61					ME- ICP61		ME- ICP61			ME- ICP61		ME- ICP61		ME- ICP61	ME- ICP61							ME- ICP61							ME- ICP61	ME- ICP61	ME- ICP61	Zn-OG62	Au-AA23
SAMPLE	Aq	AI	As	Ва	Be	Bi	Ca	Cd	Co	Cr	Cu	Fe	Ga	к	La	Mg	Mn	Мо	Na	Ni	Р	Pb	s	Sb	Sc	Sr	Th	Ti	TI	U	V	W	Zn	Zn	Au
DESCRIP	ppm	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	%	ppm	%	ppm	%	ppm	ppm	%	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm	%	ppm
282370																																			0.074
282370																																			0.074
282384	2.5	0.15	85	-10	6.5	7	16.1	717	86	1	922	19.4	20	0.01	-10	1.81	17200	41	-0.01	6	30	17	6.42	-5	-1	37	-20	0.01	-10	10	7	110	>10000		
282384	2.5	0.15	84	10	6.2	5	15.2	680	85	-1	883	18.4	20	0.01	-10	1.72	16250	39	-0.01	8	30	15	6.09	-5	-1	37	-20	0.01	-10	10	7	100	>10000		
282385																																		1.09	
282385																																		1.06	
68969																																			-0.005
68969																																			-0.005
68976																																			0.192
68976																																			0.196
68981	-0.5				3		1.44									0.81	700	10			950		1.48		9	62	30				-	1180	49		
68981	-0.5	4.53	-5	430	2.8	15	1.37	-0.5	13	49	108	3.54	10	1.87	70	0.75	649	10	0.23	16	900	4	1.4	-5	8	57	30	0.32	-10	-10	48	1120	47		
68989																																			-0.005
68989																																			-0.005
282338																																		1.515	-0.005
282338																																		1.51	-0.005
282346		0.12							37		519					3.09		98					2.79			9		-0.01				-	>10000		L
282346	1.1	0.12	16	-10	5.6	6	18.7	303	38	2	511	13.2	10	0.2	10	3.05	8640	98	0.01	-1	20	3	2.77	-5	-1	9	-20	-0.01	-10	10	1	220	>10000		

APPENDIX C:

Drill Core Geochemistry Sample

Preparation, Analysis and Assay Procedures



SAMPLE PREPARATION PACKAGE

PREP-31

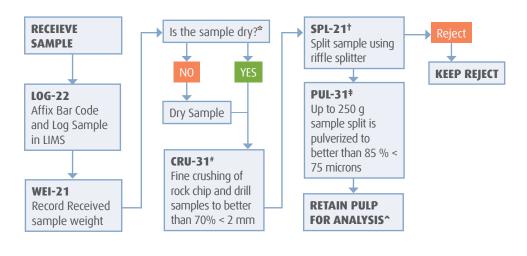
STANDARD SAMPLE PREPARATION: DRY, CRUSH, SPLIT AND PULVERIZE

Sample preparation is the most critical step in the entire laboratory operation. The purpose of preparation is to produce a homogeneous analytical sub-sample that is fully representative of the material submitted to the laboratory.

The sample is logged in the tracking system, weighed, dried and finely crushed to better than 70 % passing a 2 mm (Tyler 9 mesh, US Std. No.10) screen. A split of up to 250 g is taken and pulverized to better than 85 % passing a 75 micron (Tyler 200 mesh, US Std. No. 200) screen. This method is appropriate for rock chip or drill samples.

METHOD CODE	DESCRIPTION
L0G-22	Sample is logged in tracking system and a bar code label is attached.
DRY-21	Drying of excessively wet samples in drying ovens. This is the default drying procedure for most rock chip and drill samples.
CRU-31	Fine crushing of rock chip and drill samples to better than 70% of the sample passing 2 mm.
SPL-21	Split sample using riffle splitter.
PUL-31	A sample split of up to 250 g is pulverized to better than 85% of the sample passing 75 microns.

FLOW CHART - SAMPLE PREPARATION PACKAGE - PREP-31 STANDARD SAMPLE PREPARATION: DRY, CRUSH, SPLIT AND PULVERIZE



- *If samples air-dry overnight, no charge to client. If samples are excessively wet, the sample should be dried to a maximum of 120°C. (DRY-21)
- #QC testing of crushing efficiency is conducted on random samples (**CRU-QC**).
- †The sample reject is saved or dumped pending client instructions. Prolonged storage (> 45 days) of rejects will be charged to the client.
- ‡QC testing of pulverizing efficiency is conducted on random samples (**PUL-QC**).
- ^Lab splits are required when analyses must be performed at a location different than where samples received.



FIRE ASSAY PROCEDURE

Au-AA23 & Au-AA24

FIRE ASSAY FUSION, AAS FINISH

SAMPLE DECOMPOSITION

Fire Assay Fusion (FA-FUS01 & FA-FUS02)

ANALYTICAL METHOD

Atomic Absorption Spectroscopy (AAS)

A prepared sample is fused with a mixture of lead oxide, sodium carbonate, borax, silica and other reagents as required, inquarted with 6 mg of gold-free silver and then cupelled to yield a precious metal bead.

The bead is digested in 0.5 mL dilute nitric acid in the microwave oven, 0.5 mL concentrated hydrochloric acid is then added and the bead is further digested in the microwave at a lower power setting. The digested solution is cooled, diluted to a total volume of 4 mL with de-mineralized water, and analyzed by atomic absorption spectroscopy against matrix-matched standards.

METHOD CODE	ELEMENT	SYMBOL	UNITS	SAMPLE WEIGHT (G)			DEFAULT OVERLIMIT METHOD
Au-AA23	Gold	Au	ppm	30	0.005	10.0	Au-GRA21
Au-AA24	Gold	Au	ppm	50	0.005	10.0	Au-GRA21



GEOCHEMICAL PROCEDURE

ME- ICP61

TRACE LEVEL METHODS USING CONVENTIONAL ICP- AES ANALYSIS

SAMPLE DECOMPOSITION

HNO₃ -HClO₄ -HF-HCl digestion, HCl Leach (GEO-4ACID)

ANALYTICAL METHOD

Inductively Coupled Plasma - Atomic Emission Spectroscopy (ICP - AES)

A prepared sample (0.25 g) is digested with perchloric, nitric, hydrofluoric and hydrochloric acids. The residue is topped up with dilute hydrochloric acid and the resulting solution is analyzed by inductively coupled plasma-atomic emission spectrometry. Results are corrected for spectral interelement interferences.

NOTE: Four acid digestions are able to dissolve most minerals; however, although the term "*near- total*" is used, depending on the sample matrix, not all elements are quantitatively extracted.

ELEMENT	SYMBOL	UNITS	LOWER LIMIT	UPPER LIMIT	DEFAULT OVER- LIMIT METHOD
Silver	Ag	ppm	0.5	100	Ag-0G62
Aluminum	Al	%	0.01	50	
Arsenic	As	ppm	5	10,000	
Barium	Ва	ppm	10	10,000	
Beryllium	Ве	ppm	0.5	1,000	
Bismuth	Ві	ppm	2	10,000	
Calcium	Са	%	0.01	50	
Cadmium	Cd	ppm	0.5	500	
Cobalt	Со	ppm	1	10,000	Co-0G62
Chromium	Cr	ppm	1	10,000	
Соррег	Cu	ppm	1	10,000	Cu-0G62
Iron	Fe	%	0.01	50	
Gallium	Ga	ppm	10	10,000	
Potassium	К	%	0.01	10	
Lanthanum	La	ppm	10	10,000	
Magnesium	Mg	%	0.01	50	
Manganese	Mn	ppm	5	10,0000	



ME-ICP41a

ELEMENT	SYMBOL	UNITS	LOWER LIMIT	UPPER LIMIT	DEFAULT OVER- LIMIT METHOD
Molybdenum	Мо	ppm	1	10,000	Mo-0G62
Sodium	Na	٥/٥	0.01	10	
Nickel	Ni	ppm	1	10,000	Ni-0G62
Phosphorus	Р	ppm	10	10,000	
Lead	Pb	ppm	2	10,000	Pb-0G62
Sulphur	S	%	0.01	10	
Antimony	Sb	ppm	5	10,000	
Scandium	Sc	ppm	1	10,000	
Strontium	Sr	ppm	1	10,000	
Thorium	Th	ppm	20	10,000	
Titanium	Ti	%	0.01	10	
Thallium	TI	ppm	10	10,000	
Uranium	U	ppm	10	10,000	
Vanadium	V	ppm	1	10,000	
Tungsten	W	ppm	10	10,000	
Zinc	Zn	ppm	2	10,000	Zn-0G62

ELEMENTS LISTED BELOW ARE AVAILABLE UPON REQUEST

ELEMENT	SYMBOL	UNITS	LOWER LIMIT	UPPER LIMIT	DEFAULT OVER- LIMIT METHOD
Lithium	Li	ppm	10	10,000	
Niobium	Nb	ppm	5	2,000	
Rubidium	Rb	ppm	10	10,000	
Selenium	Se	ppm	10	1,000	
Tin	Sn	ppm	10	10,000	
Tantalum	Та	ppm	10	10,000	
Tellurium	Те	ppm	10	10,000	
Yttrium	Υ	ppm	10	10,000	
Zirconium	Zr	ppm	5	500	

APPENDIX D:

2014 Haskins Reed Diamond Drill Program

Drill Logs

14-01 to 14-06

Pacific	Bay N	linerals				2014 Mt. Reed - Brett Zone	D	DH N	lo. :	14-0	1			P	g 1 of 1		
Collar D	etails:	Note: GPS	S only, n			Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths						Star	rted:		August 1	7, 2014	
Easting						UTM NAD 83 Z09						Fini	shed:	:	August 1	<mark>8, 2014</mark>	
Northing	g					UTM NAD 83 Z09						Log	ged E	By:	Lesley Hur	ıt	
Elevatio				1		m ASL						EZS	Shot T	ests:	Depth (m)		Dip
End of H					32.0	m									32.00	227.10	-45.60
Azimuth	۱				225.0°												
Dip					<mark>- 45.0°</mark>												
From	То	Lithology	Struc	Pic.	m	Description		-		alizat					Sam		T
(m)	(m)		0	#			Py	Po	Сру	Sph	Mag	g Mo		From	То	Sample	Width
0.00	9.10	OB				Overburden											
9.10	16.80	Н				Hornfels: spotted biotite hornfels, dark grey - brown, aphanitic, 5% biotite (clusters) or porphyroblasts to 4mm are locally zoned											
						with darker rims											
						10.0 -12.4 bleached pale grey, 2cm quartz/carb vnlt sub parallel tca, local rounded frags to 4cm, py is fine grained and muddy at											
						quartz/carb vnlt selvages, Hornfels is moderately fractured, core is moderately broken, few mm scale quartz/carb vnlts with no											
						pdo.											
16.80	17.30	Н	FLT			Hornfels Fault, intense clay gouge											
17.30	22.20	н				Hornfels; as above											
22.20	25.90	QTZT		227	22.95	Quartzite, (chert?), Pale grey, very fine grained quartzite or chert, numerous black silica filled fract (or muddy py with dendritic-											
				228	23.2	tectonic "shock"-like fracture pattern (see photo 228)											
						White quartz/carb vnlt with same pyritic vnlts/frac fill, Few older quartz/carb vnlts 1-4mm avg 20° tca											
						2cm quartz/carb vnlt sub parallel to is pre pyrite mineralization (ie fract filling wtih graphite/muddy py)											
																	_
25.90	31.50	QTZT	FLT			Quartzite intensely rubbly, intensely fractured core, local iK gouge (29.4 - 30.0)	1										
						Local Mo in fractures, very fine grained, Fractures with Mo mineralization are avg. 70-80° tca, fracs are continuous, not irregular											
						like the dendritic texture fracts with graphite/muddy py above										1	
31.50	32.00	QTZT				Quartzite (chert?) (as above)											1
						32.0m EOH	1	1		Î	1						1

Collar Detail casting Iorthing Iorthing Iorthing Idevation Ind of Hole Izimuth Dip From To 0.00 7.60 .60 10.8 0.80 11.9	е Го	Note: GPS	S only, no	6	473839	Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths UTM NAD 83 Z09						Starte Finish			18, 2014	
Iorthing Ilevation ind of Hole izimuth bip From T((m) (m 0.00 7.60 10.8	Го			6												4
Idevation ind of Hole izimuth bip From To (m) (m) 0.00 7.60 0.60 10.8	Го				574697	UTM NAD 63 209						ogge		Lesley Hun	21, 2014 t	
zimuth Pip From To (m) (m 1.00 7.60 1.60 10.8	Го				1,359.0									Depth (m)	Az	Dip
Dip From To (m) (m) 0.00 7.60 0.60 10.8					96.0 130.0°	m					-			47.26 50.30	<u>127.70</u> 129.10	-45.90
(m) (m .00 7.60 .60 10.8					- 45.0°						-			80.79	130.30	
.60 10.8	,	Lithology	Struc.	Pic. #	m	Description	Ру		eraliz Cpy S		<u>`</u>) Mo	From (m)	Sam To (m)	pling Sample No.	Width (m)
		OB				Overburden										
0.80 11.9	.80 I	LMST		221	10.17	Limestone; pale grey/wht, fn to mgr, with irregular dk. grey bands and parallel elongated patches, weak pdo @ 15°tca										
0.80 111.9		0 14				LC is associated with a 2cm quartz/carb vnlt also 15°tca with cgr py at vnlt sevages, few Magnetite/Sphalerite frac fill and vnlts.										
	.95	SK	FLT			Diopside/Epidote/Magnetite Skarn; Buff Yellow in color, UC distinct @ 15°tca, mSi alteration pervasive, Locallized creamy white colored quartz/carb stockwork and sub to euhedral garnets, Local iX gouge, mod carb. filled fracts., LC @ 30°tca with associated quartz/carb 4cm with 0.5 - 1% med to fine gr. py, Tr cpy.	1		Tr				10.8) 11.95	282363	3 1.15
1.95 15.8	80	LMST		222	13.95	Limestone; as above, with few muddy py filled fracs, Noted Diopside Garnet Skarn filled vnlts or strain fracture @ 13.95 (see photo).							10.6	11.95	202304	5 1.15
5.80 24.0						Magnetite/Sphalerite/Garnet/+-Epidote Skarn; UC distinct @ 45 °tca. 20 - 30cm semi-massive to massive soft brick red in color highly effervescent, non-magnetic mineral staining at UC. Majority of the non-magnetic portion of the SK is buff to pale yellowish in color, aphanitic, mSi, Local coarse grained							15.8	17.40	282364	4 1.60
						subhedral greenish yellow Garnets associated with quartz carbonate stockwork.										
						Patches of massive Magnetite +/- black/jack Sphalerite to 5cm along side of irreg. thin 2-4mm bands with local chaotic swirl pattern of										
						Magnetite/Sphalerite. On a large scale - bands (0.5 - 1.0m) of Magnetite/Sphalerite rich skarn alternate with same size zones of buff yellow non Magnetite/Sphalerite skarn, locally Magnetite/Sphalerite stkwk of mm to cm scale network vnlts, local brick-red staining in patches and frac fill., LC										
						indiscreet with local SK stkwk 5 - 10 cm below main LC.										
4.00 26.8						Limestone: as above with numerous SK filled and partially assimilated fracture selvages		\square	\perp							\perp
6.85 27.4			стки	224	35.9	Magnetite/Sphalerite/Garnet/Epidote Skarn; as above Limestone/Skarn Stockwork, pale grey limestone hosts numerous Magnetite/Sphalerite skarn veinlets and fracture fillings.	$\left \right $				-+		_			+
7.45 40.2	20		5160	223	36.1	Limestonerstam stockwork, pare grey innestone nosis numerous magnetite/sphalence skam veiniets and nacture inlings. Notable skam zones as follows:										
				231	36.7	28.1-28.3, 28.2-29.5, 32.3 - 32.75 (less Magnetite, higher garnet content, pervasive K alt., buff yellow, aphanitic),										
						34.35 - 35.05 (massive Magnetite/Sphalerite/cpy)										
						@ 32.8 & 36.0 mSK patches are aphanitic with sub to euhedral yellow and green garnets (Grossular or Uvarovite?) to 0.5cm. also noted are vnlts of										
						green garnets +/- Magnetite/Sphalerite?										
0.20 55.7	75	IMST		232	45.2	38.6 - 40.2 SK as above with higher Epidote content to 40.2m. Limestone becomes m-cgr with numerous quartz/carb vnlts, locally fngr, local Garnet/Magnetite-Sphalerite? fract fill., one in particular parallels CA and			_	_						+
5.20 00.1	10	211101		235	46.9 47.0	Emperative because in egri man nameda qualizera man action man action in gri near can be magnetic ophalence in der mit, eine in particular part										
				235 236 238	47.0 41.2											
5.75 56.7	70	SK				Garnet/Magnetitenetite/Sphalerite Skarn: Pea yellow-green, aphanitic with black (Magnetite/Sphalerite) massive patches to 0.40 m, and veinlet stockworks to 50 % of unit.										
6.70 57.4						Fault: iK gouge, discreet contacts @ 30° tca										
7.40 60.5				0.17		Skarn, as above										
0.50 61.6	.60	LMST		247	61.10	Limestone, UC discreet with drusy cgr quartz/carb vnlt 2cm, Lmst is pale to med. grey fine to mgr, num wht quartz/carbonate, carbonate and clay filled fracts and vnlts. Patches of black (v. weak Magnetite) + Sphalerite?, +/- Epidote, red purple hue (hem?), +/- cpy (very fn. gr)+ pea yellow Garnet/Diopside skarn										
1.60 62.9	.90	SK				Magnetite/Sphalerite Skarn; 90% Magnetite/Sphalerite, num pale yellow (Garnet?) filled fracs. and vnlts, no pdo.										1
2.90 73.8	.80	SK		248	66.7	Diopside/Garnet Skarn; +/- Woll, buff to yellowish, aphanitic, with local yellow sub to euhedral yellowish Garnets (photo 248), loc. zoned (darker cores)										1
				250	71.6	63.6 - 63.8 Magnetite /Sphalerite SK, Note: If sample 282387 runs high Zn (returned 2.39%), sample remaining unit, 64.4 - 73.8m Need to sample: Few Magnetite+/- Sphalerite fracs and vnlts <5% of unit, @ 73.4 white - creamy patches becom more abundant (wollastinite?, non calcareous), Garnets										
						from 1mm to 1cm, no sorting of size, local hematite? staining, very soft, bright red @ 77.9m,										
						70.5 - 71.8 test sample 282388 to verify - is the pale yellow mineral mixed with honey Sphalerite? Note: result .024% Zn, therefore no honey sphalerite, just the black jack sphalerite seen with the magnetite. Photo 250 shows the typical skarn that is buff to yellowish in color (is this diopside?), with local										
						partie black part spirateline second menter inspirate: in the case and and a partie of the menter spirateline second is and a partie in the regulation of the menter spirate in a spirate in the spirate										
3.80 80.9	90	SK		249	74.40	Diopside/Wollastinite/Garnet Skarn; pale buff yellowish with green tinge, local pale honey colored garnets, few hem soft fract. fillings, withclay + Carb+/-					Ī					
						quartz, locally gougey, seagreen/blue patches (partially digested frags?), Local zoned Garnets esp @ 74.4 (see photo 249), fracts become gougey @ 78.3 to contact with Fault below, avg pdo of gougey fracts 30° tca										
						R0.35 - 80.9 - qtz/carb string sub parallel tca, cgr py euhedral in patches 3% +/- Sphalerite, black myrmekitic like texture with quartz carb rimmed with										
						black (fn gr. Sx) in a wormy texture.										
0.90 81.2	20	sĸ	FLT			Skam Fault, iK gouge	\vdash	+	+	+						+
1.20 82.5				1		Skarn, mK pervasive, ifract, pale buff colored, aphanitic, green/red patchy texture, map +/- Sphalerite fracts and vnlts and patches to 3%										
2.50 84.6	60	SK		261	83.85	CalcSilicate Skarn, greenish grey, iSim mod to locally ifractured, py fngr on fracs, Mo on fracts and disseminated (see photo # 261)	2		\top	1		1				1
4.60 87.5	50 I	н				Hornfels, dark brown/black grey, aphanitic with local spotted bi clusters to 2mm, few cherty (calc silicate skarn) filled vnlts (see photo), core is moderately broken, numerous guartz/mica sericite and or biotite vnlts throughout, no pdo, numerous white clay fracts, py on fracs and in vnlts throughout.										
7.50 89.0	.00	GR				Granite (or mgr Aplite Dyke), Note 50% core recovery, med to pale grey, med grained. Note: I don't think it's an aplite dyke, local porphyritic feldspar and							1			1
0.00 00 1	50	CP.	FLT			quartz - 1-2mm porphyroblasts, biotite mm scale to 2% throughout Granite. Fault, ik gouge	\square	+								
9.00 89.5 9.50 96.0			FLI			Granite, Fault, ik gouge Hornfels: as above, 90.2 - 90.5 SKCS as above			-+	+	-+					+
						96.0m End of Hole										1

Image: bit is a strength of the strengt of the strength of the strength of the strength of the	Pacific	Bay M	linerals				2014 Mt. Reed - Brett Zone	DD	DH N	lo. :	14-0	3				Pg 1 of 1		
Number of the state o	Collar De	etails:	Note: GPS	only, I	non su	rveyed	Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths						Sta	rted:		August	22, 2014	
Interview Unit with the second s	Easting												Fini	ished:				
Under labelUnder label	Northing				65	74697	UTM NAD 83 Z09						Log	iged B	By:	Lesley Hur	t	
Name Name <th< td=""><td></td><td></td><td></td><td></td><td>1</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td>EZS</td><td>Shot T</td><td>ests:</td><td>Depth (m)</td><td>Az</td><td>Dip</td></th<>					1								EZS	Shot T	ests:	Depth (m)	Az	Dip
Unit Unit </td <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>m</td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td></td> <td>83.84</td> <td>219.5</td> <td>) -44.10</td>							m									83.84	219.5) -44.10
Tem No United basis Production Matrix control in the second secon								4										┿
Image by a base by a baba base by a baba baba babase by a base by a base	Dip	1			-	-45		_							1			
Image Image <th< td=""><td>From</td><td>То</td><td>1.141 1</td><td>~</td><td>Pic.</td><td></td><td>Providentian</td><td></td><td>-</td><td>Mine</td><td>raliza</td><td>ion (</td><td>%)</td><td>-</td><td></td><td></td><td></td><td>1</td></th<>	From	То	1.141 1	~	Pic.		Providentian		-	Mine	raliza	ion (%)	-				1
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Image: Control (Control (Contro(Control (Control (Control (Control (Control (Control (Control (Co					_			_	_				_					
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Image: Section of the sectin of the section of the section	18.10	22.00	SK				Magnetite Sphalerite Skarn, Magnetite/Sphalerite avg 50%											
No. No. No. Separation framework in any object metabolishie framework in any object metabolishi any object metabolishie framework in any object metabolishie fr	22.00	26.00	SK				Diopside Garnet Skarn, Yellow buff, weak Magnetite / Spalerite Skarn stockwork, few sub parallel greasy acicular greenish (actinolite or											
Non-particle planetic wite and particle and mice-stateworks compare 0.14 with mice-stateworks compare 0.14 with mice-stateworks on the rock. Locally sphanetic, [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow to but if the rock in any planetic [3] yellow is planetic [3] yellow to but if the rock in any planetic [3] yellow is planetic																		
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28.40 28.40 QVBX Quarts Calmonics Braces Multishing Vein Multishin																		
a a b	26.00	26 45	OVEY		+				-				+	+				+
i i	20.00	20.45	QVBX	1		1		'		1	1	l l						
a a b				1		1		1		1	1	l l						
a a b																		
A A																		
Image: Control in the state of the							LC discreet with chilled margin @ 40°tca.											
Image: Control in the state of the																		
Image: Normal Section 1 Image: Sec	26.45	27.00	SK				Diopside Garnet Skarn, Red/Brown/Green, i foliated mm scale laminae, iSi, few cross cutting vnlts and fractures of blackish green mineral											
27.00 28.30 SK FL C Stamp Fault. IK gouge fracture controlled gouge_are, pot 64-60°, cole distributing fractures of Diackish graen minoral (2000). C							(Serp?)											
28.30 29.00 VVBX 200	27.00	28.30	SK	FI T														1
Image: Note: Note					286	28.6		-	-				-					-
All of the set of the	20.30	29.00	QVDA															
Image: Note Note Note Note Note Note Note Note																		
Image: Note Note Note Note Note Note Note Note	29.00	29.70	SK				Actinolite Garnet Skarn, Red/Brown/Green, i foliated mm scale laminae, iSi, few cross cutting vnlts and fractures of blackish green mineral											
28.70 38.90 SK A Assistive Sulphide Skarn, Po. Pythe Sphalente, Magnetile, chalcopytie, local punds arg 2-4mm g20*ca, lew quantification with gives and many with gives contradied by particle with and many with gives contradied by particle with and and any with gives contradied by particle with and and any and arg 2-4mm g20*ca, lew quantification with gives and many with gives contradied by particle with and with gives and many with gives contradied by particle. 20 15 30 5 10 15 2 2 2 2 10 15 2 2 2 2 10 15 2 2 2 10 15 2 2 2 2 10 15 2 2 2 10 15 2								0.5	5				1.0					
a b	00.70	00.00			_			_	_				_					
1 1 207 31.1 control of the control	29.70	39.90	SK															
i i					207	24.4												
All massive subplide zones are closes out with lise stage furgy py lited racts. Numerous local vigs to 2cm, irregular, weak pdo, local Image: Content integration of the stage in the stage furgy py lited racts. Numerous local vigs to 2cm, irregular, weak pdo, local Image: Content integration of the stage in the stage integration of t						31.4		10	30	5	10	15						
All massive subplide zones are closes out with lise stage furgy py lited racts. Numerous local vigs to 2cm, irregular, weak pdo, local Image: Content integration of the stage in the stage furgy py lited racts. Numerous local vigs to 2cm, irregular, weak pdo, local Image: Content integration of the stage in the stage integration of t						34.7		40	50	5	10	15						
39.00 43.00 SK 110 36.3 Skarn. Semi-Massive Sulphide. Calc Silicate Skarn, moderately foliated @ 40° tca, pale green-grey siliceous bands alternating with massive Po bands @ 41.3, locally pale red anhedral parailally digested garnels to 3%. 20 15 3 3 7 I <th< td=""><td></td><td></td><td></td><td></td><td></td><td>-</td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td><td></td></th<>						-												
Info							coxcomb texture (photo 104 & 105)											
Info	39.90	43.00	SK				Skarn, Semi-Massive Sulphide, Calc Silicate Skarn, moderately foliated @ 40° tca, pale green-grey siliceous bands alternating with											
43.00 43.30 SK FLT BX Skarn Fault Breccia. IK gouge, ipy, i drusy, py replaced fragments avg 1-3mm, yellow-buff soft matrix is aphanitic, very finely effervescent, vfngr py is disseminated throughout matrix. 50 61 60 61 60 61 61 61 60 61					110	36.3		20	15	3	3	7						
All and the set of the set																		
45.10 SK BX Import by a dissemination throughout matrix. A phantic locally ifoliated esp. near the top of the unit with in the start of the start in	43.00	43.30	SK	FLT			Skarn Fault Breccia, iK gouge, ipy, i drusy, py replaced fragments avg 1-3mm, yellow-buff soft matrix is aphanitic, very finely effervescent,	50										
43.30 45.10 SK BX Semi Massive Sulphide Skarn Breccia, pale orangey yellow calcareous matrix, aphanitic, locally ifoliated esp. near the top of the unit with no polo, altered pyritic bands, locally in chaotic network texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control of the unit with the properties of the unit with the properies of the unit with the properis of the unit with the properies of the unit with the properis of				ВΧ		1	vfngr py is disseminated throughout matrix.	50		1	1	l l						1
A Incomposed Add Processing Control on processing Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture grades to chaotically brecciated, local qc vnlt and weak Image: Control on provide texture (see photo), texture (see photo), dark green sepentinite (actinolite?) in mm scale Image: Control on provide texture (see photo), texture (see photo), dark green sepentinite (actinolite?) in mm scale Image: Control on provide texture (see photo), dark green sepentinite (actinolite?) in mm scale Image: Control on provide texture (see photo), dark green sepentinite (actinolite?) in mm scale Image: Control on prove (see photo), dark green sepentinite (actinolite?)	43.30	45.10	SK	ΒХ		1	Semi Massive Sulphide Skarn Breccia, pale orangey yellow calcareous matrix, aphanitic, locally ifoliated esp, near the top of the unit with	1		1	1	1	1					1
45.10 55.20 H Image: Size work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work work, noted greyish silvery Sx near bottom of unit diss with fn. gr py, LC discret at 20°tca Image: Size work with With work with With original fabric is. non angertic, numerous py filled fractures and quartz/carb vnlts with try in writes. fn grey diss throughout Hornfels, silvery grey slickensided sulphide (see photo), dark green serpentinite (actinolite?) in mm scale fracts. Size work with sig gouge fracts, few carb vnlts to 3cm, few q/c stockworks usually associated with iK gougey fracts and increased intensity of fracts. Local Biolite spotted hornfels, restricted to zones 1-2m in length - Bi porphyroblasts avg 20% of unit. Image: Size work work work work work with iK fracture controlled AS. Numerous iK gougey fracts and increased intensity of fracts. Local Biolite spotted hornfels, restricted to zones 1-2m in length - Bi porphyroblasts avg 20% of unit. Image: Size work work work work work work work work					106	121												
45.10 5.20 H $_{6.3}$ I																		
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k k	45.10	55 20	Ц		+		Harnfald Plack to brown groonich black, you four to appoprite non-magnetic summaries or filled features and quarteries with the set	+	-				+	┥ ┥	-			+
a 100	-10.10	55.20		1		1		1		1	1	l l						1
Image: Section 2010 Image: Section 2010<				1		1		1		1	1	l l						
a a a b				1	109	1		1		1	1	l l						
Section H FLT Hornfels, Fault Breccia, creamy white quartz/carb stockworks dualy associated with R gdugey flacts and increased intensity of and the associated with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and pervasive, locally volts are mostly carbonate with R gdugey flacts and increased intensity of a stock work with R gdugey flacts and pervasive, locally volts are mostly carbonate with R gdugey flacts and pervasive, locally volts are mostly carbonate with R gdugey flacts and pervasive, locally volts are mostly carbonate with R gdugey flacts and pervasive, locally volts are mostly carbonate with R gdugey flacts and pervasive, inclusive representation in the associated with R gdugey flacts and pervasive, inclusive representation in the associated with R gdugey flacts with R gdugey flacts and pervasive, inclusive representation in the associated with R gdugey flacts with R gdugey flacts and pervasive, inclusive representation in the associated with R gdugey flacts flat in the associated with R gdugey flacts flat in the associated with R gdugey flacts and pervasive, in the stock work with R gdugey flacts and pervasive, inclusive representation R gdugey flacts flat in the stock with R gdugey flacts and pervasive, in the stock work with R gdugey flacts and flacts with R gdugey flacts flat in the stock with R gdugey flacts and flacts with R gdugey flacts flat in the stock with R gdugey flacts and flacts with R gdugey flacts flat in				1		46.3		50		1	1	l l						
Image: Normal StateImage: Normal				1		1		1		1	1	l l						
55.20 56.20 H FLT Hornfels, Fault Breccia, creamy white quartz/carb onate +/- K alt., locally some Hornfels fragments in carb+/- quartz matrix are incipiently altered to Image: Constraint one constran				1		1		1		1	1	l l						
BX with very little quartz, Hornfels is paler to medium buff colored to maroon (Hematite alteration?) ipy fn to cgr in fracts with quartz/carbonate +/- K alt., locally some Hornfels fragments in carb+/- quartz matrix are incipiently altered to Image: Colored to muddy py. 56.20 65.00 H Mornfels, as above, biotite porphryoblasts more predominant Image: Colored to muddy py. Image: Colored to muddy p				1		1	Local Biotite spotted hornfels, restricted to zones 1-2m in length - Bi porphyroblasts avg 20% of unit.	1		1	1	l l						
BX with very little quartz, Hornfels is paler to medium buff colored to maroon (Hematite alteration?) ipy fn to cgr in fracts with quartz/carbonate +/- K alt., locally some Hornfels fragments in carb+/- quartz matrix are incipiently altered to Image: Colored to muddy py. 56.20 65.00 H Mornfels, as above, biotite porphryoblasts more predominant Image: Colored to muddy py. Image: Colored to muddy p	55 20	56 20	н	FLT		1	Hornfels, Fault Breccia, creamy white quartz/carb stockwork with iK fracture controlled and pervasive, locally writs are mostly carbonate	1	1	1	1		+					1
a a b	50.20	20.20	l			1		1		1	1	l l						
a a	I			57		1				1	1	l l						
56.20 65.00 H Image: Main and the state of the	I			1		1				1	1	l l						
65.00 68.60 H FLT Hornfels, Fault, iK fracts and pervasive, i broken core, numerous quartz/carb vnlts and fracts with iK I	56.00	65.00	<u> </u>		+			+	-				+	+				+
68.60 N <td></td> <td></td> <td></td> <td></td> <td></td> <td>ļ</td> <td></td> <td></td> <td>-</td> <td>_</td> <td><u> </u></td> <td>ļ</td> <td>_</td> <td> </td> <td></td> <td></td> <td></td> <td></td>						ļ			-	_	<u> </u>	ļ	_					
Mosts of unit is dark maroon brown locallized spotty Biotite porphyroblasts, K fracs are common, quartz/carb frac with Sx throughout	65.00	68.60	н	FLT		1	Hornreis, Fauit, ik tracts and pervasive, i broken core, numerous quartz/carb vnlts and fracts with ik	1		1	1	l l						
Mosts of unit is dark maroon brown locallized spotty Biotite porphyroblasts, K fracs are common, quartz/carb frac with Sx throughout	68.60	83,80	н	1		1	Hornfels, as above 68.6 - 71.0 diss pv +/- cov pervasive and especially fract controlled with guartz/carb filling	1	+	1	1	-	1					1
	30.00	20.00	1	1		1		1		1	1	l l						
				1		1		1		1	1	l l						
					+			+	+				+	┣──┤				

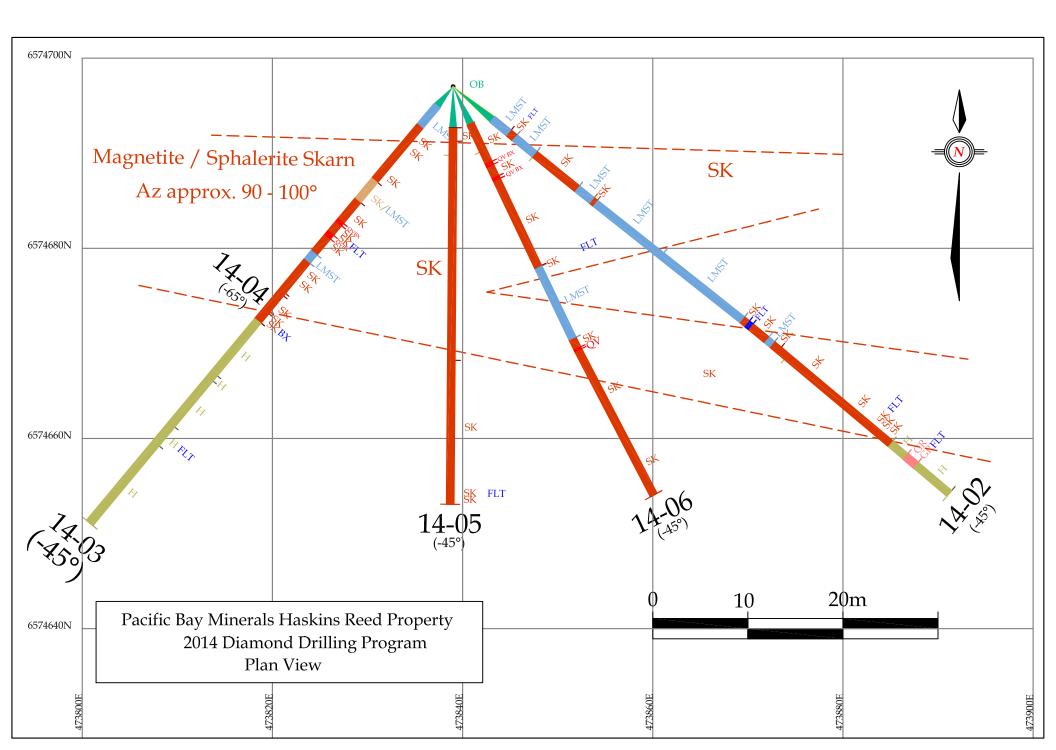
Pacific	: Bay N	linerals				2014 Mt. Reed - Brett Zone	DD	H No	o. : '	14-04				Pg 1 of 1	i							
Collar D	etails:	Note: GPS	only, non			Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths					s	started:		August 24, 2014								
Easting						UTM NAD 83 Z09				[inished: August 26, 2014		26, 2014						
Northing	g					UTM NAD 83 Z09					Logged I		gged By: Lesley Hunt		t							
Elevatio				1,		m ASL											EZSho		Tests:	Depth (m)	Az	Dip
End of H					67.1	m												67.07	219.80	0 -65.70		
Azimuth	۱				220.0																	
Dip	-		1		-65°																	
From	To Pic.		Pic.	Mineralizatio				1 (%))	_	Sampling											
(m)	(m)	Lithology	Struc.	#	m	Description	Ру	Ро	Сру	Sph M	ag I	Мо	From (m)	To (m)	Sample No.	Width (m)						
0.00	6.10	OB				Overburden																
6.10	13.70	LMST	STKWK			Limestone Skarn Contact Zone; Drill hole is skirting the contact of the limestone / Diopside Garnet Magnetite Sphalerite Skarn																
						contact. Classic Metasommatic contact ie. wavy sharp chilled margins (photo 181), Some epidote, maybe diopside with pinkish																
				181		orangey hues (altered or partially digested garnets?) with local hematite staining.																
						Contact zones are mag/sph/hem with formation of garnets further inboard, (see photo)																
						5cm iKgouge 13.3 to 13.35																
13.70	54.40	LMST				Limestone: Dark grey / Pale grey / white locally finely laminated																
						24.0 - 54.4 Mostly banded alternating dark grey light grey limestone, avg pdo of laminae low angle tca, few quartz/carb vnlts, some																
						chaotic microfaulting, a few noted quartz/carb vnlts parallel tca, avg 1cm																
54.40	57.20	LMST	STKWK			Limestone Stockwork: Few Epidote/Hematite/Diopside with Magnetite/+/- Sphalerite Skarn filled fractures and few 10 - 15 cm																
						irregular patches (either skarn selvage or larger fracture filled with skarn)																
57.20	67.10	SK				Diopside Garnet Skarn; Pale greenish yellow, aphanitic, 57.2 - 59.2 is avg. 40% Magnetite +/- sphalerite in Skarn up to 4cm																
						irregular patches.																
						@ 60.0m green becomes more bluish green (diopside?), few Magnetite/Sphalerite vnlts <5%, , pinkish orangish color patches																
						altered / partially digested garnets.																
						Local rare bornite with Magnetite/sphalerite (sample #57411)																
						67.1m End of Hole																

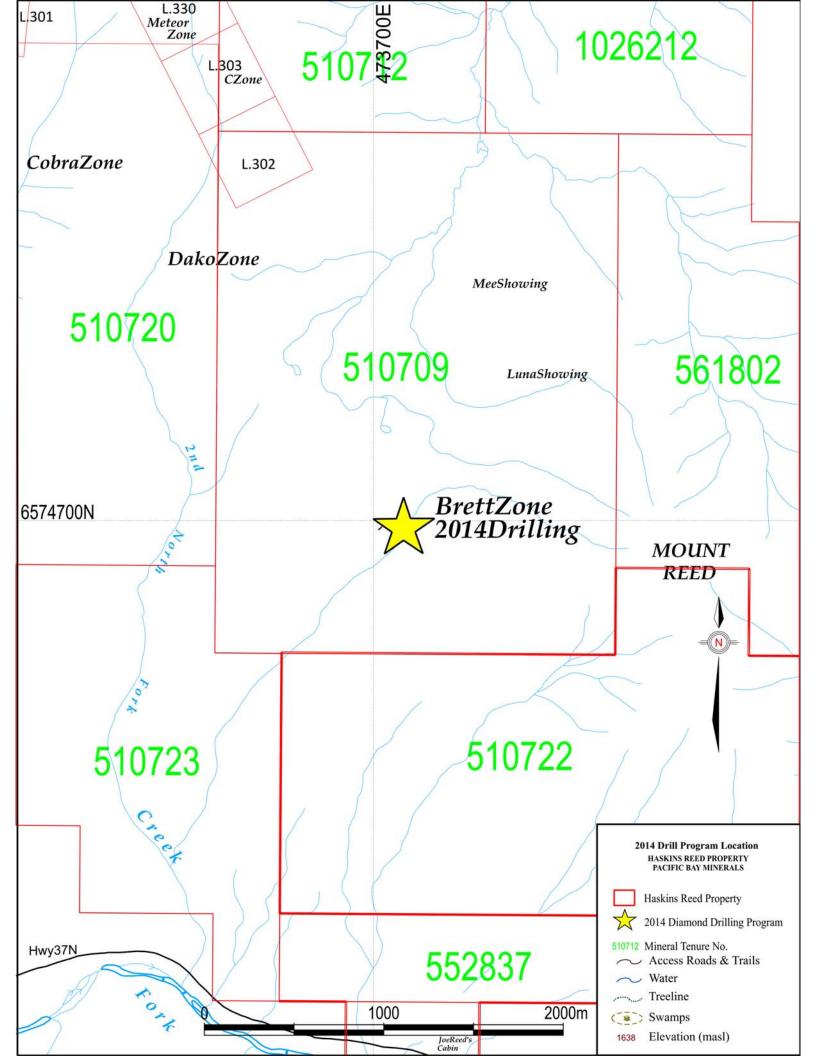
Pacific	Bay N	linerals				2014 Mt. Reed - Brett Zone	DD	H No	o.: 1	4-05	5			Pg 1 o	f 1				
Collar D	etails:	Note: GPS	ն only, r	non sur	veyed	Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths						Started: August 26, 20							
Easting						UTM NAD 83 Z09									Finish		0	27, 2014	
Northin	g					UTM NAD 83 Z09						ogge		Lesley Hu					
Elevatio						m ASL					I	EZSho	ot Tests:	Depth (m)		Dip			
End of I					61.9	m					_			61.89	180.80	<mark>-44.60</mark>			
Azimuth	1				180.0														
Dip	1				-45°														
From	То		-	Pic.		Provide the		M	inera	izati	on (%)			npling	1			
(m)	(m)	Lithology	Struc.	#	m	Description	Ру	Po Cpy Sph Ma		Mag	Mo From (m)		To (m)	Sample No.	Width (m)				
0.00		OB				Overburden													
6.10	8.10	SK				Diopside/Garnet Skarn, Buff to pale yellowish, aphanitic, local mgr to cgr garnets, esp. coarse grained at 7.8m sub to euhedral													
						garnets olive green to brownish red in color to 2mm in vnlts and irregular structures sub parallel tca													
8.10	40.60	SK				Diopside/Garnet Magnetite Sphalerite Skarn, Buff to pale yellowish, locally pale green, aphanitic, w numerous black Magnetite/Sphalerite patches, vnlts, and stkwks, local patches of py +/-po to 5 - 10 %, semi massive sulphides to locally massive sulphides, +/- cpy, +/- Bn, local quartz/carbonate fracture filling. OVERALL AVG for unit: Mag/sph patches -50%, py -20%, Cpy -5%, details below, 8.1 - 19.5 avg Mag/sph 30 - 40% 19.5 - 22.0 avg Mag/sph 5 - 7%, quartz carbonate vnlts to 3cm become more abundant 17.0 to 22.9m local drusy quartz crystals with open fractures and drusy py to 2mm, local pinkish hue hem? gnts? 25.0 -29.8 avg Mag/sph 50 up to 70%, 29.2 cpy on fracs 29.8 - 31.8 avg Mag/sph 80%, grey silverish Sx 15 - 25% Po? (Note analysis revealed nothing anomalous). 31.8 - 35.7 avg Mag/sph 40 - 60%, local 20% 35.7 - 40.6 avg Sx 30 - 40%, Po 20% LC subparallel tca, relatively discreet													
	59.95			133 135 136 137 139	41.8 42.85 41.85 43.7 44.3	Garnet Diopside Skarn: iSiliceous, mod foliated @ 25° tca, pale green, orangey pink irregular bands, late rusty red/brown late stage cross cutting fracture filling, also Mo in fractures Alteration overprinting of iSiliceous skarn, some Skarn appears to be calc silicate skarn and others are Garnet rich skarn. Alteration appears to be late stage as shown in photo 137 & 138, the color of green varies from the classic pistachio green of epidote to the blue green diopside color. Mo in patches of disseminated mm scale grains. Note if samples 282331 & 332 run high Mo, sample the remaining drill core. Patches of grainy pinkish material may be partially digested garnets or hematite. 59.45m - 3cm iK fault gouge @ 70°tca													
	60.05		FLT			Skarn; iK gouge													
60.05	61.90	SK				CalcSilicate Skarn: Pale grey green and maroon, irregular patches of iSiliceous skarn, moderately fractured with fn gr py, cpy +/- po to 5%.													
						61.9m End of Hole							T						

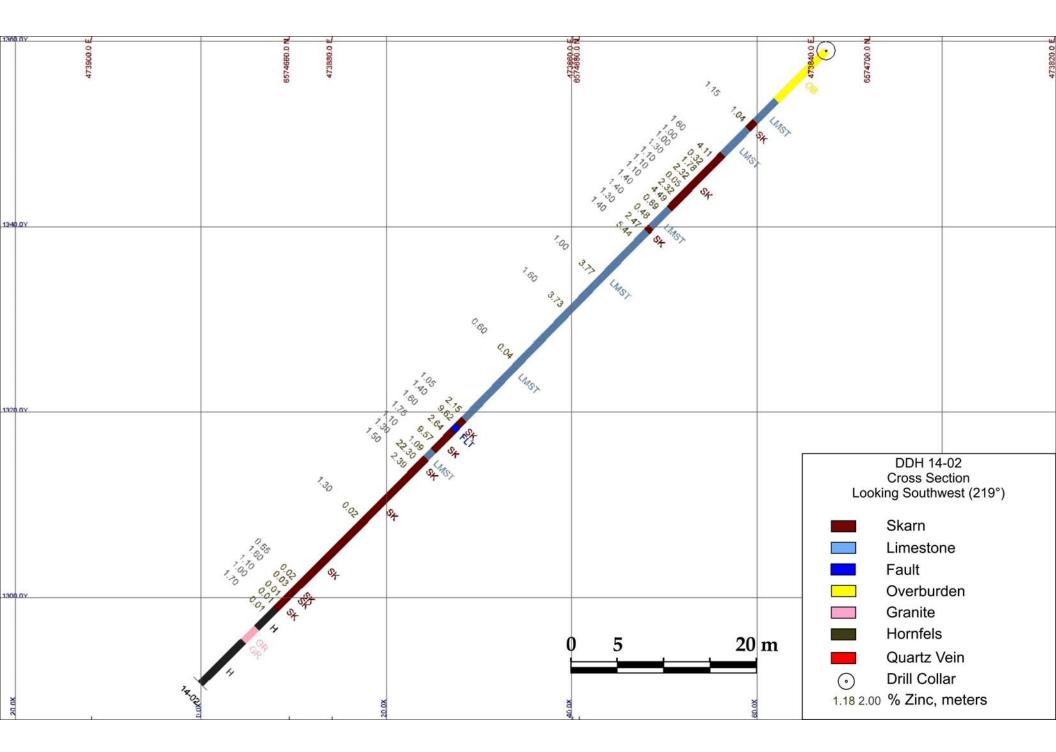
Pacific	Bay N	linerals				2014 Mt. Reed - Brett Zone	DD	H No	.: 1	4-06				Pg 1 of 1		
Collar De	tails:	Note: GP	S only	, non s		Purpose: Test Brett Zone, confirm 97-01 & 02 grades and widths						rted:			gust 27, 2014	<mark>,</mark>
Easting						UTM NAD 83 Z09						ished			<mark>gust 29, 201</mark> 4	<u> </u>
Northing				6574697 UTM NAD 83 Z09								jged E		Lesley Hu		
Elevation			1,359.0 m ASL								EZS	Shot T	ests:	Depth (m)		Dip
End of H	ole				67.8		-							44.21	154.20	-45.00
Azimuth Dip					<u>155.0</u> -45°		-									-
		Lithology	Struc	. Pic.	m			м	ineral	ization	(%)		1	Sa	nnling	
From	То		0	#		Description			1		Ť	T	From		Sample	Width
(m)	(m)						Ру	Ро	Сру	Sph Ma	ig Mo	•				(m)
0.00	6.10	OB				Overburden										
	12.50					Magnetite Sphalerite Skarn, buff yellow, aphanitic, locally iSiliceous, weakly fractured with numerous quartz/carb hairline fractures,										
		-				moderately broken core to 8.2m not a lot of FeOx.										
						6.1 - 10.5 very weak local magnetite/sphalerite patches to 2%.										
						10.5 - 12.0 magnetite sphalerite patches and stockworks avg 30%										
10.50	40.00	01/	вх	142	12.9		-									
12.50	13.00	QV	БХ	142	12.9											
10.00	44.05	01/		_		py, large irregular vugs some with iFeOx coated drusy quartz.					_	-				
	14.85		DY	1.40	45.0	Magnetite Sphalerite Skarn as above Magnetite Sphalerite avg 10%	<u> </u>				_				ļ	──
14.85	15.30	QV	вх	143 144	15.2 15.2	Quartz Vein Breccia; as above, very large drusy euhedral quartz crystals to 2cm (see photo)	1									
			1	145	15.2		1									
15.30	29.65	SK				Magnetite Sphalerite Skarn; more of a green tinge to body of skarn (diopside), Magnetite Sphalerite patches and stockwork 40%, local	Description D	1								
		-				diopside in fractures (maybe Serp?)										
						20.5 - 29.6 Magnetite Sphalerite 30%, +/- po, +/- cpy										
						@25.4, the skarn becomes a Magnetite Sphalerite Diopside Garnet Epidote Skarn, greensih hue with green garnets and pinkish										
						patches (altered red garnets or hematite?), quartz carbonate vnlts with po and epidote, locally to 2mm, avg hairline										
						LC is iK gouge 10cm.										
29.65	29.75	SK	FLT			Fault; Kqouge is a fault contact with limestone	1									+
		LMST	· - ·	148	36.1	Limestone: Pale grey, fine grained, numerous dark grey fract. (dk. grey Si?), quartz/carb vnlts irreg to 0.5cm avg 1%,	-									1
20.10				150		Note: a pdo of quartz carbonate vnlts and fracture filling and skarn filled fractures is weak sub parallel tca.										
						29.75 - 36.0 Few Diopside Garnet Epidote Magnetice Skarn filled fractures more well assimilated than further down hole towards skarn										
						contact.										
						36.0 - 41.5 Diopside Garnet Epidote Magnetite Skarn filled fractures with discreet contacts become more prominent ie.10-15%, 36.2m										
						locallized up to 4cm partially digested skarn fragments.										
						36.1m - Dark grey Sx? disseminated, rust colored mineral (not hematite or garnet looking), garnets that are there appear more orangey										
						and indiscreet boundaries and more in patches (photo 148).										
						40.0 - 41.5m Skarn patches to 25%										
41.50	43.10	SK				Magnetite Sphalerite Diopside (Epidote?) Garnet Skarn; few quartz/carb vnlts @ low angles tca to 1cm, avg mm scale with med to										++
41.50	40.10	ÖN				coarse grained py disseminated, especially at vnlt selvages.										
						Mineralization: Magnetite-Sphalerite 80%, Po 10%, Py 5%, silver grey mineral 3%, +/- cpy										
43.10	43.40	ov	-	+		Quartz Vein; UC & LC is 30°tca, 2cm band of skarn sub paralled to contact with Magnetite Sphalerite +silverish grey mineral, +/- cpy,			\vdash		_	-	\vdash			+
43.10	40.40	Q V														
						texture parallel to fractures.										
43.40	57.75	SK		152	49.35	Diopside Epidote Garnet Magnetite Sphalerite Skarn;	<u> </u>				-	+				+
-3.40	51.15			153												
						43.4 - 45.9 Magnetite/Sphalerite <5%, few quartz/carb vnlts @ 30°tca, fractures are irregular with quartz/carb +/-K, pinkish to rusty										
						colored patches local (altered garnets)										
						45.9 - 49.0 Magnetite/Sphalerite 40-50%										
						garnets disseminated and partially digested to 0.5cm.										
						53.0 - 57.75 Magnetite/Sphaletite 10-15%										
			1			Note: Green in Skarn grades from Epidote colored - (pistachio) @ 53.4m to Diopside (darker aqua teal) colored further downhole.	1									
			I				<u> </u>				_	-			ļ	\square
57.75	67.80	sĸ		154	67.08 67.08	CalcSilicate and Diopside/Garnet Skarn; (Calc Silicate overprinting on a banded Diopside Garnet Skarn)										
			1	100	07.08	Pale green/tan, siliceous, aphanitic, Calc Silicate Skarn bands alternate with softer Diopside/Garnet skarn, CS skarn is often diopside	1									
			1			rich, K +/- py on irregular mm scale fractures is common.	1									
			1			66.95 - 67.8 Patches of blackish green with siliceous aphanitic (not CS skarn) with 25 - 30% po/py/+/-cpy.	1									
			1			@ 67.08m - 2cm brecciated vnlt with trace carbonate in the matrix 30°tca, myrmekitic like texture of relatively soft hexagonal euhedral	1									
						cystals and wormlike crystals - creamy white in color, matrix is tan brown.										
						67.8m End of Hole						1				

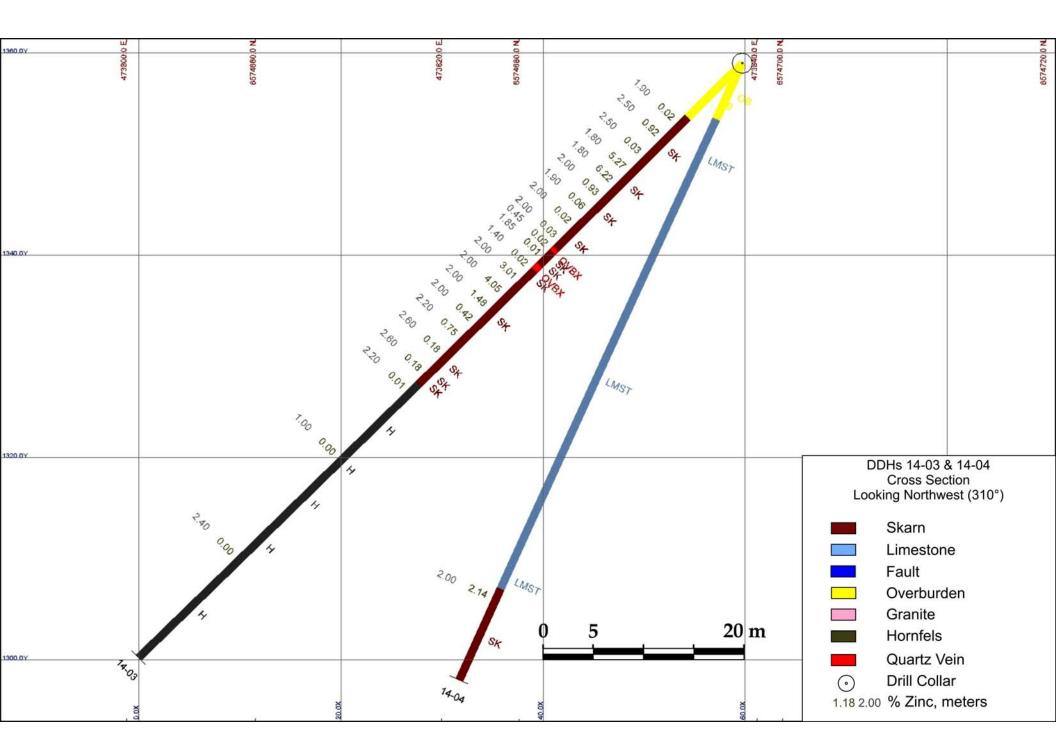
APPENDIX E

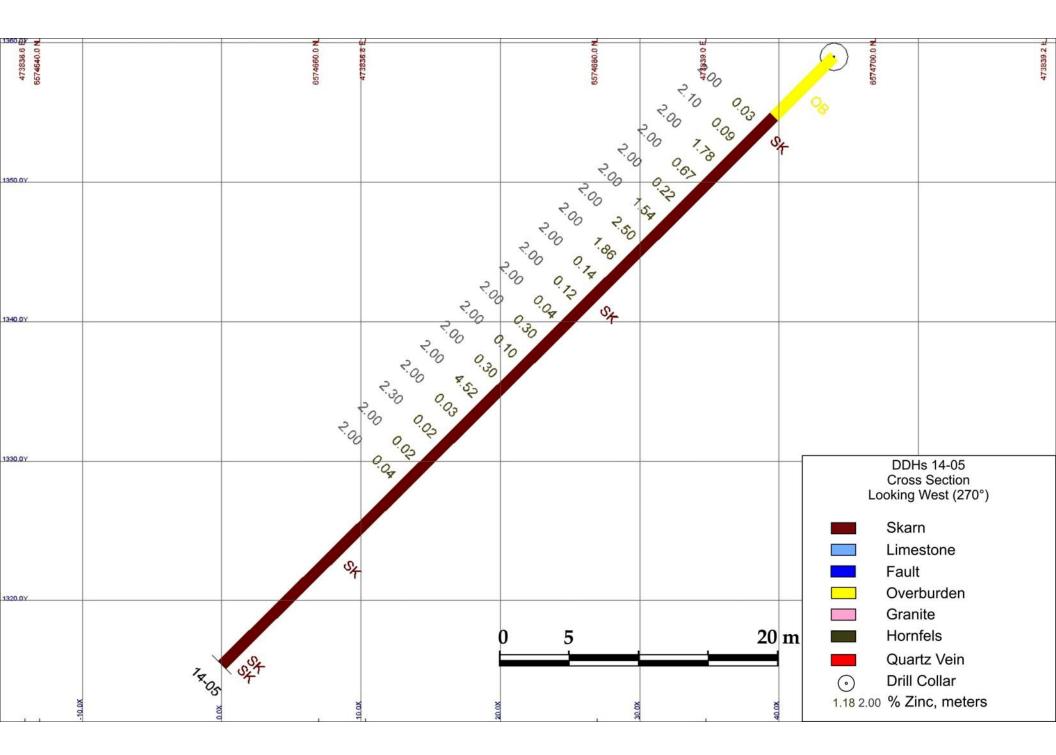
Diamond Drill Plan Views & Cross Sections

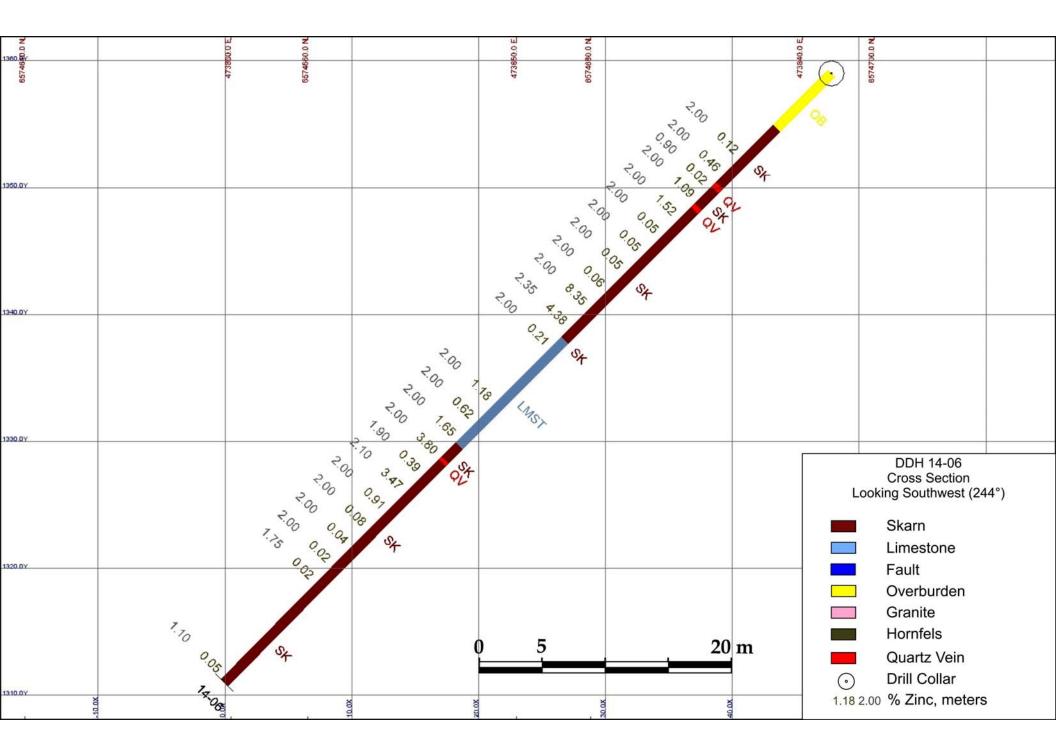












APPENDIX F:

2014 Haskins Reed Diamond Drill Program

COST STATEMEMT

2014 Haskins Reed, Diamond Drill Program Cost Statement

Exploration Work type	Comment	Days			Totals
Personnel (Name)* / Position	Field Days (list actual days)	Days	Rate	Subtotal*	
Lesley Hunt, Manager, Geologist	August 15th to September 6th, 2014		\$600.00		
Lesicy Hunt, Manager, Geologist	August Totil to September offi, 2014	23	\$000.00	\$13,800.00	\$13,800.00
Office Studies	List Personnel (note - Office only	, do not	include f		\$10,000.00
Literature search		, ao no	\$0.00		
Database compilation	Lesley Hunt	4.0	\$600.00		
Computer modelling		ч.0	\$0.00		
Reprocessing of data			\$0.00		
General research			\$0.00		
Report preparation	Lesley Hunt	4.0	\$600.00		
Other (specify)		4.0	\$000.00	\$2,400.00	
Other (specify)				\$4,800.00	\$4,800.00
Airborne Exploration Surveys	Line Kilometres / Enter total invoiced ar	mount		\$4,000.00	\$4,000.00
Aeromagnetics	Line Kilometres / Enter total invoiced a	nount	\$0.00	\$0.00	
Radiometrics	-		\$0.00		
			\$0.00		
Electromagnetics			\$0.00		
Gravity					
Digital terrain modelling			\$0.00		
Other (specify)			\$0.00	I	¢0.00
B I. C				\$0.00	\$0.00
Remote Sensing	Area in Hectares / Enter total invoiced a	mount or			
Aerial photography			\$0.00		
LANDSAT			\$0.00		
Other (specify)			\$0.00		<u> </u>
		1		\$0.00	\$0.00
Ground Exploration Surveys	Area in Hectares/List Personnel				
Geological mapping					
Regional			penditures		
Reconnaissance				in Personnel	
Prospect		пеіа ехр	enditures a	ibove	
Underground	Define by length and width			#0.00	* 0.00
Trenches	Define by length and width			\$0.00	\$0.00
Ground geophysics	Line Kilometres / Enter total amount inv	voiced list	personnel		
Radiometrics					
Magnetics					
Gravity					
Digital terrain modelling					
Electromagnetics	note: expenditures for your crew in t				
SP/AP/EP	should be captured above in Personn	nel			
IP	field expenditures above				
AMT/CSAMT					
Resistivity					
Complex resistivity					
Seismic reflection					
Seismic refraction					
Well logging	Define by total length				
Geophysical interpretation					
Petrophysics					
Other (specify)					
				\$0.00	\$0.00

2014 Haskins Reed, Diamond Drill Program Cost Statement

Exploration Work type	Comment	Days			Totals
Geochemical Surveying	Number of Samples	No.	Rate	Subtotal	
Drill (cuttings, core, etc.)			\$0.00	\$0.00	
Stream sediment			\$0.00	\$0.00	
Soil	note: This is for assays or		\$0.00	\$0.00	
Rock	laboratory costs	115.0	\$52.17	\$6,000.00	
Water			\$0.00	\$0.00	
Biogeochemistry			\$0.00	\$0.00	
Whole rock			\$0.00	\$0.00	
Petrology			\$0.00	\$0.00	
Other (specify)			\$0.00		
				\$6,000.00	\$6,000.00
Drilling	No. of Holes, Size of Core and Metres	No.	Rate	Subtotal	
Diamond	6 NQ, 408.6meters	408.6	\$141.38	\$57,767.00	
Reverse circulation (RC)			\$0.00	\$0.00	
Rotary air blast (RAB)			\$0.00		
Other (specify)			\$0.00		
			1	\$57,767.00	\$57,767.00
Other Operations	Clarify	No.	Rate	Subtotal	
Trenching	č		\$0.00	\$0.00	
Bulk sampling			\$0.00		
Underground development			\$0.00	\$0.00	
Other (specify)			\$0.00		
				\$0.00	\$0.00
Reclamation	Clarify	No.	Rate	Subtotal	
After drilling	Z		\$0.00	\$0.00	
Monitoring			\$0.00	\$0.00	
Other (specify)				\$0.00	
			1		
Transportation		No.	Rate	Subtotal	
Airfare			\$0.00	\$0.00	
Taxi			\$0.00	\$0.00	
truck rental			\$0.00	\$0.00	
kilometers			\$0.00	\$0.00	
ATV			\$0.00	\$0.00	
fuel			\$0.00	\$0.00	
Helicopter (hours)			\$0.00	\$0.00	
Fuel (litres/hour)			\$0.00	\$0.00	
Other					
			1	\$0.00	\$0.00
Accommodation & Food	Rates per day				
Hotel			\$0.00		
Camp Rental		15.00	\$300.00		
Meals	day rate or actual costs-specify		\$0.00	l	
				\$4,500.00	\$4,500.00
Miscellaneous			+ ·		
Telephone		_	\$0.00	\$0.00	
Other (Specify)					• • •
				\$0.00	\$0.00
Equipment Rentals			+ ·		
Field Gear (Specify)			\$0.00		
General Camp Supplies			\$0.00		
			.	\$588.00	\$588.00
Freight, rock samples	1 day round trip to Whitehorse	1.00	\$400.00	l	
				\$400.00	\$400.00
TOTAL Expenditu		1	1	+	\$87,855.00