

REPORT OF CORE DRILLING
APRIL – MAY, 2014
ON THE
SHUNSBY VMS Cu-Zn PROJECT
CUNNINGHAM TOWNSHIP
NORTH CENTRAL, ONTARIO

OF
BLACK WIDOW RESOURCES INC.
FOR
BLACK WIDOW RESOURCES INC.

By
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August 29, 2014

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INTRODUCTION

Black Widow Resources Inc. (BWR) owns a 59.8% interest to the Shunsby volcanogenic massive sulphide (VMS) copper-zinc property located in Cunningham Township, north central Ontario. Previously, extensive exploration, including 23,460m of core drilling in 214 drill holes, has been conducted by several companies on this 20 claim property over a 110 year period. Between April 24 and May 4, 2014 BWR completed a short, 6 hole, 750m core drilling program with holes designed to twin holes from previous drill campaigns. The Author, an independent geological consultant, acted as the on-site supervisor for the BWR program, logged the core and selected the samples for assay. Mr. M. Peplinski, an independent prospector / explorationist, cut the core and otherwise capably assisted.

In order to apprise shareholders in a timely manner as to BWR's activities, Mr. Neil Novak, President and CEO of BWR, retained the Author to prepare this report. For detailed information regarding the Shunsby property the reader is referred to the NI 43-101 report dated September 30, 2013, and titled "Technical Report on the Shunsby Base Metal Property, Cunningham Township, Ontario, for Black Widow Resources Inc." by Paul Sobie, P. Geo of MPH Consulting Ltd., Toronto (the **Shunsby Report**) which was filed on <http://www.sedar.com> and on the BWR website at www.blackwidowresources.com.

PROPERTY DESCRIPTION AND LOCATION

Twenty (20) contiguous, surveyed, patented crown claims comprise the property. These claims cover 314.43 hectares and include both surface and mineral rights.

The property is situated some 190km NW of Sudbury and 145km SW of Timmins, Ontario (**Figure 1**). Geographical coordinates for the approximate property centre are 82° 39' 20" west longitude and 47° 43' 10" north latitude or in UTM coordinates 375840E and 5286415N (WGS 84, Zone 17). The NTS designation is 41 O/10.

ACCESS AND INFRASTRUCTURE

There is road access onto and across the property. From the juncture of Provincial Highways 144 (which links the cities of Sudbury and Timmins) and 560, follow the Sultan industrial road (an all weather gravel road) westward for 64km, and then the Blamey and Cunningham roads northward for another approximate 25km to the property. Bush roads (former logging and exploration access roads) provide access across the property. Neither the Blamey nor Cunningham roads are maintained and are not open during the winter months.

Other than the bush roads, there is no infrastructure on the property. The main CP rail line passes in a general NNW / SSE direction some 25km SW of the property. Airports with several daily flights to Toronto, as well as goods and services for the mining and exploration industry are readily available in both Sudbury and Timmins.

EXPLORATION HISTORY

The Shunsby property was first staked for its iron ore potential in 1904, but it was not until 1927 that its base metal (Cu-Zn-Pb) potential was realized, and exploration intensified. Since 1904, work conducted on the property includes prospecting, geological mapping, soil geo-chemical surveys, geophysical surveys (magnetic and horizontal loop electromagnetic), stripping & trenching, 23,460m of core drilling in 214 drill holes, and data compilation. For a more detailed account of the work carried out by individual companies the reader is referred to the “**Shunsby Report**”.

GEOLOGICAL SETTING

The property is situated in the Swayze greenstone belt that is interpreted to be the southwest extension of the Abitibi greenstone belt that hosts the famous gold and / or base metal camps of Timmins, Detour Lake, Kirkland Lake and Larder Lake in Ontario, as well as the Noranda, Malartic, Val d’Or, Mattagami and Chibougamau located in Quebec. Several north to northwest striking faults as well as granodiorite / monzonite intrusions partially separate the Swayze from the Abitibi belt.

MPH Consulting Ltd. completed the latest geological mapping for the property and immediate area in 1989 - 1991, and the reader is referred to the “**Shunsby Report**” for a detailed summary of the property geology. The following description has been condensed from the Shunsby Report.

Three geological domains (**Figure 2**) are recognized on the Shunsby property;

- A **Lower** (oldest) mafic volcanic-gabbro-iron formation domain exposed in the extreme western portion of the property.
- A **Central** pyroclastic, chemical sedimentary – clastic sedimentary domain with minor basalt that underlies the central and western parts of the property. { This domain hosts the Shunsby copper-zinc mineralization. } and
- An **Upper** (youngest) mafic (basalt-gabbro) domain in the easterly and northeasterly property quadrants.

Rock units strike in a generally northerly direction, dip shallowly - moderately westward, and are overturned with tops to the east. Folds, faults as well as late intrusions have complicated the volcanic stratigraphy.

Two copper – zinc mineralized areas are known on the property; the **Main Zone** located central to the property and the **South Zone** in the south central portion. The Main zone displays higher copper values while the South Zone is more zinc rich.

From west to east the salient stratigraphy relevant to the mineralized zones, as per the Shunsby Report, comprises

- **Upper Chert Complex** (which are in actuality stratigraphically lower)
- **Variolitic basalt** (unmineralized)
- **Lower Chert Complex** (stratigraphically upper)

HISTORIC DRILLING DATA

Introduction

Between 1954 and 1993 some 10 companies drilled 214 core holes on the Shunsby property. The bulk of these holes were drilled in the Main and South zone areas (**Figure 3**). Due to variations in core size, sampling methodology, laboratories used, assay methods employed, etc., as detailed below, there is the potential for the assay data from the various drill programs to be suspect.

Core Size

The Author reviewed the reports and drill logs available for past drill programs, and was able to determine only that the holes drilled by Placer Development, MW Resources and Phelps Dodge were of BQ size (36.5mm dia.). In the Author's experience, most holes drilled prior to the late 1960's to mid 1970's (the transition period from standard to wireline drilling) were probably of size AXT (core diameter 28.6 mm) although some might have been EXT (core diameter 22.2mm) or even smaller. It was only in the late 1970's –early 1980's that the company norm became BQ.

Depending upon the drill contractor, the size of the contract, the availability of equipment and the preference of the exploration company it is expected that the size of the core drilled varied considerably. Since the larger the core size means a greater sample volume the difference in core size is important most particularly for gold exploration whereby the gold is irregularly distributed. Since, for the Shunsby project since much of the mineralization is unevenly distributed, and is present in multiple settings - along fractures in thin quartz-carbonate veins, disseminated and in semi massive to massive sulphide beds - core size may also be relevant.

Sampling Methods

For most drill programs the common practice is to cut the core in half and send one half for assay and retain the other half for future reference. Prior to the mid 1970's, the usual way to cut the core was with a mechanical splitter. Post the 1970's, cutting with a saw equipped with a diamond impregnated blade became the norm, if power was available at the core logging /core cutting site. For both methods it is imperative that the sampler be consistent and unbiased. No sampling methodology description is available for any of the drill programs.

Assay Method and Laboratory

Commercial laboratories are reliable enterprises, but regardless, mistakes do happen. It was not until 1998 that laboratories began to be ISO accredited to international standards. Also, prior to that date it is probable that most laboratories did not take part in round-robins exercises as a means to check their results against other labs. It is also unlikely that the laboratories reported their in-house QA/QC results to the client company unless specifically requested by the company.

Assay certificates are not available for the historic holes. (Prior to 1991 companies submitting assessment reports for exploration projects were not required to include assay results, let alone assay certificates.) Therefore, for none of the drill programs, neither the laboratory at which the assaying was performed nor the assay methodology(s) is (are) known.

Although there is no reason to believe that the assay data from previous drill programs may be suspect, there is no means to check the validity of the assay data.

APRIL – MAY, 2014 CORE DRILLING PROGRAM

Purpose

Due to the potential for problematic assay data from previous drill programs, BWR needed to determine the reliability of the historic assay results for possible usage in any future resource estimation exercise. To that end, BWR selected 6 historic holes to twin (drill a second hole in close proximity to the original, and at the same azimuth and dip).

Program Description

The criteria upon which the holes to be twinned were selected were **age** (2 were 1956 vintage one of which one was extended in 1966, 2 were drilled in 1968 and 2 were from 1981), **mineralogy** (whether the assay results for the original hole were copper, zinc or copper & zinc rich), and **location** (2 holes were situated in the Main Zone, 2 in the South Zone and 2 between the two zones).

In 1990, MPH Consulting Limited (acting as contractors to Kirkton Resources Corporation) located the majority of the historic drill holes, and tied these to a surveyed grid. Their locations were transposed later into the UTM grid system, WGS 84 datum. The drill sites for the BWR holes were located with a hand held GPS unit (with an accuracy of +/- 5m) according to the UTM coordinates for the historic holes being twinned. For some BWR holes evidence for the historic hole being duplicated was found, but for others snow conditions at the time of the BWR drill program covered any potential evidence, and also prevented an assessment as to whether or not the topography at the site had been altered as a result of the stripping and trenching conducted by MPH Consulting during the 1989 – 1991 exploration program.

Chenier Drilling Services from Hamner, Ontario was awarded the drill contact. The core produced was B thin wall (BTW) with a 42mm diameter. Core recovery was excellent.

The core for all 6 holes was logged by the Author in a consistent manner, and to industry best practice standards. Samples were selected based upon lithology, mineralogy, alteration and structure, and where possible did not cross contacts. Sample size was nominally 1.5m, but varied from 0.50 to 2.40m. Most samples were cut with a saw, but in a few cases, where the core was extremely friable, the whole core was taken for assay. To ensure unbiased sampling, for each sample the core cutter (Mr. M. Peplinski)

meticulously drew a line on the core, and after cutting placed in a sample bag the pieces of core from the same side of the line. Each sample was individually numbered, and secured with a plastic tie wrap. Approximately 8 samples (depending upon weight) were placed into woven plastic bags, and secured with an individually numbered security seal. Samples were delivered at the end of the drill program by Mr. M. Peplinski to the Actlabs facility in Sudbury, Ontario for sample preparation and for analysis / assay either there or at the main Actlabs facility in Ancaster, Ontario. All Actlabs facilities are ISO accredited.

The samples were treated as follows;

- Prepared according to Actlabs' code RX1 (crush the entire sample so that 90% passes a 2mm screen, split out a 250gm sub-sample, pulverize the sub-sample so that 95% passes a 105 micron screen).
- The sub-sub-sample was analyzed for multi elements according to Actlabs' code UT-5 Total Digestion (a combination of INAA and 4 acid digestion with ICP finish).
- All samples that assayed over limits for Cu, Zn and Pb (>1% Cu, >1% Zn & >0.5% Pb) by the UT-5 Total Digestion process were re-analyzed according to Actlabs' code 8-Peroxide Fusion-ICP.

Complete descriptions for the Actlabs procedures may be found at www.actlabs.com. Actlabs regularly inserts blanks, standards and duplicates into the sample stream as an in-house QA / QC check. BWR did not employ any field blanks, standards or duplicates.

Results

Included in this report are the drill logs as **Appendix 1**, and a comparison chart of the assays results from BWR's drill program versus the comparable historic drill hole as **Appendix II**. Note that for Appendix II the sample intervals for the twinned historical drill holes were in Imperial measurements, and these have been converted to metric units. In the discourse for the sets of holes that follows only the values for Cu & Zn are included in the discussions as these two elements are the only ones considered germane to the economics of the zones. Also, for many of the historic holes only Cu and Zn values are available.

BWR 14-01 (Figures 3 & 4), located in the Main Zone, was drilled on a 190° azimuth with an 80° dip to 100.5m depth, and was designed to replicate MW Resources hole 81-08 which had been drilled on the equivalent azimuth and dip to 69.6m depth. Hole 81-08 is considered to be a "Zn rich" hole. No evidence for 81-08 was found at site at the time of drilling.

A 13.65m rhyolite tuff interval cored between 44.65m and 58.30m contained variable amounts of pyrite, chalcopyrite, sphalarite and galena mineralization in fractures, disseminations and semi massive bands (to 40cm) which produced 2 significant assay intercepts as follows;

- 44.65m – 48.10m (3.45m) assayed 0.14% Cu and 2.07% Zn
- 53.00m – 58.30m (5.30m) assayed 0.08% Cu and 1.86% Zn.

The entire lithological unit, including the intervening low grade section, averaged

- 44.65m – 58.30m (13.65m), at 0.08% Cu and 1.29% Zn.

By comparison, hole 81-08 cored the same lithological unit

- 51.82 – 63.40m (11.58m) and assayed 1.58% Cu and 4.68% Zn.

Although the thicknesses for the corresponding intervals are similar, the assays for both Cu & Zn are markedly lower for BWR 14-01. The difference in the depths of the mineralized units might in part be due to the stripping and trenching conducted by MW Resources plus a potential 5m difference in the locations for the 2 holes, whereas the differences in the Cu and Zn assays may reflect the irregular distribution of the mineralization.

A weakly mineralized rhyolite tuff / graphitic ash unit separated by a variolitic basalt unit from the intercept mentioned above was cored between 68.20 and 71.70m. Elevated but non significant Zn values were obtained for the two samples selected. Hole 81-08 was not drilled sufficiently deep to cut this unit.

BWR 14-02 (Figures 3 & 5) was collared in the Main Zone in close proximity to MW Resources hole 81-01, and drilled vertically to 100.5m. 81-01 had been drilled vertically to 39.3m and is considered to be a “Cu rich” hole. Evidence for prior drilling was a drill rod found at the BWR 14-02 drill site.

In BWR 14-02 a mineralized section was cored from 3.80m to 36.50m comprising rhyolite tuff, ash, and slump breccia (a chaotic mixture of angular to sub rounded rhyolite tuff, ash, basalt, chert and semi massive sulphide fragments) lithologies. Significant intercepts cored;

- 4.10m – 22.00m (17.90m) assayed 1.06% Cu and 2.15% Zn including
 - 10.50m – 16.45m (5.95m) assayed 1.49% Cu and 3.24 Zn
- 30.00m – 36.00m (6.00m) assayed 0.86% Cu and 1.41% Zn.

The entire mineralized section, including the low grade material, averaged

- 4.10m to 36.00m (31.90m) at 0.83% Cu and 1.53% Zn.

The equivalent lithological unit(s) in hole 81-01ran

- 4.88m – 37.49m (32.61m) at 0.92% Cu and 1.36% Zn.

Note that the low grade mineralization was not assayed by MW Resources and was given a zero value for the average.

Note also that Hole 81-01 cut higher grade material, but not at depths similar to the higher grade mineralization in BWR 14-02. This fact is not unexpected given the irregular and multiple occurrences of the mineralization.

The two holes compare very favourably in terms of the average thickness for the mineralized lithological unit(s) cored, and the average grade thereof.

BWR 14-03 (Figures 3 & 6), located in the South Zone, was drilled vertically to 150m depth. Its intent was to duplicate hole 56-57 (a Zn rich” hole) which was drilled vertically to 164.0m by Nipiron Mines Ltd. who had optioned the property from Shunsby

Mines Ltd. The drill collar for 56-57 was not located, but an old drill rod was found nearby.

A variably mineralized, continuous section comprising rhyolite agglomerate, rhyolite tuff, ash, rhyolite breccia, plus a thin bed of semi massive sulphides was cored in BWR 14-03 from the bedrock interface at 0.90m to the contact with a diorite intrusion at 46.95m. Core angle measurements indicate that lithologies are shallow dipping. Within this interval there is a continuous higher grade intercept of

- 10.50m – 45.95m (35.45m) at 0.06% Cu and 2.37% Zn including
 - 24.00m – 30.20m (6.20m) at 0.06% Cu and 6.07% Zn.

Note also that the 2.85m semi massive sulphide unit from 25.65m to 28.50m averaged **11.16% Zn**.

Below the diorite there is another variably mineralized, continuous section of rhyolite tuff, graphitic ash, and minor massive sulphides from 96.20m to 139.05m within which there are modestly interesting Zn grades plus two significant intercepts.

- 96.20m – 99.50m (3.30m) at 0.32% Cu & 2.18% Zn
- 116.40m – 121.70m (5.30m) at 1.37% Cu & 3.38% Zn.

The descriptions given in hole 56-57 drill log indicate that Zn and Cu mineralization was cored over substantial intervals. Unfortunately, sampling was limited to short sections of better mineralized material. A visual inspection of the assays for the 56-57 samples vis a vis the assays for samples at corresponding depths in BWR 14-03 indicate that they are of the same order of magnitude.

Even though the assay data for hole 56-57 appears to be acceptable, the use of the assay data for hole 56-57 and similar holes that were insufficiently sampled in a resource estimation would result in an understatement of the resource so estimated.

BWR 14-04 (Figures 3 & 7), collared in the South Zone area, was drilled at a 067° azimuth at a 45° dip to 130.5m depth. The hole was designed to twin “Zn rich” hole 56-62 which had been drilled to 64.9m by Teck Exploration Ltd. (Teck and partners had optioned the property from Shunsby Mines Ltd.), and deepened in 1966 to 132.0m by the FR Joubin Prospecting Syndicate. The 56-62 drill hole collar was located, and is situated some 3m SE from BWR 14-04.

A variably mineralized, continuous section comprising rhyolite agglomerate, rhyolite tuff, ash, slump breccia plus a thin bed of semi massive sulphides lithologies was cored in BWR 14-04 from the bedrock interface at 4.35m to the contact with a variolitic basalt unit at 40.70m. Within this interval there is a continuous higher grade intercept of

- 10.00m – 40.70m (30.70m) at 0.29% Cu and 2.81% Zn including
 - 22.20m – 28.50m (6.30m) at 0.07% Cu and 5.31% Zn.

Note also that a 1.15m semi massive sulphide unit from 27.35m to 28.50m assayed **8.36% Zn**.

The comparable interval for hole 56-62 extended from 2.13m to 47.85m, but was cut by a porphyry and a basic dyke (which were not mineralized) from 15.08m to 20.88m. The remaining lithologies, which were mineralized, were incompletely sampled although the log indicates the presence of Cu & Zn mineralization. A visual comparison of the assays available for 56-62 vis a vis the assays for samples at corresponding depths in BWR 14-04 indicate that they are of the same general order of magnitude.

Below the basalt in BWR 14-04 there is another variably mineralized, continuous section of rhyolite tuff, graphitic ash, and chert from 91.15m to 122.85m (31.70m) in which

- 91.15m – 109.95m (18.80m) assayed 0.36% Cu and 1.33% Zn and included two higher grade intercepts
 - 91.15m – 100.20m (9.05m) at 0.29% Cu & 2.11% Zn
 - 105.00m – 109.95m (4.95m) at 0.73% Cu & 0.95% Zn.

The comparable interval for hole 56-62 extended from 90.03m to 125.67m (35.64m) of which only 20.06m was sampled. A visual comparison of the assays available for hole 56-62 vis a vis the assays for samples at corresponding depths in BWR 14-04 indicate that they are of the same general order of magnitude.

BWR 14-05 (Figures 3 & 8), collared in the northern portion of the South Zone, was drilled vertically to 140.0m. Its purpose was to duplicate hole 68-16, a “Cu-Zn rich” hole, drilled vertically to 141.1m by Consolidated Shunsby Mines Ltd. Drill piping was found nearby.

A variably mineralized, continuous section comprising rhyolite agglomerate, rhyolite tuff, ash, rhyolite breccia was cored in BWR 14-05 from the bedrock interface at 4.00m to the contact with a variloitic basalt unit at 49.85m. Core angle measurements suggest that that lithologies are moderate to shallow dipping. No significant intercepts were cored. From the equivalent depths in hole 68-16 only one 0.55m sample was taken which assayed an interesting 0.95% Zn.

Below the basalt a second interval of variably mineralized rhyolite tuff, ash plus a thin layer of massive sulphides (0.65m) was cut from 88.65m to the contact with a diorite intrusion at 129.50m within which there are three significant intercepts;

- 94.50m to 97.80m (3.30m) assayed 0.29% Cu & 0.97% Zn
- 104.00m to 108.00m (4.00m) assayed 0.12% Cu & 0.98% Zn and
- 115.50m to 121.70m (6.20m) assayed 1.51% Cu & 1.61% Zn.

Although the equivalent units in hole 68-16 were not completely sampled, three mineralized intercepts were recorded;

- 96.01m – 100.58m (4.57m) at 0.27% Cu & 0.88% Zn
- 106.68m - 110.34m (3.66m) at 0.32% Cu & 1.75% Zn and
- 115.82m – 131.67m (15.85m) at 0.95% Cu & 1.50% Zn.

All three pairs of intercepts are of the same order of magnitude, but the length for the third in 68-16 is ~2.5 times that of BWR 14-05.

BWR 14-06 (Figures 3 & 9) was collared mid way between the South and Main Zones, and drilled vertically to 170m. It was designed to replicate historic hole 68-06, which was drilled vertically to 176.5m by Consolidated Shunsby Mines Ltd., and which is considered to be a “Zn rich” hole. Other than the fact that the drill site is in the middle of an old road/trail, there was no evidence for past drill activities.

In BWR 14-06, a continuous section comprising rhyolite agglomerate, rhyolite tuff and ash, was intersected from bedrock surface at 2.50m to the contact with a variolitic basalt at 80.80m. Only those intervals in which Cu or Zn mineralization was visible were sampled. There was only one intercept of note.

- 72.00m – 75.50m (3.50m) assayed 0.03% Cu & 0.82% Zn.

In hole 68-06 there was as a similar intercept (depth, core length and assay values) of

- 67.36m – 69.49m (2.13m) of 0.04% Cu & 0.84% Zn.

A suspected rip-up clast of mineralized ash and rhyolite agglomerate within the stratigraphically lower portion of the basalt in BWR 14-06 assayed

- 86.00m – 87.45m (1.45m) at 0.20% Cu & 4.74% Zn.

In hole 68-06 a longer intercept but of similar grade intercept was cored

- 81.69m – 89.31m (7.62m) at 0.06% Cu & 5.34% Zn.

CONCLUSION

From the results for the twinned pairs of holes the following are concluded;

- Although there are differences for the assay interval lengths and Cu & Zn values between the twinned holes, results are generally of the same order of magnitude. Consistent sample by sample elevated or depressed results for one or the other of the twinned holes was not noted. Thus, it is believed that the assay data from the previous drill programs are reliable.
- For previous drill programs there are gaps in the assay data even though Cu and / or Zn mineralization was noted in the drill logs. This fact is not surprising since, at least for the earliest programs, the focus for the exploration program would have been to locate high grade massive sulphide mineralization that could be mined by underground methods. The lower grade material would not have been considered significant. Unfortunately, the core boxes containing the historic core have rotted to the extent that recovery of the core for re-sampling purposes is impossible. Any resource estimation using the historic assay data will result in an under estimation of the resource.
- Notwithstanding the point above, the results for the historic holes are invaluable for the design of future exploration programs.
- The better mineralization, as represented by the thickness and grades of the significant intercepts, appears to be hosted by the stratigraphically lower “upper chert complex” stratigraphically below the variolitic basalt. It is also at this

stratigraphic level that the thin beds of massive sulphides are located. For the BWR drill holes, in the stratigraphically upper “lower chert complex” there are elevated Cu & Zn values but only a few significant intercepts.

- The assay values for the thin beds of semi massive or massive sulphides, (as in BWR 14-03 & 04) are substantial and definitely warrant further follow-up efforts.

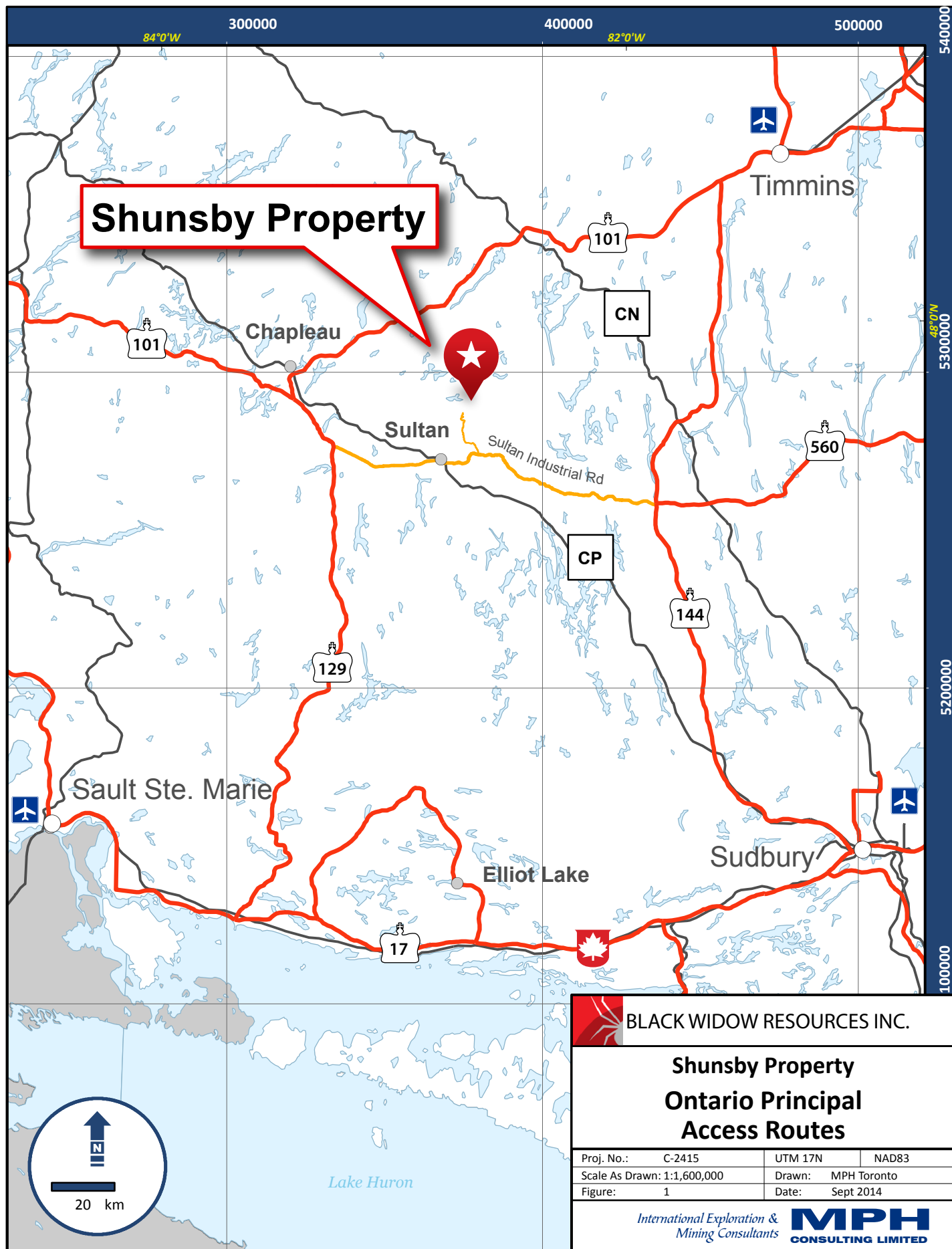
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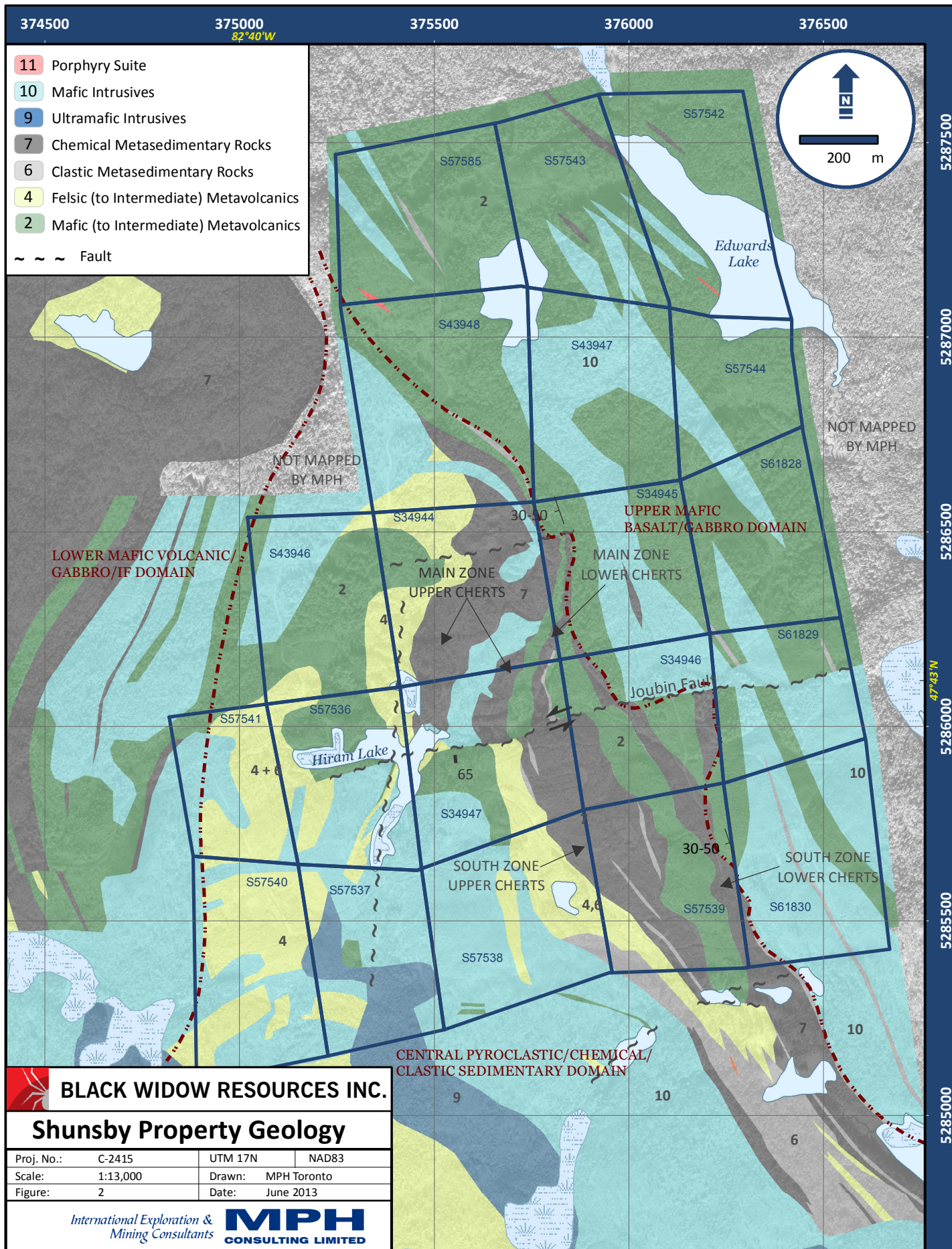
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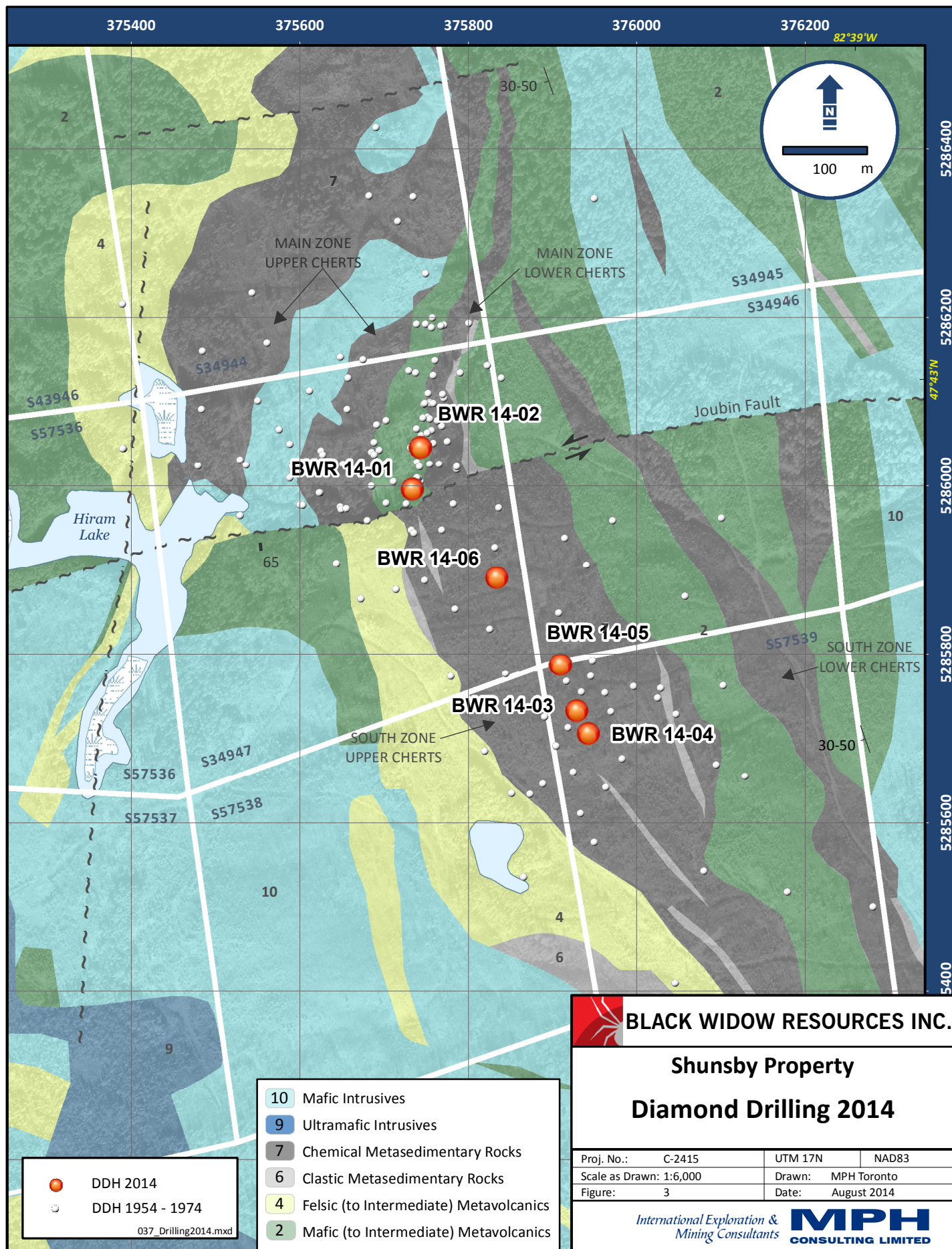
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August 29, 2014

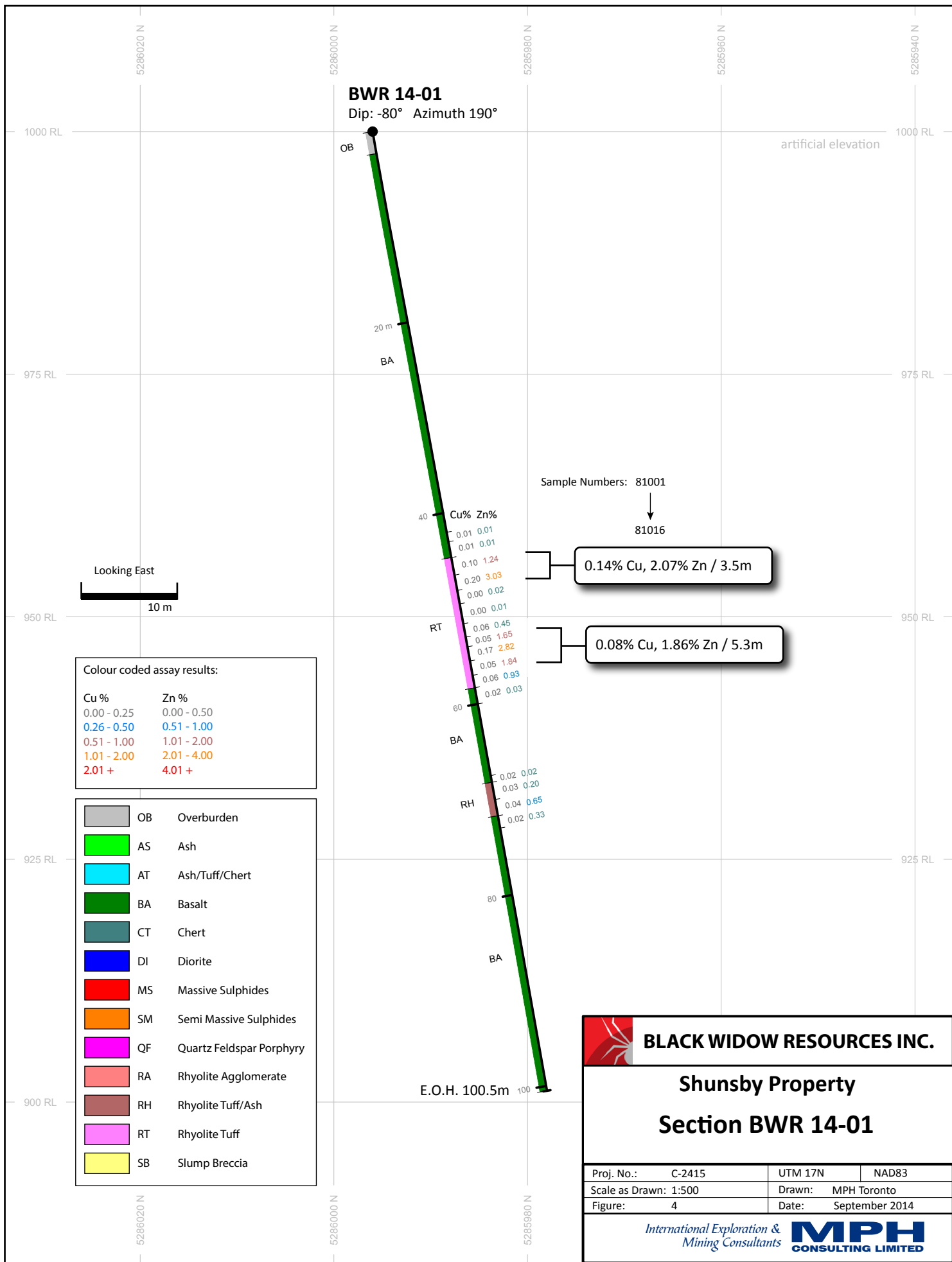
FIGURES

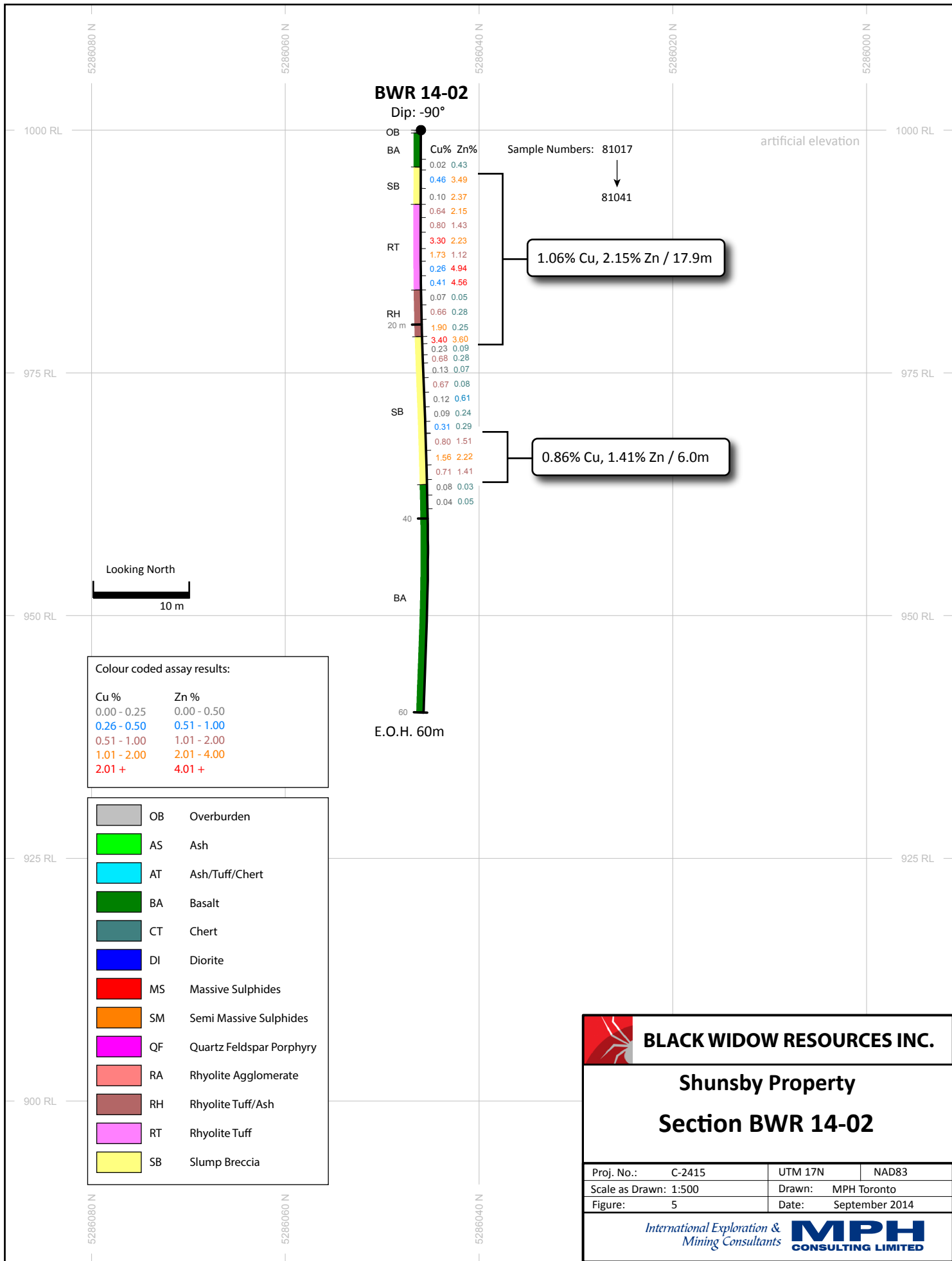
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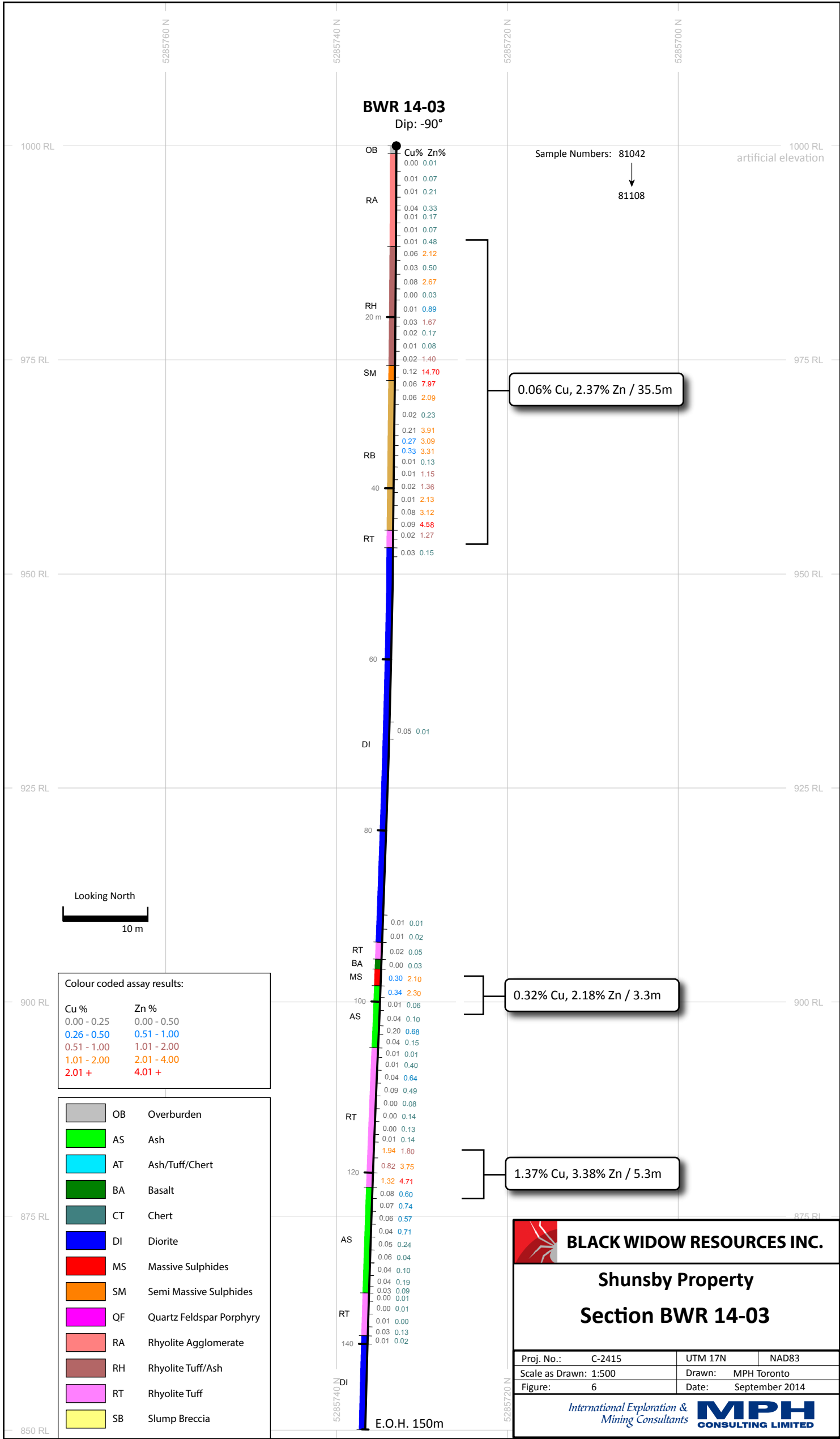












BWR 14-04

Dip: -45° Azimuth 67°

Sample Numbers: 81109

81159

0.29% Cu, 2.81% Zn / 30.7m

0.29% Cu, 2.11% Zn / 9.0m

0.73% Cu, 0.95% Zn / 5.0m

Looking Northwest
10 m

Colour coded assay results:

Cu %	Zn %
0.00 - 0.25	0.00 - 0.50
0.26 - 0.50	0.51 - 1.00
0.51 - 1.00	1.01 - 2.00
1.01 - 2.00	2.01 - 4.00
2.01 +	4.01 +

OB	Overburden
AS	Ash
AT	Ash/Tuff/Chert
BA	Basalt
CT	Chert
DI	Diorite
MS	Massive Sulphides
SM	Semi Massive Sulphides
QF	Quartz Feldspar Porphyry
RA	Rhyolite Agglomerate
RH	Rhyolite Tuff/Ash
RT	Rhyolite Tuff
SB	Slump Breccia

BLACK WIDOW RESOURCES INC.

Shunsby Property

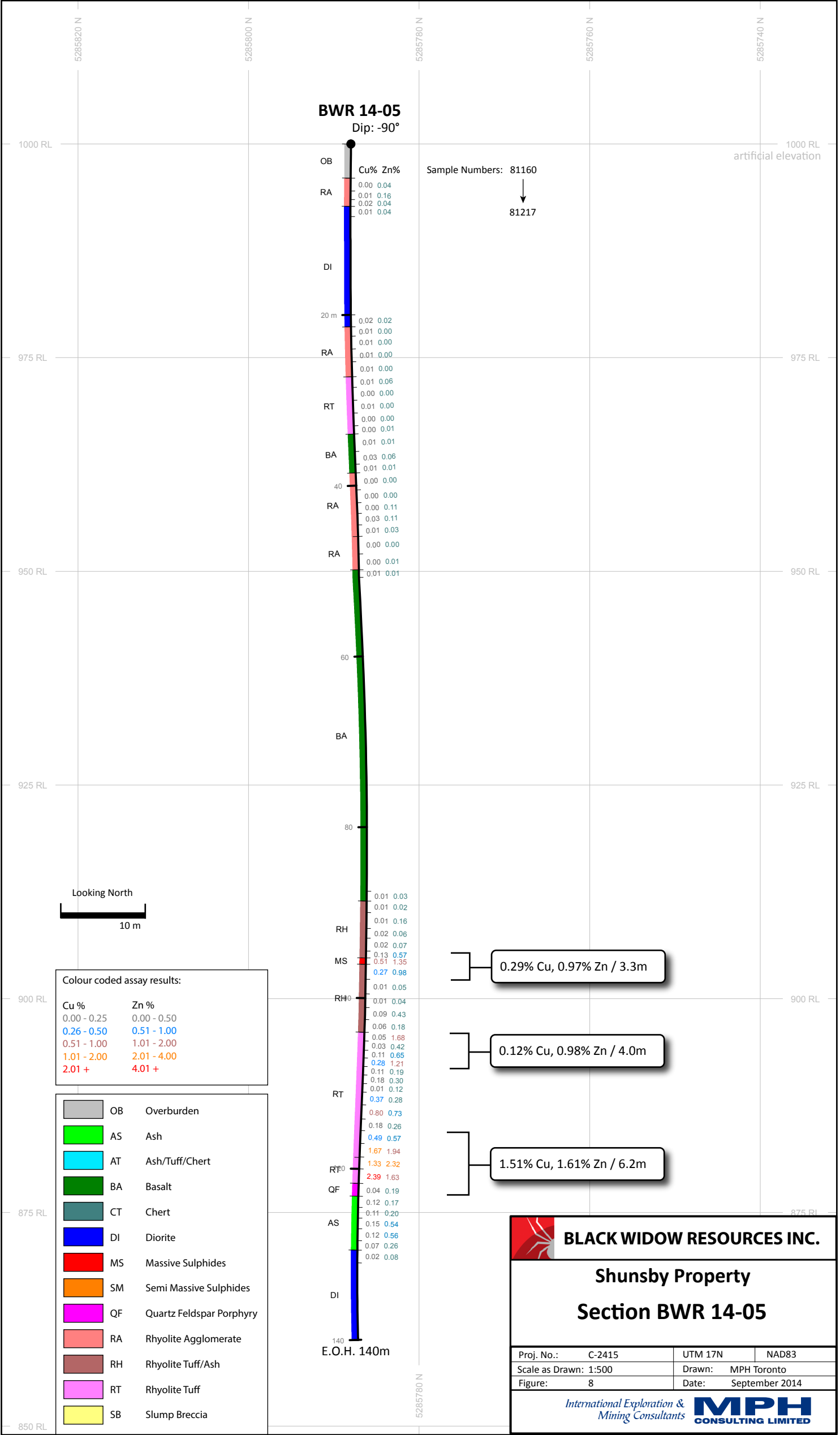
Section BWR 14-04

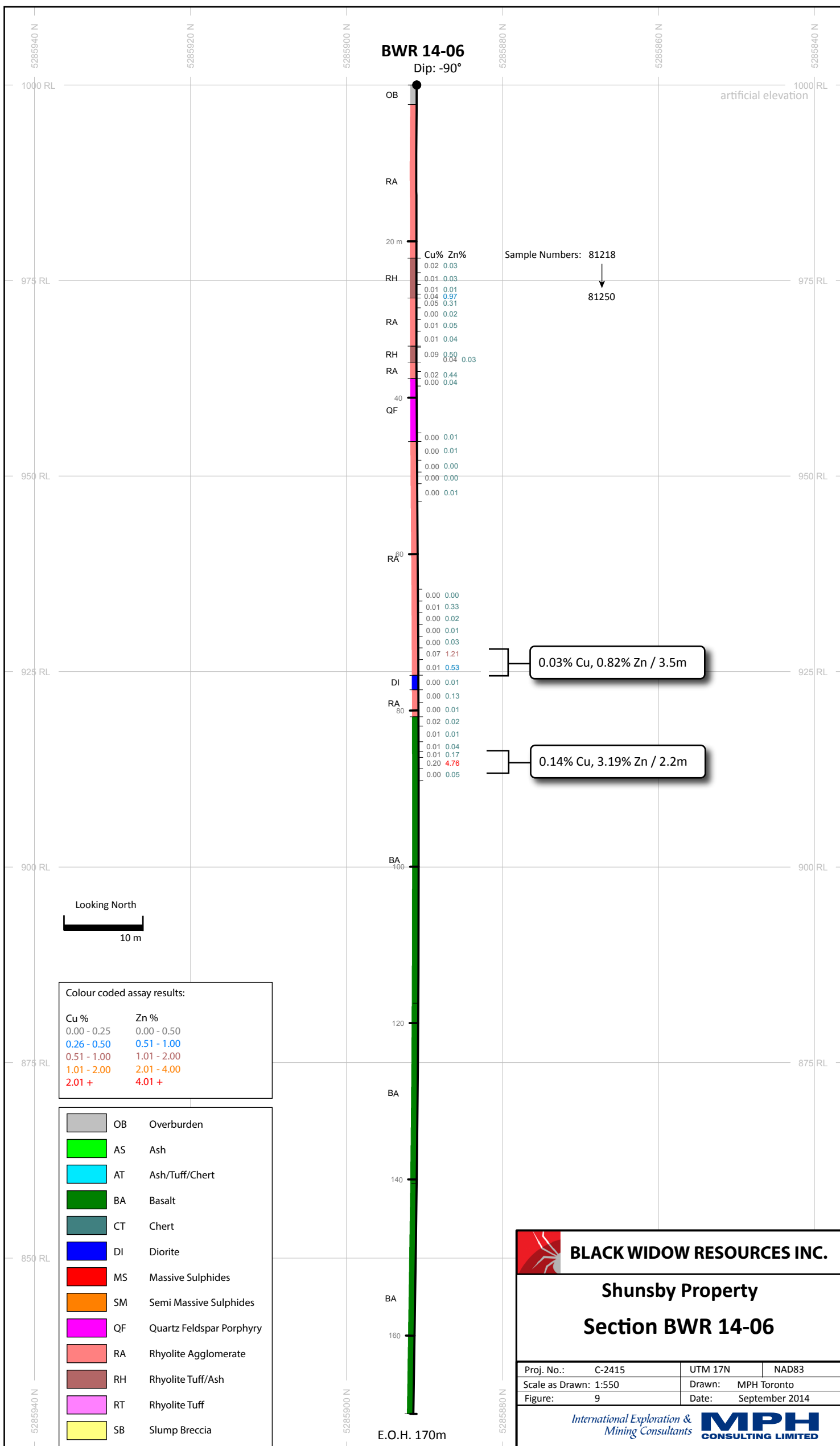
Proj. No.:	C-2415	UTM 17N	NAD83
Scale as Drawn:	1:500	Drawn:	MPH Toronto
Figure:	7	Date:	September 2014

International Exploration &
Mining Consultants

MPH
CONSULTING LIMITED

E.O.H.
130.5m





APPENDIX I

DRILL LOGS

BWR 14-01

BWR 14-02

BWR 14-03

BWR 14-04

BWR 14-05

BWR 14-06

Black Widow Resources Inc.

Project:	Shunsby	
Hole Number:	BWR 14-01	
Units of Measurement:	metric	
Location	NTS	41 O/10
	Twp	Cunningham Twp.
	Claim #	S34947
	Grid:	
	Easting:	
	Northing:	
	Elevation:	
GPS Co-ordinates: (if applicable)	Zone:	17T
	Datum:	WGS 84
	Easting:	375733
	Northing:	5285996
Collar Dip:	80°	
Collar Azimuth:	190°	
Hole Length:	100.5	
Core Size:	BTW	
Recovery:		
Logged By:	J. G. Burns	
Date:	Start:	Apr. 26, 2014
	Finish:	Apr. 26, 2014
Drilled By:	Chenier Drilling Services	
	Start:	Apr. 24, 2014
	Finish:	Apr. 25, 2014

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Page 1 of 4

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Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-02		PAGE: 2 of 4			
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
0.00	0.30	Overburden: 1.5m casing left in hole									
0.30	3.80	Basalt: light grey, very fine to fine grained; weakly sericitic; 5% very irregular qtz-chlorite filled fractures with minor py;									
		<3% py, <1/2% sph; (sulphides all in the last 30cm - sample inadvertently crossed the contact)	81017	3.00	4.10	1.10	0.02	0.43	0.09	0.51	0
	3.80	Contact: sharp @ 55°									
3.80	7.65	Slump Breccia: chaotic mix of angular to sub rounded felsic clasts to 5mm in a very fined grained, black (Ash) matrix; clasts of Rhyolite Tuff, Ash, Basalt, & Semi Massive Sulphides ; rough graded bedding with fining down hole; sulphides occur as <1cm clasts, disseminated in the matrix & in some clasts, and in fractures and late veinlets									
		1-3% py, 3% sph, 1% cpy, <1% gal	81018	4.10	6.00	1.90	0.46	3.49	1.11	2.43	25
		1-3% py, 3-5% sph, 1% cpy, <1% gal	81019	6.00	7.65	1.65	0.10	2.37	0.63	1.55	14
	7.65	Contact: at disappearance of large clasts									
7.65	16.45	Rhyolite Tuff: light grey, fine grained, occassional lapilli to 1cm; poor to mod bedded; sulphides occur disseminated along bedding, in small clasts, and in fractures and veinlets									
		8.3m; bedding @ 40°									
		<1% py, 1% cpy, <1/2% sph	81020	7.65	9.00	1.35	0.64	2.15	0.40	2.80	32
		<1% py, <1/2% cpy, tr sph	81021	9.00	10.50	1.50	0.80	1.43	0.27	3.11	21
		<1% py, 3% cpy, <1/2% sph;	81022	10.50	12.20	1.70	3.30	2.23	0.68	11.20	48
		2% py, 1% cpy, <1/2% sph; 12.2 - 12.7 basalt fragment	81023	12.20	13.50	1.30	1.73	1.12	0.21	6.53	39
		2% py, 1% cpy, 3% sph	81024	13.50	15.00	1.50	0.26	4.94	1.40	3.28	50

Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-02			PAGE: 3 of 4		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		2% py, <1% cpy, 3-5% sph; 16.0-16.45 basalt fragment	81025	15.00	16.45	1.45	0.41	4.56	1.35	3.510	47
	16.45	Contact: sharp at 40° at first appearance of graphitic argillite									
16.45	21.25	Rhyolite Tuff & Ash: interbedded; approx 50/50; mod to well bedded, beds contorted (folded, faulted & slumped); Ash as graphitic beds 2mm - 0.5m thick, occassional Rhyolite Tuff lapilli & basalt frags to 18cm; interbedded with light grey, very fine grained felsic tuff beds; py cubes to 5mm plus py, cpy, sph & gal in fractures more so in the ash sections									
		3% py, tr cpy: at 17.5 bedding @ 30°	81026	16.45	18.00	1.55	0.07	0.05	0.02	1.080	0
		7% py	81027	18.00	19.50	1.50	0.66	0.28	0.05	1.990	19
		7% py, <1/2% cpy, tr sph; basalt clasts of 12 & 18cm	81028	19.50	21.25	1.75	1.90	0.25	0.13	4.480	34
		at 21.25 contact @ 30° at end of well bedded ash bed									
21.25	36.50	Slump Breccia: chaotic mixture of rhyolite, rhyolite tuff, ash, basalt and possible chert frags; sulphides occur disseminated, in fractures and as semi massive veins to 5cm thick; better mineralization in chert sections. intensity of brecciation increases down hole									
		foliation @ 45°; 1-3% py, 5-7% cpy, 1% sph, tr gal	81029	21.25	22.00	0.75	3.40	3.60	0.07	7.660	25
		mainly basalt clast; 1% py, <1.2% cpy, tr sph concentrated in a 5cm rhyolite tuff at 22.2m.	81030	22.00	23.15	1.15	0.23	0.09	0.01	1.590	0
		basalt clast 23.55-23.65; 3-5% py, 2-3% cpy, tr sph	81031	23.15	24.00	0.85	0.68	0.28	0.02	1.180	0
		basalt fragment 24.9-25.5; 1% py, <1/2% cpy	81032	24.00	25.50	1.50	0.13	0.07	0.01	0.590	0
		graphitic ash 25.5-26.0; bedding at 26.5m @ 25°, 1% py, <1% cpy tr sph	81033	25.50	27.00	1.50	0.67	0.08	0.05	2.090	22
		mainly tuff clasts: tectonic breccia with qtz vein 27.35 - 27.45 1% py, tr cpy, tr sph	81034	27.00	28.50	1.50	0.12	0.61	0.01	0.810	0

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Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-03			PAGE: 4 of 8		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		to be slumped; sulphides generally interstial to the fragments but there are a few small <1cm frags of massive sulphides in heterolithic slump sections									
		1-2% py+po, 3% sph, tr gal	81060	28.50	30.20	1.70	0.06	2.09	0.09	0.98	0
		1-2% po+py, tr sph	81061	30.20	32.50	1.30	0.02	0.23	0.05	0.52	0
		mainly graphitic ash, tuff & agglomerate: 3% po+py, 3% sph, tr cpy, tr gal	81062	32.50	33.85	1.35	0.21	3.91	0.90	2.88	20
		5% po+py, 3% sph	81063	33.85	35.00	1.15	0.27	3.09	0.28	2.98	11
		3% po+py, 7% sph, <1/2% cpy	81064	35.00	36.20	1.20	0.33	3.31	0.16	2.79	0
		1-3% py	81065	36.20	37.50	1.30	0.01	0.13	0.03	0.60	0
		1-2% po+py, 3-5% sph; 38.5-39.0 slump bx - heterolitic including massive sulphide frags to 3mm	81066	37.50	39.00	1.50	0.01	1.15	0.01	0.36	0
		39.7 - 39.8 tectonic breccia with qtz vein: 1% py, 3% sph; slump bx 39.85-40.3	81067	39.00	40.50	1.50	0.02	1.36	0.02	0.34	7
		1% py, 5% sph	81068	40.50	42.00	1.50	0.01	2.13	0.01	0.35	0
		1.2% py, 3-5% sph, tr cpy, tr gal	81069	42.00	43.50	1.50	0.08	3.12	0.15	0.81	9
		1-2% py, 5% sph, tr cpy, <1.2% gal	81070	43.50	44.90	1.40	0.09	4.58	0.71	1.43	0
	44.90	Contact sharp @ 35°									
44.90	46.95	Rhyolite Tuff: very fine grained, light grey, fining down hole; minor ash; sulphides occur disseminated, and on fractures									
		5% py, 3-5% sph, tr cpy, tr gal	81071	44.90	45.95	1.05	0.02	1.27	0.33	1.12	0
	46.95	Contact: sharp @ 85°									
46.95	93.05	Diorite: light to mid green, fine to med grained; structurally featureless; 5-7% leucoxene flecks decreasing down hole; weak pervasive chlorite and carb alteration; 5-7% qtz-carb veinlets decreasing down hole; variably altered so as to obliterate rock texture; possibly a flow but no rip up clasts or other features									
		46.95-47.3: chill margin									
			81072	46.95	48.00	1.05	0.03	0.15	0.08	0.52	0

Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-03			PAGE: 5 of 8		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		2% py+po, <1/2% cpy in qtz-carb veinlet semi parallel core axis	81073	67.30	69.30	2.00	0.05	0.01	0.00	0.23	0
		89.85-93.05: variably silicified with epidote 89.85-91.1 & 92.6-93.05									
			81074	89.85	91.50	1.65	0.01	0.01	0.00	0.17	0
			81075	91.50	93.05	1.55	0.01	0.02	0.01	0.25	0
	93.05	Contact: sharp @ 40°									
93.05	95.05	Rhyolite Tuff: fine grained, light grey, faintly bedded @ 55°; minor sulphides in fractures, & disseminated along bedding									
		<1% py+po	81076	93.05	95.05	2.00	0.02	0.05	0.00	0.41	29
	95.05	Contact: sharp at a fracture @ 15°									
95.05	96.20	Basalt: variolitic, fine grained									
			81077	95.05	96.20	1.15	0.00	0.03	0.00	0.19	0
	96.20	Contact: sharp @ 55°									
96.20	98.15	Massive Sulphides: 90+% po, cpy; strongly magnetic, fine grained, dark brownish-grey; minor late fractures with py & sph; fine banding; po dominates									
		1% cpy, < 1/2% sph	81078	96.20	98.15	1.95	0.30	2.10	0.18	6.88	196
	98.15	Contact: sharp @ 90°									
98.15	105.40	Ash: very fine grained, black and graphitic with fine grained grey tuff interbeds; well bedded at 103 @ 70°; variable sulphide content along bedding, in cross cutting fractures and qtz-carb veinlets and as py blebs to 5mm									
		5-7% py, <1/2% sph (mainly near contact in qtz-carb veinlets)	81079	98.15	99.50	1.35	0.34	2.30	0.47	7.32	207
		3% py, tr sph	81080	99.50	101.00	1.50	0.01	0.06	0.01	1.01	0

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Black Widow Resources Inc.

Project: Shunsby

Hole Number: BWR 14-04

Units of Measurement: metric

Location	NTS	41 O/10
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Twp	Cunningham
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Claim #	S57539
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Grid: _____

Easting: _____

Nothing: _____

Elevation: _____

GPS Co-ordinates: Zone: 17 T

(if applicable)	Datum:	WGS 84
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Easting: 375942

Northings: 5285706

Collar Dip: 45°

Collar Azimuth: 067°

Hole Length:	130.5
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Core Size:	BTW
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Recovery: _____

Logged By: J.G. Burns

Date:	Start:	April 30, 2014
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Finish: May 01/2014

Drilled By: Chenier Drilling Services

Start: April 28, 2014

Finish: April 29, 2014

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Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-04		PAGE: 2 of 6			
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
0.00	4.35	Overburden									
		Casing to 4.5m									
4.35	10.00	Rhyolite Agglomerate: rounded clasts to 5cm, moderately to strongly brecciated with a later qtz-carb veining overprint; sulphides mainly in fractures;									
		2% po+py, <1/2% sph, tr gal	81109	4.35	6.00	1.65	0.01	0.31	0.05	0.21	0
		1% po+py, 2% sph, tr gal	81110	6.00	7.50	1.50	0.02	0.66	0.07	26.40	0
		3% po+py, 2% sph, tr cpy, tr gal	81111	7.50	9.00	1.50	0.03	0.44	0.07	1.52	14
		1% po+py, 2% sph, tr gal	81112	9.00	10.00	1.00	0.02	0.71	0.09	0.74	12
	10.00	Contact: sharp @ 55°; obvious colour change with the appearance of graphite									
10.00	16.30	Ash: graphitic; interbeds of tuff, fine to coarse and mod-well bedded; sulphides parallel to bedding in fractures and rimming clasts; local soft sediment slump texture									
		mainly ash with lesser tuff; ~5% lapilli to 5cm;m 1% po+py, 1-2% sph, tr gal	81113	10.00	12.00	2.00	0.04	1.62	0.45	2.25	0
		1% po+py, 3% sph, <1/2 gal; mainly ash with 3% lapilli; bedding at 13.5 at 65°	81114	12.00	13.50	1.50	0.05	3.26	0.94	2.84	0
		mainly ash; <1/2% sph	81115	13.50	14.80	1.30	0.03	2.01	0.46	4.89	12
		50/50 ash/tuff; 16.15-16.3 50% lapilli; 1-2% po+py, 3-5% sph, tr cpy tr gal	81116	14.80	16.30	1.50	0.10	4.34	0.74	4.41	17
	16.30	Contact: ground core									
16.30	22.20	Rhyolite Tuff; lesser ash beds; fine to coarsely & moderately to well bedded; sulphides in fractures, disseminated along bedding and rimming clasts									
		diorite; pervasively carbonated	81117	16.30	17.00	0.70	0.00	0.14	0.13	0.92	0

Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-04			PAGE: 3 of 6		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		mainly very fine grained tuff; <1% py+po, 3% sph, tr cpy, tr gal	81118	17.00	18.00	1.00	0.09	2.90	0.64	2.64	18
		50/50 tuff/ash; at 18.4 bedding @ 75°; 1% py+po, 3% sph, tr cpy, tr gal	81119	18.00	19.40	1.40	0.05	5.36	0.56	3.90	23
		coarse tuff/lapilli tuff; 2% py, 3-5% sph, tr gal	81120	19.40	21.00	1.60	0.01	2.06	0.20	1.08	0
		3% py, tr sph	81121	21.00	22.20	1.20	0.00	0.18	0.03	0.41	0
	22.20	Contact: sharp @ 65°									
22.20	27.35	Ash; weakly graphitic, fine grained tuff interbeds; moderate-well & fine to coarsely bedded; local sections with lapilli clasts; sulphides on fractures, disseminated along bedding and rimming clasts									
		1-2% py, 3% sph	81122	22.20	24.00	1.80	0.04	2.75	0.62	1.86	0
		<1% py, 2% sph	81123	24.00	25.50	1.50	0.05	3.69	0.48	2.71	7
		<1% py, 1% sph	81124	25.50	27.35	1.85	0.04	2.13	0.28	1.46	9
	27.35	contact: 65°									
27.35	28.50	Semi Massive Sulphides: bands of very fine grained sulphides to 15cm; 20% py, 10% sph, tr gal	81125	27.35	28.50	1.15	0.09	8.36	0.50	3.43	0
28.50	40.70	Slump Breccia: heterolithic frags to 10 cm of rhyolite, tuff; ash and some massive sulphides in an ash matrix; some frags display bedding and may be chert; sulphides interstial to fragments, in fractures and as semi massive beds and as fragments									
		1% py, 7-10% sph, tr cpy	81126	28.50	30.00	1.50	0.07	6.28	0.34	2.09	23
		3% py, 2% sph, tr cpy	81127	30.00	31.50	1.50	0.01	0.90	0.02	0.68	80
		2% py, 7% sph, <1/2% gal	81128	31.50	33.00	1.50	0.02	2.07	0.01	0.36	26
		2% py, 5% sph, tr gal	81129	33.00	34.50	1.50	0.06	3.71	0.03	0.65	32
		2% py, 5% sph, tr cpy	81130	34.50	36.00	1.50	0.27	2.09	0.03	1.32	14

Black Widow Resources Inc.

Project:	Shunsby		
Hole Number:	BWR 14-05		
Units of Measurement:	metric		
Location	NTS	41 O/10	
	Twp	Cunningham	
	Claim #	S57539	
	Grid:		
	Easting:		
	Northing:		
	Elevation:		
GPS Co-ordinates: (if applicable)	Zone:	17 T	
	Datum:	WGS 84	
	Easting:	375909	
	Northing:	5285788	
Collar Dip:	90°		
Collar Azimuth:			
Hole Length:	140		
Core Size:	BTW		
Recovery:			
Logged By:	J.G. Burns		
Date:	Start:	May 2, 2014	
	Finish:	May 3, 2014	
Drilled By:	Chenier Drilling Services		
	Start:	April 30, 2014	
	Finish:	May 1, 2014	

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[illegible]

Black Widow Resources Inc.

Project:	Shunsby		
Hole Number:	BWR 14-06		
Units of Measurement:	metric		
Location	NTS	41 O/10	
	Twp	Cunningham	
	Claim #	S34947	
	Grid:		
	Easting:		
	Northing:		
	Elevation:		
GPS Co-ordinates: (if applicable)	Zone:	17 T	
	Datum:	WGS 84	
	Easting:	375833	
	Northing:	5285891	
Collar Dip:	90		
Collar Azimuth:			
Hole Length:	170		
Core Size:	BTW		
Recovery:			
Logged By:	J.G. Burns		
Date:	Start:	May 3, 2014	
	Finish:	May 4, 2014	
Drilled By:	Chenier Drilling Services		
	Start:	May 2, 2014	
	Finish:	May 3, 2014	

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Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-06		PAGE: 2 of 5			
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
0.00	2.50	Overburden casing to 4.5m									
2.50	22.15	Rhyolite Agglomerate: rounded clasts to 5 cm+ set in a sericitic tuff matrix; general fining down hole; unit somewhat brecciated; limonite on fractures to ~7.5m; sulphides generally in fractures									
		<1% po+py		2.50	4.15	1.65					
		basalt; massive; 1% py		4.15	6.80	2.65					
		3% po+py		6.80	9.00	2.20					
		3% po+py		9.00	10.50	1.50					
		3% po+py		10.50	12.00	1.50					
		3% po+py		12.00	13.50	1.50					
		3% po+py		13.50	15.00	1.50					
		2% po+py		15.00	16.50	1.50					
		2% po+py		16.50	18.00	1.50					
		3% po+py		18.00	19.50	1.50					
		5-7% po+py		19.50	21.00	1.50					
		5-7% po+py, <1/2% sph		21.00	22.15	1.25					
	22.15	Contact: sharp @ 50°									
22.15	27.25	Rhyolite Tuff /Ash: moderate to well bedded, graded bedding with tops down hole, occasional lapilli, sulphides disseminated along bedding & in fractures									
		3-5% py+po, tr sph	81218	22.15	24.00	1.25	0.02	0.03	0.00	0.31	0
		3-5% py+po, tr sph	81219	24.00	25.50	1.50	0.01	0.03	0.00	0.27	0
		1% po+py	81220	25.50	26.75	1.25	0.01	0.01	0.00	0.33	0
		Semi Massive Sulphides 27.0-27.15; 50% po+py, 2% sph (sph mainly above & below SMS)	81221	26.75	27.25	0.50	0.04	0.97	0.19	3.22	6
	27.25	Contact: sharp @ 55°									
27.25	33.40	Rhyolite Agglomerate: clasts to 15cm in sericitic tuff matrix; sulphides generally in tuff matrix									
		3-5% po+py, <1% sph	81222	27.25	28.60	1.25	0.05	0.31	0.06	1.53	0

Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-06			PAGE: 3 of 5		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		3% po+py, tr sph	81223	28.50	30.00	1.25	0.00	0.02	0.00	0.26	0
		3% po+py	81224	30.00	31.50	1.50	0.01	0.05	0.00	0.35	0
		3% po+py	81225	31.50	33.40	1.90	0.01	0.04	0.00	0.36	0
	33.40	Contact: gradational marked by appearance of Ash									
33.40	35.55	Rhyolitre Tuff / Ash: ~ 50/50; moderate to well bedded; at 34.9 bedding @ 45°									
		10% po+py, <1% sph, tr cpy	81226	33.40	35.55	2.15	0.09	0.50	0.01	2.41	38
	33.55	Contact: sharp @ 50°									
33.55	37.55	Rhyolite Agglomerate: as previously									
		15% po+py	81227	33.55	36.65	1.10	0.04	0.03	0.00	0.61	0
		7% po+py, <1/2% sph, tr cpy	81228	36.65	37.55	0.90	0.02	0.44	0.08	0.68	4
	37.55	Contact: sharp @ 45°									
37.55	45.60	Quartz-Feldspar Porphyry: light grey, weak pervasive carb alteration; quartz & feldspar phenos to 2mm, 10cm chill margin; 3cm qtz vein at 37.7 @ 30°									
			81229	37.55	38.50	0.95	0.00	0.04	0.01	0.26	0
			81230	44.50	45.60	1.10	0.00	0.01	0.00	0.24	0
	45.60	Contact: sharp @ 50° rotated an estimated 20° from alignment of agglomerate clasts									
45.60	75.50	Rhyolite Agglomerate: very siliceous clasts to 5cm+ in a sericitic tuff matrix; sulphides (principally po with lesser py) in matrix; unit chaotic with no apparent sorting; occasional tuff clast									
		5% po+py	81231	45.60	48.00	2.40	0.00	0.01	0.00	0.38	0
		5% po+py	81232	48.00	49.50	1.50	0.00	0.00	0.00	0.25	0
		5% po+py	81233	49.50	51.00	1.50	0.00	0.00	0.00	0.31	0
		5% po +py, tr sph in cross cutting veinlet	81234	51.00	53.30	2.30	0.00	0.01	0.00	0.35	7

Black Widow Resources Inc.			PROJECT: Shunsby			HOLE NO: BWR 14-06			PAGE: 4 of 5		
FROM	TO	DESCRIPTION	ANALYTICAL RESULTS								
			SAMPLE	FROM	TO	LENGTH	Cu %	Zn %	Pb %	Ag g/t	Au ppb
		~53.3 ~72.0 increase in siliceous clasts and a decrease in sulphides									
		<1% po+py		53.30	55.50	2.30					
		<1% po+py		55.50	57.00	1.50					
		<1% po+py		57.00	58.50	1.50					
		1% po+py		58.50	60.00	1.50					
		1% po+py		60.00	61.50	1.50					
		1% po+py		61.50	63.00	1.50					
		1% po+py		63.00	64.50	1.50					
		1% po+py	81235	64.50	66.00	1.50	0.00	0.00	0.00	0.11	0
		3%po+py, 1% sph, tr cpy (sph & cpy in chloritic tuff)	81236	66.00	67.50	1.50	0.01	0.33	0.05	0.25	7
		1% po+py, <1/2% sph	81237	67.50	69.00	1.50	0.00	0.02	0.00	0.16	0
		1% po+py, tr sph	81238	69.00	70.50	1.50	0.00	0.01	0.00	0.16	0
		3% po+py, tr sph	81239	70.50	72.00	1.50	0.00	0.03	0.01	0.45	0
		3%po+py, 3% sph, tr cpy, tr gal	81240	72.00	73.50	1.50	0.07	1.21	0.13	1.03	0
		3% po+py, 1% sph	81241	73.50	75.50	2.00	0.01	0.53	0.13	0.72	14
	75.50	Contact: sharp @ 60°									
75.50	77.35	Diorite: fine grained, structureless, mod - strong pervasive carb;									
			81242	75.50	77.35	1.85	0.00	0.01	0.00	0.17	0
	77.35	Contact: sharp @ 50°									
77.35	80.80	Rhyolite Agglomerate: as previously									
		5% po+py, 1/2% sph	81243	77.35	79.00	1.65	0.00	0.13	0.04	0.18	0
		3% po+py, tr sph	81244	79.00	80.80	1.80	0.00	0.01	0.00	0.08	0
	80.80	Contact: sharp @ 60°									
80.80	117.45	Basalt: variolitic,									
		80.8~87.45: silicic, variable mixture of Rhyolite Agglomerate and Ash									
		rip up clasts; 10-15% chlorite filled fractures & veins									
			81245	80.80	82.00	1.20	0.02	0.02	0.01	0.12	0
			81246	82.00	84.00	2.00	0.01	0.01	0.01	0.10	0

APPENDIX II

Assay Intercepts with Comparison to Previous Assay Data for Twinned Holes

BWR 14-01								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81001	42.00	43.00	1.00	0.01	0.01	0.00	0.18	0
81002	43.00	44.65	1.65	0.01	0.01	0.00	0.12	0
81003	44.65	46.50	1.85	0.10	1.24	0.27	1.22	30
81004	46.50	48.10	1.60	0.20	3.03	0.66	3.99	35
81005	48.10	49.50	1.40	0.00	0.02	0.00	0.77	0
81006	49.50	51.60	2.10	0.00	0.01	0.00	0.40	0
81007	51.60	53.00	1.40	0.06	0.45	0.13	1.20	46
81008	53.00	54.00	1.00	0.05	1.65	0.37	1.26	0
81009	54.00	55.50	1.50	0.17	2.82	0.80	3.14	37
81010	55.50	57.00	1.50	0.05	1.84	0.43	1.62	0
81011	57.00	58.30	1.30	0.06	0.93	0.25	1.20	14
81012	58.30	60.00	1.70	0.02	0.03	0.01	0.51	0
81013	67.50	68.20	0.70	0.02	0.02	0.00	0.36	13
81014	68.20	70.00	1.80	0.03	0.20	0.05	1.25	49
81015	70.00	71.70	1.70	0.04	0.65	0.15	1.75	70
81016	71.70	73.00	1.30	0.02	0.33	0.08	0.59	0
A	44.65	48.10	3.45	0.14	2.07	0.45	2.50	32
B	53.00	58.30	5.30	0.08	1.86	0.48	1.88	14
	44.65	58.30	13.65	0.08	1.29			

81-08							
Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
3366	51.82	54.86	3.04	3.57	7.68	9	
3367	54.86	56.66	1.80	1.89	2.99		
3368	56.66	57.87	1.21	1.23	1.63		
3369	57.87	58.79	0.92	0.08	0.37		
3370	58.79	61.22	2.43	0.38	6.51	2	
3371	61.22	62.75	1.53	0.89	3.85	2	
3372	62.75	63.36	0.61	0.19	1.24		
	51.82	63.36	11.58	1.58	4.68		

BWR 14-02

Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81017	3.00	4.10	1.10	0.02	0.43	0.09	0.51	0
81018	4.10	6.00	1.90	0.46	3.49	1.11	2.43	25
81019	6.00	7.65	1.65	0.10	2.37	0.63	1.55	14
81020	7.65	9.00	1.35	0.64	2.15	0.40	2.8	32
81021	9.00	10.50	1.50	0.80	1.43	0.27	3.11	21
81022	10.50	12.20	1.70	3.30	2.23	0.68	11.2	48
81023	12.20	13.50	1.30	1.73	1.12	0.21	6.53	39
81024	13.50	15.00	1.50	0.26	4.94	1.40	3.28	50
81025	15.00	16.45	1.45	0.41	4.56	1.35	3.51	47
81026	16.45	18.00	1.55	0.07	0.05	0.02	1.08	0
81027	18.00	19.50	1.50	0.66	0.28	0.05	1.99	19
81028	19.50	21.25	1.75	1.90	0.25	0.13	4.48	34
81029	21.25	22.00	0.75	3.40	3.60	0.07	7.66	25
81030	22.00	23.15	1.15	0.23	0.09	0.01	1.59	0
81031	23.15	24.00	0.85	0.68	0.28	0.02	1.18	0
81032	24.00	25.50	1.50	0.13	0.07	0.01	0.59	0
81033	25.50	27.00	1.50	0.67	0.08	0.05	2.09	22
81034	27.00	28.50	1.50	0.12	0.61	0.01	0.81	0
81035	28.50	30.00	1.50	0.09	0.24	0.01	0.62	5
81036	30.00	31.20	1.20	0.31	0.29	0.01	1.12	0
81037	31.25	33.00	1.75	0.80	1.51	0.03	2.43	61
81038	33.00	34.50	1.50	1.56	2.22	0.01	4.16	35
81039	34.50	36.00	1.50	0.71	1.41	0.01	1.92	32
81040	36.00	37.60	1.60	0.08	0.03	0.01	0.83	31
81041	37.60	39.00	1.40	0.04	0.05	0.01	0.31	0
C	4.10	22.00	17.90	1.06	2.15	0.56	3.99	29
incl	10.50	16.45	5.95	1.49	3.24	0.92	6.31	46
D	30.00	36.00	6.00	0.86	1.41	0.02	2.45	35
	4.1	36	31.9	0.83	1.53			

81-01

Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
3206	4.88	6.40	1.52	0.02	0.07		
3207	7.01	8.53	1.52	0.08	1.00	0.69	69
3208	8.53	10.97	2.44	0.07	1.33	1.03	69
3209	10.97	12.80	1.83	0.39	3.08		
3210	18.58	19.81	1.23	0.25	2.83		
3211	19.81	21.03	1.22	1.84	12.59	5.49	69
3212	21.03	22.25	1.22		0.21		
3214	22.25	24.38	2.13		1.86		
3213	24.38	26.37	1.99	0.17	0.88		
3215	26.37	27.74	1.37	1.11	0.44		
3216	27.74	29.72	1.98	0.03	2.24		
3217	29.72	30.48	0.76	3.65	4.78	13.37	
3218	31.39	33.22	1.83	1.81	0.05		
3219	33.22	34.75	1.53	1.32	0.03		
3220	34.75	35.81	1.06	3.75	0.05	9.60	69
3221	35.81	36.42	0.61	19.38	0.07	45.26	171
3222	36.42	37.49	1.07	0.55	0.02	1.37	
7.01	12.80	5.79	0.17	1.80			
18.58	30.48	11.90	0.61	2.81			
31.39	37.49	6.10	3.56	0.04			
4.88	37.49	32.61	0.92	1.36			

BWR 14-03								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81042	0.90	3.00	2.10	0.00	0.01	0.00	0.14	6
81043	3.00	4.60	1.60	0.01	0.07	0.02	0.22	0
81044	4.60	6.00	1.40	0.01	0.21	0.04	0.16	0
81045	7.00	7.50	0.50	0.04	0.33	0.09	0.31	21
81046	7.50	9.00	1.50	0.01	0.17	0.04	0.18	0
81047	9.00	10.50	1.50	0.01	0.07	0.01	0.23	0
81048	10.50	11.75	1.25	0.01	0.48	0.18	0.58	4
81049	11.75	13.35	1.60	0.06	2.12	0.71	2.66	0
81050	13.35	15.00	1.65	0.03	0.50	0.14	1.88	0
81051	15.00	16.70	1.70	0.08	2.67	0.39	2.63	0
81052	16.70	18.00	1.30	0.00	0.03	0.04	0.76	0
81053	18.00	20.00	2.00	0.01	0.89	0.14	0.77	0
81054	20.00	21.05	1.05	0.03	1.67	0.26	2.98	16
81055	21.05	22.60	1.55	0.02	0.17	0.04	0.89	0
81056	22.60	24.00	1.40	0.01	0.08	0.03	0.34	0
81057	24.00	25.65	1.65	0.02	1.40	0.07	0.51	0
81058	25.65	27.00	1.35	0.12	14.70	0.40	3.23	32
81059	27.00	28.50	1.50	0.06	7.97	0.32	2.74	0
81060	28.50	30.20	1.70	0.06	2.09	0.09	0.98	0
81061	30.20	32.50	2.30	0.02	0.23	0.05	0.52	0
81062	32.50	33.85	1.35	0.21	3.91	0.90	2.88	20
81063	33.85	35.00	1.15	0.27	3.09	0.28	2.98	11
81064	35.00	36.20	1.20	0.33	3.31	0.16	2.79	0
81065	36.20	37.50	1.30	0.01	0.13	0.03	0.6	0
81066	37.50	39.00	1.50	0.01	1.15	0.01	0.36	0
81067	39.00	40.50	1.50	0.02	1.36	0.02	0.34	7
81068	40.50	42.00	1.50	0.01	2.13	0.01	0.35	0
81069	42.00	43.50	1.50	0.08	3.12	0.15	0.81	9
81070	43.50	44.90	1.40	0.09	4.58	0.71	1.43	0
81071	44.90	45.95	1.05	0.02	1.27	0.33	1.12	0
81072	46.95	48.00	1.05	0.03	0.15	0.08	0.52	0
81073	67.30	69.30	2.00	0.05	0.01	0.00	0.23	0
81074	89.85	91.50	1.65	0.01	0.01	0.00	0.17	0
81075	91.50	93.05	1.55	0.01	0.02	0.01	0.25	0
81076	93.05	95.05	2.00	0.02	0.05	0.00	0.41	29
81077	95.05	96.20	1.15	0.00	0.03	0.00	0.19	0
81078	96.20	98.15	1.95	0.30	2.10	0.18	6.88	196
81079	98.15	99.50	1.35	0.34	2.30	0.47	7.32	207

56-57							
Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
458	27.13	28.96	1.83	0.90	9.08		
459	28.96	32.00	3.04	0.00	0.30		
460	32.00	35.05	3.05	0.00	5.00		
461	42.06	44.50	2.44	0.00	5.61		
462	44.50	46.63	2.13	0.00	5.81		
463	46.63	49.07	2.44	0.00	2.75		
464	49.07	52.12	3.05	0.20	5.56		
607	98.45	101.50	3.05	0.18	1.06		

BWR 14-03								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81080	99.50	101.00	1.50	0.01	0.06	0.01	1.01	0
81081	101.00	102.80	1.80	0.04	0.10	0.01	0.81	7
81082	102.80	103.80	1.00	0.20	0.68	0.01	2.89	96
81083	103.80	105.40	1.60	0.04	0.15	0.01	0.99	19
81084	105.40	106.50	1.10	0.01	0.01	0.00	0.33	0
81085	106.50	108.00	1.50	0.01	0.40	0.10	0.5	6
81086	108.00	109.50	1.50	0.04	0.64	0.12	0.56	0
81087	109.50	111.00	1.50	0.09	0.49	0.04	0.38	33
81088	111.00	112.50	1.50	0.00	0.08	0.02	0.15	0
81089	112.50	114.00	1.50	0.00	0.14	0.04	0.15	15
81090	114.00	115.50	1.50	0.00	0.13	0.01	0.16	9
81091	115.50	116.40	0.90	0.01	0.15	0.03	0.13	9
81092	116.40	118.25	1.85	1.94	1.80	0.09	4.54	66
81093	118.25	120.00	1.75	0.82	3.75	0.13	3.8	23
81094	120.00	121.70	1.70	1.32	4.71	0.50	5.17	28
81095	121.70	123.00	1.30	0.08	0.60	0.14	1.66	41
81096	123.00	124.50	1.50	0.07	0.74	0.18	1.76	43
81097	124.50	126.00	1.50	0.06	0.57	0.15	1.7	50
81098	126.00	127.50	1.50	0.04	0.71	0.16	1.46	0
81099	127.50	129.00	1.50	0.05	0.24	0.06	1.25	57
81100	129.00	130.50	1.50	0.06	0.04	0.01	0.8	26
81101	130.55	132.00	1.45	0.04	0.10	0.02	0.52	35
81102	132.00	133.30	1.30	0.04	0.19	0.02	0.8	43
81103	133.30	134.05	0.75	0.03	0.09	0.01	0.62	63
81104	134.05	135.00	0.95	0.00	0.01	0.00	0.42	0
81105	135.00	136.50	1.50	0.00	0.01	0.00	0.61	0
81106	136.50	138.00	1.50	0.01	0.00	0.00	0.59	3
81107	138.00	139.05	1.05	0.03	0.13	0.04	0.37	29
81108	139.05	140.00	0.95	0.01	0.02	0.00	0.14	0
E	10.50	45.95	35.45	0.06	2.37	0.22	1.40	3
incl.	24.00	30.20	6.20	0.06	6.07	0.21	1.47	7
F	96.20	99.50	3.30	0.32	2.18	0.30	7.06	201
G	116.40	121.70	5.30	1.37	3.38	0.23	4.50	40

56-57						
Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t Au ppb
608	101.50	103.93	2.43	0.07	0.29	
	98.45	101.5	5.48	0.13	0.72	

BWR 14-04								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81109	4.35	6.00	1.65	0.01	0.31	0.05	0.21	0
81110	6.00	7.50	1.50	0.02	0.66	0.07	26.4	0
81111	7.50	9.00	1.50	0.03	0.44	0.07	1.52	14
81112	9.00	10.00	1.00	0.02	0.71	0.09	0.74	12
81113	10.00	12.00	2.00	0.04	1.62	0.45	2.25	0
81114	12.00	13.50	1.50	0.05	3.26	0.94	2.84	0
81115	13.50	14.80	1.30	0.03	2.01	0.46	4.89	12
81116	14.80	16.30	1.50	0.10	4.34	0.74	4.41	17
81117	16.30	17.00	0.70	0.00	0.14	0.13	0.92	0
81118	17.00	18.00	1.00	0.09	2.90	0.64	2.64	18
81119	18.00	19.40	1.40	0.05	5.36	0.56	3.9	23
81120	19.40	21.00	1.60	0.01	2.06	0.20	1.08	0
81121	21.00	22.20	1.20	0.00	0.18	0.03	0.41	0
81122	22.20	24.00	1.80	0.04	2.75	0.62	1.86	0
81123	24.00	25.50	1.50	0.05	3.69	0.48	2.71	7
81124	25.50	27.35	1.85	0.04	2.13	0.28	1.46	9
81125	27.35	28.50	1.15	0.09	8.36	0.50	3.43	0
81126	28.50	30.00	1.50	0.07	6.28	0.34	2.09	23
81127	30.00	31.50	1.50	0.01	0.90	0.02	0.68	80
81128	31.50	33.00	1.50	0.02	2.07	0.01	0.36	26
81129	33.00	34.50	1.50	0.06	3.71	0.03	0.65	32
81130	34.50	36.00	1.50	0.27	2.09	0.03	1.32	14
81131	36.00	37.50	1.50	2.29	0.77	0.01	9.13	25
81132	37.50	39.00	1.50	2.38	0.56	0.02	8.52	0
81133	39.00	40.70	1.70	0.22	3.85	0.02	1.28	0
81134	40.70	42.00	1.30	0.01	0.69	0.02	0.32	0
81135	42.00	43.20	1.20	0.01	0.06	0.01	0.17	0
81136	87.00	88.50	1.50	0.03	0.05	0.01	0.17	0
81137	88.50	90.00	1.50	0.02	0.02	0.01	0.19	0
81138	90.00	91.15	1.15	0.01	0.18	0.04	0.24	0
81139	91.15	93.00	1.85	0.33	2.43	0.68	5.47	93
91140	93.00	94.50	1.50	0.17	2.23	0.74	2.07	67
91141	94.50	95.60	1.10	0.68	1.85	0.05	2.41	41
81142	95.60	97.50	1.90	0.46	1.59	0.07	1.64	20
81143	97.50	99.00	1.50	0.05	2.46	0.09	0.78	17
81144	99.00	100.20	1.20	0.04	2.06	0.05	0.56	0
81145	100.20	101.40	1.20	0.07	0.40	0.01	0.32	7
81146	101.40	102.45	1.05	0.08	0.08	0.01	0.31	3
81147	102.45	103.50	1.05	0.15	0.14	0.00	0.56	0

56-62 and extension in 1966								Ag g/t	Au ppb
Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb		
6732	2.13	5.18	3.05	0.08	1.29				
6733	5.18	8.23	3.05	0.14	1.22				
6734	8.23	11.28	3.05	0.04	1.19				
6735	20.88	22.40	1.52	0.05	7.23				
6736	31.09	34.14	3.05	0.03	7.96				
6737	34.14	37.19	3.05	0.00	8.17				
6738	37.19	40.23	3.04	0.03	2.93				
6739	40.23	43.28	3.05	0.00	0.44				
6740	43.28	46.33	3.05	0.00	2.43				
6741	46.33	47.85	1.52	0.15	2.10				
MS122	90.03	91.96	1.93	0.50	3.43				
MS123	91.96	93.33	1.37	0.50	6.26				
MS124	96.90	98.42	1.52	0.74	5.89				
MS125	98.42	100.10	1.68	3.72	1.09				
MS126	100.10	102.84	2.74	0.18	2.38				
MS127	102.84	106.19	3.35	0.34	1.24				
MS128	106.19	109.55	3.36	0.19	1.30				
1	109.55	111.07	1.52	0.18	4.00				

BWR 14-04

Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81148	103.50	105.00	1.50	0.12	0.17	0.00	0.46	0
81149	105.00	106.75	1.75	0.18	0.53	0.01	0.57	9
81150	106.75	108.00	1.25	1.82	2.36	0.03	4.93	97
81151	108.00	109.95	1.95	0.53	0.55	0.02	1.72	43
81152	109.95	111.00	1.05	0.05	0.02	0.01	0.57	35
81153	111.00	112.25	1.25	0.02	0.02	0.02	0.36	18
81154	112.25	114.00	1.75	0.00	0.00	0.00	0.11	0
81155	117.00	118.95	1.95	0.00	0.01	0.00	0.11	0
81156	118.95	119.75	0.80	0.03	0.03	0.00	0.33	56
81157	119.75	121.55	1.80	0.07	0.07	0.01	1.47	66
81158	121.55	122.85	1.30	0.09	0.10	0.01	0.5	17
81159	122.85	124.50	1.65	0.03	0.12	0.00	0.12	0
H incl J K	10.00	40.70	30.70	0.29	2.81	0.31	2.71	14
	22.20	28.50	6.30	0.07	5.31	0.54	2.73	10
	91.15	100.20	9.05	0.29	2.11	0.31	2.30	42
	105.00	109.95	4.95	0.73	0.95	0.02	2.12	28
	91.15	109.95	18.8	0.36	1.33			

56-62 and extension in 1966

Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
2	111.07	113.66	2.59	2.56	2.13		
	90.03	113.66	23.63	0.77	2.24		

BWR 14-05									68-16							
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb	Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
81160	4.00	5.50	1.50	0.00	0.04	0.01	0.12	0								
81161	5.50	6.50	1.00	0.01	0.16	0.04	0.17	0								
81162	6.50	7.30	0.80	0.02	0.04	0.01	0.42	0								
81163	7.30	8.50	1.20	0.01	0.04	0.00	0.08	0								
81164	20.00	21.40	1.40	0.02	0.02	0.00	0.08	0								
81165	21.40	22.50	1.10	0.01	0.00	0.00	0.09	5								
81166	22.50	24.00	1.50	0.01	0.00	0.00	0.05	0								
81167	24.00	25.50	1.50	0.01	0.00	0.00	0.09	0								
81168	25.50	27.25	1.75	0.01	0.00	0.00	0.17	14								
81169	27.25	28.50	1.25	0.01	0.06	0.00	0.31	39								
81170	28.50	30.00	1.50	0.00	0.00	0.00	0.28	3								
81171	30.00	31.50	1.50	0.01	0.00	0.00	0.17	10								
81172	31.50	33.00	1.50	0.00	0.00	0.00	0.14	6								
81173	33.00	33.95	0.95	0.00	0.01	0.00	0.16	0								
81174	33.95	36.00	2.05	0.01	0.01	0.00	0.12	0								
81175	36.00	37.50	1.50	0.03	0.06	0.00	0.22	0								
81176	37.50	38.50	1.00	0.01	0.01	0.00	0.13	0								
81177	38.50	40.50	2.00	0.00	0.00	0.01	0	0	595	39.62	40.17	0.55	0.12	0.95		
81178	40.50	42.00	1.50	0.00	0.00	0.00	0	0								
81179	42.00	43.25	1.25	0.00	0.11	0.00	0.09	0								
81180	43.25	44.60	1.35	0.03	0.11	0.01	0.92	35								
81181	44.60	45.95	1.35	0.01	0.03	0.00	0.13	7								
81182	45.95	48.00	2.05	0.00	0.00	0.00	0.12	0								
81183	48.00	49.85	1.85	0.00	0.01	0.00	0.08	0								
81184	49.85	50.75	0.90	0.01	0.01	0.00	0.28	0								
81185	87.50	88.65	1.15	0.01	0.03	0.01	0.14	0								
81186	88.65	90.00	1.35	0.01	0.02	0.00	0.27	9								
81187	90.00	91.85	1.85	0.01	0.16	0.04	0.53	10								
81188	91.85	93.00	1.15	0.02	0.06	0.00	0.43	0								
81189	93.00	94.50	1.50	0.02	0.07	0.00	0.41	0								
81190	94.50	95.30	0.80	0.13	0.57	0.03	3.1	63								
81191	95.30	96.05	0.75	0.51	1.35	0.05	8.68	66								
81192	96.05	97.80	1.75	0.27	0.98	0.11	7.81	345	596	96.01	97.84	1.83	0.04	0.21		
81193	97.80	99.50	1.70	0.01	0.05	0.01	0.63	11	597	97.84	99.06	1.22	0.30	1.40		
81194	99.50	101.10	1.60	0.01	0.04	0.00	0.38	0	598	99.06	100.58	1.52	0.51	1.27		
81195	101.10	102.50	1.40	0.09	0.43	0.02	1.93	78								
81196	102.50	104.00	1.50	0.06	0.18	0.03	1	26								
81197	104.00	105.00	1.00	0.05	1.68	0.44	1.02	13								

BWR 14-05

Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
81198	105.00	106.10	1.10	0.03	0.42	0.05	0.4	119
81199	106.10	107.00	0.90	0.11	0.65	0.12	0.98	15
81200	107.00	108.00	1.00	0.28	1.21	0.17	1.16	109
81201	108.00	109.00	1.00	0.11	0.19	0.02	0.34	9
81202	109.00	110.00	1.00	0.18	0.30	0.05	0.7	9
81203	110.00	111.00	1.00	0.01	0.12	0.03	0.18	9
81204	111.00	112.50	1.50	0.37	0.28	0.04	0.9	7
81205	112.50	114.15	1.65	0.80	0.73	0.07	3.6	5
81206	114.15	115.50	1.35	0.18	0.26	0.01	0.59	4
81207	115.50	117.00	1.50	0.49	0.57	0.01	1.25	8
81208	117.00	118.60	1.60	1.67	1.94	0.04	4.01	18
81209	118.60	120.00	1.40	1.33	2.32	0.48	3.52	35
81210	120.00	121.70	1.70	2.39	1.63	0.19	4.6	35
81211	121.70	123.20	1.50	0.04	0.19	0.03	0.27	0
81212	123.20	124.50	1.30	0.12	0.17	0.07	1.98	40
81213	124.50	125.75	1.25	0.11	0.20	0.02	1	22
81214	125.75	127.00	1.25	0.15	0.54	0.07	0.87	16
81215	127.00	128.40	1.40	0.12	0.56	0.04	0.73	35
81216	128.40	129.50	1.10	0.07	0.26	0.04	0.83	18
81217	129.50	131.00	1.50	0.02	0.08	0.02	0.18	0
L	94.50	97.80	3.30	0.29	0.97	0.08	6.90	213
M	104.00	108.00	4.00	0.12	0.98	0.19	0.09	39
N	115.50	121.70	6.20	1.51	1.61	0.17	3.40	14

68-16

Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Ag g/t	Au ppb
599	106.68	110.34	3.66	0.32	1.75		
600	115.82	116.74	0.92	0.78	1.22		
601	116.74	119.48	2.74	0.52	0.95		
602	119.48	121.77	2.29	4.54	6.31		
603	121.77	123.44	1.67	0.25	0.17		
604	123.44	126.49	3.05	0.39	0.42		
605	126.49	129.54	3.05	0.17	1.09		
606	129.54	131.67	2.13	0.19	0.37		
	96.01	100.58	4.57	0.27	0.88		
	106.68	110.34	3.66	0.32	1.75		
	115.82	131.67	15.85	0.95	1.50		

BWR 14-06									68-06								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb	Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Pb %	Ag g/t	Au ppb
									527	20.12	21.95	1.83	0.02	0.11			
81218	22.15	24.00	1.85	0.02	0.03	0.00	0.31	0									
81219	24.00	25.50	1.50	0.01	0.03	0.00	0.27	0	528	24.38	26.52	2.13	0.04	0.44			
81220	25.50	26.75	1.25	0.01	0.01	0.00	0.33	0									
81221	26.75	27.25	0.50	0.04	0.97	0.19	3.22	6									
81222	27.25	28.60	1.35	0.05	0.31	0.06	1.53	0									
81223	28.50	30.00	1.50	0.00	0.02	0.00	0.26	0	529	29.26	30.48	1.22	0.07	0.57			
81224	30.00	31.50	1.50	0.01	0.05	0.00	0.35	0									
81225	31.50	33.40	1.90	0.01	0.04	0.00	0.36	0	530	31.09	32.61	1.52	0.06	0.66			
81226	33.40	35.55	2.15	0.09	0.50	0.01	2.41	38	531	32.61	34.14	1.52	0.04	0.84			
81227	33.55	36.65	3.10	0.04	0.03	0.00	0.61	0									
81228	36.65	37.55	0.90	0.02	0.44	0.08	0.68	4									
81229	37.55	38.50	0.95	0.00	0.04	0.01	0.26	0	532	39.01	41.15	2.13	0.03	0.49			
81230	44.50	45.60	1.10	0.00	0.01	0.00	0.24	0									
81231	45.60	48.00	2.40	0.00	0.01	0.00	0.38	0									
81232	48.00	49.50	1.50	0.00	0.00	0.00	0.25	0									
81233	49.50	51.00	1.50	0.00	0.00	0.00	0.31	0									
81234	51.00	53.30	2.30	0.00	0.01	0.00	0.35	7									
81235	64.50	66.00	1.50	0.00	0.00	0.00	0.11	0									
81236	66.00	67.50	1.50	0.01	0.33	0.05	0.25	7									
81237	67.50	69.00	1.50	0.00	0.02	0.00	0.16	0									
81238	69.00	70.50	1.50	0.00	0.01	0.00	0.16	0	533	67.36	69.49	2.13	0.04	0.84			
81239	70.50	72.00	1.50	0.00	0.03	0.01	0.45	0	534	69.49	70.71	1.22	0.03	0.05			
81240	72.00	73.50	1.50	0.07	1.21	0.13	1.03	0									
81241	73.50	75.50	2.00	0.01	0.53	0.13	0.72	14									
81242	75.50	77.35	1.85	0.00	0.01	0.00	0.17	0									
81243	77.35	79.00	1.65	0.00	0.13	0.04	0.18	0									
81244	79.00	80.80	1.80	0.00	0.01	0.00	0.08	0									
81245	80.80	82.00	1.20	0.02	0.02	0.01	0.12	0	535	80.16	81.69	1.53	0.02	0.11			
81246	82.00	84.00	2.00	0.01	0.01	0.01	0.1	0	536	81.69	83.21	1.52	0.03	4.00			
81247	84.00	85.25	1.25	0.01	0.04	0.01	0.11	0	537	83.21	84.73	1.52	0.04	4.27			
81248	85.25	86.00	0.75	0.01	0.17	0.03	0.35	0	538	84.73	86.26	1.53	0.09	7.32			
81249	86.00	87.45	1.45	0.20	4.76	1.10	2.42	23	539	86.26	87.78	1.52	0.08	8.12			
81250	87.45	89.00	1.55	0.00	0.05	0.02	0.11	0	540	87.78	89.31	1.53	0.05	3.21			
									541	93.88	95.71	1.83	0.01	0.74	0.05		
									542	95.71	97.54	1.83	0.01	0.82	0.11		
									543	126.19	127.10	0.91	0.06	0.05			

BWR 14-06								
Sample #	From (m)	To (m)	Length (m)	Cu (%)	Zn (%)	Pb (%)	Ag ppm	Au ppb
P	72.00	75.50	3.50	0.03	0.82	0.13	0.85	8
Q	85.25	87.45	2.20	0.14	3.19	0.74	1.71	15

68-06								
Sample #	From (m)	To (m)	Length (m)	Cu %	Zn %	Pb %	Ag g/t	Au ppb
544	138.68	140.21	1.53	0.12	0.29			
545	140.21	142.95	2.74	0.62	0.07			
546	142.95	144.48	1.53	0.22	0.74			
547	151.49	153.01	1.52	0.10	0.56			
	67.36	69.49	2.13	0.04	0.84			
	81.69	89.31	7.62	0.06	5.34			