

NI 43-101 Technical Report on Resources West Pequop Gold Exploration Project Nevada, USA

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Summary (Item 3)

The West Pequop Gold Exploration Project (West Pequop) is an advanced stage Carlin-Type gold exploration property located in eastern Nevada, with established resources. AuEx Ventures, Inc. (AuEx) has been conducting exploration at West Pequop since 2004 and its venture participant Agnico Eagle Mines Ltd. (Agnico) has been conducting drilling for the benefit of the Joint Venture since 2006. West Pequop is located within the newly emerging Pequop Gold District in eastern Elko County, Nevada, which includes AuEx interests in the Long Canyon gold deposit, on an adjoining property east of West Pequop. This report presents the initial resource estimates for three gold deposits within a 4 square mile area of the West Pequop property, the Section 34, Acrobat, and Mountain Top gold deposits.

Current West Pequop resources for the combined classified Mineral Resource estimate for all three deposits, at a cut-off grade of 0.3 grams per tonne are shown in Table 1 and Table 2.

Table 1: Total In-situ Mineral Resources for West Pequop at 0.3 g/t Au cutoff grade (SRK 2010).

Deposit	Classification	Tonnes	Au (g/t)	Au Grams	Au Ounces
Total	Indicated	1,227,000	1.63	1,995,000	64,140
Total	Inferred	5,504,000	1.41	7,759,000	249,430

The state of exploration for West Pequop is too early to justify estimating in-pit resources. SRK has estimated and is reporting an in-situ total resource rather than an in-pit resource, primarily because the resource is predominantly of Inferred classification. In reporting the resource at a 0.3 g/t Au cutoff, SRK has examined potential mineability and can state the mineralization has the potential for economic extraction at current gold prices.

Exploration potential exists to increase the confidence of the classification by targeted in-fill drilling; and there is potential to expand the resources and define additional mineralized zones at West Pequop.

Property Description and Location

The West Pequop property is located in the Pequop Mountain range, Elko County, approximately 23 miles by road to the southeast of the town of Wells, Nevada. The total land package is a contiguous block approximately 48 square miles in size, consisting of 1361 unpatented lode mining claims, 160 acres of private lands, and 3660 acres of leased/optioned lands; a block of land that is approximately five miles wide, from the valley floor east to the range crest, and traverses north to south for nine miles. The Long Canyon property adjoins the West Pequop property on the east side of the Pequop range.

Ownership

AuEx Ventures, Inc.(www.auex.com), through its wholly owned US affiliate AuEx, Inc., (AuEx) controls the lands that comprise the West Pequop Gold Exploration Project. The lands are held by Pittston Nevada Gold Company (PNGC) a wholly owned subsidiary of AuEx, Inc., a Nevada corporation.

PNGC entered into a Joint Venture with Agnico-Eagle (USA) limited (Agnico), a Colorado corporation, on May 9, 2006, whereby Agnico can earn a 51% vested initial interest in the West

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Pequop project by expending \$5.0 million by May 9, 2010. Agnico completed the 51% earn-in during 2008, and elected to increase its participating interest by 19% to a total 70% vested interest. The election to proceed to 70% requires Agnico to carry all exploration/development costs to completion of a "bankable feasibility study" within a five-year period. Agnico can elect at any time during the five-year period to stay at the current 51% vested interest. Agnico continues to operate the Joint Venture with AuEx (49% vested interest) through the West Pequop Project LLC, Operating Agreement, a Nevada limited liability company.

Geology and Mineralization

The mineralization defined at West Pequop Property is Carlin-Type gold mineralization. Gold is hosted in Cambrian age silty carbonate sedimentary rocks, in association with geochemically anomalous arsenic, antimony, mercury, and tungsten. The mineralization is typically found in outcrop with decalcification, hematitic alteration, and locally jasperoid silicification along bedding planes as replacement or bedded breccias, and in fault zones. Gold mineralization in soils, rock chips, and in drill samples is present over an area at least 4 square miles in extent and includes drilling that has defined gold resources for the Acrobat, Section 34, and Mountaintop deposits, with exploration targets elsewhere on the property.

Gold mineralization grades in drilling range from low-grades of 0.1 g/t to over 20.0 g/t Au. Controls to mineralization are stratigraphic as well as structural. Controls to high-grade intercepts (+ 3.0 to 5.0 g/t Au) are not easily identifiable and traceable in drilling. Mineralization drilled is essentially all in oxidized rock; very little sulfides are noted in drill samples, except for remnant sulfides associate with some of the high-grade intercepts. Mineralization occurs at surface in all three deposits, and most of the mineralization is within 150 to 200m of surface; extending to depths below surface of 190m at Mountain Top, 250m at Acrobat, and 350m at Section 34.

Exploration

Prior to 1994, gold mineralization was not known to exist in the Pequop Mountains. In that year, a regional dry stream sediment sampling program by Pittston Nevada Gold Company Ltd. (PNGC) defined anomalous gold samples; and upon follow-up soil grid sampling and rock chip sampling, high grade gold (+5.0 g/t) mineralization was defined in several target areas. During the period of 1994 to 2000, PNGC consolidated the land position, conducted significant ground exploration work, completed an airborne geophysical survey, and drilled 53 exploration drill holes on four separate targets. That work culminated, at the end of 2000, and resulted in 25 of the 53 holes intersecting greater than 20 ft @ 0.03 oz/Ton gold; with the highest grade intercept being 20 ft @ 0.47 oz/Ton in a core hole at the Acrobat Target. PNGC discontinued gold exploration activities in 2001, due to a change in corporate objectives and the low gold prices at that time.

AuEx acquired the property from PNGC in 2004 and conducted limited follow-up drilling in 2005. Agnico began drilling on behalf of the Joint Venture in 2006, and has been focused largely on the three outcropping target areas, Acrobat, Section 34 and Mountain Top. Agnico has been conducting RC and core drilling since 2006, using industry standard drilling, sampling, and assaying procedures.

The West Pequop property is an excellent example of a grass-roots discovery of Carlin-Type gold mineralization in Nevada, which has progressed to the point of initial resources on three

deposits, and forms part of the newly emerging Pequop Gold District that includes the adjacent Long Canyon Gold Exploration Property. Drilling through 2009 totals 206 drillholes for 49,228m of which 183 holes at 42,962m are in the three resource areas.

Mineral Resources

SRK examined the geology and drillhole database available for West Pequop in concert with AuEx and Agnico staff input and geological interpretations. SRK has determined there is sufficient data to support resource estimation by industry standard methods.

The West Pequop mineral resource estimate was completed by SRK with drillhole and project data current through March 23, 2010 (the effective date of this report). The resources were modeled and estimated by evaluating the drill data statistically, and utilizing two-dimensional lithological shapes (strings) provided by Agnico-Eagle to interpret mineral domains on cross sections spaced at 40-meter intervals throughout the extent of the Section 34, 25 meters on Acrobat and variably through the Mountain Top mineralization. The modeled mineralization database was analyzed statistically to establish estimation parameters. Gold grades were estimated by inverse-distance methods into a block model with 10 meter (width) x 10 meter (length) x 6 meter (height) blocks that were constrained to the mineral domains using Datamine Studio3® mining software. The resources are classified according to CIM definitions for reporting mineral resources.

SRK determined that a cutoff of 0.3 g/t (approximately 0.01 ounces per short ton) is appropriate for oxide-gold mineralization in Nevada to define the resource as having the potential for economic extraction. Resource sensitivity tables at various cut-off grades are presented in Section 15 of this report to represent grade distributions. Table 2 presents the current resource estimate, by classification and by deposit for West Pequop.

Table 2: Total In-situ Mineral Resources by Deposit and Classification, at 0.3 g/t Au cut-off grade (SRK 2010).

Deposit	Classification	Tonnes	Au (g/t)	Au Grams	Au Ounces
Section 34	Indicated	527,000	1.64	865,000	27,810
Section 34	Inferred	2,883,000	1.41	4,063,000	130,620
Acrobat	Indicated	581,000	1.32	767,000	24,640
Acrobat	Inferred	2,178,000	1.22	2,655,000	85,360
Mountain Top	Indicated	119,000	3.06	363,000	11,690
Mountain Top	Inferred	443,000	2.35	1,041,000	33,450

As mineralization is at surface and in-part shallow in all three deposits, a portion of the mineralization in all three deposits is potentially amenable to open pit mining. The resources stated in this report are insitu total resources, not in-pit resources, primarily due to the majority of the resource being classified an Inferred. SRK recommends future resource estimations should seek to develop in-pit potentially mineable resources.

At this stage of the project, there is only some very limited information on cyanide-soluble gold assays, which indicate the oxide gold mineralization is potentially amenable to industry-standard processing. There has been no relevant metallurgical testing to determine potential metallurgical

recoveries. SRK recommends a program of metallurgical testing to define the potential for economic extraction, methods of recovery, and estimated recoveries.

Conclusions and Recommendations

West Pequop is an advanced-stage exploration property with current resources in three gold deposits within a 4-square mile area. The project warrants follow-up in-fill drilling in an effort to increase the confidence in the resource classification, and step-out and exploration drilling with the goal of increasing the total project resource base.

A Recommended Phase I program for 2010 includes continued drilling at the same level as the 2009 drilling program, approximately 50 drillholes for 13,500m of total RC and core drilling, with a minimum of three targeted sets of twin-holes (RC-vs-Core) in higher-grade mineralization. Metallurgical testing and follow-up resource estimation is also recommended. The proposed program envisions a budget of \$2.64 million, assuming drilling commenced in May and is completed by mid-November.

Continued successful drilling from the recommended Phase I program would justify and additional Phase II program of perhaps \$3.5 million in the following year to include further definition drilling, geotechnical drilling for potential open pit slope determinations, a definitive metallurgical program and a Scoping Study to determine the project potential for economic development.

1 Introduction (Item 4)

1.1 Terms of Reference and Purpose of the Report

This Technical Report for the West Pequop Gold Exploration Project (West Pequop), Elko County, Nevada, is a summary of the technical merits of the gold exploration project, based upon current and historical geologic information. AuEx and its venture participant Angico have identified gold mineralization of potential economic interest in a number of drill holes on a portion of the West Pequop property. Current drilling has defined three separate gold deposit which are similar to oxide Carlin-Type gold deposits elsewhere in Nevada. The West Pequop property is categorized as an advanced-stage exploration property by virtue of the estimated \$13 expenditures million exploration to date. including extensive geological/geochemical/road access work and 206 drill holes completed. This technical report presents initial mineral resources estimates for three deposits, Acrobat, Section 34, and Mountain Top.

SRK Consulting (U.S.), Inc. (SRK) was commissioned by AuEx, Inc., a wholly owned U.S. subsidiary of AuEx Ventures Inc., in early 2010, to prepare a report compliant with the Canadian National Instrument 43-101 (NI 43-101) requirements on AuEx interests in West Pequop. This Technical Report on the West Pequop Gold Exploration Project is prepared according to NI 43-101 guidelines. Form NI 43-101F1 was used as the format for this report.

For the purpose of this report, "AuEx" will be used interchangeably to refer to AuEx, Inc, a Nevada corporation, and to AuEx Ventures Inc, the parent company; with respect to the West Pequop Gold Exploration Project. AuEx Ventures Inc. is a Toronto Venture Exchange(TSX-V) listed company with the stock symbol "XAU". AuEx has executive offices located at 940 Matley Lane, Suite 17, Reno, Nevada, U.S.A. 89502; and corporate offices at Unit 1 - 15782 Marine Drive, White Rock, BC, Canada V4B 1E6. For the purpose of this report, "Agnico" will be used to refer to Agnico Eagle Mines, Limited, a Toronto-listed (TSX) company, and its US wholly owned subsidiary Agnico-Eagle (USA) Limited, the venture participant with AuEx at West Pequop.

This report is prepared using the industry accepted Canadian Institute of Mining Metallurgy and Petroleum (CIM) "Best Practices and Reporting Guidelines" for disclosing mineral exploration information and the Canadian Securities Administration revised regulations (2005) in NI 43-101 (Standards of Disclosure For Mineral Projects), and Companion Policy 43-101CP.

The purpose of this report is to describe the basic data available and exploration work conducted to date that supports the exploration target concepts and the resource estimates for the West Pequop Gold Exploration Project. The author understands that AuEx may use this report, as a reporting issuer, in any filings it deems necessary to comply with Canadian National Instrument 43-101, or any other jurisdictional or financial requirement for disclosure of material mineral exploration information.

1.2 Reliance on Other Experts (Item 5)

The authors, as Qualified Persons, have relied on the available data to prepare this report, and have not independently verified the drillhole analyses, location of all historical drill holes, or examined the validity of the chain of title on the mining claims and private lands comprising the West Pequop Gold Exploration Project. It is the author's opinion, based on field observations in 2004, and review of the drilling results from 2005 through 2009, that the exploration data for West Pequop is complete, credible, and verifiable in the field.

Geologic interpretations and opinions presented in this report are the author's, and may not entirely coincide with the opinions of the management of AuEx.

SRK's opinion is that the data for the West Pequop Gold Exploration Project was collected in sufficient and acceptable detail to establish the exploration concepts and interpretations that are discussed in this technical report. Sufficient information is available to prepare this report, and any statements in this report related to deficiency of information are directed at information that, in the opinion of the authors, has not yet been gathered.

The authors have relied upon the work of others to describe the land tenure and land title in Nevada. The author is not qualified with respect to environmental laws in Nevada, as regarding issues addressed in Section 2.5 of this report – Environmental Liabilities and Permitting. Information in these regards was supplied by AuEx and/or Agnico. SRK has relied upon the drillhole database provided by Agnico, and has not independently verified the entire database. SRK has selectively reviewed some of the basic data that support the drillhole database upon which resources are estimated.

The author and SRK are not insiders, associates, or affiliates of AuEx. The results of this Technical Report are not dependent upon any prior agreements concerning the conclusions to be reached, nor are there any undisclosed understandings concerning any future business dealings between AuEx and the authors or SRK. SRK will be paid a fee for its work in accordance with normal professional consulting practice.

1.2.1 Sources of Information

Information available for the West Pequop property is a combination of public documents and private company exploration data in the possession of AuEx. The data is present in the offices of AuEx, Inc. in Reno, Nevada. The majority of the project information prior to 2005 is exploration data generated by Pittston Nevada Gold Company Ltd. (PNGC); which conducted exploration on the property from 1994 through 2000. Since 2005, the project data is a combination of AuEx generated drill data and that generated by Agnico since 2006.

Data reviewed includes reports, memoranda, assays, drill hole logs, drill core, maps and cross-sections, as well as land and environmental data pertaining to past exploration activities. The various exploration data examined are primarily hard copy information and reports for which many are in digital format. The surface and drill hole assay data are in digital computer files, with most of the maps, sections and data plots being in digital format. The list of references (Section 19) is a partial list of reference material for the West Pequop property.

Tables and Figures in this report are numbered consecutively and referenced to the major sections of the report.

1.3 Qualifications of Consultants (SRK)

The SRK consultants and authors of this report are Allan V. Moran, Principal Geologist with SRK Consulting in Tucson, Arizona, and Frank A. Daviess, Associate Principal Resource Geologist with SRK Consulting in Denver, Colorado.

Allan V. Moran, R.G., C.P.G.

Mr. Moran is a "Qualified Person" as defined by NI 43-101, is the primary author, and is the Qualified Person responsible for all sections of this report. The author, an independent geological consultant, previously completed the NI 43-101 technical report titled "*Pequop Exploration Property, Nevada, USA*", *dated January 05*, 2005, for AuEx, Inc. The author is a Qualified Person for the purpose of this report and NI 43-101 requirements. He has visited the project site.

The author has completed several NI 43-101 technical reports on exploration properties, and has over 20 years of relevant industry experience in gold exploration; including approximately 10 years experience directly with Nevada geology and Nevada gold exploration.

Frank A. Daviess, FAusIMM., Resource Geologist

Frank Daviess is a "Qualified Person" as defined by NI 43-101, and is the Qualified Person responsible for the of the resources reported for the West Pequop Gold Exploration Property, in Section 15 of this report. He has visited the project site.

Mr. Daviess has been involved with geological modeling and resource estimation for over 25 years, including resource estimation for Nevada gold deposits.

1.3.1 Site Visit

Mr. Moran first visited the site in 2004 which included two days of field review by the author on August 25 and 26, 2004, a data review in the Reno offices of AuEx on August 27, 2004, and a review of drill core in the offices of AuEx on September 17, 2004.

A second field visit was conducted by Mr. Moran and Mr. Daviess while drilling was in progress on June 30, 2009, and drill core was examined in Agnico's field offices in Wells, Nevada on July 1, 2009. Field investigations consisted of verification of access, geology, drill hole locations, and drill core lithologies and alteration.

1.4 Effective Date

The effective date of this report, March 23, 2010, is the date at which the drillhole database and sectional mineralized shapes were completed and presented by Agnico to AuEx, and provided to SRK for use in resource estimation. SRK understands the project data as of the effective date includes all project drillhole data through the end of drilling in late 2009.

1.5 Units of Measure

The data described in this report are a combination of Imperial units of measure; miles, feet, acres, etc., as these are the common units of measure in the United States for the land/legal descriptions; and Metric units of measure for the resource model; meters, metric tonnes, and g/t Au. Metric units are provided as AuEx and Agnico are Canadian listed public companies and Metric units of measure are common internationally. All currency references are US dollars (US\$) unless specified otherwise. Geochemical values are expressed in either parts per million

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(ppm) or part per billion (ppb) for geochemical values of gold, silver, arsenic, antimony and mercury; with ppm or grams per tonne (g/t) for drill hole gold values. Drillhole coordinates are UTM (metric) coordinates. See Section 20 for a Glossary of Terms and Abbreviations.

2 Property Description and Location (Item 6)

2.1 Property Location

The West Pequop property is located in the Pequop mountain range, Elko County, northeastern Nevada; approximately 23 miles by road southeast from the town of Wells, Nevada (Figure 2-1), and approximately 10 miles south of Interstate Highway 80. It consists of approximately 48 square miles of land that lies on the west side of the mountain range. The approximate geographic center of the West Pequop property is located at 110,703,000 m East and 4,538,000 m N (UTM coordinates; Zone 7, NAD 27 Datum).

The West Pequop property consists of 1361 unpatented lode mining claims, 160 acres of owned private lands and 3605 acres of leased/optioned private lands in a contiguous block of land as shown in Figure 2-2. The total land package includes most of T36N, R65E, and most of the north half of T35N, R65E, Mount Diablo Baseline and Meridian.

2.2 Legal Surveys

Copies of MT Plats (master title plats) are available from the BLM, and they provide information on the basic ownership (private versus public) of lands within each Township (six square mile legal land subdivision). Copies of MT plats are also present in the office files of AuEx, Inc. Copies of individual unpatented mining claim notices and the detailed map showing their locations are on file with the BLM office in Reno, Nevada, and with the Elko County Recorder's office in Elko Nevada. The claim notices and maps on file with the BLM and County constitute the legal descriptions of the unpatented mining claims; and the locations in the field take precedent should there be a discrepancy with descriptions or maps (none are known). Due to the large number of individual unpatented claim comprising the West Pequop property, a detailed map of the claims is not provided in this report, just a map of the outer boundary of the area claimed as shown in Figure 2-2. The BLM serial numbers (NMC numbers) for each claim or claim group are listed in the table in Appendix B, and are sufficient information to identify specific claims and their detailed description and map which are on file with the BLM.

2.3 Mineral Titles

There are 1361 unpatented active lode mining claims, a complete list of which is listed in Appendix B. The unpatented mining claims are located in the field with 2-inch square wooden posts (as per Nevada staking regulations), and while validity and location of the unpatented mining claims has not been independently verified in the field, several claim location monuments and claim corners were noted by the authors during the field visits. AuEx represents that the list of unpatented claims in Appendix B is complete and accurate as of September 1, 2009; and the claims are valid through August 31, 2010.

Ownership of unpatented mining claims is in the name of the holder (locator), with ownership of the minerals belonging to the United States of America, under the administration of the U.S. Bureau of Land Management (BLM). Under the Mining law of 1872 which governs the location of unpatented mining claims on Federal lands, the locator has the right to explore, develop, and mine minerals on unpatented mining claims without payments of production royalties to the Federal government. It should also be noted that in recent years there have been U.S.

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Congressional efforts to change the 1872 mining law, to include the provision of federal production royalties; however, currently annual claim maintenance fees are the only federal encumbrances to unpatented mining claims. Nevada BLM records of mining claims can be searched on-line at http://www.blm.gov/lr2000/.

In 2004, AuEx Ventures, Inc.(www.auex.com), through its wholly owned US affiliate AuEx, Inc., (AuEx) acquired control of most of the lands that comprise the West Pequop Gold Exploration Project. The lands, including unpatented mining claims, owned private lands, and leased private lands, were held by Pittston Nevada Gold Company (PNGC) a wholly owned subsidiary of AuEx, Inc., a Nevada corporation. AuEx, Inc. completed a Members' Interest Purchase Agreement dated August 18, 2004, as amended, between MPI Gold (USA) Ltd. and PMV Gold Company (together the sellers), the owners of the outstanding membership interests in PNGC, and AuEx, Inc. (purchaser), a private Nevada corporation. Upon completion of the terms of Members' Interest Purchase Agreement, AuEx, Inc. acquired all of the outstanding ownership interests in PNGC.

PNGC (AuEx) entered into a Joint Venture with Agnico-Eagle (USA) limited (Agnico), a Colorado corporation, on May 9, 2006. The Joint Venture Agreement provided for Agnico to earn a 51% vested initial interest in the West Pequop project by expending \$5.0 million by May 9, 2010. Agnico completed the 51% earn-in on June 17, 2008, and elected, through the Joint Venture terms, to increase its participating interest by 19% to a total 70% vested interest. The election to proceed to 70% requires Agnico to carry all exploration/development costs to completion of a "bankable feasibility study" within a five-year period (by June 17, 2013). Agnico can elect at any time during the five-year period to stay at the current 51% vested interest, and co-fund the project with AuEx at 49% vested interest.

Agnico continues to operate the Joint Venture with AuEx through the West Pequop Project LLC, Operating Agreement, a Nevada limited liability company. The Operating Agreement has an effective date of June 17, 2008. All lands within the joint venture Area of Interest have been transferred by deed into the West Pequop Project LLC with the exceptions of Sections 5 and 7, T36N, R65E, which by agreement must be included in the agreement.

The Joint Venture and Operating Agreements have standard provisions for property maintenance, an area of interest (Figure 2-2) to allow for property additions, dilution of interests, funding at the participating interests (51/49) except during the current option period to earn additional interests, project management, programs and budgets, accounting, and other standard provisions.

There is a provision that allows for conversion of a participating interest that fall below 10% to a 2.5% NSR royalty, and elimination of that participant. And either member to the Agreements can transfer their participating interest, subject to Preemptive Rights of the other participant.

As of the December 31, 2009, land holdings for West Pequop consist of the following:

1. 5 sections of leased/optioned private lands from Nevada Land and Resources Company, LLC (NLRC), a Nevada corporation, totaling approximately 1459 hectares (3605 acres), in Section 5, 7, 21, and 33, T36N R65E. Sections 5 and 7 are held under option to AuEx from NLRC, are subject to the Joint Venture, and have separate private surface ownership;

- 2. 1 partial section of leased private mineral rights, 186 hectares (460 acres) in Section 29, T36N, R65E, from the McMullen family;
- 3. 1 partial section of 100% owned private surface land without mineral rights, 65 hectares (160 acres) in Section 29, T36N, R65E. Control of the mineral rights to these lands are included in the McMullen family lease;
- 4. 1,361 unpatented lode mining claims in T35N and T36N, R65E as listed in Appendix B

The total annual land holding costs are approximately \$251,881, as further defined below:

Table 2.1: West Pequop Land Holding Costs

Land	Due Date	Amount
1361 Unpatented mining claims	BLM: annual claim fees	\$204,834.50
	Elko County: (annual recording fees)	
1481 hectares leased private lands	2010 fees	\$46,906
65 hectares of owned private lands	Annual taxes	\$140,48
TOTAL		\$251,880.98

In March 2010, the Nevada State Assembly passed a bill implementing the collection of a fee for active unpatented lode mining claims. The cost to the West Pequop Project LLC will be \$85 per claim for a total of \$115,685 due in full on November 1, 2010, or, alternatively, half of the total by November 1 and the remainder by June 1, 2011. The fee, as mandated by the new law, is a one-time levy,

There is no work obligation on the part of AuEx, with respect to any of the unpatented lands, only a commitment to maintain the mining claims by making annual claim maintenance fees to the BLM. The lease in Section 29 from McMullen has a work commitment of \$200,000 by the end of 2010 and \$450,000 by the end of 2011. NLRC leased lands have no work commitment requirement. All land holding fees, lease payment, and work obligations are being paid by Agnico, with no costs accruing to AuEx until a definitive participating interest is elected by Agnico; either the 70%-30% split interests upon completion of a feasibility study, or an election by Agnico to proceed at the vested 51%-49% split interests (49% AuEx).

2.4 Location of Mineralization

The West Pequop property in general is a contiguous block of land that is approximately 5 miles in width east-west, extending from the valley bottom on the west side of the Pequop Mountains to the range crest, and approximately 9 miles in north-south length along the west side of the Pequop Mountains. The land position is immediately contiguous on the east with lands of the Long Canyon Gold Exploration Project, which is co-owned and Joint Venture operated by AuEx and its venture participant Fronteer Development (Fronteer). The property traverses rugged mountainous terrain from elevations of 5500 to over 9000 ft (1680 to 2750 m). Figures 3-1 and 3-2 show the property outline, access routes, and exploration target areas within the permitted exploration area.

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2.5 Royalties, Agreements and Encumbrances

The 1361 unpatented lode mining claims comprising the West Pequop property have no third party royalties or encumbrances. There are no federal government production royalties applicable to unpatented lode mining claims.

The five parcels of private land leased/optioned from Nevada Land and Resource Company, LLC, are subject to a 3% NSR production royalty on any gold, silver and platinum production, and 2% for any other mineral production from that parcel of land. There is an option to purchase the lands, and the leases are initial 5 year terms, renewable for up to 20 years. All the exploration targets thus far defined, and exploration drill holes to date, have been outside these leased parcels of land.

The McMullen leased private lands have a 4% NSR production royalty on any gold, silver and platinum production, is a ten-year lease initiated in 2007, and is available for renewal. The 65 hectare parcel of private surface land purchased by PNGC and conveyed into the West Pequop Project LLC is subject to the McMullen production royalty.

Should any gold production occur from the West Pequop property, such production would be subject to the State of Nevada Net Proceeds of Mine Tax, which is limited to 5% of the production net proceeds (similar to a 5% net profits tax). This is a tax that is levied by the State on Nevada on all mine production in the state.

The only other encumbrance to the West Pequop property relates to reclamation obligations as defined in section 2.6.

2.6 Environmental Liabilities and Permitting

Environmental liabilities all relate to reclamation liabilities for exploration activities conducted by PNGC during the period of 1994 to 2000, and by AuEx and Agnico from 2005 through 2009; as there were no previous mineral exploration activities, or any other ground disturbances in the area. The primary environmental liability is covered by bonding in place relating to an Environmental Assessment that is tied to the BLM Plan of Operation under which exploration activities have been conducted. There are no historical mineral prospects or mines on the property, except for a small historical barite claim on the southern edge of the Area of Interest boundary, for which minimal if any work was done, and there are no archeological or historical sites of consequence.

2.6.1 Required Permits and Status

There are several permits in place which provide for continuing exploration activities at the West Pequop property.

Exploration activities for the part of the West Pequop property that envelopes the Acrobat, Section 34, and Mountain Top resource areas are permitted under a revised Plan of Operations with the BLM, dated May 24, 2000; Plan number NV071287. That plan allows for up to 100 acres of total ground disturbance for the purposes of mineral exploration, in phases, within the permitted project area. Some of the earliest drill access roads/drill sites have been reclaimed and re-seeded.

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As part of the requirement for an approved Plan of Operations, an Environmental Assessment is required to address the magnitude of proposed disturbance and the methods and costs required to mitigate the proposed disturbance. The details of the approved Plan of Operations/Permit for Reclamation are contained in the original Plan of Operations dated September 1999 and revised in March, 2000, then approved on May 24, 2000 as referenced above; and in the Pequop Environmental Assessment BLM/EK/PL-2000/011,3809, N17-99-002P, N-66237 (BLM May, 2000). The Environmental Assessment resulted in a "Finding of No Significant Impact" (FONSI) on May 24, 2000, which means that proposed disturbance could proceed with appropriate bonding and reclamation plans in place. Both the BLM Plan of Operations and the Environmental Assessment are public documents that are available for review in the Elko, Nevada office of the BLM. The BLM Plan of Operations/Permit for Reclamation (PoO)was issued a new BLM serial number, NV071287, on February 7, 2001. An amendment to the existing Plan (PoO) was submitted by Agnico in April 2010 and is in the process of review. The amended Plan, upon approval, will greatly expand the area within the AoI eligible for construction of access and drilling.

Reclamation is regulated through the BLM in concert with the State of Nevada, Department of Conservation and Natural Resources, Division of Environmental Protection (NDEP). The NDEP issued Reclamation Permit No. 0193 on May 21, 2000. Reclamation bonding is currently set at \$443,755, for which PNGC has implemented a Statewide Reclamation bond with the Nevada State Office of the BLM for that amount. An annual disturbance summary report was prepared in April 2010 by Enviroscientists, Inc., Reno, Nevada, and submitted to the BLM and NDEP. Such reports are required prior to April 15, annually, to describe the total acreage of disturbance. The total acreage of disturbed public and private lands under the Plan of Operations (PoO) is currently at 66.86 acres disturbed, 2.92 acres reclaimed but not released from bonding, and 17.10 acres proposed disturbance; for a total of 86.88 acres of the 100 acres allowable under the PoO. There is a minimal annual fee for submission of annual reports to the BLM based on public and private acreages of disturbance; the fee being less than \$1000 for 2010.

Two Notices of Intent filed with the BLM are bonded for disturbance related to small early-stage drilling programs in areas removed from the resource areas bonded under the Plan of Operations. These notices include: N-83262 bonded for approximately \$11,000 and N-83264 bonded for \$43,669. Notice N-83262 is closed and awaiting release.

A water well permit is in place and allows for the pumping of 35,000 gallons per day for 252 days per year. The permit, serial # 62041, was issued in 1996 by the Division of Water Resources, Department of Conservation and Natural Resources of the State of Nevada. In 1996, PNGC completed a water well in Section 20, T36N, R65E, and has used the well to support exploration drilling activities since then. Annual costs to maintain the permit are approximately \$500. Water rights in Nevada are typically granted through appropriation from the State of Nevada. And after a water well is in place, the appropriation process requires the application for "beneficial use" based upon the water that is actually used, rather than the amount of water that is requested. Application for beneficial use has been postponed until such time as the water is being used to maximum capacity in support of exploration activities. In 2008, Agnico filed a water right application for an additional 4 cfs to support future exploration, development, and production. Development of this water right has not occurred yet.

2.6.2 Compliance Evaluation

SRK did not conduct a compliance evaluation of permits and environmental compliance for the West Pequop project; however, SRK is not aware of any environmental or permitting issues that would prevent or hinder continued exploration at West Pequop. AuEx reports that Agnico is operating the exploration program in compliance with existing permits and authority.

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Figure 2-1: West Pequop Location Map

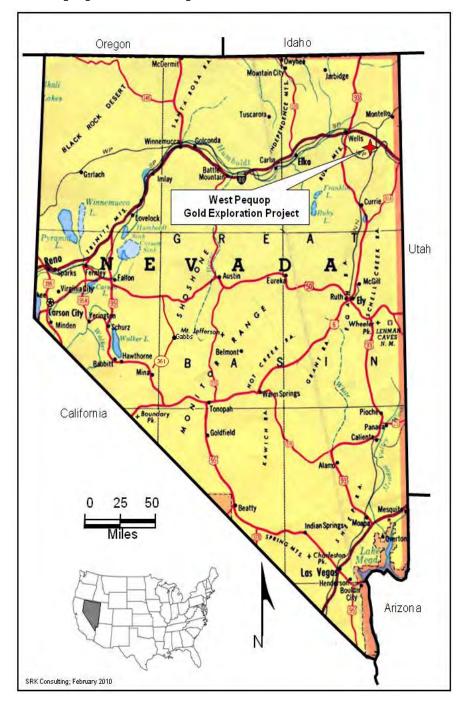
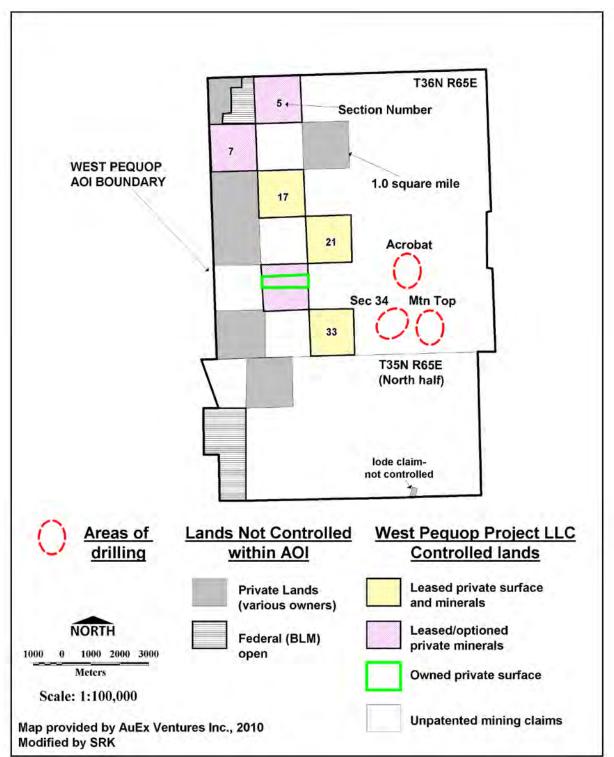


Figure 2-2: West Pequop Land Status and areas of Drilling



3 Accessibility, Climate, Local Resources, Infrastructure and Physiography (Item 7)

3.1 Topography, Elevation and Vegetation

The West Pequop property lies in the Basin and Range physiographic province of Nevada and western Utah, which is a series of northerly trending mountain ranges of typically 2000 to 5000 feet of topographic relief with broad relatively flat intervening valleys. The West Pequop property occurs in the south-central portion of the Pequop Mountain range of Eastern Nevada. Elevations range from 5500ft (1675m) above sea level in the valleys on the west side of the Pequop Mountains, to elevations of over 9000ft (2740m) on the ridge tops. Elevations for the Acrobat and Section 34 deposits on the western side off the range are at about 7700-8000ft (2350-2440m), and the Mountain Top deposit is about 8700ft (2650m) in elevation near the top of the range.

The lower slopes of the project area are covered by sage brush, progressing up-slope to Pinion Pine and Juniper woodlands; typical high desert mountain vegetations. Locally scattered Sub-Alpine Fir, Limber Pine, and Mountain Mahogany woodland stands are present at higher slope elevations; giving way to sage brush and grasses on the otherwise barren ridge tops.

3.2 Climate and Length of Operating Season

Climate is typical for the high-desert regions of Northern Nevada, with usually hot dry summers and cold snowy winters. Summer high temperatures can peak at 100 degrees Fahrenheit (F), with winter low temperatures typically at 0 to 15 degrees F, and winter high temperatures of only 30-40 degrees. Most of the precipitation for the region falls as snow in the winter months, with lesser precipitation as rainfall in the Spring and as thunderstorms during the late summer. Winter storms can deposit many feet of snow in the upper mountains, with elevations above 7000ft (2130m) being continually snow covered from November through April. The highest elevations can have snow accumulation in the tens of feet.

In the absence of all-weather road access, and the on-site presence of equipment necessary to keep roads open, the typical exploration operating season for the West Pequop project would be from mid-late May through mid-November. Drilling activities are commonly conducted during June through November. Improved road access and road maintenance/snow removal equipment can extend the exploration operating season through the winter months if necessary; however, Agnico drilled through the winter in 2007/2008, with attendant decreased productivity during periods of snow storms and drifting snow.

3.3 Access to Property

Access to the west side of the West Pequop property is via Interstate Highway 80 east 13.0 miles from Wells, Nevada to exit 365 (Independence Valley exit), then proceeding east on the south frontage road for 2.0 miles, and then south on an improved gravel road 8.5 miles to the west edge of the property (Figures 3-1 and 3-2). A graded road extends for 3 miles to the east, to the Acrobat deposit, and from there up the mountain to the southeast to the Mountain Top deposit. A second graded road accesses the Section 34 deposit, and a drill access road connects Section 34 with Acrobat.

3.4 Surface Rights

The 160 acres of private surface land owned by West Pequop Project LLC, in Section 29, T36N, R65E provides access to the target areas from the main public gravel road on the west side of the property. However, the current access road crosses from the owned private land onto private lands owned by others, for a short section before continuing onto BLM land held by unpatented mining claims. Currently there are two access or surface use agreements in place for these lands.

3.5 Local Resources and Infrastructure

Wells, Nevada is the nearest town with services and a population of less than 3000; however, Elko Nevada is located another 50 west of Wells on Interstate 80. Elko has a population base of 30,000 to 40,000, and is a support community for many major gold mining operations in northern Nevada. As such, Elko has all the services available to support gold exploration and development activities.

3.5.1 Power Supply

There is no power grid in the Independence Valley on the west side of the project. The nearest major power grid is near the east-west rail line; both located approximately 9-10 miles north of the West Pequop property, north of Interstate Highway 80.

3.5.2 Water Supply

Water for drilling is available to the West Pequop exploration targets from the water well, pump, and generator developed by PNGC and owned by the West Pequop Project LLC.

3.5.3 Buildings and Ancillary Facilities

There are no buildings on the project site lands under exploration. Agnico operates project exploration offices and core logging/core storage facilities in Wells, Nevada.

3.5.4 Manpower

Reverse circulation (RC) and core drilling contractors, heavy equipment contractors, and field technical personnel are all available from service companies and contractors in Elko, Nevada and numerous other locations in the intermountain west. Furthermore, should an economic gold deposit be developed on the West Pequop property, experienced mining personnel and equipment suppliers are available in Elko.

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Figure 3-1: West Pequop Gold Exploration Property – Access Map

Source: Map by Google Earth 2008, modified by SRK, 2010

Figure 3-2: West Pequop Gold Exploration Property – Local Access Map

Approximate Property Boundary 3.2 km

Source: Map by Google Earth 2010, modified by SRK, 2010

4 History (Item 8)

4.1 Ownership

Prior to 1991, there were no mining claims of record for the West Pequop property. In 1991, ACM Gold Mgt Inc (ACM) staked 36 claims in response to gold anomalies in sediment and outcrop samples from the area that would come to be known as the Acrobat target. ACM did not renew the claims in 1993 and allowed the claims to lapse. In 1994, PNGC expanded upon the earlier work, with additional regional BLEG (see section 6.2) and rock chip sampling. Prior to PNGC activities, the project was part of the "checkerboard" lands of Nevada, which is a checkerboard pattern of alternating sections being public and private. In this case, the private lands were held by Big Springs Associates, a California holding company, as private ranch lands, with the alternating sections being BLM administered public lands.

PNGC staked some initial unpatented mining claims in 1994, on BLM lands in the checkerboard. PNGC pursued exploration on staked mining claims during the period of 1994 to 1999, while monitoring the progress on a land exchange between the BLM and Big Springs Associates. The land exchange between the BLM and Big Springs Associates became final on May 20, 1999, and following a 90-day segregation period, the BLM acquired lands became open for location of mining claims. PNGC located mining claims on alternate sections in August of 1999 thus consolidating a contiguous block of land in the range. Additional exploration identified the adjacent Long Canyon target area, and additional mining claims were located there in 2000.

The mining claims, and additionally acquired private and leased lands described in Section 2 were held by Pittston Nevada Gold Company Ltd. (PNGC). PNGC was formed in 1993 as a jointly owned, private gold exploration company. The joint owners were MPI Gold (USA) Ltd., a US subsidiary of Mining Project Investors Pty Ltd of Melbourne, Australia; and PMV Gold Company, a subsidiary of the Pittston Company of Richmond, Virginia (now the Brinks Company). In August of 2002, PMV Gold Company acquired all the shares of MPI Gold (USA) Ltd., thereby consolidating 100% interest in PNGC and the West Pequop and Long Canyon gold properties with PMV Gold Company.

AuEx, Inc. completed a Members' Interest Purchase Agreement dated August 18, 2004, as amended, between MPI Gold (USA) Ltd. and PMV Gold Company (together the sellers), the owners of the outstanding membership interests in PNGC. Upon completion of the terms of the Members' Interest Purchase Agreement, AuEx, Inc. acquired all of the outstanding ownership interests in PNGC, which became a wholly-owned subsidiary of AuEx.

AuEx conducted exploration on the West Pequop and adjacent Long Canyon gold exploration properties in 2005. In 2006, AuEx subdivided the West Pequop and Long Canyon properties, entering into separate joint venture agreements for each.

AuEx (PNGC) entered into a Joint Venture with Agnico-Eagle (USA) limited (Agnico) in 2006, whereby Agnico could earn a 51% vested initial interest in the West Pequop project by expending \$5.0 million by May 9, 2010. Agnico completed the 51% earn-in in June 2008, and elected to increase its participating interest by 19% to total 70% interest, by continuing to fund all exploration costs thorough a feasibility study within a five-year period. Agnico continues to operate the Joint Venture with AuEx (49% vested interest) through the West Pequop Project

LLC, Operating Agreement, a Nevada limited liability company, which controls all lands initially acquired by PNGC and, subsequently, by AuEx and Agnico.

4.2 Past Exploration and Development

Aside from a couple of small historical lead-zinc prospects located north of the West Pequop property, and a historical barite claim on the southern property boundary, there is no evidence of historical mineral prospecting, mining, or modern day mineral exploration prior to 1991 for the West Pequop property.

PNGC first became interested in the region in 1994, by way of a regional BLEG geochemical sampling program. BLEG (bulk leach extractable gold) is a modified form of stream sediment sampling and analysis that is used to detect low levels of gold in surficial material. PNGC's BLEG sampling identified samples with anomalous gold from the dry wash drainages on the flanks of the Pequop Range.

Follow-up sampling into the range led to the discovery outcrop of jasperoid on the current Acrobat exploration target; an outcrop with +10 g/t gold. Additional geochemical work including stream sediment sampling, ridge and spur soil sampling, detailed soil sampling grids, and selective rock chip sampling of outcrops has identified the five areas of anomalous gold and associated trace elements that comprise the West Pequop property exploration targets.

Exploration activities conducted by PNGC between 1994 and 2000 included regional and detailed geologic mapping, a limited IP/resistivity geophysical survey at the Acrobat target, an airborne multi-channel Electromagnetic-Magnetic-Spectrometer/Radiometric survey over a portion of the project area by Fugro, biostratigraphic and petrographic examinations, and both reverse circulation (RC) and core drilling. This work was fairly extensive, and included the collection of 602 BLEG samples, 2050 ridge and spur samples, 4397 grid soils samples, 639 rock chip samples, and 5150 drill hole samples (RC and core) from 49 drill holes in four separate targets. In addition to gold, most of the samples were also analyzed for 30 additional elements by ICP methods.

PNGC terminated exploration activities in the US in November 2000, due to a combination of corporate and strategic issues for the principal partners; however; the core property has been maintained to this date. The West Pequop property suffered from the checkerboard ownership in the early years of PNGC's exploration programs, even though there was early drilling success. And the project exploration activities were terminated late in 2000, in spite of land ownership consolidation in 2000, and additional encouraging drill results, at the bottom of the gold-price commodity cycle. Exploration at the West Pequop property was not terminated in 2000 due to a lack of exploration targets or a lack of success.

AuEx secured the property in 2004 and continued with exploration efforts. AuEx completed 8 exploration drillholes for 1193 m in 2005 on West Pequop.

Agnico has been drilling since 2006 and continues to this date. Agnico has drilled 149 drillholes on the property since 2006. In addition, Agnico has conducted detailed mapping, rock chip sampling, and soil sampling across broad areas of the property.

4.3 Historic Mineral Resource and Reserve Estimates

Although PNGC originally indentified the current three resource areas, Acrobat, Section 34, and Mountain Top, there are no historical resource estimates that were prepared by PNGC or AuEx, as the drilling information was too premature. The resource estimates presented in Section 15 of this report are the initial resource estimates for the West Pequop Gold Exploration Property.

Total exploration expenditures amount to over \$16 million. PNGC (1994-2000) expenditure total approximately \$2.8 million. AuEx expenditures in 2005 total \$1.1 million. Agnico has spent \$5.0 million from 2006 through June 2008, and approximately \$12.0 million additional through December 2009.

5 Geologic Setting (Item 9)

5.1 Regional Geology

The Pequop Mountains are an uplifted block of regionally east dipping lower Paleozoic carbonate sedimentary rocks. They were uplifted and tilted as a result of the Ruby-East Humboldt metamorphic core complex located to the west in the Ruby and East Humboldt mountain ranges. Lower Paleozoic rocks to west of the Ruby Mountains have undergone Paleozoic age thrusting, most notably by the Roberts Mountain Thrust of the Antler Orogeny. Many of the Carlin-Type gold deposits on north-central Nevada are associated with Ordovician age host rocks located immediately below the Roberts Mountain Thrust.

East of the Ruby Mountains, including the Pequop range, the lower Paleozoic rocks are east of the paleo-continental margin, and have experienced little of the Paleozoic orogenic events recognized to the west. The most important structural event affecting the Pequop range is the late Jurassic-Cretaceous Sevier Orogeny, and the thrust faulting associated with it. Coincident with thrusting and prior to subsequent Tertiary uplift, the lower Paleozoic stratigraphy underwent deep burial and consequential metamorphism from upper greenschist to lower amphibolite facies rocks.

A nearly complete section of Paleozoic miogeosynclinal sedimentary rocks is exposed in the Pequop range, with an aggregate thickness of approximately 10 km (Figure 5.1) Lowermost Cambrian rocks in the project area exhibit metamorphism in the form of recrystallization in limestone units, and a foliation with WNW to ESE stretching lineation. This fabric overprints aplite intrusive dikes which are dated at 153 Ma. Folding in the rocks is limited, perhaps in part due to semi-plastic deformation at burial depths with bedding plane slippage and thinning of units.

The regional structure is dominated by the Independence Thrust (Camilleri, 1994), and is a southeast-vergent thrust that offsets metamorphic isograds. While the Independence Thrust has been mapped by Camilleri over the length of the West Pequop project, its precise trace is somewhat problematic in relation to detailed mapping by PNGC, AuEx and Agnico. A total of six structural events have affected the rocks in the Pequop range beginning with synmetamorphic shear, thinning, and lineation development in late Jurassic to early Cretaceous. This was followed by two late Cretaceous folding and thrusting events including the Independence Thrust. A period of late Cretaceous to Late Eocene extension resulted in west-dipping low angle normal faults. Oligocene to Miocene age extension related to the Ruby Mountains core complex resulted in moderately dipping normal faults And finally, Miocene to Recent Basin and Range extension resulted in high-angle normal faults and the present north trending range topography. Gold mineralization may be related to stage four or five extension, and is offset by late Basin and Range Faulting.

There are no major mineral occurrences or other known mineral deposits in the Pequop range; aside from the gold mineralization at West Pequop and the adjacent Long Canyon Gold Project.

5.2 Local Geology

The West Pequop property geology has been mapped in detail and is comprised of an east-dipping normal stratigraphic sequence of lowermost Paleozoic units that has been cut by dominantly north-trending normal faults with down to the east displacements. The rocks are a nearly complete section of Cambrian to Ordovician continental shelf to slope environment sedimentary rocks; predominantly limestones and dolomites with lesser thicknesses of interbedded clastic units. Limestone units observed in the field include both thin bedded to thinly laminated silty (dirty) carbonates that commonly are recessively weathered and intermittent resistant massive limestone and dolomite units that form bold outcrops. The lowermost unit is the Cambrian Prospect Mountain Quartzite, with an overlying thick sequence of Cambrian carbonates topped by the Notch Peak Formation. The Ordovician Pogonip Group and Eureka Quartzite overlie the Cambrian units, and occur near the range crest (Figure 5.1).

Gold mineralization commonly occurs in the silty limestone units, particularly where these rocks have been subjected to carbonate dissolution with associated solution and karst breccia development. Locally, gold-bearing zones lie in close proximity to the major mapped faults.

Two types of intrusive rocks are mapped. Finely crystalline equigranular aplite dikes and sills are common as thin lenses or small pods intruding the units below the Pogonip. They are commonly bleached, oxidized, and foliated in similar fashion to the surrounding units. They are clearly premineral in age.

A mapped unit of diorite is present southeast of the Mountaintop exploration target, as a sill-like body in the Pogonip Group. And similar medium crystalline equigranular to porphyritic diorite dikes and sills are present in the Long Canyon area to the east (Gustin et al, 2009). Very low silica contents reported by Gustin et al (2009) suggest that some of these intrusions are lamprophyres, a common occurrence in Carlin-Type gold districts. Intrusive rocks at Long Canyon and West Pequop appear to predate gold mineralization.

The sequence of stratigraphy in the Pequop range is tabulated in the simplified stratigraphic column of Figure 5.2; and the detailed stratigraphic column for the West Pequop property is shown in Figure 5-5.

5.2.1 Local Lithology

Units mineralized at West Pequop are several, across the stratigraphic column (Figure 5.2 an Figure 5.5). At Acrobat the mineralization is hosted in several units including the Upper Oasis Formation, Candland Shale, and the Notch Peak Limestone. The banded marble unit distinguished in Figs. 5.1 through 5.4 appears to be a zone of strong regional metamorphism affecting the Oasis Formation and, possibly, parts of the Notch Peak Limestone, Candland Shale, and Shafter Formation. At Section 34, the hosts to mineralization are the Morgan Pass formation and the lower Shafter Formation, which are separated by barren dolomite of the Decoy Limestone. Gold also occurs in parts of the Upper Shafter and Oasis Formations. Gold mineralization at Mountain Top is hosted in brecciated limestone units comprising the middle part of the Pogonip Group.

There has been no detailed mapping of the range between the Acrobat-Section 34-Mountaintop targets on the West Pequop property, and the Long Canyon target on the east side of the mountain range. Detailed mapping at Long Canyon suggests a similar stratigraphy and north trending major faults, but with downward displacement to west on the west dipping northerly faults. Generally regional east dips to stratigraphy are also present. This suggests that the core of the range may be graben-like in character, as shown in Cross-section A-A' (Figure 5.3).

5.2.2 Geochemistry

Extensive surface geochemical work had been done by PNGC. BLEG sampling was the initial work that defined anomalous gold for the West Pequop property. BLEG sampling along the range front in 1994 produced gold values of 2.2 to 3.1 ppb Au, against a regional background threshold of 0.93 ppb Au, which was enough to warrant follow-up sampling in the area that is now the Acrobat-Juggler target area.

Ridge and spur soils sampling, and grid soil sampling was done by PNGC on a 200 by 200 ft grid for the main project area of Acrobat-Section34-Mountaintop. In addition to gold, multi-element ICP geochemical analyses show anomalous arsenic, antimony, tungsten, and mercury to be present in areas of anomalous gold. Rock chip sampling and road cut sampling were also done

Soil gold anomalies are typically hundreds of feet across, composed of multiple samples in the range of 50 to 300 ppb Au for the West Pequop areas of resource drilling. In addition to the three primary areas of drilling, there are additional lesser magnitude and size geochemical anomalies, which point to perhaps yet additional exploration targets.

Rock chip sampling of outcrops and road cuts in the soil-gold anomalies show strongly elevated gold and associated trace element numbers. Silver values are generally low, less than 30 ppm, but shows a positive correlation with gold values (+0.9 correlation coefficient). Arsenic and mercury have a +0.8 correlation coefficient with gold at Acrobat.

At Section 34, gold has a strong correlation with arsenic, and a moderate correlation with antimony, although antimony values are an order of magnitude lower than at Acrobat. There is also anomalous tungsten (to 607 ppm in rock chips), although it is not directly correlative with gold.

The Mountaintop soil anomaly does not outcrop. Six samples from two shallow pits show maximum values for gold, arsenic, and mercury, respectively at 8,400 ppb Au, 1,108 ppm AS, and 1,500 ppb Hg. There are also traces of antimony. Values at this level and above, for the same elements, have been found in rock samples collected from trenches completed by Agnico.

The suite Au-As-Sb-Hg-W is the most common geochemical association on the West Pequop property. This is a typical Carlin-Type gold geochemical suite. Also noted in the multi-element geochemistry is a negative correlation of gold with calcium content; denoting calcium depletion or carbonate dissolution or leaching, sometimes with silica addition (as jasperoid development).

Similar geochemical associations are found in drill sample assays; however, the only element of potential economic interest is gold

5.2.3 Structure

Structural controls to mineralization are difficult to define, although structure certainly is important to the mineralization. Structural geology of the mapped area, Figure 5-4, shows several north-south trending normal faults at Acrobat: the Feeder Fault, the QM Fault, and the Ridgetop Fault. All are moderately to steeply east dipping normal faults with downward displacement to the east. The "Feeder" fault appears to offset mineralization that exists east of the fault at lower elevation than the mineralization at surface on the west side of the fault, with mineralization contained in the fault – the Feeder fault is not interpreted as the primary feeder to mineralization. The Curvilinear Fault crosses the north-south faults, with a general northwest-southeast trend, and intersects the Ridgetop Fault.

Many of the stratigraphic units have general recessive weathering patterns and form soil covered hill slopes; therefore, much of the details of structure may not be mapped. Road cut exposures and cross-sections through drill holes in the Acrobat area demonstrate there are unmapped low angle faults or bedded breccias and a more complex nature to high angle faults such as the Feeder Fault.

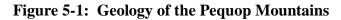
At Section 34, the overall orientation to the the mineralization is northeast (approximately N20E to N30). There are two mapped structures at Section 34, the F45 and F47 faults. Some higher grade mineralized intercepts in Section 34 appear to related to these fault structures. Other areas of higher-grade mineralization in Section 34 may also be related to structures that have not been identified in the drilling (perhaps due to RC drilling rather than core). The F45 fault is located on the west side of mineralization, strikes nearly north-south, and dips approximately 45 degrees to the west. The F47 fault strikes N20E and dips 85 degrees to northwest.

At Mountain Top, clear structural controls to mineralization are not apparent; however, the overall trend to mineralization is a shallow plunge to the N30 East.

General NE strike and shallow dips to mineralization at Section 34 and Mountain Top are generally parallel to the trend of mineralization at Long Canyon (A. Moran, 2008). However, the Long Canyon geological model of gold occurring in breccias at the boundaries of boudin (pull-apart) blocks comprised of competent dolomite strata (Gustin et al, 2009), is not recognized at West Pequop.

5.2.4 Geophysics

Historically, geophysical surveys were conducted at West Pequop and are described in the the NI 43-101 report on the Pequop Exploration Property (A. Moran, 2005). Exploration since 2005 by AuEx and Agnico has been focused on drilling known areas of mineralization and other geological targets.



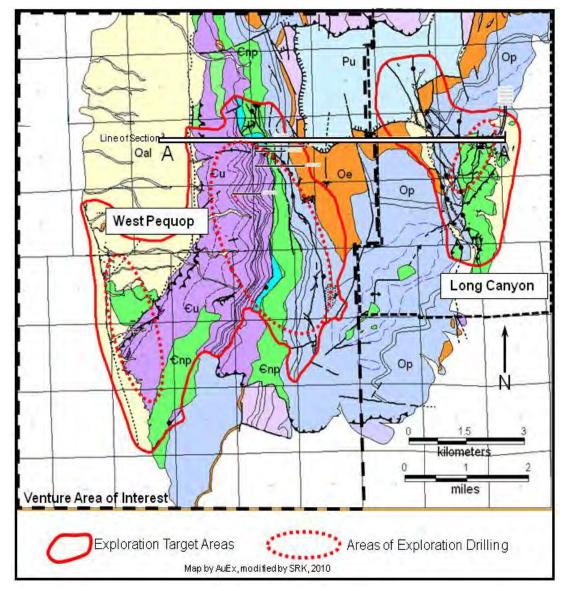


Figure 5-2: Geology Map Explanation

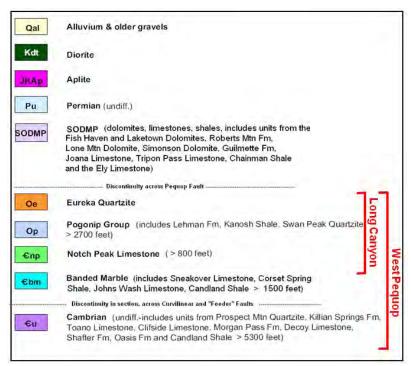
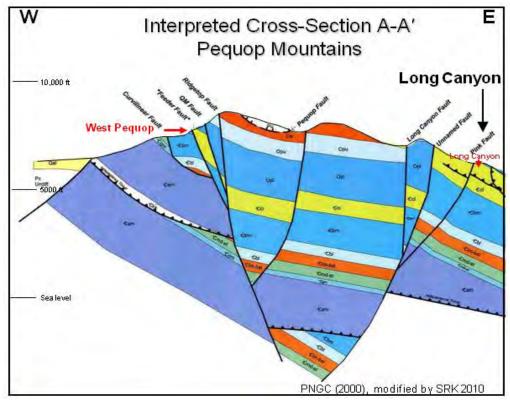
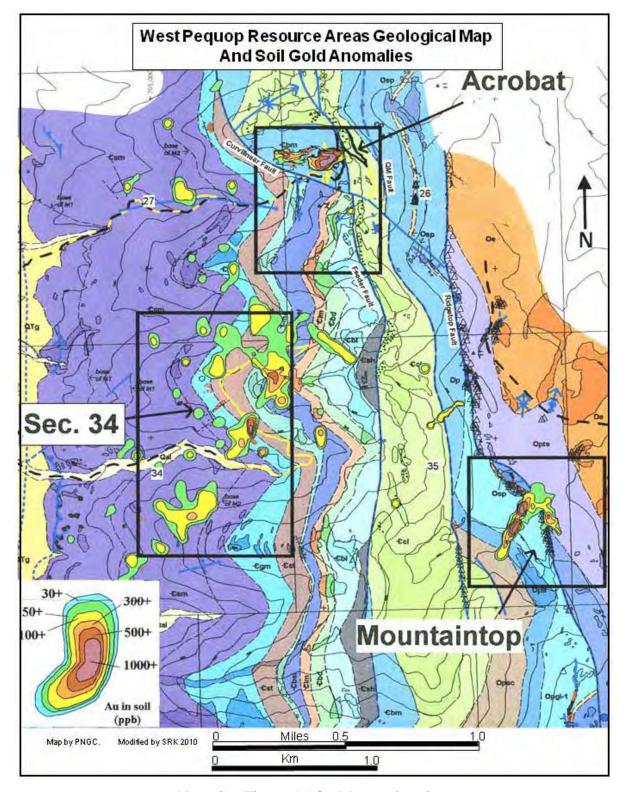


Figure 5-3: Regional Geology Cross Section A-A'



Note: Individual Cambrian and Ordovician units on the Cross Section are lumped together in figures 5-1 and 5-2

Figure 5-4: West Pequop Project Geology – Mineral Resource Areas



Note; See Figure 5-5 for Map explanation

Figure 5-5: West Pequop Project Geology – Stratigraphic Column

	Formation / Group	Thickness (m) [ft.]	Mineralization
Q	Quaternary		
Oe	Eureka Quartzite	uncertain	
Ор	Pogonip Group (multiple units/colors)	835m [2740ft]	Mtn. Top
Cel	Notch Peak Limestone	244 [800]	Acrobat
Cbm	banded Marble	457 [1500]	Acrobat
Csh	Candland Shale	76 [250]	Acrobat
СЫ	upper Oasis Fm.	152 [500]	Acrobat
Cbd	lower Oasis Fm.	91 [300]	
Clm	upper Shafter Fm.	?	
Chsl	lower Shafter Fm.	107 [350]	Section 34
Cmd	Decay Limestone.	30 [100]	Section 34
Cst	Morgan Pass Fm.	107 [350]	Section 34
Cgm	Clifside Limestone	67 [220]	Section 34
Csm	To ano Limestone	731 [2400]	

6 Deposit Type (Item 10)

There are three drill-defined gold deposits on the West Pequop property that are best described as similar to oxide Carlin-Type gold deposits. This section describes the deposit type (Carlin-Type gold deposits), as it relates to the mineralization encountered to date in exploration drill holes. The West Pequop property gold mineralization is described in Section 9.0 (Mineralization).

Carlin-Type gold deposits are a class of gold deposits that are perhaps not unique in the world, but exist in far greater numbers and total resource size in Northern Nevada than anywhere else in the world. They are characterized by concentrations of gold very finely disseminated in silty carbonaceous calcareous rock, such as carbonaceous calcareous siltstone, calcareous mudstone, and silty limestone. The gold is present as micron size to sub-micron size disseminations, often internal to iron sulfide minerals (pyrite is most common) or with carbonaceous material in the host rock. Free particulate gold, and particularly visible free gold, is not a common characteristic of these deposits; therefore, significant placer alluvial concentrations of gold are not commonly produced when Carlin-Type gold deposits are eroded. This is the case for the Carlin trend, and the reason why gold exploration (principally by gold panning) in the 1800's and early 1900's did not discover the Carlin gold district. The Carlin name comes from the Carlin mine, the first of this type gold deposit put into operation along the Carlin trend, in 1965. Since 1965, the Carlin trend has produced well over 70 million ounces of gold from an area 35 miles long by five miles wide. Total production and reserves is well over 100 million ounces.

Productive Carlin-Type gold deposits along the Carlin trend have gold concentrations as low as 0.02 oz/Ton Au, to a high of +1.0 oz/Ton. All the Carlin-Type deposits in Nevada have some general characteristics in common, although there is a wide spectrum of variants. Typically associated with the gold mineralization are anomalous concentrations of arsenic, antimony, and mercury. Other trace elements which can be pathfinder elements associated with Carlin-Type gold deposits are thallium, tungsten, and molybdenum. The magnitudes of trace element concentrations and their spatial relationship to gold are not necessarily direct, or 1:1. Alteration of the gold bearing host rock is typically exhibited by decalcification of the host, often with the addition of silica (jasperoidal silica is common), addition of fine-grained disseminated sulfide minerals, re-mobilization and or the addition of carbon to the rock, and with late barite and/or calcite veining. Small amounts of white clays (illite) can also be present. Decalcification of the host produces volume loss, with incipient collapse brecciation, which increases the fluid channel ways allowing for mineralization. Visible signs of alteration and mineralization can be subtle to the inexperienced observer. In any given hand sample an ore-grade sample can look much like barren waste rock. Assaying the rock is the key to defining mineralization, as with all gold deposits, but due to the lack of free particulate gold, Carlin-Type gold deposits generally do not have the "nugget effect" assay problem common in many other types of gold deposits.

Deposit configurations and shapes are quite variable, and are controlled by the overall host rock lithology, major plumbing structures and fractures, and by the porosity and permeability of the host rock as they may have been affected by decalcification or brecciation. Typically Carlin-Type gold deposits are somewhat stratiform, mineralizing characteristics being best exhibited in specific stratigraphic units or sub-units. Breccias can also be primary hosts to mineralization. Edges of deposits can be bounding faults, edges of brecciation, or imperceptible

porosity/permeability changes on a microscopic level. Orebody shapes can therefore be pod-like, tabular, flat lying or high-angle, or highly irregular and amoeboid in plan or section. Grades can vary greatly over short distances. Deposit size and grade vary greatly. Carlin-Type deposits in Nevada are mined from open pits as well as underground. Carlin-Type ores can be either oxide ores that are amenable to a direct cyanidation gold recovery process, or "refractory ores" which are either sulfide or carbon rich ores that require a pre-oxidation process (roasting or pressure-leach autoclaving) prior to cyanidation and gold recovery.

6.1 Geological Model

It is the author's opinion that the mineralization thus far identified on surface and in drill holes at the West Pequop Gold Exploration Property has many of the characteristics of oxidized Carlin-Type gold mineralization, and thus respresents a variant of the typical Carlin-Type gold deposit model. In particular the mineralization is associates with oxides (hematitic), has the As, Sb and Hg geochemical association, and the mineralization is associated with thinly-laminated silty calcareous units and fault structures.

7 Mineralization (Item 11)

Mineralization at the West Pequop property was first defined in surface rock and soil samples, with assays greater than 2.0 g/t Au in soils and greater than 10.0 g/t in rock samples, at the Acrobat target area. The exposures that are anomalous in gold are weakly oxidized (hematitic) and locally decalcified silty limestone. Multi-element geochemistry indicates an association of the gold mineralization with arsenic, antimony, mercury, and tungsten. Carlin-Type gold targets were the basis for the regional exploration efforts that led to the discovery of gold in jasperoidal silica at the Acrobat target; and the geological and geochemical data gathered to date confirms the mineralization defined at the West Pequop property as Carlin-Type gold mineralization. The Carlin-Type characteristics present at West Pequop are listed below:

- Associated As, Sb, Hg which is typical of most all Carlin-Type deposits;
- Associated low levels of anomalous W, which is present in some Carlin-type systems;
- Low to very low levels of Ag;
- Low levels of base metals;
- Hosted in-thin bedded silty carbonate sediments;
- Evidence of decalcification of carbonate units (sanding);
- Presence of jasperoidal silica replacement of the carbonate units. Jasperoids associated with gold bearing systems often exhibit a sugary granular texture in hand sample that is an interlocking mosaic (jigsaw) texture in thin-section microscopic analysis; as do the samples from West Pequop;
- Alteration and mineralization associated with both high-angle faults and selected strata;
- Lack of alluvial gold down-drainage from gold bearing outcrops lack of coarse grained gold;
- Stratiform and structurally controlled solution collapse breccias; and
- Weak to moderate oxidation and calcite veining.

The West Pequop property has a few different characteristics from typical Carlin-Type gold deposits.

- The common association of hematite with gold mineralization at West Pequop is not a common characteristic among most Carlin-Type deposits, and the intensity of hematitiic oxidation at West Pequop can be strong.
- The general location of the property is outside known gold deposit trends in Nevada.
- Host rocks here are Cambrian-Ordovician carbonates, whereas the majority of Nevada Carlin-Type deposits are in Ordovician-Devonian rocks.

None of these last three points are critically important in defining the type of mineralization present at West Pequop.

In the author's opinion, the mineralization encountered at West Pequop is most analogous to a Carlin-Type gold setting.

7.1 Mineralized Zones

7.1.1 Acrobat-Juggler-Section 34-Mountain Top Targets

The Acrobat target is the area of the original discovery of gold mineralization. Based on gold in soils, an anomaly is present at the Acrobat target that is at least 1000 ft in east-west length and 300 ft in width. Soil gold values exceed 2.0 g/t and rock chip and road cut samples within the anomaly are strongly mineralized with sample values up to 14.0 g/t Au. Road cut sampling in the anomaly returned 160 continuous feet averaging 4.1 g/t Au; in the vicinity of drill holes WN-6, 7, and 33. This is a high magnitude gold anomaly that has been the focus of drilling. The soil gold anomaly is relatively confined and would appear to be in part resulting from down-slope movement of material from the Feeder Fault. However, similar scattered gold in soil anomalies throughout Section 34 and at the Mountaintop target indicate a gold mineralizing system that covers a minimum of two miles in strike length and nearly the same distance east-west.

Drilling in the Acrobat area has defined significant gold mineralization in a number of drill holes. Mineralization is related to hematitic oxide zones in both stratiform zones parallel to bedding and/or bedding parallel solution collapse breccia, as well as adjacent to moderate to high angle structural zones with similar hematitic staining. Silicification in the form of jasperoidal silica replacement is present at surface and locally in drilling, but generally silicification is limited in extent. Mineralization at Acrobat is at surface and extends to depths of 250m, much of the mineralization is at depths of less than 150m from surface.

The Juggler target, an extension to Acrobat, is located less than 1000 ft to the south of the Acrobat target, and is of similar geology. Geologic units at the Acrobat-Juggler targets are thinly bedded silty limestones that dip east, into the mapped Feeder Fault. Outcrops and road cuts demonstrate that alteration and mineralization (correlative with hematite staining) is selective to bedding planes, suggesting that the mineralization at surface is leakage up structure and along favorable bedding horizons. Drilling in 2005 and 2009 demonstrated a connection to mineralization between Acrobat and Juggler.

At Section 34, the similar geologic setting exists, although the gold mineralization is hosted in different sub-units of the Cambrian carbonate sequence, and here the soil anomalies are at least 0.5 miles distant from the mapped Feeder Fault. In addition, there is a northeast alignment to the gold-in-soil anomalies in Section 34 that generally parallels the mapped F47 fault. Here as well, selective decalcification along bedding planes and bedding parallel solution collapse breccias has provided fluid pathways; evident by hematitic coloration and gold/trace element mineralization and locally bedding parallel jasperoidal silica replacement. Mineralization at Section 34 comes to surface and extends to depths of 350m, much of the mineralization is at depths of less than 250m from surface (see Figure 8.4).

The Mountain Top mineralization is hosted in the shallow sub-surface in Ordovician Pogonip limestone units. Here a strong gold soil anomaly is present on a steep slope immediately downhill from the mapped Mountaintop Fault. There are no outcrops of mineralization; however, shallow trenches have encountered high-grade gold values in oxidized rocks. The Mountaintop drilling has defined a small deposit of gold mineralization that is oriented with a shallow apparet dip to the northeast, and has a greater proportion of higher grades (+3.0 g.t Au) than at Section 34 and Acrobat. Mineralization at Mountain Top is essentially at surface, and has

been exposed in trenches comes, and extends to depths of 190m, much of the mineralization is at depths of less than 100m from surface.

Alteration and mineralization have been examined petrographically by Dr. L.T. Larson of Carson City, Nevada, including thin-section analysis as well as X-ray diffraction (XRD) and Scanning Electron Microscopy (SEM). At Acrobat, the rocks are marbles, silty limestones, and jasperoid. There is a strong correlation of gold with jasperoids which commonly display interlocking "jigsaw" textures and multiple stages of silicification. Silty carbonates exhibit weak foliations and micaceous minerals consistent with weak metamorphism of a dirty carbonate protolith. Fault breccias and gouge contain quartz, calcite, muscovite, and rare kaolinite in addition to the obvious hematite staining. Pyrite (and iron-oxide pseudomorphs) is rarely present in mineralized samples. Although rare, pyrite has been identified encapsulated in jasperoid, in limestone, and rarely encapsulated in calcite. Stibnite has been identified in core from Section 34.

7.2 Surrounding Rock Types

The surrounding rock types that bound mineralization are typically more massive limestone units such as the Decoy limestone at Section 34. The more massive limestone units are rarely or only locally mineralized.

7.3 Relevant Geological Controls

Relevant geological controls are both stratigraphic and structural. Stratigraphic controls are the typically thin bedded silty limestone lithological units, which include from Morgan Pass Formation (oldest), Shafter Formation, Oasis Formation, Candland Shale, and the Notch Peak limestone of Cambrian age, and the Pogonip limestone of Ordovician age.

Structural controls to mineralization are the north and northeast trending faults, and likely bedding plane shears and bedded breccias sub-parallel to bedding which generally is of shallow easterly dip into the range. Karst breccias, extensively developed in limestone at unconformities in the Paleozoic sequence, locally, are important hosts for gold mineralization.

8 Exploration (Item 12)

In the author's opinion, the historical work by PNGC at the West Pequop property was well planned, detailed in execution and documentation, and meets or exceeds industry standards for early to mid-stage exploration efforts. Historical exploration also included RC and core drilling by industry standard methods. That exploration was previously described in Section 4 (History). This section describes the current exploration method in use by Agnico.

8.1 Exploration Methods

Agncio has been exploring West Pequop since 2006 primarily with both RC and core drilling techniques. Earlier drilling was RC using Lang Drilling, and Eklund drilling was used in 2009. Core drilling was by Boart Longyear.

Exploration efforts on the West Pequop property have been successful in encountering potentially ore-grade gold mineralization in multiple drill holes in three separate targets, within a combined area extent of approximately 4 square miles. The three gold deposits are Acrobat (formerly Acrobat and adjacent Juggler targets), Section 34, and Mountain Top. Table 8.1 shown the extent of drilling that has been completed on the various targets at West Pequop.

Table 8.1: West Pequop Property Exploration Drillholes (SRK 2010)

ible 6:1: West I equop 1 toperty Exploration Diffinities (SIXX 2010)								
				Range of depths (m)				
Area	No. RC	No. Core	Total (m)	min.	max.	Average Depth (m)		
Acrobat	59	21	16,490	12	627	206		
NE Section 34	54	16	18,538	52	431	265		
Mt. Top	11	22	7,934	37	519	240		
Range Front	14	4	4,837	50	685	269		
SW Sec 34	4	1	1,430	242	305	286		
	1		1	1				

Individual drillhole intercepts are not here listed, as a summary of the 183 drillhole used in the resource database would by necessity be selective, extensive, and perhaps misleading. AuEx has published drilling results as news releases on its website (www.auex.com) The distribution of gold grades, is best shown in the cumulative frequency plots of all gold assays for Acrobat, Section 34, and Mountain Top (Figures 15-9, 15-10,and 15-11), as discussed in Section 15 (Mineral Resources). A discussion of core versus rotary drilling results is also presented in Section 15.

Figure 8-1 shows the plan map of drillholes for the three gold deposits. Figure 8-2 shows the drillhole plan map for Section 34; individual gold assay values greater than 0.0 g/t Au are shown color-coded. Figure 8-3 shows the plan map of drillholes for Section 34 with lines of sections 40m apart. Figure 8-4 shows a the cross-section through Section 34, and demonstrates the location of mineralized intercepts with respect to 2D sectional interpretations of mineralization limits – 2D string files provided by Agnico as mineralized domain boundaries. Similarly defined mineralized domains constrain mineralization at Section 34 and Mountain Top.

8.2 Surveys and Investigations

Early drillholes were relatively shallow drilling at Acrobat and Section 34, and downhole surveys were not done. A total of 49 early shallow and more recent shallow holes were not surveyed down-hole. For those holes, the collar azimuth and dip are assumed for the entire hole. Early surveying contractors were Scientific Drilling International, Elko, Nevada, and Silver State Surveys Inc, Elko, Nevada. Some holes show more than 10 degrees of deflection in azimuth and/or inclination. More recent holes were surveyed down-hole by International Directional Services LLC, of Chandler Arizona (IDS). IDS performed down-hole surveys for hole deviations using a tool that measures magnetic bearing and inclinometer readings. Typically, readings were taken every 50 ft (15.2m) down-hole. The bottom survey depth was corrected to match the total depth of the drillhole in the database.

Drillhole collars are surveyed in the field with GPS instrumentation and have been re-surveyed in 2009 to verify collar coordinates. There are differences in collar elevations with the digital topographic map that have not yet been rectified; typically a number of drillholes are above topo by up to 10 m in elevation (Figure 8-4). This discrepancy has a negative effect on a very small amount mineralization from surface to shallow depth in drilling that currently is above topo at Acrobat; SRK does not consider this to have any material effect on the total resource. SRK understands that Agnico will re-fly the project area in 2010 and prepare a new topographic map with which to compare collar elevations, and make appropriate corrections to the database, as necessary. Some of the collar elevation discrepancies for Acrobat may be due to the original collar site having been excavated (cut-down) by new drill roads, resulting in newer adjacent drillhole collars being 3 to 5 meters lower that an original nearby collar.

Drillhole collars have been surveyed in the field with differential GPS equipment.

8.3 Interpretation

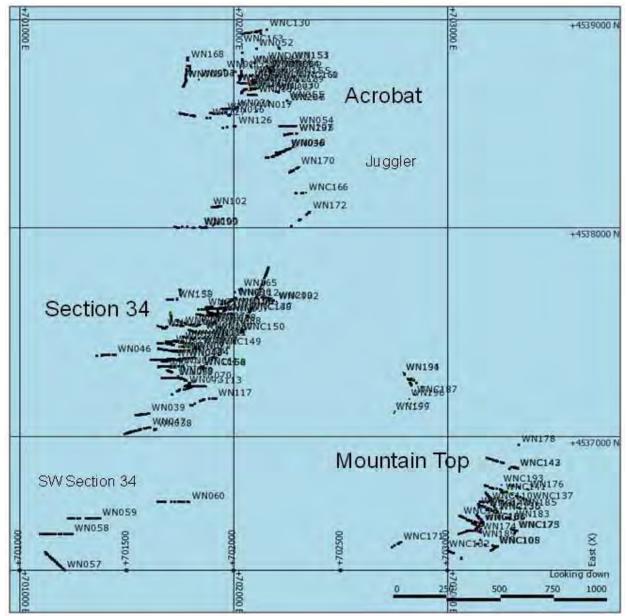
The mineralization at all three deposits begins at surface or very shallow sub-surface; therefore, they have the potential to be accessed by open pit mining methods.

Other exploration targets exist on the property, including the SW Section 34 and the Range Front target. As the three know gold deposits are essentially outcropping, there is obvious additional exploration potential for non-outcropping (blind) deposits at West Pequop

SRK conducted a site visit in 2009 during Agnico's active drilling program. SRK's opinion is that the methods employed for RC and core drilling are appropriate drilling methods for the mineralization, drillholes are oriented correctly to intersect the mineralization at West Pequop, and drilling and surveying methods are industry standard techniques for drilling, typical of exploration for Nevada gold deposits.

The exploration programs have generated information that is adequate to support the initial resource estimates presented in this report.

Figure 8-1: Drillhole Trace Map for Acrobat, Section 34 and Mountain Top Gold Deposits



Source: SRK 2010

Looking down

July 15, 2010

200

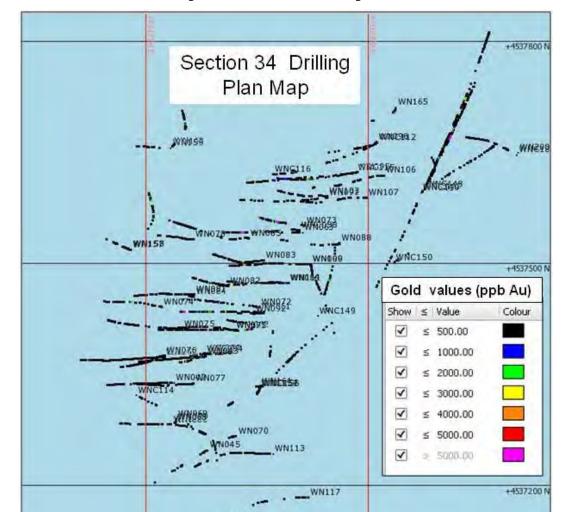


Figure 8-2: Drillhole Trace Map for Section 34 Gold Deposit

Source: SRK 2010

100

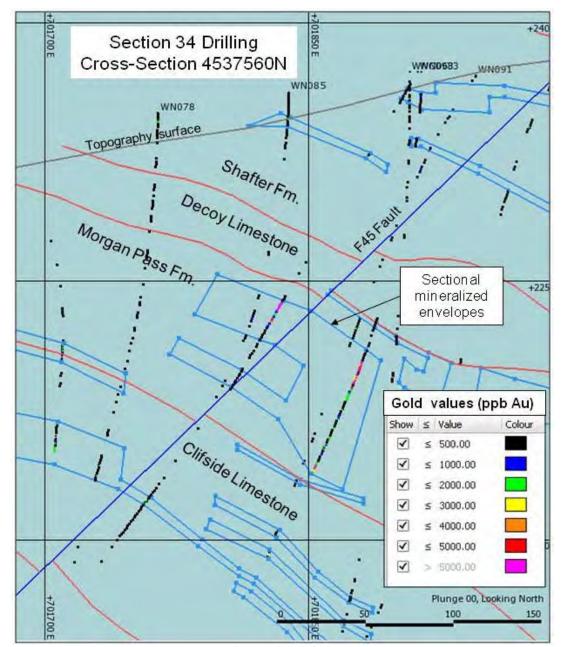
WN039

+4537800 N Section 34 Drilling Plan Map WN165 WANG 96 2 WNHS WN2982 IGTEN106 WNG116 Section 4537560N 40m spaced sections WN08 WN083 WNC150 +4537500 N Gold values (ppb Au) Show ≤ Value Colour ≤ 500.00 ≤ 1000.00 WNOWN077 ≤ 2000.00 WNEES ≤ 3000.00 ≤ 4000,00 ≤ 5000.00 5000,00 +4537200 N WN117 Looking down WN039

Figure 8-3: Drillhole Trace Map for Section 34; Showing Cross-Section lines

Source: SRK 2010

Figure 8-4: Section 34, Cross-Section 4537560N



Source: SRK 2010

9 Drilling (Item 13)

9.1 Type and Extent of Drilling

Drilling methods employed at the West Pequop property have been a combination of reverse circulation (RC) and core; both of which are typical of standard gold exploration practices in Nevada.

RC drilling has been conducted at all target areas. Drilling procedures have been to collect cutting samples from each hole on 1.52m (5 ft) sample intervals, utilizing a rotary sample splitter attached to the output from the cyclone sample collector on the drill rig. Most of the drilling was done with water injection to reduce dust at the drill site, and prevent loss of material as fines in the dust coming out of the cyclone stack. The rotary splitter, whether wet or dry, uniformly splits the sample volume per five-foot interval to a manageable size, approximately 10-20 lbs in average weight. Samples were bagged and sent directly to the analytical labs for preparation and analysis. The RC drilling contractor used by PNGC was Eklund Drilling Co. Inc., Elko, Nevada. Agnico has used Lang Exploratory Drilling and Eklund Drilling

Core drilling has been done for some of the holes on all three gold deposits. The PNGC core drilling contractors were Connors Drilling Inc. Montrose, Colorado, and Tonto Drilling Services Inc., Salt Lake City, Utah. Drill core is HQ size, 6.5 cm (2.5) inch diameter. Core was boxed, moved offsite, and cut in half with a diamond saw. Sample intervals were commonly at five-foot intervals, but occasionally geology dictated differently; so sample intervals can vary from 0.5m to 3.0m (1.5 to 7.0 ft). Half-core sample intervals were bagged and delivered to the analytical lab, with the remaining half-core sample retained in storage. Prior to cutting, boxes of whole core were photographed, although RQD (rock quality derivation) structural information was not gathered, based on the several core logs examined. Agnico has used Boart Longyear for core drilling.

Both core and RC samples are geologically logged for lithology, alteration, mineralization and structure (core). Logs are adequate to describe the rocks for use in deposit modeling.

9.2 Results

It is the author's opinion, and with personal knowledge of the various contractors, that the drilling methods employed by PNGC, AuEx, and Agnico at the West Pequop property are adequate to define the mineralization, and appropriate industry standard practices.

10 Sampling Method and Approach (Item 14)

10.1 Sample Methods

Drilling has been exploration drilling, attempting to determine the controls on and limits to mineralization.

The approach in drill sampling was to sample at reasonable intervals to adequately define the mineralization at West Pequop. A 1.52m (5 ft) drill sample interval is industry standard practice for RC drilling in Nevada. Variable core sample intervals are also an industry standard practice in Nevada. The mineralized intervals encountered in drilling at the West Pequop property are commonly composed of several 1.52m (5 ft) sample intervals; suggesting that the sample interval is indeed sufficient to determine average grades over the widths of mineralization encountered.

Section 9.1 describes the samples taken during drilling. The resource estimation is based entirely on drillhole sampling and the sampling methods and approach are acceptable

Surface sampling methods were several. Rock chip sampling was conducted as random chip sampling of selective rock outcroppings, and as continuous chip samples along the outcrop or sub-outcrops in road cuts. Various sample intervals were used.

Soil samples were collected initially on 200 by 200 foot grids over target areas, with in-fill to 100 by 100 foot spacing where required to further define geochemical anomalies. In Nevada, there is typically very little to no A-horizon soil development, so soil samples are B-horizon mixed rock and soil, and represent a fairly uniform sample medium for hill slopes, as is the case for the West Pequop property.

BLEG and stream sediment sampling were initially done to define the areas for follow-up soils and rock chip sampling. Both are typically taken in dry wash drainages or base of slope settings, and the -30 and-80 mesh fractions were collected for analysis.

It is the author's opinion that the sampling methods and approach used by PNGC, AuEx, and Agnico at West Pequop are appropriate for the mineralization of interest, the topography, and the geologic setting in northern Nevada; resulting in reasonably good sample quality.

10.2 Factors Impacting Accuracy of Results

RC sampling can sometimes be prone to sample loss or sample contamination, depending upon drilling conditions, and SRK recommends further work to verify RC sampling provided sufficient sample quality in comparison to core drilling.

10.3 Sample Quality

Core recovery is typically +90%, resulting is overall good to excellent sample quality. RC sample quality is difficult to determine. RC sample quality depends upon the sample volume recovered, wet or dry, and can be measured by weighing each sample, a practice that is rarely done unless sample quality is a known concern. RC samples for West Pequop were not weighed.

10.4 Conclusions

One measure of RC sample quality is to compare RC versus core sample results for drilling in the same deposit. SRK has shown a comparison of RC versus core sample assays globally for the West Pequop project in Section 15.2 (Drilling database). Further investigations on each deposit and comparing nearby holes is recommended prior to future resource estimation, as Figure 15-8 indicates a relative high bias of RC over core assays on a cumulative frequency distribution curve.

At this point in time, there is nothing to indicate the RC samples are indeed biased high, until further detailed work is done. The differences may be due simply to where the hole were placed relative to mineralization (location bias). The quality of RC and core samples are considered adequate for the initial resource estimate presented in this report.

11 Sample Preparation, Analyses and Security (Item 15)

11.1 Sample Preparation and Assaying Methods

Specific documentation relating to PNGC sample preparation is lacking; however, personal communication with the former PNGC drill project coordinator (S. Green) provided the following information:

- RC drill samples were collected at the drill site by PNGC staff, as splits from a rotary wet splitter.
- RC samples, approximately 15-20 pounds, were bagged and transported to a staging area
 on the project, where samples were picked up by AAL and transported to the AAL
 sample preparation lab in Elko, Nevada.
- B-splits of RC samples were not collected, but coarse rejects (a split at the assay lab prior to pulverization) were retained for many RC holes. Pallets of coarse rejects from some holes are in the possession of Agnico at their field office in Wells, Nevada.

Core samples were collected at the drill site by PNGC staff, transported to a core cutting facility in either Wells or Reno, Nevada, logged, marked for sampling, photographed, and diamond-saw cut as half core to those marked sample intervals. Half core has been retained in original core boxes; and those core boxes are now in the possession of Agnico at their field office in Wells, Nevada. Cut and bagged samples of core were shipped to American Assay Lab's sample preparation facility in Elko, Nevada.

Work by AuEx and Agnico follow similar procedures for sample collection and preparation. Agnico is collecting core at the rig, and transporting it to secure facilities in Wells, Nevada for core logging, photographing, core cutting, sample tagging, and shipment to the analytical lab. RC samples are bagged at the rig and shipped directly to the analytical lab for processing.

Sample security was maintained from sample collection in the field by PNGC, AuEx, and Agnico staff, to delivery of samples to the various analytical labs.

11.1.1 Testing Laboratories

Analyses for early geochemical samples during the period 1994 to 1996 utilized Acme Analytical Lab, 852 E. Hastings St., Vancouver, B.C., Canada; and Rocky Mountain Geochemical Corp. 1323W 2900S, West Jordan, Utah. Both are reputable analytical labs, known to the mineral exploration industry. Geochemical analyses were typically for gold by fire-AA analytical techniques, and 30-element ICP analyses for major and trace elements.

In 1996, going forward, all samples were analyzed at American Assay Laboratories Inc. (AAL), 1500 Glendale Avenue, Sparks, Nevada. AAL's standard sample preparation procedure is to crush the entire sample (1/2 core or RC chips) to 8-10 mesh size, and split the material to 1/16th volume in a riffle splitter. The 1/16th split is pulverized to a nominal 150 mesh size fraction. AAL's standard gold assay procedure is a 30 gram charge fire assay with an AAS finish. This is a standard fire assay preparation, with AAS (atomic absorption spectro-photometery) analysis of the resultant fire-prepared bead. As well, AAL performed 30-element ICP analyses for major and

trace elements. AAL has a Certificate of Compliance to the ISO/IEC 17025 Standard of Quality; issued through Global QA Inc. AAL is a reputable analytical lab that has been servicing the mining and exploration industry of Nevada for the past 18 years.

Agnico is using the Inspectorate America Corporation (Inspectorate) analytical lab, 605 Boxington Way, Sparks, Nevada. Inspectorate's sample preparation protocols are presented in Figure 11-1. ABS Qualtiy Evalutations, Inc., certified Inspectorate with ISO-9001:2000 certification. Inspectorate is an internationally known analytical laboratory that provides assay services to the exploration and mining industry.

All samples were analyzed for gold and the majority of drill holes have either 30-element ICP analyses for individual samples, or ICP analyses at 20 ft intervals.

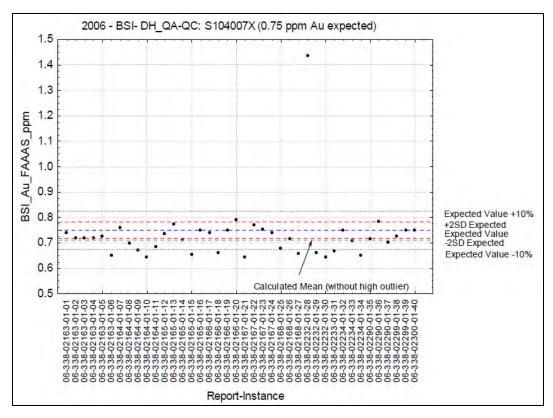
11.2 Quality Controls and Quality Assurance (QA/QC)

Agnico has in place a program of insertion of standard, blank, and duplicate samples for assay along with the standard stream of drillhole samples. An independent evaluation of the data has been completed by an outside consultant (Blair, 2010), examining the data from drilling campaigns in 2006, 2007, 2008, and 2009. Data are examined statistically and with scatter plots and box plots to show the variances. Excerpts from the Blair report, by year, are presented here along with a few sample plots:

2006

The primary assay laboratory during 2006 was Inspectorate Precious Metals Laboratory (BSI) with sample preparation and precious-metal analyses in Reno, Nevada. Gold (Au) assay methods were standard fire-assay with AAS (FAAAS) metal determination on a 30g sample charge. Samples with original assay greater than 3 ppm Au were re-assayed using a fire assay with gravimetric metal determination (FAGRAV). A total of 12 assay reports from 11 drill holes were completed during 2006. These reports contained 1812 samples including 230 quality control samples (standards and duplicates). Reference sample results show a slight low bias for the 2006 period. Check assays show some low grade bias to the original assays for grades greater than 1.5 ppm Au. Additional check assaying is recommended.

Standards submitted during the period were gold standards supplied by Shea Clark Smith and Minerals Exploration and Environmental Geochemistry (MEG) of Reno, Nevada. Blank material was also supplied by MEG and is coarse (~0.5 inch) crushed landscaping rock (welded rhyolite tuff) packaged in 3Kg bags. The MEG standard results used by the project show much variability when compared to the characterization data supplied by MEG. For this reason, both the +/- 2 standard deviation limits on the expected value, supplied by MEG, and +/-10% of the expected value are plotted on the control charts and used as performance measures. Results for the Blank for the period are acceptable with only one of 75 instances of the blank with anomalous metal.

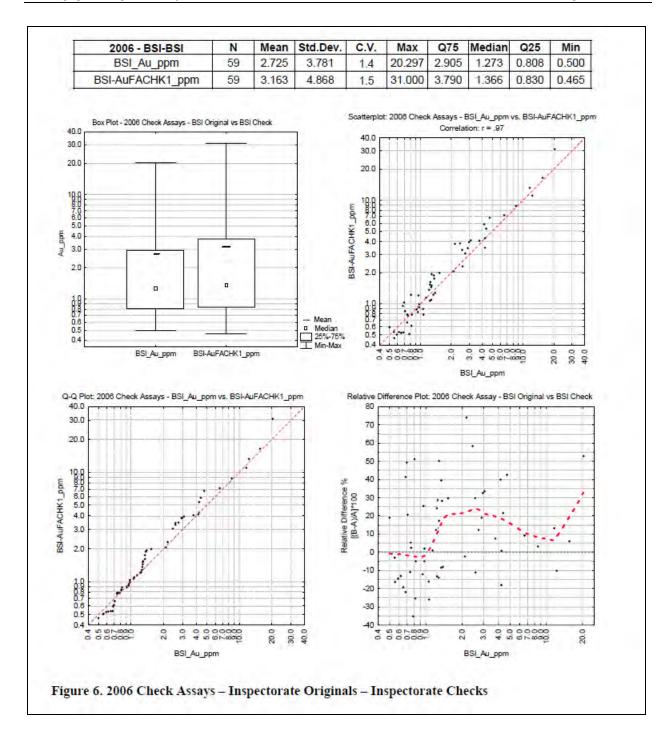


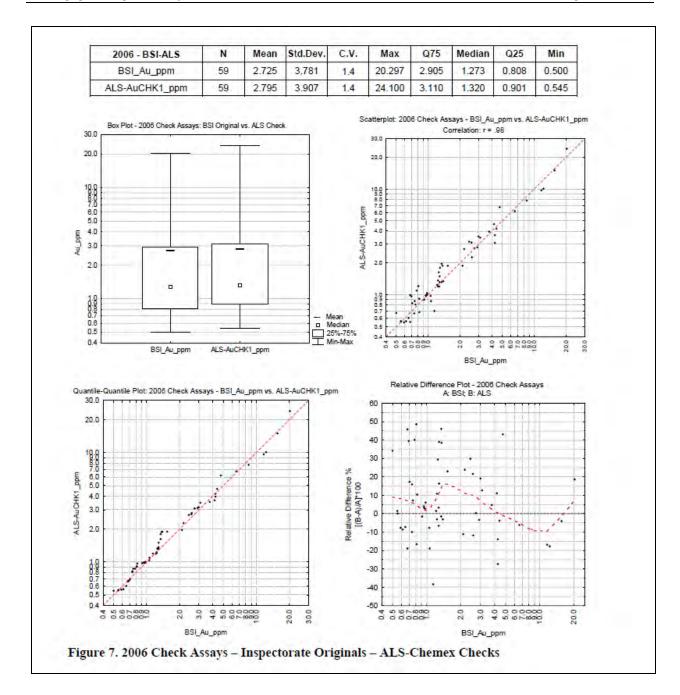
Source: Blair, 2010

As part of the reverse circulation drill sampling process, a second split is collected at regular intervals and labeled as the "A" and "B" duplicates. There were 72 duplicate samples collected in 2006; all were sent to Inspectorate for analysis. Comparative sample statistics are summarized in Table 6 and a scatter plot of the data is presented in Figure 5.

The original and duplicate sample assays agree fairly well; however, there are relatively few duplicate samples at significant Au grades (> 0.2 ppm Au). A definitive statement regarding split sampling and assaying cannot be made.

There were few check analyses completed for the 2006 drilling. Fifty-nine samples from 6 drill holes (WN58, WN59, WN61, WN62, WN63, and WN64) and six original reports (06-338-02164-01, 06-338-02165-01, 06-338-02167-01, 06-338-02168-01, and 06-338-02232-01, 06-338-02234-01) were checked. These samples were sent to Inspectorate and ALS-Chemex (ALS) in Reno, Nevada for check assay using the same assay protocol (1AT/FAAAS/FAGRAV). The results for the Inspectorate check assays are summarized in Figure 6; results for the ALS checks are summarized in Figure 7





The Inspectorate check assays show good agreement with the original assays for grades up to approximately 1.5 ppm Au; above this level the check assays show a high systematic bias of 10% to 20%. The ALS check assays also show similar results to the originals up to approximately 1.5 ppm Au with a "spike" of high check samples at and around the 1.5 ppm level. Above approximately 2 ppm Au, the assays agree well with the expected variability for this grade level. Both sets of check assays suggest a possible low bias to some of the original assays greater than 1.5 ppm Au. Additional check analyses are recommended for the 2006 assay information.

SRK notes that Similarly, QA/QC data were evaluated for 2007 through 2009, and the results presented with similar data plots. A brief summary of the conclusions are presented here as excerpts from the Blair report, without the supporting plots.

2007

The primary assay laboratory during 2007 was Inspectorate Precious Metals Laboratory with sample preparation and precious-metal analyses in Reno, Nevada. Gold assay methods were standard fire-assay with AAS metal determination on a 30g sample charge. Samples with original assay greater than 3 ppm Au were re-assayed using a fire assay with gravimetric metal determination. A total of 75 assay reports from 39 drill holes were completed at Inspectorate during 2007. These reports contained 7466 samples including 707 quality control samples (standards and duplicates). All assay reports and associated drill holes are listed in Appendix I. Reference sample results for 2007 show a slight high bias. Check assays show satisfactory agreement with original assays. Additional check assaying for the 2007 assay reports is recommended.

Results for the blank are acceptable with 7 of 346 instances with detectable metal; 3 instances were greater than 3x the detection limit. All reports with blanks with detectable metal were examined and only one appears to be related to possible contamination: sample 479530 from report 07-338-03165-01. This sample occurs just following 5 samples with grades greater than 10 ppm Au at the end of the report. The contamination, if any, is not significant at these levels. Samples were selected from this mineralized zone, including the blank, for check analysis (Appendix II). All other anomalous instances are from low-grade reports with few samples with assays greater than the detection limit. No other action is recommended.

There were 320 duplicate samples collected 2007The original and duplicate sample assays agree well with a slight tendency for the "B" sample to be higher grade in the 0.05 to 0.3 ppm Au range, which is not significant from a grade estimation point of view. Above 0.3 ppm Au level, there are relatively few sample pairs (13); however, the assays agree well for the grade ranges represented.

2008

Assay laboratories used during 2008 were Inspectorate Precious Metals Laboratory with sample preparation and precious-metal analyses in Reno, Nevada and American Assay Laboratories (AAL) with sample preparation in Elko, Nevada and precious-metal analysis in Reno, Nevada. There were many standards during 2008 that were not labeled or incorrectly labeled in the QC sample database received; the corresponding standard identifications were inferred from the gold and multi-element analyses. All modified standard identification records are listed in Appendix III. Reference sample results for the 2008 period show slight low biases for both primary laboratories. Check assays show satisfactory agreement with the original assays. Additional check assaying for the 2008 assay reports is recommended.

Gold assay methods at Inspectorate were standard fire-assay with AAS metal determination on a 30g sample charge. Samples with original assay greater than 3 ppm Au were re-assayed using a fire assay with gravimetric metal determination. A total of 48 assay reports from 23 drill holes were completed during 2008 at Inspectorate. These reports contained 4686 samples including 581 quality control samples (standards and duplicates).

Results for the blanks are acceptable with no instances with detectable metal.

Gold results by FAAAS for S104007X (SRM) show a slight low bias to the calculated mean and 11 of 60 instances outside the +/-10% limits (3 above and 8 below). All reports were examined and all were found to be low-grade with few samples greater than the detection limit or other reference samples in the same report were within tolerance. With the extreme low outliers removed, the results are acceptable with a tight spread of assays around the expected value. No action is recommended.

The duplicate sample assays show satisfactory agreement; although there are few duplicate samples at significant grades (>0.2 ppm Au).

Replicate assays are repeat assays done by the primary assay laboratory on the original pulp as part of their internal QC procedures. Replicate assays were reported for the FAGRAV assays only.

Samples were selected from the 2007 and 2008 Inspectorate reports for check analyses. Check sample batches were sent to Inspectorate and ALS-Chemex.It appears that the original assays were selected by a filter of greater than 0.3 ppm Au; the check assays show an anomalous group of samples where the check assay is less than 0.3 ppm. Above the 0.3 ppm level, the assays agree fairly well with some other anomalous assay pairs that should be reviewed. A subset of these original check samples were also sent to ALS-Chemex for check analysis. These checks agree well with the original assays for grades greater than approximately 0.2 ppm Au. Additional significantly mineralized intercepts should be selected from the 2007-2008 drill holes and BSI reports for further check assaying.

2009

The primary assay laboratory for 2009 was American Assay Laboratories with sample preparation in Elko, Nevada and assaying in Reno, Nevada. Gold assay methods were standard fire-assay with AAS metal determination on a 30g sample charge. Samples with original assay greater than 6 ppm Au were re-assayed using a fire assay with gravimetric metal determination. A total of 117 assay reports from 57 drill holes were completed during 2009 at AAL. These reports contained 11478 samples including 1068 quality control samples (standards and duplicates). Reference sample results show a slight low bias to the 2009 standard assays. There are very few check assays for the 2009 assay reports. The valid check assay pairs show satisfactory agreement; however, many more check assays are required for final database approval.

In 2009, the "A" and "B" duplicate samples were sent to separate laboratories. The "A" sample was sent to American Assay and the "B" sample was sent to ALS-Chemex. The duplicate sample results show good agreement with the original AAL assays showing a slight low bias to the ALS assays on sample B up to approximately 0.06 ppm Au. This difference is not significant when looking at cross-laboratory duplicate sample information.

SRK notes that the conclusion from the Blair report on QC suggests additional check assays be done; but no significant variances were noted to cause concerns with the assay database.

Inspectorate has internal lab QC sample insertions as well, as further stated in their overview of analytical procedures as shown on Figure 11-2.

11.3 Interpretation

It is the author's opinion that the sample preparation procedures, analytical procedures, and analytical labs used are all appropriate for the mineralization at the West Pequop property, and they are typical industry standard procedures. Quality control procedures are in place with Agnico's field sample collection, and internally within Inspectorate's lab, which are deemed sufficient to provide assurance of the assay data quality of the West Pequop drillhole database.

In SRK's opinion, the sample preparation, sample security, sample assays and QA/QC procedures are adequate, and validate the drillhole database.

Figure 11-1: Inspectorate Sample Preparation Procedures Flow Chart

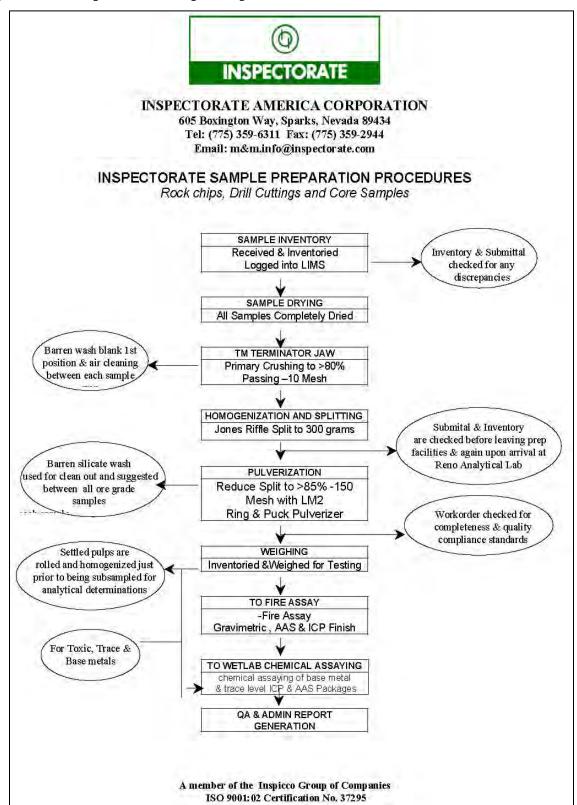


Figure 11-1: Inspectorate Sample Preparation Procedures Flow Chart



Inspectorate America Corporation 605 Boxington Way, Suite 101, Sparks, Nevada Tel: (775) 359-6311 Fax: (775) 359-2944

Overview Inspectorate's Preparation & Analytical Protocol

Sample Preparation:

The entire sample will be thoroughly dried prior to crushing. Samples will then be reduced to >80% -10 mesh using a TM Terminator Jaw Crusher. A 300 gram split will then be obtained using a Jones riffle splitter and reduced to >85% -150 mesh with a LM2, R&P pulverizer. Clean sand is utilized in pulverization for cleaning and can be employed between each and every sample, which is recommended for ore grade materials, further reducing the possibility of any cross sample contamination.

Analytical Procedures:

All of the Inspectorate's analytical determinations will be performed at our new 30,000 sq. ft. facility in Sparks, Nevada. Our technical and professional staff is located here and available to discuss your questions, comments and methodology options. A minimum fifteen percent of all analyses performed are directly run for quality control. Every tenth sample is repeated and for every 20 samples run, a standard or blank is also analysed. For gold determinations a total of 9 certified gold standards purchased through 2 separate manufacturers (Rocklabs and CDN Resources) are implemented into our fire assay QC program for gold analyses finished with both gravimetric and AAS methodologies. Six standards ranging from 0.79 g/ton to 20.77 g/ton were purchased from CDN and three different standards from Rocklabs varying in matrix from oxides to sulfides and ranging from 0.651 g/ton to 2.643 g/ton. These CRM's are utilized in addition to our manufactured internal gold standard running approximately 1 gram per metric ton. Inspectorate's trace element determinations will be performed using optimal acid digestions followed by Atomic Absorption Spectroscopy & Inductively Coupled Plasma, Our wet-chemical assaying procedures finished with either AAS or ICP similar QC applies with insertion of a CANMET poly-metallic standard inserted in place of the gold standard utilized in fire assay. This wet-chemical CANMET standard is deemed representative across poly-metallic environments inclusive of Ag, Cu, Pb, Zn, Mo, Cd as well as several other toxic and associated elements. In addition to this standard quality control, selected high and low values are rerun as checks. Of course, the number of checks or reruns is highly dependent upon each individual batch of samples. However, it is safe to say that your quality control will always exceed 15 percent and mineralized work-orders average 20-30 percent. Our internal quality control values can be reported along side your mainstream samples in certificates, upon your request. Analytical results may be reported in ppb/ppm, grams, opt and %. All limits of detections can be found in our geochemical & assay schedule of fees. Additionally, priority is always given to any samples that exceed our upper limit of detection for a particular procedure. Thus, "over-limit" samples are quickly identified, reanalysed using assay grade procedures and reported within 24-48 hours of any preliminary data. We have adopted this procedure in order to better service our clients as well as reduce costs and overall turnaround time. If further information is required on any of our services, please consult our Sparks, NV laboratory

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12 Data Verification (Item 16)

12.1 Quality Control Measures and Procedures

Assay data verification has been accomplished by several methods.

- Standards and blanks submitted by PNGC;
- Assay re-runs by the labs;
- Spot sampling checks by AuEx, Inc.;
- Spot sampling checks by the author of this report;
- Confirmation drillhole gold assays by multiple companies with different drilling programs;
- A comprehensive quality assurance/quality control (QA/QC) program by Agnico involving evaluation of standard reference material (SRM), blank sample assays, duplicate assays, and outside lab checks

PNGC submitted into their sample submission stream, standards and blanks, beginning with the earliest BLEG sampling programs through drilling. This effort did not identify any major problems with the commercial lab analyses. During the early drilling programs, most submitted sample standards were within 5% of the known values. Duplicates submitted to outside labs were within 5% of the original assay.

AAL routinely and randomly re-runs analyses within sample batches, as internal duplicate assays; and it was noted that for the assays certificates examined, the sample analyses compare reasonably well (visual spot check).

AuEx, Inc. has collected 51 surface samples to verify gold mineralization, confirming gold up to 19.95 ppm (g/t) Au, as did the author (A. Moran 2005) collect spot samples of outcrops to verify gold and trace element analyses, verifying the presence of gold mineralization.

Verification of drill hole collar locations was done through spot checks in the field with handheld GPS units; only a few holes were checked with reasonable accuracy in verifying drillhole locations. Recent drill hole collars are marked in the field with a rebar and marked aluminum cap set in concrete as part of the hole plugging procedures. During the field visit, the author noted drill hole collars were located on the ground where they appear to be on the maps.

There is no specific documentation regarding the procedures used to verify the accuracy of the assay database, such as assay double entry. Spot checking of the assay database against the original assay certificates for several drill holes in three separate targets, showed no database errors. SRK did not perform a complete audit of the drillhole database against the assay certificates; it was deemed not necessary based on spot checks.

Probably the best data verification has been the drilling of gold at Acrobat, and Section 34 by PNGC was verified by AuEx drilling in 2005, and has been confirmed by Agnico drilling since 2006. SRK did not conduct a detailed examination of the various drilling programs; however, it

has been visually confirmed that similar grade drill holes have been drilled by each company in both resource areas.

Agnico has in place a QA/QC sample data verification program, which has been ongoing since 2006, and includes:

- Standards (SRM);
- blanks;
- duplicate RC samples collect at the rig;
- and checks assays between AAL

A summary of the Agnico QA/QC program is described in section 11.2. Some assay bias was noted between AAL and Inspectorate at partial grade ranges, and there were occasional sample outliers values in the standard, blank, and duplicate assays; however, the data do not warrant any corrections to the database. Additional check assays have been recommended to further validate the database.

12.2 SRK Conclusions and Recommendations

It is the author's opinion that the assay data for drillhole database at West Pequop is accurately represented. A QA/QC program using standards, blanks, replicate samples, duplicate assays, and secondary lab check assays, has been in place by Agnico since 2006. There are no known issues with the QA/QC data. As there are now three defined mineral resources at West Pequop, albeit dominantly of Inferred classification, SRK has the following QA/QC recommendations for the project in advance of future resource estimations, to the extent that they may not have already been done by Aginco:

- Conduct a thorough assay database verification to include:
 - Comparison of each drillhole in the database with the assay certificate. Blind double entry of the data from the assay sheets can be done as a means to accomplish this
 - o It is recommended this work be done independently for certification
 - o Document all procedure, database errors, and corrections made
- Verify drillhole collars with topography; adjust as necessary and document adjustment
- Verify down-hole survey data are entered and plotted correctly, and document verification
- Conduct a thorough evaluation of RC versus core drilling assays from the existing drill database. (see recommendation(Section 18.1) for RC-Core drilling to examine this in twin-hole data)
- Conduct statistical evaluation of the various drilling campaigns, PNGC, AuEx, and Agnico for any potential bias, particularly in RC drilling, and document
- Further evaluate the identified bias of AAL versus Inspectorate assay data, after additional check assays are compeleted.

Complete database verification as above is recommend to provide confidence in future resource estimations which may become the basis for potential mine development. Fully documenting all aspects of the process allows for ease of third party reviews.

13 Adjacent Properties (Item 17)

The West Pequop property is immediately adjacent to and west of the Long Canyon Gold Exploration Project; a property that is currently undergoing continued gold exploration and predevelopment drilling. The Long Canyon property is held by Fronteer Group(51%) and AuEx (49%): a property that was unified with the West Pequop property under AuEx 100% control until 2006. The Long Canyon Property immediately adjoins the West Pequop property on the east. Long Canyon contains established mineral resources that are compliant with CIM definitions of mineral resources.

The Long Canyon resources as of May 2010 (AuEx News Release, May 19, 2010) are presented here as the mineralization at Long Canyon is similar to that at West Pequop. The Classified Mineral Resource estimate for Long Canyon is quoted at a cut-off grade 0.2 grams per tonne ("g/t") and consists of:

- A Measured and Indicated resource of 672,000 ounces of gold at an average grade of 1.71 g/t gold (12,240,000 tonnes); and
- An Inferred resource of 552,000 ounces of gold at an average grade of 1.65 g/t gold (10,394,000 tonnes)

The Long Canyon mineral resources are hosted in the Notch Peak limestone, and overlying Pogonip limestone, and in breccias of both units. The Long Canyon mineral resources are located approximately 3 miles east-northeast of the Mountain Top deposit, which is also located in Pogonip limestone. Both the Long Canyon and West Pequop gold resource areas constitute the newly emerging Pequop Gold District.

Is it important to note that while the Long Canyon gold deposit is immediately adjacent to West Pequop and of similar geology, the resource stated above for Long Canyon, and any comparison to West Pequop, is in no way indicative that a mineral deposit of similar size or grade does occur or will be found on the West Pequop Property.

14 Mineral Processing and Metallurgical Testing (Item 18)

There has been no preliminary metallurgical testing done for the gold mineralization at West Pequop, other than a few cyanide-soluble gold assays by AuEx in 2005. While the West Pequop property is still an exploration property, it is not too early to consider some preliminary metallurgical testing on gold mineralized intercepts. This section therefore provides some recommendations for preliminary metallurgical work to be contemplated as the project moves forward.

As part of the AuEx drilling program in 2005, several intervals from the Acrobat deposit were analyzed for cyanide-soluble gold (Au-CN) compared to 30 gram Fire-AA assay (Au-FA30) by AAL labs. The ratio provides an initial impression as to the cyanide solubility of the gold in drillhole samples. The indicated cyanide-soluble gold to total-gold ratio ranges from 73% to 98% in sample pulps. The results are shown in Table 14.1. Therefore, it is presumed that the highly oxidized (hematitic) brecciated and/or decalcified silty limestone that carries gold values at West Pequop may indeed be amenable to cyanide leach solutions. Similar demonstrated recoveries are noted from similar rock types, alteration, and mineralization at the adjacent Long Canyon gold deposit; however, cyanide-soluble gold assays and ratios need to be verified with dedicated metallurgical lab tests.

Table 14.1: AuEx Cyanide-Soluble Au

Hole ID	From (m)	To (m)	Interval (m)	Total-Au Au-FA30	Cyanide-Soluble Au Au-CN	Au-Cn/Au-FA30 Ratio(%)**
WN050	0.0	15.2	15.2	2483	2201	89%
	3.0	6.1	3.0	5195	4491	83%
WN053	118.9	143.3	24.4	1234	963*	78%
	118.9	120.4	1.5	5440	4720	87%
WN056	3.0	15.2	12.2	2812	2565	91%
	7.6	9.1	1.5	7000	6430	92%
	22.9	39.6	16.8	4015	3238	81%
	25.9	27.4	1.5	9800	9580	98%
	29.0	30.5	1.5	7300	7000	96%
WN054	146.3	167.6	21.3	926	815*	88%
WN055	132.6	161.5	29.0	1598	1167*	73%
	134.1	137.2	3.0	6120	5700	93%
	87%					

Notes:

- * missing samples for Au-Cn used the average for the rest of the interval
- ** using arithmetic average of the sample values

SRK recommends the following as initial metallurgical work that should be considered:

 A suite of drill sample rejects and or pulps should be re-run for cyanide-soluble gold, for each of the three deposits. This can be done for selected samples that have a range of anomalous gold values. The initial work by AuEx in 2005 was primarily for samples with gold grades greater than 1.0 g/t;

- Bottle roll cyanide leach tests should be done for a series of samples or sample composites of pulps from each of the three gold deposits, to verify cyanide solubility and thus the potential for gold recovery by standard industry processing techniques. Core intervals of gold mineralization can be split again (quartered) to provide sufficient coarse material for simple bottle roll cyanide soluble tests; while still maintaining some of the original core interval for future reference;
- Carbonaceous material in any mineralized zone should be examined for active carbon "preg-robbing" characteristics. Loss in Ignition (LOI) analyses can provide information on carbon content which may have metallurgical importance.
- Column leach tests can also be done for an approximation of the recovery rate and the ultimate recovery that might be expected from a heap leach processing of West Pequop gold mineralization; run on each deposit's primary host lithologies and at more than one grade range.

The lack of significant metallurgical test data is not a particular concern at this early stage of the project.

15 Mineral Resources and Mineral Reserve Estimates (Item 19)

15.1 Background

In August 2009 SRK estimated resources for the West Pequop deposits using the data available at that time and provided AuEx with an internal "Report on Resources through August 18, 2009 West Pequop Gold Exploration Project" – the report was dated December 16, 2009. At that time it was determined that there were database issues that rendered the resources Inferred at best in classification, primarily the lack of accurate Z values (elevations) for drillhole collars. A significant number of drillholes were noted to be ± 10 m off (above and below) topography, typically higher, and some were significantly above topography. SRK prepared a separate memo identifying some of the collar elevation problems.

SRK understands that a re-survey of drill collars has been completed and these database issues have been addressed in that all "located" holes have been re-surveyed. Many of the collar elevation problems have been resolved; however, the issue of drillhole collars being ±10m off (above and below) topography, typically higher, remains a problem. SRK understands this will be rectified and collars will be more accurately tied to topography when a new topographic surface from a new aerial survey is constructed this year. SRK used the topographic surface, as provided, to create the 2010 models recognizing the inherent potential volumetric errors at the topographic interface, which affects a small amount of mineralization that is essentially at surface (particularly for Acrobat).

Since SRK's August 2009 estimate, new assays have been made available for drilling through the end of the 2009 field season. In addition, Agnico has provided "mineralization strings", created as section interpretations, as discussed in section 15.2 below.

The deposits have variable amounts of additional assay data newly available since August 2009. Section 34 has two new holes with intercepts (WNC182 and WN200), encountering minor incremental additional mineralization in the northeast. Acrobat has seven new holes three of which (WN195, WN197 & WN201) serve to further define the Juggler extension to the south and two of which (WN204 & WN205) help in-fill delineation between Juggler and Acrobat. WN202 also provides in-fill information. Mountain Top has four new holes however, three of which, WNC193, WN185 and WN189 only encountered lower grade mineralization.

15.2 Geological Controls, Mineralization Envelopes

The mineralization strings constructed by Agnico identify significant mineralized material on two-dimensional sections. Agnico reports that these were constructed in the following manner:

"The mineralized shapes were created by connecting reasonably apparent mineralized drill-hole intervals together. The shapes were further fashioned to be concordant with lithology where the mineralization is strata bound or to be shaped/restricted by folding or faulting. In the cases of mineralization restricted to structural zones the drill information pretty much dictated that mineralization has a restricted zonation about a structure and whether that placement was in the footwall or hanging wall of the structure. At times mineralization might also appeared to cross through a structure without regard to anything more than lithology. The mineralized shapes were further modified by carrying them outward – in all directions possible, the equivalent of the

distance between sections (25, 40 and 47m) unless restricted by other drill hole information or structures.

Controls on higher-grade mineralization: Steeply dipping northwest to northeast trending (~N25W – N25E) structures seem to be important higher-grade feeder structures. They tend to not only carry higher grades but also tend to be more silicified and to carry significant amounts of stibnite (Sb) – far in excess to other zones of mineralization. Other structures, typically low angle structures appear to play an important role in dispersing the mineralization – they may have acted as principal channel ways more than as feeders but this is still not clearly defined. A low angle feature in Acrobat seems to have acted as a channel way for gold mineralization but the location of the structure in or along the Candland shale may have also been a factor in restricting the distribution of the gold and thus concentrating it along the Candland shale – lower Notch Peak limestone contact. A low angle feature in Section 34 may have played an important role in localizing gold in the Morgan Pass Formation. The structure does not show evidence of it being a high-grade feeder but gold is localized about it and two large pockets of gold seem to be separated by the structure.

Minor features such as fold axis may also tend to host higher grade gold mineralization.

Sediment-filled karst breccias also tend to host higher-grade gold mineralization. These karst features are ill-defined and irregular."

For each of the deposits, two-dimensional mineralized envelopes were constructed on cross sections to represent the overall limits of potential possible mineralization. The West Pequop deposits appear to be a complex mix of relatively narrow higher-grade and broader lower-grade intercepts with interspaced barren zones. The controls to this mineralization, particularly the high-grade, are not well understood; therefore, it is not possible to create valid accurate three-dimensional representations of the mineralization such as grade shells for specific grade ranges. The SRK approach was to define the total extent of possible mineralization, as an overall-encompassing mineralized shape, and the Agnico two-dimensional mineralized sectional envelopes are used for this purpose.

Varying amounts of drilling and levels of geologic interpretation and information are available, with Section 34 and Acrobat having the most detail and Mountain Top the least. Figure 15-1 is a perspective view of the mineralization strings provided for Section 34 and Figures 15-5 and 15-7 are those for Acrobat and Mountain Top respectively. Figures 15-2 and 15-6 are single sections for Section 34 and Acrobat. The strings effectively identify significant mineralization in the two dimensions of the sections.

As noted in section 15.6 below the "best" range (maximum) achieved from isotropic variograms is about 12m to 15m implying a continuity of grades more or less of this magnitude. With drillhole sections on Section 34 spaced at 40m and on Acrobat at 25m, the correlation of mineralized shapes from section to section is problematic. In some cases, the strings on a given section can be correlated to strings on adjacent sections and the creation of a linked wireframe between would be straightforward. However, especially in complex higher-grade zones, the correlation from section to section is not at all obvious and the linking between sections becomes complicated. Zones on a given section coalesce or break apart on adjoining sections, the splitting and linking of zones becomes quite arbitrary. The wireframe volumes achieved are very sensitive to the manner of this linking and without some form of geological or interpretive guidance; these volumes can be quite unrealistic. Therefore, SRK choose to load the two-dimensional

representation of potentially mineralized zones into the block models and to collect the mineralized assay population via a half way to the next section process for both Section 34 and Acrobat. With the lack of sectional interpretation for Mountain Top SRK chose to construct a crude wireframe to represent the mineralized envelope; given the correlation problems noted above the volume estimated was further constrained during grade assignment.

In August of 2009, SRK created three-dimensional, "grade" shells using the available iron ("Fe") geochemical data at Section 34. The Fe grade shell was used as it appeared to represent alteration associated with mineralization, and it encompasses essentially all gold values in drilling, and mostly excludes the generally barren Decoy limestone. Figure 15-3 displays the 2010 sections and a 2009 representation of the Decoy and shows that the barren material continues to be appropriately excluded. Figure 15-4 has the 2009 mineralization envelope superimposed on the 2010 strings; the 2009 envelope is broader. The additional detail of the 2010 sections provides a better control over the mineralized versus non-mineralized assignment, at least within the two dimensions of the sections.

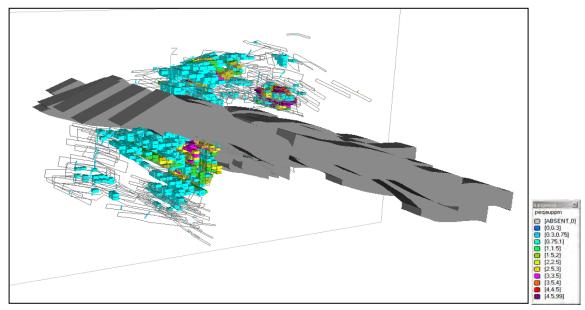
Figure 15-1: Oblique View of Section 34 Mineralization Sections showing Mineralized Shapes (2-D Strings)

Note: Looking northwest downward at approximately -30 degrees (SRK, 2010)

Figure 15-2: Section 34 Mineralization Section 4537480N, Looking North

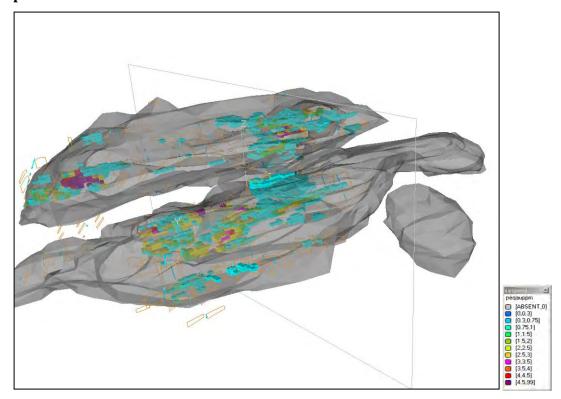
Note: 50m square grid for scale (small grid squares), Looking North at 0 degrees inclination, (SRK, 2010)

Figure 15-3: Oblique View of Section 34 Mineralization Sections with intervening barren Decoy Limestone



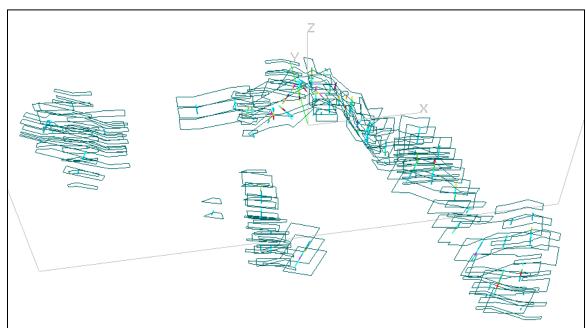
Note: Section looking N15E and down at -15 degrees. Decoy Limestone shown as dark gray colored solid, dipping easterly, (SRK, 2010)

Figure 15-4: Oblique View of Section 34 Mineralization Sections with 2009 Mineralization envelope and 2010 Mineralized Blocks



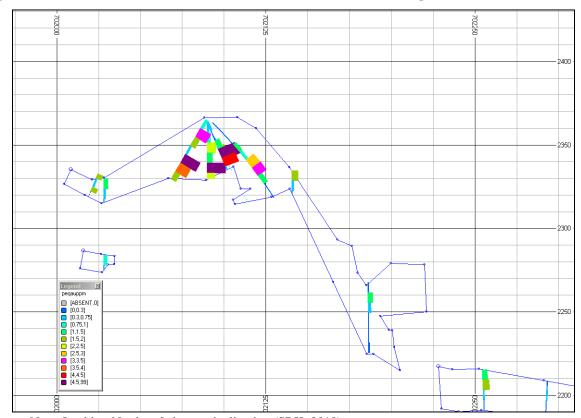
Note: Looking N45E and downward at -20 degrees, (SRK, 2010).

Figure 15-5: Oblique View of Acrobat Mineralization Sections



Note: Looking N10 E and downward at -30 degrees, (SRK, 2010)

Figure 15-6: Acrobat Mineralization Section 4539700N, Looking North



Note: Looking North at 0 degrees inclination (SRK, 2010).

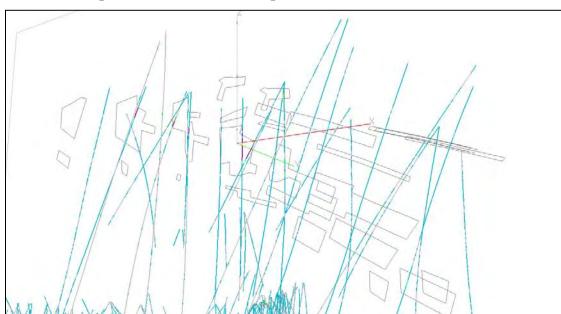


Figure 15-7: Oblique View of Mountain Top Mineralization Sections

Note: Looking N45W and upward at +30 degrees; showing drillhole traces and sectional shapes, (SRK, 2010)

15.3 Database

The West Pequop drillhole database consists of 206 drillholes (combination of RC and Core), for which 183 holes are in the three resource areas; as shown in Table 15-1:

Table 15.1: West Pequop Drillholes

		Acrobat	5	Section 34	Mtn Top		Mtn Top Other		Other
Period	No.	m	No.	m	No.	m	No.	m	
pre-2009	61	10,633.1	57	14,655.4	18	4,013.7	13	4,272.0	
2009	19	5,856.5	13	3,882.9	15	3,920.5	10	1,994.3	
sub-totals	80	16,489.6	70	18,538.3	33	7,934.2	23	6,266.3	

Source: SRK 2010

Due to the preliminary nature of the initial resource estimate, and the bulk of the drilling as RC (see Table 8.1), SRK did not exhaustively examine core versus RC drilling for any potential bias; this is recommended for future resource estimation. However, the following can be said for core versus RC drilling at West Pequop:

- For all three deposits, there are a total of 59 core holes for 15,669.4m and 124 RC holes for 27,292.7m;
- All RC samples (except 2) are on 1.52m (5ft) intervals, whereas core assays are variable in length with only 67% of all core intervals at 1.52m;
- The total database has 19,680 RC assays and 12,434 core assays;
- There are 703 core assays and 1,577 RC assays greater than 100pbb Au: and
- A CF plot of all core and RC assays greater than 100 ppb Au is shown in Figure 15-8. The curves indicating a bias low in RC assays relative to core; however, the analysis is not a direct comparison, as in twin-hole analysis. The apparent bias does indicate that the further study is needed of RC versus core sampling, particularly for grades above 3.0 g/t Au (3000 ppb).

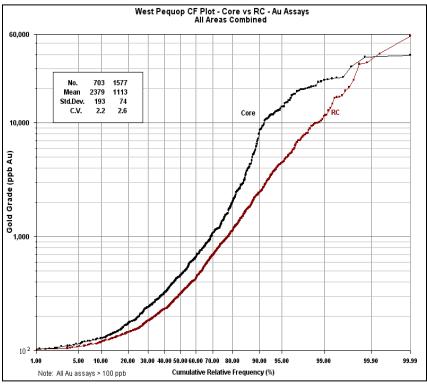


Figure 15-8: Core versus RC Gold Assays >100 ppb Au – All Deposits

Source: SRK 2010

SRK has used the RC and core assay data, as received, to create a composite file for each deposit. Clearly, a closer look at potential bias of RC sampling is required, and it should be done on a deposit specific basis. SRK suggests that for this resource estimate, it is acceptable to use the combined database for core and RC assays, for the following reasons:

- It is assumed that RC assays may in fact be biased low, as much of the mineralization is fracture controlled and associated with fine grained hematitic oxidation, and the fine grained oxides could conceivably be partially lost in RC drill sampling;
- Core is considered a more definitive sampling method, and has been proven to be so in the adjacent Long Canyon gold deposit as well. It is assumed that core assays are a better representation of actual assays, and not likely biased high (no logical reason known for core being biased high relative to true assay);
- The resource is predominantly an Inferred confidence classification; and
- If the RC sampling is in fact biased low, then the overall deposit grades are also biased low and thus the resource estimate is conservative with respect to grade.

A potential problem is that RC sampling may tend to not only understate actual grade, but also smear grade down-hole from a narrow high-grade intersection. This has been shown to be the case for RC drilling of high-grade intersections at the adjacent Long Canyon gold deposit, by means of core-hole twins, and down-hole systematic analysis of RC assays patterns. SRK

recommends this type work and/or analysis be carried out at West Pequop to determine if RC sampling does indeed incur bias.

15.4 Block Models

SRK constructed three separate block models using the Datamine Studio3® mining software package for the West Pequop Deposits with data provided by AuEx. The models have the following spatial limits:

Table 15.2: Model Limits

Section 34 Model Limits (m)				
	Minimum	Maximum		
Easting Limits	701,500	702,300		
Northing Limits	4,537,000	4,538,000		
Elevation Limits	1,900	2,620		
Block Size East-West	10			
Block Size North-South	10			
Block Size Elevation	6			
Acrobat Mode	el Limits (m	1)		
	Minimum	Maximum		
Easting Limits	701,700	702,400		
Northing Limits	4,538,300	4,538,900		
Elevation Limits	2,000	2,480		
Block Size East-West	10			
Block Size North-South	10			
Block Size Elevation	6			
Mountain Top M	odel Limits	s (m)		
	Minimum	Maximum		
Easting Limits	703,000	703,400		
Northing Limits	4,536,500	4,536,900		
Elevation Limits	2,400	2,760		
Block Size East-West	10			
Block Size North-South	10			
Block Size Elevation	6			

Source: SRK 2010

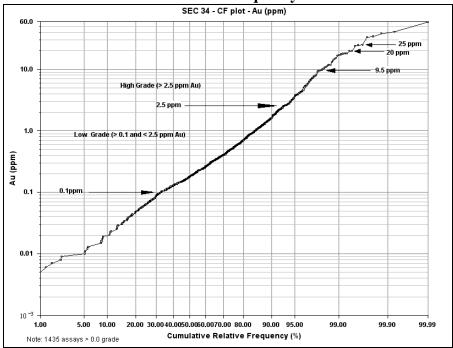
The block size of ten meters in plan and six meters vertically is considered appropriate for Section 34 and Acrobat given the drilling density and the approximation of a twenty-foot bench height (expected for open pit mining in the area), and marginally appropriate for Mountain Top with its lower drilling density.

15.5 Assay & Composite Data – Population Domain Analysis

The mineralization strings provided by Agnico were used to select drill-hole assays for each deposit as being part of the "mineralized population". In general, these strings worked well for the purpose of this selection. Some minor editing was also done to bring assays on the margins of strings (which were obviously intended to be selected) into the mineralized population data set (3 assays for Section 34, 9 assays for Acrobat and 3 for Mountain Top). All subsequent data

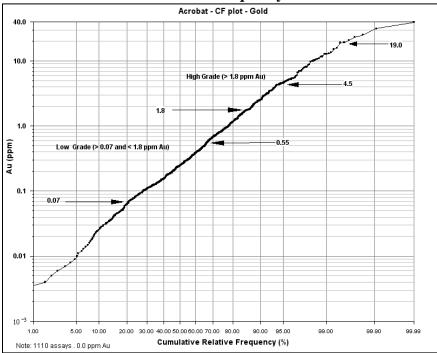
analysis was restricted to this mineralized population. Assay data was capped prior to compositing. Figures 15-9 through 15-11 below are cumulative relative distribution diagrams for the Section 34, Acrobat, and Mountain Top raw gold assay data distributions, respectively. Table 15.3 shows the basic statistics for capped raw assays and 6-meter composites by deposit.

Figure 15-9: Section 34 Cumulative Relative Frequency Distribution



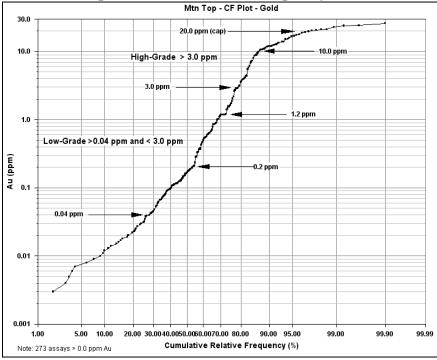
Source: SRK 2010

Figure 15-10: Acrobat Cumulative Relative Frequency Distribution



Source: SRK 2010

Figure 15-11: Mountain Top Cumulative Relative Frequency Distribution



Source: SRK 2010

Table 15.3: Deposit Assay & Composite Summary Statistics

Raw Assays Capped					
Basic Statistics (Au - ppm)	Acrobat	Section 34	Mountain Top		
Threshold cutoff	0.07	0.1	0.04		
Number of Assays	1110	1438	276		
Maximum	19	17	20		
Mean	1.0	0.8	2.6		
Standard Deviation	2.3	2.2	5.2		
C.V.	2.3	2.9	2.0		
6 Me	eter Comp	osites			
Basic Statistics (Au -			Mountain		
ppm)	Acrobat	Section 34	Top		
Threshold cutoff	0.07	0.1	0.04		
Number of Composites	306	464	85		
Maximum	10.7	17	20		
Mean	0.9	0.7	2.1		

Source: SRK 2010

Standard Deviation

C.V.

While the West Pequop deposits may be considered to be related, the distributions of grades are markedly different. Mountain Top in particular appears to have higher grades. As was done in 2009 using the cumulative frequency distribution diagrams (CF plot) as a guide, in conjunction with an examination of the distribution of drillhole data, three "thresholds" were selected for each deposit. Firstly, a minimum threshold was selected distinguishing lower grade "mineralized" versus non-mineralized material based, subjectively, by choosing an inflection point on the lower grade tail of the CF plot. Secondly, a threshold was selected above which grades would be considered part of a "higher grade" population, which would require separate grade estimation constraints. Lastly, an inflection point was selected to identify assays that are to be considered "outliers" to the general distribution and "capped" or setback to a defined threshold. The thresholds identified are tabulated below on Table 15.4 and shown on the respective CF plots.

1.5

1.6

1.7

2.5

4.1

1.9

Table 15.4: Grade Population Cutoff Thresholds

Threshold (ppm)	Section 34	Acrobat	Mountain Top
Lower Grade Population	0.1	0.1	0.0
Higher Grade Population	2.5	1.8	3.0
High Grade Cap	20.0	19.0	20.0

Source: SRK 2010

While the selection of the thresholds described above is somewhat subjective, SRK is of the opinion that the mineralization on these deposits is not at all homogeneous and that this

differentiation needs to be made. Inspection of the core supports this; relatively lower grade mineralization is intersected abruptly with higher-grade values. The nature of the style of the higher-grade mineralization is extremely complex, and/or is the result of more than one geological structure, and should be further investigated.

SRK recognized in 2009 that at West Pequop there is more than one style of mineralization; a narrow higher-grade component bracketed by a sometimes not so broad lower grade component. In 2009 an attempt was made to assign the percentage of higher grade and lower grade material to each of the potentially mineralized blocks. The capped raw data was once again composited this time into 1.5m intervals; this interval length seemed to correspond to the most frequently occurring higher-grade intercept in the RC assays. Fractional probabilities of one were assigned to composited 1.5m intervals whose values exceeded the higher-grade thresholds identified on Table 15.4 above for each deposit respectively. All other composites received a value of zero; these fractional probabilities were interpolated recognizing that it was unlikely that an entire 10mX10mX6m block would be composed of higher-grade material. In 2010, SRK reiterated this modeling procedure but the global resource results were not significantly different from those achieved via simple grade estimation and this attempt to differentiate mineralized versus non-mineralized fractions within model blocks was abandoned. Likewise attempts to model an intermediate population ("higher grade population", as done in 2009) were abandoned in favor of a more straightforward single population approach.

For each deposit, the raw assays were capped or "set back" to the respective threshold values, prior to compositing, as noted in Table 15.4 above.

15.6 Geological Controls, Anisotropy and Search Orientation Domains

15.6.1 Preferential Orientation Planes of Mineralization

SRK examined the two-dimensional strings (2-D closed polygon shapes) provided by Agnico, of sectional mineralized zones, using Leapfrog® 3-D visualization software. SRK created preferred geological orientations to mineralization by creating curvilinear strings (2-D lines) through the Agnico two-dimensional strings and the drillhole grades, section-to-sections, and linking those strings to create planes through the preferred orientation of high-grade mineralization. This was also done for the lower grade mineralization where areas of two-dimensional strings could be approximately linked section-to-section. The result is a set of plans that pass through the drillhole grades, and preferentially follow stratigraphy, and in the case of Acrobat also follows major structural directions. For Section 34 the set of defined faults provided additional preferred orientation planes. Geologically, these planes represent the "preferred orientation of mineralization" in each deposit.

The preferred orientation planes created in Leagfrog® mimic the linking of sectional mineralized shapes, approximately through the center of those shapes, and through the location of drill hole assays.

For Acobat, SRK generated 6 curvi-planar surfaces that represent the preferred orientation of mineralization and include the northerly trending mapped Feeder Fault. For section 34, 10 planar surfaces represent preferred orientations of mineralization, as do the F45 and F47 defined

fault structures. Similarly, 6 planes of preferred orientation of mineralization were developed for Mountain top.

The preferred orientation planes for mineralization and for faults in Section 34 were exported as surfaces for use in Datamine software, and used for anisotropy determinations as described in 15.6.3.

15.6.2 Variography

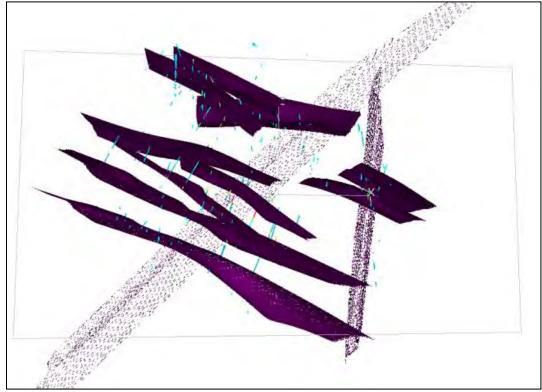
Variograms, indicator variograms and correlograms were constructed for raw and composited assay values for each deposit. Given the variation of lower and higher grade values and the lack of closely spaced values (except down hole), even within Section 34, only very erratic results were obtained, with very high nugget values relative to sills. To achieve a reasonably well behaved variogram SRK used combined 6 meter composites from all of the deposits (within the respective mineralized envelope) to achieve "isotropic" (all directional) variograms with ranges on the order of 12 to 15 meters. While, no preferential orientations (anisotropies) of the continuity of mineralization could be observed from the variography, SRK is of the opinion from general geologic inspection that broad orientation trends do exist.

15.6.3 Anisotropy

The dynamic anisotropy option in Datamine Studio3® allows the anisotropy rotation angles for defining the search volume to be defined individually for each cell in the model. The search volume is oriented precisely and follows the trend of the mineralization. The rotation angles are assigned to each cell in the model; and it is assumed that the dimensions of the ellipsoid, the lengths of the three axes, remain constant. A point file, where each point has a value for dip and dip direction, is created from the mineralization wireframes and is intended to represent the preferential "down dip" direction, which varies locally, over the vertical and horizontal extent of the wireframes (or digital terrain model). Since the three axes of the search volume are orthogonal and only two rotations are used (dip and dip direction) the orientation of all axes are explicitly defined. The point values are taken from the orientation of the triangular facets that comprise the surface of the wireframe.

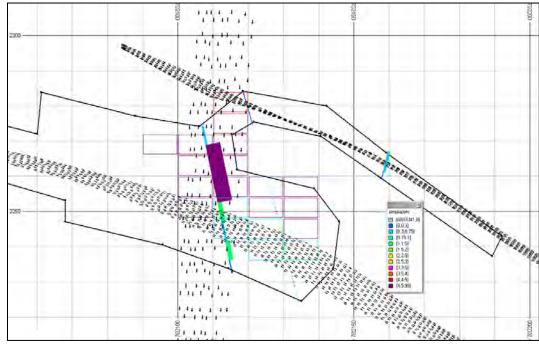
For Section 34 the anisotropy points were established from the facets of various Au digital terrain model triangles and have a mean azimuth of 92 degrees and dip of 25 degrees, which supports a general interpretation of a trend of the overall mineralization as plunging shallowly to the NE (Figure 15-12). Each arrow is a locally interpreted dip and dip direction. Multiple planes (domains) were created for both a lower and a higher-grade population, as defined in Section 15.6.1 above. The steeply dipping planes are from the F45 and F47 fault surfaces provided by Agnico. To model the assumed steep anisotropies associated with the faults these orientations were interpolated into the block model using very short search distances while the general model matrix received values from the shallowly dipping surfaces (Figure 15-13). In this manner, the effects of the fault anisotropies are intended to be constrained to the domains of areas immediately adjacent to the fault intersections.

Figure 15-12: Oblique View of Section 34 Dynamic Anisotropy Surfaces



Note: Looking N30E and downward at -20 degrees, (SRK, 2010)

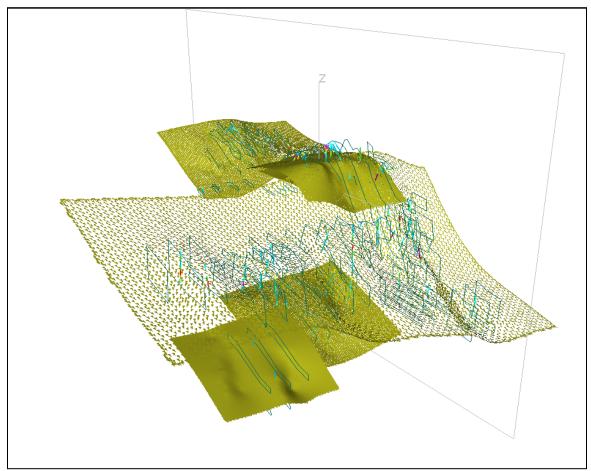
Figure 15-13: Section 34 Anisotropy Section 4537680N, Looking North



Note: Small squares are 10m for scale, Looking North at Azimuth 0 degrees (SRK, 2010)

Acrobat has a complex situation where multiple trends, orientations, or controlling structures probably exist, have yet to be defined or fully understood, and probably vary across the deposit. Digital terrain model "planes" (preferred orientation planes as described in Section 15.6.1) based on the Au grade shell were created as can be seen on Figure 15-14. The steeply dipping structures to the north contrasts sharply with the much shallower dips to the south and west, all of which suggests multiple "domains" of orientation that should be investigated, further geologically defined, and modeled separately in more detail.

Figure 15-14: Oblique View of Acrobat Anisotropy Surfaces



Note: Looking N60W and downward at -30 degrees, (SRK, 2010)

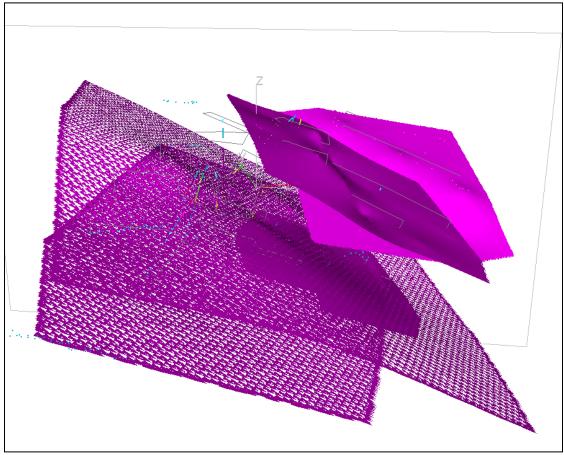


Figure 15-15: Oblique View of Mountain Top Anisotropy Surfaces

Note: Looking N20E and downward at -20 degrees

Similar methods were used on Mountain Top (Figure 15-15). For all deposits, the anisotropy point data was assigned to model block positions using a modified nearest neighbor methodology and used to control the orientation of all subsequent interpolations.

15.7 Grade Assignment

With the limited sample set available (and erratic variography) an inverse to the distance power of two (ID²) was chosen to weight grades selected in the search ellipse. The orientation of the search ellipse was controlled by the dynamic anisotropies as discussed in section 15.6. Table 15-5 below summarizes the interpolation parameters.

Table 15.5: Search Neighborhood & Grade Interpolation Parameters

Section 34 Lower and Higher Grade Populations					
	Search Distance (meters)			Minimum number of composites	Maximum from one drillhole
SVOL	X	Y	Z	•	
1	15	15	7.5	3	2
2	30	30	15	3	2
3	42	42	21	3	2
	Acro	bat l	ower and Hig	gher Grade Popula	ations
	Search Distance (meters)			Minimum number of composites	Maximum from one drillhole
SVOL	X	Y	Z		
1	15	15	7.5	3	2
2	30	30	15	3	2
3	42	42	21	3	2
	N	/loun	tain Top Low	er Grade Population	on
			h Distance neters)	Minimum number of composites	Maximum from one drillhole
SVOL	X	Y	${f z}$		
1	15	15	7.5	3	2
2	33	33	16.5	3	2
	N	loun	tain Top High	er Grade Populati	
	Search Distance (meters)			Minimum number of composites	Maximum from one drillhole
SVOL	X	Y	${f Z}$		
1	15	15	7.5	2	2

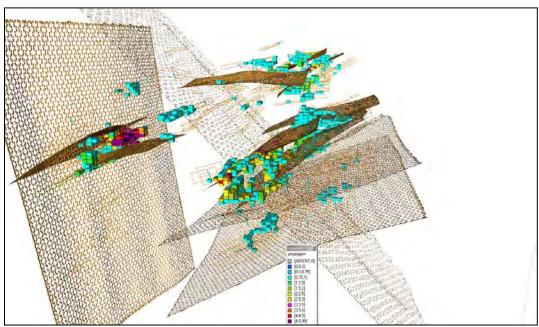
Source: SRK 2010

To preserve local grade variation, a search neighborhood strategy with three search ellipse (SVOL) volumes was used for Section 34 and Acrobat. For Mountain Top, with its' minimally defined limits to mineralization only two search volumes were used for lower grade (less than 10 ppm) and only one for higher grade (greater than 10 ppm) assignment.

Only blocks not estimated with the first set of parameters were estimated with a subsequent expanded search. In order to preserve this local variation of grades and also have a requirement for grade assignment using data from more than one drillhole, a minimum of three 6m composites was required, with a maximum of 2 from any given hole, for estimation. Except for the higher grade population on Mountain Top all block estimation required data from at least two drillholes. For that higher grade population the single estimation used a single short search distance. For future models, alternative methods could be adopted. With considerably more data, multiple indicator kriging or conditional simulation methodologies should be examined.

Figures 15-16 and 15-17 are intended to show the contrast between Section 34 blocks interpolated using the high angle "fault" anisotropies and those using the general anisotropy plains. Figures 15-18 and 15-19 displays the grade patterns achieved with the Acrobat anisotropy surfaces.

Figure 15-16: Oblique View of Section 34 Block Model Projection with Anisotropy Surfaces



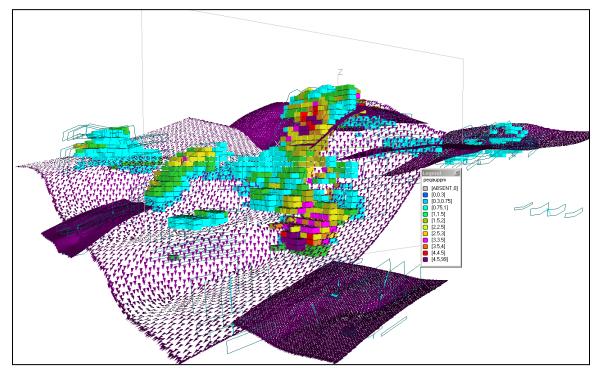
Note: Looking S20E and downward at -20 degrees, (SRK, 2010)

| Market | M

Figure 15-17: Section 34 Block Model Projection with Anisotropy Surfaces

Note: Looking S20E and downward at -20 degrees

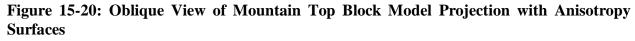


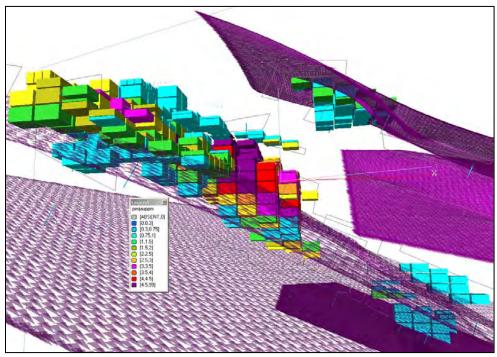


Note: Looking N60E and downward at -10 degrees, (SRK, 2010)

Figure 15-19: Acrobat Block Model Section 4538725N, Looking North

Note: Note: Looking North at 0 degrees inclination, (SRK, 2010)





Note: Looking N20W and upward at +20 degrees, (SRK, 2010)

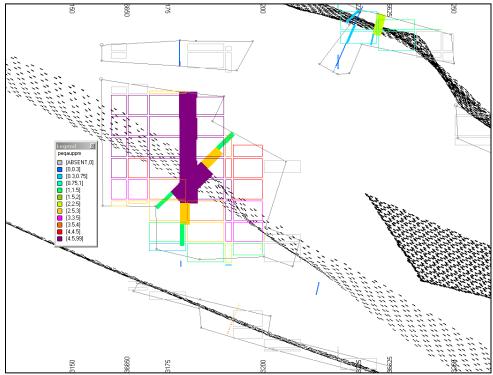


Figure 15-21: Mountain Top Block Model Section, Looking N20E,

Note: Looking N20E, at 0 degrees inclination, (SRK, 2010)

15.8 Specific Gravity Measurements (Bulk Density)

A density of 2.7 was used for reporting all resources. This is the average density for mineralized material; densities for waste were not modeled.

Density data is available, by lithology, consisting of a total of 225 sample data results from core sample density measurements for Section 34, 50 sample results for Acrobat, and 134 for Mountain Top. Density values range from 2.4 to 2.9, with a slightly tighter range for lithologies that host mineralization. SRK did not develop a lithological model with associated densities; this is recommended for future resource estimations.

15.9 Resource Confidence Classification

The majority of the resource for each of the three deposits has been classified as Inferred with Inferred constituting 74% (Mountain Top) to 82% (Section 34) of the resources. A requirement was made that a 10m X 10m X 6m potentially mineralized block must be within 10 meters of a 6 meter composite for that block to be considered "Indicated". Indicated blocks are then either penetrated by a drillhole or are immediately adjacent to one.

While from the point of view of data distribution this classification may be reasonable, it ignores any criteria regarding potential mineability. With pockets or pods of gold mineralization, and the use of solely the distribution of data for confidence classification, a "spotted dog" effect of non-continuous (especially between sections) Indicated zones has been achieved. Targeted in-fill

drilling (targeted to intercept mineralization between sections) will be required to upgrade additional resources classified from Inferred to Indicated. In addition, to upgrade resource classification will require some form of in-pit resource calculation where simple designs (possibly the result of optimizations) are intersected with the models to ensure Indicated blocks are potentially mineable under some conceivable economic scenario; that is they fall within the limits of a preliminary pit design. The state of exploration for West Pequop gold deposits is too early to justify estimating in-pit resources. SRK has estimated and is reporting an in-situ total resource rather than the preferred option of reporting an in-pit resource, primarily because the resource is predominantly at an Inferred classification.

15.10 Block Model Validation & Mineral Resource Sensitivity

The criteria noted above where "Indicated" blocks must be within 10 meters of a 6 meter composite ensures that the Indicated resource closely approximates the grade of nearby composites; a comparison of estimated Indicated grades and composites validates this. With Inferred, the resource is most sensitive to the mineralization envelopes. As noted above the correlation of these zones between sections spaced at 25 or 40 meters apart is ambiguous. Infill drilling and envelopes correlated over shorter distances will be required to improve on this delineation.

The resources estimated in 2010 for Section 34 (which had marginal additional drilling from the 2009 estimate) are globally very similar to those estimated in 2009 using "fractional probability" and 1.5-meter composites. This demonstrates that alternative methodologies produce more or less similar global totals. .

15.11 Resource Statement

The resources stated below in Tables 15-6 through 15-12 for West Pequop are classified as Inferred and Indicated according to CIM classification criteria. The current density of drilling information and the degree of known geological controls to the gold mineralization are among the primary factors limiting the classification of Indicated.

SRK has highlighted the 0.3 g/t cutoff grade in Tables 15-6 through 15-12, as a reasonable grade for which resources could be reported at West Pequop. Mine Development Associates has reported insitu resources at a 0.30 g/t Au cutoff grade for the adjacent Long Canyon Gold Project, without respect to mineability or processing. The 0.3g/t cutoff grade equates to 0.01 ounce per short ton (oz/st) in-situ mineralized material. At a price of US\$1000 per ounce gold, the cutoff grade equals an value of approximately \$10.00/st; which is deemed by SRK to be at or near an expected mining cutoff for oxide heap leach gold operations in Nevada. Therefore SRK deems the cutoff grade sufficient to define a "reasonable prospect for economic extraction", a necessary condition to satisfy CIM resource definitions in future resource reporting.

Table 15.6: Total In-situ Mineral Resources by Deposit and Classification, at 0.3 g/t Au cutoff grade (SRK 2010).

Deposit	Classification	Tonnes	Au (g/t)	Au Grams	Au Ounces
Section 34	Indicated	527,000	1.64	865,000	27,810
Section 34	Inferred	2,883,000	1.41	4,063,000	130,620
Acrobat	Indicated	581,000	1.32	767,000	24,640
Acrobat	Inferred	2,178,000	1.22	2,655,000	85,360
Mountain Top	Indicated	119,000	3.06	363,000	11,690
Mountain Top	Inferred	443,000	2.35	1,041,000	33,450

Source: SRK 2010

Tables 15-7 through 15-12 present the sensitivity of the resource estimates to cutoff grade for each deposit, by classification.

Table 15.7: Section 34 Inferred Resources by Cutoff Grade (SRK 2010).

Cutoff Au g/t	Tonnes (k)	AU	Grams	Oz
0.1	4,747	0.93	4,424,049	142,238
0.2	3,714	1.15	4,264,848	137,119
0.3	2,883	1.41	4,062,857	130,625
0.4	2,491	1.58	3,926,309	126,235
0.5	2,162	1.75	3,778,732	121,490
0.6	1,867	1.94	3,616,809	116,284
0.7	1,651	2.11	3,477,204	111,796
0.8	1,452	2.29	3,328,012	106,999
0.9	1,276	2.49	3,178,482	102,191
1	1,135	2.68	3,044,976	97,899

Table 15.8: Section 34 Indicated Resources by Cutoff Grade (SRK 2010).

	- J			
Cutoff Au g/t	Tonnes (k)	AU	Grams	Oz
0.1	840	1.10	922,112	29,647
0.2	642	1.39	892,608	28,698
0.3	527	1.64	864,962	27,809
0.4	481	1.77	849,054	27,298
0.5	426	1.93	824,308	26,502
0.6	334	2.31	773,913	24,882
0.7	294	2.54	748,095	24,052
0.8	259	2.79	721,373	23,193
0.9	244	2.90	708,881	22,791
1	221	3.10	687,308	22,098

Table 15.9: Acrobat Inferred Resources by Cutoff Grade (SRK 2010).

Cutoff Au g/t	Tonnes (k)	AU	Grams	Oz
0.1	2,871	0.98	2803230	90,127
0.2	2,570	1.07	2755704	88,599
0.3	2,178	1.22	2655025	85,362
0.4	1,827	1.39	2535082	81,505
0.5	1,604	1.52	2435452	78,302
0.6	1,470	1.61	2361739	75,932
0.7	1,351	1.69	2284781	73,458
0.8	1,210	1.80	2179592	70,076
0.9	1,104	1.89	2089521	67,180
1	992	2.00	1983167	63,761

Table 15.10: Acrobat Indicated Resources by Cutoff Grade (SRK 2010).

Cutoff Au g/t	Tonnes (k)	AU	Grams	Oz
0.1	770	1.04	804,438	25,863
0.2	676	1.17	789,888	25,396
0.3	581	1.32	766,501	24,644
0.4	496	1.49	737,668	23,717
0.5	450	1.59	716,439	23,034
0.6	411	1.69	695,067	22,347
0.7	366	1.82	666,153	21,417
0.8	340	1.90	646,382	20,782
0.9	303	2.03	615,758	19,797
1	258	2.22	572,525	18,407

Table 15.11: Mountain Top Inferred Resources by Cutoff Grade (SRK 2010).

Cutoff Au g/t	Tonnes (k)	AU	Grams	Oz
0.1	492	2.13	1,050,614	33,778
0.2	470	2.23	1,047,266	33,671
0.3	443	2.35	1,040,519	33,454
0.4	422	2.45	1,032,947	33,210
0.5	408	2.52	1,026,972	33,018
0.6	384	2.64	1,013,194	32,575
0.7	356	2.80	995,226	31,998
0.8	329	2.97	975,069	31,349
0.9	312	3.08	961,323	30,907
1	293	3.22	943,459	30,333

Table 15.12: Mountain Top Indicated Resources by Cutoff Grade (SRK 2010).

Cutoff Au				
g/t	Tonnes (k)	AU	Grams	Oz
0.1	129	2.84	365,762	11,760
0.2	126	2.91	365,325	11,746
0.3	119	3.06	363,450	11,685
0.4	104	3.44	358,316	11,520
0.5	99	3.59	356,137	11,450
0.6	91	3.86	351,680	11,307
0.7	85	4.10	347,515	11,173
0.8	83	4.15	346,504	11,140
0.9	81	4.24	344,765	11,085
1	81	4.24	344,765	11,085

15.12 Conclusions and Recommendations

SRK's initial resource estimate for West Pequop did not include a comprehensive database audit; SRK relied upon the drillhole database provided by Agnico, based on spot checks of the data. A complete audit of the Agnico database audit is recommended for future resource estimates, to verify that the database does not have other areas of error aside from the known collar survey errors, which SRK understands have been corrected for drillhole collars that can be located. SRK has reviewed the existing Agnico QA/QC data, and SRK assumes Agnico will continue with adequate survey and assay QA/QC practices for future drilling. SRK understands that a resurvey of the topography is planned for 2010 as well. With a new and hopefully more accurate topographic surface, the remaining discrepancy between drillhole collar elevations and topography can hopefully be resolved or compensated for.

The geological interpretation available for these deposits is in the form of the two dimensional cross section mineralization envelopes. In lower grade areas the correlation of these shapes from section to section is straightforward. Complex structures, and possibly the intersection of complex structures, appear to control the location of higher-grade mineralization and the correlation of these shapes from section to section is not at all straightforward. In-fill drilling will be required to derive a valid three-dimensional picture. SRK recommends targeted in-fill drilling to define controls to mineralization, particularly higher-grade mineralization. For example, targeting holes on intermediate (20m offset) sections for portions of Section 34 will provide sufficient data to: a) validate the variogram analysis of 15-20m ranges, and b) provides some definition of higher grade and/or structurally controlled mineralization.

The variograms constructed (of any kind) have a maximum interpreted range of 12 to 15 meters which would suggest that these deposits will require targeted delineation on something of the order of magnitude of a 15 to 20 meter grid to be fully classified as indicated. Better defined geological interpretations and/or geostatistical interpretations could relax this requirement.

To "upgrade" additional resources to an Indicated classification will require some form of "inpit" resource calculation where simple designs (possibly the result of optimizations) are intersected with the models to ensure "Indicated" blocks are potentially mineable under some "reasonable" economic scenario; that is they fall within the limits of a preliminary pit design.

15.13 Reserve Estimation

There are no current established reserves for the West Pequop gold deposits

16 Other Relevant Data and Information (Item 20)

The West Pequop Gold Exploration Property is part of a developing gold district, the "Pequop Gold District", which encompasses the property and the adjacent Long Canyon Gold Exploration Project, as further described in Section 13 (Adjacent Properties). Therefore, exploration is early for the district's development, and the exploration potential is still developing. Attendant with additional exploration will be increased road access and infrastructure to the area, which will aid future possible development.

There are no known other relevant data that will materially affect continued exploration on the West Pequop Gold Exploration Project.

17 Interpretation and Conclusions (Item 21)

The West Pequop Gold Exploration Project is an advanced stage exploration property with established mineral resources. Combined total Indicated resources at a 0.3g/t cutoff are 1,227,000 metric tonnes grading 1.63 g/t Au for 64,140 contained ounces gold, and a combined Inferred resource of 5,504,000 metric tonnes grading 1.41 g/t Au for 249, 430 contained ounces gold.

The style of gold mineralization defined at the West Pequop property is analogous to Carlin-Type gold mineralization. Gold is found in association with decalcified and locally jasperoid-silicified silty carbonate sediments, in association with geochemically anomalous arsenic, antimony, mercury, tungsten, and thallium; as mineralization in both high-angle fault/breccia structures, and in altered and locally brecciated stratiform horizons in several lower Paleozoic carbonate stratigraphic units, spread over a large area of the Pequop Mountain range. The mineralization appears to be dominantly oxidized gold mineralization; very little sulfide mineralization is associated with the gold grades in drilling.

17.1 Field Surveys

The work performed by PNGC at the West Pequop property, during the period of 1994 to 2001, was exploration work that conforms to industry standard practices and procedures. PNGC's work is an excellent example of a grass-roots discovery of significant gold mineralization in a region of northern Nevada not previously known for gold mineralization. The program progressed in a logical way from regional BLEG and steam sediment sampling, through soil sampling, to discovery of high-grade gold in outcrop, definition of several gold-in-soil anomalies of size, and confirmation of gold mineralization by RC and core drilling.

Continued exploration work on the West Pequop property by AuEx and its venture participant Agnico from 2005 through present has also been by industry standard procedures, primarily as a mix of RC and core drilling. That drilling has defined sufficient gold mineralization to allow for initial resource estimation by industry-standard block modeling techniques, according to CIM Definitions of mineral resource classification. The project is now at an advanced exploration stage, with gold resources defined by a combined 206 drillholes in three deposits, Acrobat, Section 34, and Mountain Top.

17.2 Analytical and Testing Data

Analytical data are industry standard gold assays, and have been confirmed by multiple drilling campaigns by several different companies. The project drill data are sufficient to support initial resource estimation according to CIM definitions for estimation of mineral resources.

17.3 Exploration Conclusions

Additional drilling, as in-fill drilling will be required to achieve an upgrading of resource classification from Inferred to Indicated. As well, to advance the resource to a significant percentage as Indicated resources will require the assessment of potential mineablilty by determination of "in-pit" resources.

Further database investigations are required, such as the correlations of core and rotary drill assays, and a reconciliation of drill collars to topographic surveys. However, the project database is sufficiently well defined to support the current resource estimation.

Additional metallurgical test data are recommended to determine the potential for mineral processing and gold recovery.

Exploration potential is present to expand the known deposits and to define additional mineralization

18 Recommendations (Item 22)

18.1 Recommended Work Programs

Recommendations for the next phase of exploration at the West Pequop property are several:

- Continued exploration drilling in an attempt to expand current resources and define new areas of mineralization:
- Targeted in-fill drilling to establish a better confidence in the geological model of mineralization, and to confirm grade continuity section-to-section;
- Re-examine variography after in-fill drilling to determine optimal drill density for Indicated classification;
- Re-model the geology incorporating lithological units and densities;
- As part of a complete database verification, evaluate in detail the existing RC versus core assays for potential sample bias. And, if necessary for further analysis, conduct three core and RC twin hole pairs in higher grade material to examine the possibility of low-bias to RC samples, and the possibility of down-hole contamination from RC drilling;
- Determine potential mineability by application of pit optimization and determination "inpit" resources; and
- Initiate a metallurgical test program to determine the potential for mineral processing and gold recovery.

A Recommended Phase I budget is here presented for the above-recommended program. Continued exploration drilling is recommended at the level of 2009 drilling, approximately 50 drillholes for 13, 500m of total RC and core drilling, with a minimum of three targeted as twin sets of higher-grade mineralization.

Phase I Exploration Costs:

	Total Phase I Costs	\$2	.638,750
•	Metallurgical test program	\$	80,000
•	In-pit resource estimate by pit optimization	\$	15,000
•	Re-estimation of resources at end of drilling program	\$	40,000
•	Re-model the geology	\$	20,000
•	Detailed audit of RC versus core assays, and variography	\$	20,000
•	25 Core holes for 6750m @ \$230/m all-in costs	\$1	,552,500
•	25 RC drillholes for 6750m @ \$135/m all-in costs	\$	911,250

The recommended Phase I program and budget is intended for a field season, assuming drilling commences in May and is completed by mid-November.

Continued successful drilling from the recommended Phase I program would justify and additional Phase II program of perhaps \$3.5 million in the following year to include further definition drilling, geotechnical drilling for potential open pit slope determinations, a definitive metallurgical program and a Scoping Study to determine the project potential for economic development.

19 References (Item 23)

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20 Glossary

20.1 Mineral Resources and Reserves

20.1.1 Mineral Resources

The mineral resources and mineral reserves have been classified according to the "CIM Standards on Mineral Resources and Reserves: Definitions and Guidelines" (December 2005). Accordingly, the Resources have been classified as Measured, Indicated or Inferred, the Reserves have been classified as Proven, and Probable based on the Measured and Indicated Resources as defined below.

A Mineral Resource is a concentration or occurrence of natural, solid, inorganic or fossilized organic material in or on the Earth's crust in such form and quantity and of such a grade or quality that it has reasonable prospects for economic extraction. The location, quantity, grade, geological characteristics and continuity of a Mineral Resource are known, estimated or interpreted from specific geological evidence and knowledge.

An 'Inferred Mineral Resource' is that part of a Mineral Resource for which quantity and grade or quality can be estimated on the basis of geological evidence and limited sampling and reasonably assumed, but not verified, geological and grade continuity. The estimate is based on limited information and sampling gathered through appropriate techniques from locations such as outcrops, trenches, pits, workings and drillholes.

An 'Indicated Mineral Resource' is that part of a Mineral Resource for which quantity, grade or quality, densities, shape and physical characteristics can be estimated with a level of confidence sufficient to allow the appropriate application of technical and economic parameters, to support mine planning and evaluation of the economic viability of the deposit. The estimate is based on detailed and reliable exploration and testing information gathered through appropriate techniques from locations such as outcrops, trenches, pits, workings and drillholes that are spaced closely enough for geological and grade continuity to be reasonably assumed.

A 'Measured Mineral Resource' is that part of a Mineral Resource for which quantity, grade or quality, densities, shape, physical characteristics are so well established that they can be estimated with confidence sufficient to allow the appropriate application of technical and economic parameters, to support production planning and evaluation of the economic viability of the deposit. The estimate is based on detailed and reliable exploration, sampling and testing information gathered through appropriate techniques from locations such as outcrops, trenches, pits, workings and drillholes that are spaced closely enough to confirm both geological and grade continuity.

20.1.2 Mineral Reserves

A Mineral Reserve is the economically mineable part of a Measured or Indicated Mineral Resource demonstrated by at least a Preliminary Feasibility Study. This Study must include adequate information on mining, processing, metallurgical, economic and other relevant factors that demonstrate, at the time of reporting, that economic extraction can be justified. A Mineral Reserve includes diluting materials and allowances for losses that may occur when the material is mined.

A 'Probable Mineral Reserve' is the economically mineable part of an Indicated, and in some circumstances a Measured Mineral Resource demonstrated by at least a Preliminary Feasibility Study. This Study must include adequate information on mining, processing, metallurgical, economic, and other relevant factors that demonstrate, at the time of reporting, that economic extraction can be justified.

A 'Proven Mineral Reserve' is the economically mineable part of a Measured Mineral Resource demonstrated by at least a Preliminary Feasibility Study. This Study must include adequate information on mining, processing, metallurgical, economic, and other relevant factors that demonstrate, at the time of reporting, that economic extraction is justified.

20.2 Glossary

Table 20.2.1: Glossary

Assay:
Capital Expenditure: Composite: Combining more than one sample result to give an average result over a larger distance. Comentrate: A metal-rich product resulting from a mineral enrichment process such as gravity concentration or flotation, in which most of the desired mineral has been separated from the waste material in the ore. Initial process of reducing ore particle size to render it more amenable for further processing. Cut-off Grade (CoG): Dilution: Dilution: Dilution: Waste, which is unavoidably mined with ore. Angle of inclination of a geological feature/rock from the horizontal. The surface of a fracture along which movement has occurred. Footwall: The underlying side of an orebody or stope. Mon-valuable components of the ore. The measure of concentration of gold within mineralized rock. Haudiage: A horizontal underground excavation which is used to transport mined ore. A process whereby material is graded according to size by exploiting centrifugal forces of particulate materials. Igneous: Primary crystalline rock formed by the solidification of magma. Kriging: An interpolation method of assigning values from samples to blocks that minimizes the estimation error. Level: Horizontal tunnel the primary purpose is the transportation of personnel and materials. LRP: Horizontal tunnel the primary purpose is the transportation of personnel and materials. LRP: Horizontal tunnel the primary purpose is the transportation of personnel and materials. LRP: Material Properties: Mining Assets: Ongoing Capital: One Range Plan. Mineral/Mining Lease: Mining Assets: Ongoing Capital: One Reserve: Pillar: Rok: Rok: left behind to help support the excavations in an underground mine. Row: See Mineral Reserve. Pertaining to rocks formed by the accumulation of sediments, formed by the erosion of other rocks. Shaft: An opening cut downwards from the surface for transporting personnel, equipment, supplies, ore and
Composite: Concentrate: A metal-rich product resulting from a mineral enrichment process such as gravity concentration or flotation, in which most of the desired mineral has been separated from the waste material in the ore. Crushing: Cut-off Grade (CoG): Initial process of reducing ore particle size to render it more amenable for further processing. The grade of mineralized rock, which determines as to whether or not it is economic to recover its gold content by further concentration. Dilution: Dip: Angle of inclination of a geological feature/rock from the horizontal. The surface of a fracture along which movement has occurred. The underlying side of an orebody or stope. Grade: Hangingwall: Haulage: A horizontal underground excavation which is used to transport mined ore. A process whereby material is graded according to size by exploiting centrifugal forces of particulate materials. Igneous: Primary crystalline rock formed by the solidification of magma. Kriging: An interpolation method of assigning values from samples to blocks that minimizes the estimation error. Level: Horizontal tunnel the primary purpose is the transportation of personnel and materials. Lith-ological: Lith-ological: Lidh-ological: Lidh-ological: A general tunnel the primary purpose is the transportation of personnel and materials. Long Range Plan. Material Properties: Mineral/Mining Lease: Mineral/Mining Lease: Mineral/Mining Lease: Ongoing Capital: Ore Reserve: Pullar: Rook: Rock left behind to help support the excavations in an underground mine. Rook: See Mineral Reserve. Rock left behind to help support the excavations in an underground mine. Rook: Seedimentary: Pratining to rocks formed by the accumulation of sediments, formed by the erosion of other rocks. An opening cut downwards from the surface for transporting personnel, equipment, supplies, ore and
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1 7
waste.
Sill: A thin, tabular, horizontal to sub-horizontal body of igneous rock formed by the injection of magma
into planar zones of weakness.
Smelting: A high temperature pyrometallurgical operation conducted in a furnace, in which the valuable metal
is collected to a molten matte or doré phase and separated from the gangue components that
accumulate in a less dense molten slag phase.
Stope: Underground void created by mining.
Stratigraphy: The study of stratified rocks in terms of time and space.
Strike: Direction of line formed by the intersection of strata surfaces with the horizontal plane, always
perpendicular to the dip direction.
Sulfide: A sulfur bearing mineral.
Tailings: Finely ground waste rock from which valuable minerals or metals have been extracted.
Tailings: Finely ground waste rock from which valuable minerals or metals have been extracted. Thickening: The process of concentrating solid particles in suspension.
Tailings: Finely ground waste rock from which valuable minerals or metals have been extracted.

Abbreviations

The metric system has been used throughout this report unless otherwise stated. All currency is in U.S. dollars. Market prices are reported in US\$ per troy oz of gold and silver. Tonnes are metric of 1,000kg, or 2,204.6lbs. The following abbreviations are used in this report.

Table 20.2.2: Abbreviations

Abbreviation	Unit or Term
A	ampere
AA	atomic absorption
A/m^2	amperes per square meter
ANFO	ammonium nitrate fuel oil
Ag	silver
Au	gold
AuEq	gold equivalent grade
°C	degrees Centigrade
CCD	counter-current decantation
CIL	carbon-in-leach
CoG	cut-off grade
cm	centimeter
cm ²	square centimeter
cm ³	cubic centimeter
cfm	cubic feet per minute
ConfC	confidence code
CRec	
CSS	core recovery closed-side setting
CTW	calculated true width
C1 W	degree (degrees)
dia.	diameter
EIS	
EMP	Environmental Impact Statement
FA	Environmental Management Plan fire assay
ft	foot (feet)
ft^2	square foot (feet)
ft ft ³	cubic foot (feet)
g gal	gram gallon
gal g/I	gram per liter
g/L g mol	gram-mole
g-mol	gallons per minute
gpm g/t	grams per tonne
ha	hectares
HDPE	Height Density Polyethylene
hp	horsepower
HTW	horizontal true width
ICP	induced couple plasma
ID2	inverse-distance squared
ID3	inverse-distance cubed
IFC	International Finance Corporation
ILS	International Finance Corporation Intermediate Leach Solution
kA	kiloamperes
kg	kilograms
km	kilometer
km ²	square kilometer
koz	thousand troy ounce
kt	thousand tonnes
kt/d	thousand tonnes per day
kt/y	thousand tonnes per day
kV	kilovolt
kW	kilowatt
±±,11	1

kWh kilowatt-hour

kWh/t kilowatt-hour per metric tonne

L lite

L/sec liters per second

L/sec/m liters per second per meter

lb pound

LHD Long-Haul Dump truck

LLDDP Linear Low Density Polyethylene Plastic

LOI Loss On Ignition
Life-of-Mine
m
meter
m²
square meter
m³
cubic meter

masl meters above sea level

MARN Ministry of the Environment and Natural Resources

MDA Mine Development Associates

mg/L milligrams/liter millimeter mm square millimeter mm^2 mm^3 cubic millimeter **MME** Mine & Mill Engineering Moz million troy ounces million tonnes Mt MTW measured true width MW million watts million years m.y.

NGO non-governmental organization
NI 43-101 Canadian National Instrument 43-101
OSC Ontario Securities Commission

oz troy ounce oz/Ton Ounces per short ton

% percent

PLC
PLS
Programmable Logic Controller
Pregnant Leach Solution
PMF
ppb
Probable maximum flood
parts per billion

ppb parts per billion parts per million

QA/QC Quality Assurance/Quality Control

RC rotary circulation drilling

RoM Run-of-Mine

RQD Rock Quality Description

SEC U.S. Securities & Exchange Commission

sec second specific gravity

SPT standard penetration testing st short ton (2,000 pounds)

tonne (metric ton) (2,204.6 pounds)

 $\begin{array}{ccc} t/h & & tonnes \ per \ hour \\ t/d & tonnes \ per \ day \\ t/y & tonnes \ per \ year \end{array}$

T Ton (short ton) (200 pounds)
TSF tailings storage facility
TSP total suspended particulates

µm micron or microns, micrometer or micrometers

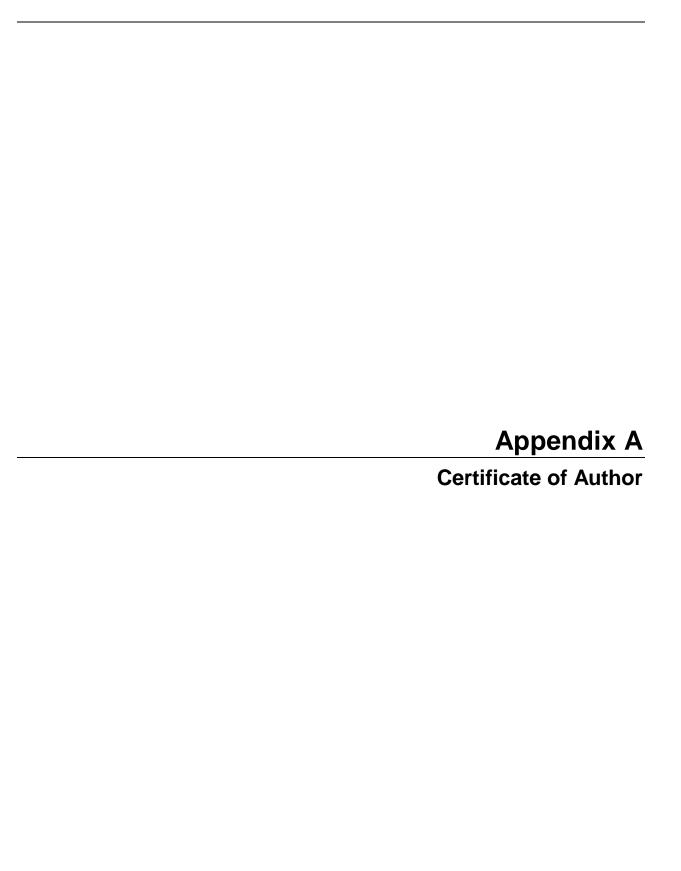
V volts

VFD variable frequency drive

W watt

XRD x-ray diffraction

year



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CERTIFICATE of AUTHOR

Allan V. Moran

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CERTIFICATE of AUTHOR

- I, Allan V. Moran, a Registered Geologist and a Certified Professional Geologist, do hereby certify that:
 - 1. I am currently employed as a consulting geologist to the mining and mineral exploration industry, as Principal Geologist with SRK Consulting (U.S.) Inc, with an office address of 3275 W. Ina Rd., Tucson, Arizona, USA, 85741.
 - 2. I graduated with a Bachelors of Science Degree in Geological Engineering from the Colorado School of Mines, Golden, Colorado, USA; May 1970.
 - 3. I am a Registered Geologist in the State of Oregon, USA, # G-313, and have been since 1978. I am a Certified Professional Geologist through membership in the American Institute of Professional Geologists, CPG 09565, and have been since 1995.
 - 4. I have been employed as a geologist in the mining and mineral exploration business, continuously, for the past 39years, since my graduation from university.
 - 5. I have read the definition of "qualified person" set out in National Instrument 43-101 ("NI 43-101") and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfill the requirements to be a "qualified person" for the purposes of NI 43-101. The Technical Report is based upon my personal review of the information provided by the issuer. My relevant experience for the purpose of the Technical Report is:
 - Vice President and U.S. Exploration Manager for Independence Mining Company, Reno, Nevada, 1990-1993
 - Manager, Exploration North America for Cameco Gold Inc., 1998-2002
 - Exploration Geologist for Freeport McMoRan Gold, 1980-1988
 - Gold exploration experience in Nevada from 1980 to 2000 with Freeport Exploration, Freeport McMoRan Gold, Independence Mining Company, Vista Gold, and Cameco Gold Inc.
 - As a consultant, I have completed several NI 43-101 Technical reports, 2003-2010.
 - 6. I am responsible for the content, compilation, and editing of all sections of the technical report titled "NI 43-101 Technical Report on Resources, *West Pequop Gold Exploration Project, Nevada, USA*", and dated July 15, 2010 (the "Technical Report") relating to AuEx Venture, Inc.'s West Pequop gold exploration project in Elko County. I have personally visited the Project in the field during the period August 25-27, 2004, and on June 31 and July 01, 2009.
 - 7. I have had prior involvement with the property that is the subject of the Technical Report, as author of the initial Technical Report titled Pequop Exploration Property, dated January 05, 2005., and as an author of an internal "Report on Resources" dated December 16, 2009 (private internal report for AuEx)

- 8. I am not aware of any material fact or material change with respect to the subject matter of the Technical Report that is not reflected in the Technical Report, for which the omission to disclose would make the Technical Report misleading.
- 9. I am independent of the issuer applying all of the tests in Item 1.4 of National Instrument 43-101.
- 10. I have read National Instrument 43-101 and Form 43-101F1, and the Technical Report has been prepared in compliance with that instrument and form.
- 11. I consent to the filing of the Technical Report with any stock exchange and other regulatory authority and any publication by them, including electronic publication in the public company files on their websites accessible to the public, of the Technical Report

Dated in Tucson, Arizona, July 15, 2010.

contact

This signer gray year scanged for the exclusive

Allan V. Moran

(Signed (Sealed)

CERTIFICATE OF AUTHOR

Frank A. Daviess

Principal Resource Geologist SRK Consulting (U.S.), Inc. Email: fdaviess@comcast.net

I, Frank A Daviess do hereby certify that:

- I am currently employed as a consulting resource geologist to the mining and mineral exploration industry and I am currently under contract as an associate Principle Resource Geologist with SRK Consulting (U.S.) Inc, with an office address of 7175 W. Jefferson Avenue, Suite 3000 Lakewood, Colorado, U.S. 80235.
- 2. I graduated from the University Of Colorado, Boulder, Colorado, USA with a B.A. in Geology in 1971 and a M.A. in Natural Resource Economics and Statistics in 1975
- 3. I am a Member of the Australasian Institute of Mining and Metallurgy (Registration No. 226303).
- 4. I am a Registered Member of the Society for Mining, Metallurgy and Exploration, Inc. (Registration No. 0742250).
- 5. I have been employed as a geologist in the mining and mineral exploration business, continuously, for the past 31 years, since my graduation from university.
- 6. I have read the definition of "qualified person" set out in National Instrument 43-101 ("NI 43-101") and certify that by reason of my education, affiliation with professional associations (as defined in NI 43-101) and past relevant work experience I fulfill all the requirements to be a "qualified person" for the purposes of NI 43-101. I have authored sections of the Technical Report. The Technical Report is based upon my personal review of the information provided by the issuer. My relevant experience for the purpose of input to the Technical Report is:
 - Specialization in the estimation, assessment and evaluation of mineral resources including gold since 1975.
 - Specialization in uranium resource estimation experience as an Ore Reserve Analyst, US Department of Energy, Resource Division, Grand Junction, CO, 1975-1978
- 7. I am responsible for the Mineral Resource and Mineral Reserve Estimates section of the technical report titled "NI 43-101 Technical Report on Resources, *West Pequop Gold Exploration Project, Nevada, USA*", and dated July 15, 2010 (the "Technical Report") relating to AuEx Venture, Inc.'s West Pequop gold exploration project in Elko County. I have personally visited the Project in the field during the period June 30 to July 01, 2009.
- 8. I have had prior involvement with the property that is the subject of the Technical Report, as an author of an internal "Report on Resources" dated December 16, 2009 (private internal report for AuEx).
- 9. As of the date of this certificate, to the best of my knowledge, information and belief, the Technical Report contains all the scientific and technical information that is required to be disclosed to make the Technical Report not misleading.
- 10. I am not aware of any material fact or material change with respect to the subject matter of the Technical Report that is not reflected in the Technical Report, for which the omission to disclose would make the Technical Report misleading.
- 11. I am independent of the issuer applying all of the tests in Item 1.4 of National Instrument 43-101.
- 12. I have read National Instrument 43-101 and Form 43-101F1, and the Technical Report has been prepared in compliance with that instrument and form.
- 13. I consent to the filing of the Technical Report with any stock exchange and other regulatory authority and any publication by them, including electronic publication in the public company files on their websites accessible to the public, of the Technical Report.

Dated in Denver, Colorado, July 15, 2010

Signature of Co-Author		
Frank Daviess Principal Resource Geologist		
("Signed")	("Sealed")	
		(Sealed)



SRK Consulting (US), Inc.
NI43-101_Technical_Report_Resources_176700 040_West-Pequop_FNL_20100715 July, 2010

1361 Unpatented Lode Mining Claims situated in T. 35 & 36 N., R. 65 E., MDB&M.

Elko County, Nevada

Claims are Sorted by BLM Serial Number

	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	Elko
		Serial	County			Serial	County			Serial	County
	Names	(NMC #)	Doc#		Names	(NMC #)	Doc #		Names	(NMC #)	Doc#
1	IND 2	704738	358339	45	IND 83	705768	360145	81	IND 218	742437	389009
				46	IND 84	705769	360146	82	IND 219	742438	389010
2	IND 4	704740	358341	47	IND 85	705770	360147	83	IND 220	742439	389011
3	IND 5	704741	358342	48	IND 86	705771	360148	84	IND 221	742440	389012
4	IND 6	704742	358343	49	IND 87	705772	360149	85	IND 222	742441	389013
5	IND 7	704743	358344	50	IND 88	705773	360150	86	IND 223	742442	389014
6	IND 8	704744	358345	51	IND 89	705774	360151	87	IND 224	742443	389015
7	IND 9	704745	358346	52	IND 90	705775	360152	88	IND 225	742444	389016
8	IND 10	704746	358347	53	IND 91	705776	360153	89	IND 226	742445	389017
9	IND 11	704747	358348	54	IND 92	705777	360154	90	IND 227	742446	389018
10	IND 12	704748	358349	55	IND 93	705778	360155	91	IND 228	742447	389019
11	IND 13	704749	358350	56	IND 94	705779	360156	92	IND 229	742448	389020
12	IND 14	704750	358351	57	IND 95	705780	360157	93	IND 230	742449	389021
13	IND 15	704751	358352	58	IND 96	705781	360158	94	IND 231	742450	389022
14	IND 16	704752	358353	59	IND 97	705782	360159	95	IND 232	742451	389023
15	IND 17	704753	358354	60	IND 98	705783	360160	96	IND 233	742452	389024
16	IND 18	704754	358355	61	IND 99	705784	360161	97	IND 234	742453	389025
17	IND 19	704755	358356	62	IND 100	705785	360162	98	IND 235	742454	389026
18	IND 20	704756	358357	63	IND 101	705786	360163	99	IND 236	742455	389027
19	IND 21	704757	358358	64	IND 102	705787	360164	100	IND 237	742456	389028
20	IND 22	704758	358359	65	IND 103	705788	360165	101	IND 238	742457	389029
21	IND 23	704759	358360	66	IND 104	705789	360166	102	IND 239	742458	389030
22	IND 24	704760	358361	67	IND 105	705790	360167	103	IND 240	742459	389031
23	IND 25	704761	358362	68	IND 106	705791	360168	104	IND 241	742460	389032
24	IND 26	704762	358363	69	IND 107	705792	360169	105	IND 242	742461	389033
25	IND 27	704763	358364	70	IND 108	705793	360170	106	IND 243	742462	389034
26	IND 28	704764	358365					107	IND 244	742463	389035
27	IND 29	704765	358366	71	IND 127	705812	360189	108	IND 245	742464	389036
28	IND 30	704766	358367					109	IND 246	742465	389037
29	IND 31	704767	358368	72	IND 129	705814	360191	110	IND 247	742466	389038

	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	Elko
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30	IND 32	704768	358369					111	IND 248	742467	389039
31	IND 33	704769	358370	73	IND 131	705816	360193	112	IND 249	742468	389040
32	IND 34	704770	358371					113	IND 250	742469	389041
33	IND 35	704771	358372	74	IND 133	705818	360195	114	IND 251	742470	389042
34	IND 36	704772	358373					115	IND 252	742471	389043
				75	IND 135	705820	360197				
35	IND 73	705758	360135					116	PNG-221	755559	398909
36	IND 74	705759	360136	76	IND 137	705822	360199				
37	IND 75	705760	360137					117	PNG-223	755561	398911
38	IND 76	705761	360138	77	IND 139	705824	360201				
39	IND 77	705762	360139					118	PNG-225	755563	398913
40	IND 78	705763	360140	78	IND 141	705826	360203				
41	IND 79	705764	360141					119	PNG-227	755565	398915
42	IND 80	705765	360142	79	IND 143	705828	360205				
43	IND 81	705766	360143					120	PNG-229	755567	398917
44	IND 82	705767	360144	80	IND 217	742436	389008				
121	PNG-231	755569	398919	153	SM-20	806948	449999	198	SM-65	806993	450044
				154	SM-21	806949	450000	199	SM-66	806994	450045
122	PNG-233	755571	398921	155	SM-22	806950	450001	200	SM-67	806995	450046
				156	SM-23	806951	450002	201	SM-68	806996	450047
123	PNG-235	755573	398923	157	SM-24	806952	450003	202	SM-69	806997	450048
				158	SM-25	806953	450004	203	SM-70	806998	450049
124	PNG-237	755575	398925	159	SM-26	806954	450005	204	SM-71	806999	450050
				160	SM-27	806955	450006	205	SM-72	807000	450051
125	PNG-240	755578	398928	161	SM-28	806956	450007				
				162	SM-29	806957	450008	206	SM-82	807010	450061
126	PNG-242	755580	398930	163	SM-30	806958	450009				
				164	SM-31	806959	450010	207	SM-84	807012	450063
127	PNG-244	755582	398932	165	SM-32	806960	450011				
				166	SM-33	806961	450012	208	SM-86	807014	450065
128	PNG-246	755584	398934	167	SM-34	806962	450013				
				168	SM-35	806963	450014	209	SM-88	807016	450067
129	PNG-248	755586	398936	169	SM-36	806964	450015				
				170	SM-37	806965	450016	210	SM-90	807018	450069
130	PNG-250	755588	398938	171	SM-38	806966	450017				
								. '			

	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	Elko
		Serial	County			Serial	County			Serial	County
	Names	(NMC #)	Doc#		Names	(NMC #)	Doc#		Names	(NMC #)	Doc #
				172	SM-39	806967	450018	211	SM-99	807027	450078
131	PNG-252	755590	398940	173	SM-40	806968	450019	212	SM-100	807028	450079
				174	SM-41	806969	450020	213	SM-101	807029	450080
132	PNG-254	755592	398942	175	SM-42	806970	450021	214	SM-102	807030	450081
				176	SM-43	806971	450022	215	SM-103	807031	450082
133	PNG-256	755594	398944	177	SM-44	806972	450023	216	SM-104	807032	450083
				178	SM-45	806973	450024	217	SM-105	807033	450084
134	SM-1	806929	449980	179	SM-46	806974	450025	218	SM-106	807034	450085
135	SM-2	806930	449981	180	SM-47	806975	450026	219	SM-107	807035	450086
136	SM-3	806931	449982	181	SM-48	806976	450027	220	SM-108	807036	450087
137	SM-4	806932	449983	182	SM-49	806977	450028				
138	SM-5	806933	449984	183	SM-50	806978	450029	221	SM-254	810873	453282
139	SM-6	806934	449985	184	SM-51	806979	450030				
140	SM-7	806935	449986	185	SM-52	806980	450031	222	SM-256	810875	453284
141	SM-8	806936	449987	186	SM-53	806981	450032				
142	SM-9	806937	449988	187	SM-54	806982	450033	223	SM-258	810877	453286
143	SM-10	806938	449989	188	SM-55	806983	450034				
144	SM-11	806939	449990	189	SM-56	806984	450035	224	SM-260	810879	453288
145	SM-12	806940	449991	190	SM-57	806985	450036				
146	SM-13	806941	449992	191	SM-58	806986	450037	225	SM-262	810881	453290
147	SM-14	806942	449993	192	SM-59	806987	450038				
148	SM-15	806943	449994	193	SM-60	806988	450039	226	SM-264	810883	453292
149	SM-16	806944	449995	194	SM-61	806989	450040				
150	SM-17	806945	449996	195	SM-62	806990	450041	227	SM-266	810885	453294
151	SM-18	806946	449997	196	SM-63	806991	450042				
152	SM-19	806947	449998	197	SM-64	806992	450043	228	SM-268	810887	453296
229	SM-270	810889	453298	273	PQ 26	917746	545823	318	PQ 71	917791	545869
230	SM-271	810890	453299	274	PQ 27	917747	545824	319	PQ 72	917792	545870
231	SM-272	810891	453300	275	PQ 28	917748	545825	320	PQ 73	917793	545871
232	SM-273	810892	453301	276	PQ 29	917749	545826	321	PQ 74	917794	545872
233	SM-274	810893	453302	277	PQ 30	917750	545827	322	PQ 75	917795	545873
234	SM-275	810894	453303	278	PQ 31	917751	545828	323	PQ 76	917796	545874
235	SM-276	810895	453304	279	PQ 32	917752	545829	324	PQ 77	917797	545875
236	SM-277	810896	453305	280	PQ 33	917753	545830	325	PQ 78	917798	545876
237	SM-278	810897	453306	281	PQ 34	917754	545831	326	PQ 79	917799	545877

	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	Elko
	Claim	Serial	County		Claim	Serial	County		Claim	Serial	County
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238	SM-279	810898	453307	282	PQ 35	917755	545832	327	PQ 80	917800	545878
239	SM-280	810899	453308	283	PQ 36	917756	545833	328	PQ 81	917801	545879
240	SM-281	810900	453309	284	PQ 37	917757	545834	329	PQ 82	917802	545880
241	SM-282	810901	453310	285	PQ 38	917758	545835	330	PQ 83	917803	545881
242	SM-283	810902	453311	286	PQ 39	917759	545836	331	PQ 84	917804	545882
243	SM-284	810903	453312	287	PQ 40	917760	545837	332	PQ 85	917805	545883
244	SM-285	810904	453313	288	PQ 41	917761	545838	333	PQ 86	917806	545884
245	SM-286	810905	453314	289	PQ 42	917762	545839	334	PQ 87	917807	545885
246	SM-287	810906	453315	290	PQ 43	917763	545840	335	PQ 88	917808	545886
247	SM-288	810907	453316	291	PQ 44	917764	545841	336	PQ 89	917809	545887
				292	PQ 45	917765	545842	337	PQ 90	917810	545888
248	PQ 1	917721	545798	293	PQ 46	917766	545844	338	PQ 91	917811	545889
249	PQ 2	917722	545799	294	PQ 47	917767	545845	339	PQ 92	917812	545890
250	PQ 3	917723	545800	295	PQ 48	917768	545846	340	PQ 93	917813	545891
251	PQ 4	917724	545801	296	PQ 49	917769	545847	341	PQ 94	917814	545892
252	PQ 5	917725	545802	297	PQ 50	917770	545848	342	PQ 95	917815	545893
253	PQ 6	917726	545803	298	PQ 51	917771	545849	343	PQ 96	917816	545894
254	PQ 7	917727	545804	299	PQ 52	917772	545850	344	PQ 97	917817	545895
255	PQ 8	917728	545805	300	PQ 53	917773	545851	345	PQ 98	917818	545896
256	PQ 9	917729	545806	301	PQ 54	917774	545852	346	PQ 99	917819	545897
257	PQ 10	917730	545807	302	PQ 55	917775	545853	347	PQ 100	917820	545898
258	PQ 11	917731	545808	303	PQ 56	917776	545854	348	PQ 101	917821	545899
259	PQ 12	917732	545809	304	PQ 57	917777	545855	349	PQ 102	917822	545901
260	PQ 13	917733	545810	305	PQ 58	917778	545856	350	PQ 103	917823	545902
261	PQ 14	917734	545811	306	PQ 59	917779	545857	351	PQ 104	917824	545903
262	PQ 15	917735	545812	307	PQ 60	917780	545858	352	PQ 105	917825	545904
263	PQ 16	917736	545813	308	PQ 61	917781	545859	353	PQ 106	917826	545905
264	PQ 17	917737	545814	309	PQ 62	917782	545860	354	PQ 107	917827	545906
265	PQ 18	917738	545815	310	PQ 63	917783	545861	355	PQ 108	917828	545907
266	PQ 19	917739	545816	311	PQ 64	917784	545862	356	PQ 109	917829	545908
267	PQ 20	917740	545817	312	PQ 65	917785	545863	357	PQ 110	917830	545909
268	PQ 21	917741	545818	313	PQ 66	917786	545864	358	PQ 111	917831	545910
269	PQ 22	917742	545819	314	PQ 67	917787	545865			•	
270	PQ 23	917743	545820	315	PQ 68	917788	545866	359	PQ 287	936145	561203
271	PQ 24	917744	545821	316	PQ 69	917789	545867	360	PQ 288	936146	561204
272	PQ 25	917745	545822	317	PQ 70	917790	545868	361	PQ 289	936147	561205

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362	PQ 290	936148	561206	407	PQ 335	936193	561251	451	PQ 449	937149	561913
363	PQ 291	936149	561207	408	PQ 336	936194	561252	452	PQ 450	937150	561914
364	PQ 292	936150	561208	409	PQ 337	936195	561253	453	PQ 451	937151	561915
365	PQ 293	936151	561209	410	PQ 338	936196	561254	454	PQ 452	937152	561916
366	PQ 294	936152	561210	411	PQ 339	936197	561255	455	PQ 453	937153	561917
367	PQ 295	936153	561211	412	PQ 340	936198	561256	456	PQ 454	937154	561918
368	PQ 296	936154	561212	413	PQ 341	936199	561257	457	PQ 455	937155	561919
369	PQ 297	936155	561213	414	PQ 342	936200	561258	458	PQ 456	937156	561920
370	PQ 298	936156	561214	415	PQ 343	936201	561259	459	PQ 457	937157	561921
371	PQ 299	936157	561215	416	PQ 344	936202	561260	460	PQ 458	937158	561922
372	PQ 300	936158	561216	417	PQ 345	936203	561261	461	PQ 459	937159	561923
373	PQ 301	936159	561217	418	PQ 346	936204	561262				
374	PQ 302	936160	561218	419	PQ 347	936205	561263	462	PQ 574	937160	561924
375	PQ 303	936161	561219	420	PQ 348	936206	561264	463	PQ 575	937161	561925
376	PQ 304	936162	561220	421	PQ 349	936207	561265	464	PQ 576	937162	561926
377	PQ 305	936163	561221	422	PQ 350	936208	561266	465	PQ 577	937163	561927
378	PQ 306	936164	561222	423	PQ 351	936209	561267	466	PQ 578	937164	561928
379	PQ 307	936165	561223	424	PQ 352	936210	561268	467	PQ 579	937165	561929
380	PQ 308	936166	561224	425	PQ 353	936211	561269	468	PQ 580	937166	561930
381	PQ 309	936167	561225	426	PQ 354	936212	561270	469	PQ 581	937167	561931
382	PQ 310	936168	561226	427	PQ 355	936213	561271	470	PQ 582	937168	561932
383	PQ 311	936169	561227	428	PQ 356	936214	561272	471	PQ 583	937169	561933
384	PQ 312	936170	561228	429	PQ 357	936215	561273	472	PQ 584	937170	561934
385	PQ 313	936171	561229	430	PQ 358	936216	561274	473	PQ 585	937171	561935
386	PQ 314	936172	561230	431	PQ 359	936217	561275	474	PQ 586	937172	561936
387	PQ 315	936173	561231	432	PQ 360	936218	561276	475	PQ 587	937173	561937
388	PQ 316	936174	561232	433	PQ 361	936219	561277	476	PQ 588	937174	561938
389	PQ 317	936175	561233	434	PQ 362	936220	561278	477	PQ 589	937175	561939
390	PQ 318	936176	561234	435	PQ 363	936221	561279	478	PQ 590	937176	561940
391	PQ 319	936177	561235	436	PQ 364	936222	561280	479	PQ 591	937177	561941
392	PQ 320	936178	561236	437	PQ 365	936223	561281	480	PQ 592	937178	561942
393	PQ 321	936179	561237	438	PQ 366	936224	561282	481	PQ 593	937179	561943
394	PQ 322	936180	561238	439	PQ 367	936225	561283	482	PQ 594	937180	561944
395	PQ 323	936181	561239	440	PQ 368	936226	561284	483	PQ 595	937181	561945
396	PQ 324	936182	561240	441	PQ 369	936227	561285	484	PQ 596	937182	561946
397	PQ 325	936183	561241	442	PQ 370	936228	561286	485	PQ 597	937183	561947

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398	PQ 326	936184	561242	443	PQ 371	936229	561287	486	PQ 598	937184	561948
399	PQ 327	936185	561243	444	PQ 372	936230	561288	487	PQ 599	937185	561949
400	PQ 328	936186	561244	445	PQ 373	936231	561289	488	PQ 600	937186	561950
401	PQ 329	936187	561245	446	PQ 374	936232	561290	489	PQ 601	937187	561951
402	PQ 330	936188	561246	447	PQ 375	936233	561291	490	PQ 602	937188	561952
403	PQ 331	936189	561247	448	PQ 376	936234	561292	491	PQ 603	937189	561953
404	PQ 332	936190	561248	449	PQ 377	936235	561293	492	PQ 604	937190	561954
405	PQ 333	936191	561249					493	PQ 605	937191	561955
406	PQ 334	936192	561250	450	PQ 448	937148	561912	494	PQ 606	937192	561956
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495	PQ 607	937193	561957	539	PQ 560	937828	562743	583	PQ 408	941398	564075
496	PQ 608	937194	561958	540	PQ 561	937829	562744	584	PQ 409	941399	564076
497	PQ 609	937195	561959	541	PQ 562	937830	562745	585	PQ 410	941400	564077
498	PQ 610	937196	561960	542	PQ 563	937831	562746	586	PQ 411	941401	564078
499	PQ 611	937197	561961	543	PQ 564	937832	562747	587	PQ 412	941402	564079
500	PQ 612	937198	561962	544	PQ 565	937833	562748	588	PQ 413	941403	564080
501	PQ 613	937199	561963	545	PQ 566	937834	562749	589	PQ 414	941404	564081
502	PQ 614	937200	561964	546	PQ 567	937835	562750	590	PQ 415	941405	564082
503	PQ 615	937201	561965	547	PQ 568	937836	562751	591	PQ 416	941406	564083
504	PQ 616	937202	561966	548	PQ 569	937837	562752	592	PQ 417	941407	564084
505	PQ 617	937203	561967	549	PQ 570	937838	562753	593	PQ 418	941408	564085
506	PQ 618	937204	561968	550	PQ 571	937839	562754	594	PQ 419	941409	564086
507	PQ 619	937205	561969	551	PQ 572	937840	562755	595	PQ 420	941410	564087
508	PQ 620	937206	561970	552	PQ 573	937841	562756	596	PQ 421	941411	564088
509	PQ 621	937207	561971			•		597	PQ 422	941412	564089
510	PQ 622	937208	561972	553	PQ 378	941368	564045	598	PQ 423	941413	564090
511	PQ 623	937209	561973	554	PQ 379	941369	564046	599	PQ 424	941414	564091
512	PQ 624	937210	561974	555	PQ 380	941370	564047	600	PQ 425	941415	564092
513	PQ 625	937211	561975	556	PQ 381	941371	564048	601	PQ 426	941416	564093
514	PQ 626	937212	561976	557	PQ 382	941372	564049	602	PQ 427	941417	564094
515	PQ 627	937213	561977	558	PQ 383	941373	564050	603	PQ 428	941418	564095
516	PQ 628	937214	561978	559	PQ 384	941374	564051	604	PQ 429	941419	564096
Ì				560	PQ 385	941375	564052	605	PQ 430	941420	564097
517	PQ 538	937806	562721	561	PQ 386	941376	564053	606	PQ 431	941421	564098
518	PQ 539	937807	562722	562	PQ 387	941377	564054	607	PQ 432	941422	564099
519	PQ 540	937808	562723	563	PQ 388	941378	564055	608	PQ 433	941423	564100

ſ	Claim	BLM	Elko	1	Claim	BLM	Elko		Claim	BLM	Elko
	Claim	Serial	County		Claim	Serial	County		Ciaiii	Serial	County
	Names	(NMC #)	Doc #		Names	(NMC #)	Doc #		Names	(NMC #)	Doc #
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522	PQ 543	937811	562726	566	PQ 391	941381	564058	611	PQ 436	941426	564103
523	PQ 544	937812	562727	567	PQ 392	941382	564059	612	PQ 437	941427	564104
524	PQ 545	937813	562728	568	PQ 393	941383	564060	613	PQ 438	941428	564105
525	PQ 546	937814	562729	569	PQ 394	941384	564061	614	PQ 439	941429	564106
526	PQ 547	937815	562730	570	PQ 395	941385	564062	615	PQ 440	941430	564107
527	PQ 548	937816	562731	571	PQ 396	941386	564063	616	PQ 441	941431	564108
528	PQ 549	937817	562732	572	PQ 397	941387	564064	617	PQ 442	941432	564109
529	PQ 550	937818	562733	573	PQ 398	941388	564065	618	PQ 443	941433	564110
530	PQ 551	937819	562734	574	PQ 399	941389	564066	619	PQ 444	941434	564111
531	PQ 552	937820	562735	575	PQ 400	941390	564067	620	PQ 445	941435	564112
532	PQ 553	937821	562736	576	PQ 401	941391	564068	621	PQ 446	941436	564113
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534	PQ 555	937823	562738	578	PQ 403	941393	564070				
535	PQ 556	937824	562739	579	PQ 404	941394	564071	623	PQ 482	941438	564115
536	PQ 557	937825	562740	580	PQ 405	941395	564072	624	PQ 483	941439	564116
537	PQ 558	937826	562741	581	PQ 406	941396	564073	625	PQ 484	941440	564117
538	PQ 559	937827	562742	582	PQ 407	941397	564074	626	PQ 485	941441	564118
627	PQ 486	941442	564119	671	PQ 659	947506	569184	715	PQ 703	954790	573448
628	PQ 487	941443	564120	672	PQ 660	947507	569185	716	PQ 704	954791	573449
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630	PQ 489	941445	564122	674	PQ 662	947509	569187	718	PQ 706	954793	573451
631	PQ 490	941446	564123					719	PQ 707	954794	573452
632	PQ 491	941447	564124	675	PQ 663	954750	573408	720	PQ 708	954795	573453
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634	PQ 493	941449	564126	677	PQ 665	954752	573410	722	PQ 710	954797	573455
635	PQ 494	941450	564127	678	PQ 666	954753	573411	723	PQ 711	954798	573456
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637	PQ 496	941452	564129	680	PQ 668	954755	573413	725	PQ 713	954800	573458
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640	PQ 499	941455	564132	683	PQ 671	954758	573416	728	PQ 716	954803	573461
				684	PQ 672	954759	573417	729	PQ 717	954804	573462
641	PQ 629	947476	569154	685	PQ 673	954760	573418	730	PQ 718	954805	573463

Names	Ī	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	Elko
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	764							580165	841			
- /co PQ /co 954840 573498 803 PQ 1697 963054 580166 842 PQ 1736 963093 580205	765	PQ 753	954840	573498	803	PQ 1697	963054	580166	842	PQ 1736	963093	580205

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767	PQ 755	954842	573500	805	PQ 1699	963056	580168	844	PQ 1738	963095	580207
768	PQ 756	954843	573501	806	PQ 1700	963057	580169	845	PQ 1739	963096	580208
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770	PQ 1664	963021	580133	809	PQ 1703	963060	580172	848	PQ 1742	963099	580211
771	PQ 1665	963022	580134	810	PQ 1704	963061	580173	849	PQ 1743	963100	580212
772	PQ 1666	963023	580135	811	PQ 1705	963062	580174	850	PQ 1744	963101	580213
773	PQ 1667	963024	580136	812	PQ 1706	963063	580175	851	PQ 1745	963102	580214
774	PQ 1668	963025	580137	813	PQ 1707	963064	580176	852	PQ 1746	963103	580215
775	PQ 1669	963026	580138	814	PQ 1708	963065	580177				
776	PQ 1670	963027	580139	815	PQ 1709	963066	580178	853	PQ 1781	981723	591438
777	PQ 1671	963028	580140	816	PQ 1710	963067	580179		PQ 1781 (Amdmt)	981723	597005
778	PQ 1672	963029	580141	817	PQ 1711	963068	580180				
779	PQ 1673	963030	580142	818	PQ 1712	963069	580181	854	PQ 1783	981724	591439
780	PQ 1674	963031	580143	819	PQ 1713	963070	580182		PQ 1783 (Amdmt)	981724	597005
781	PQ 1675	963032	580144	820	PQ 1714	963071	580183				
782	PQ 1676	963033	580145	821	PQ 1715	963072	580184	855	PQ 1785	981725	591440
783	PQ 1677	963034	580146	822	PQ 1716	963073	580185		PQ 1785 (Amdmt)	981725	597005
784	PQ 1678	963035	580147	823	PQ 1717	963074	580186				
785	PQ 1679	963036	580148	824	PQ 1718	963075	580187	856	PQ 1787	981726	591441
786	PQ 1680	963037	580149	825	PQ 1719	963076	580188		PQ 1787 (Amdmt)	981726	597005
787	PQ 1681	963038	580150	826	PQ 1720	963077	580189				
788	PQ 1682	963039	580151	827	PQ 1721	963078	580190	857	PQ 1789	981727	591442
789	PQ 1683	963040	580152	828	PQ 1722	963079	580191		PQ 1789 (Amdmt)	981727	597005
790	PQ 1684	963041	580153	829	PQ 1723	963080	580192				
791	PQ 1685	963042	580154	830	PQ 1724	963081	580193	858	PQ 1790	981728	591443
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793	PQ 1687	963044	580156	832	PQ 1726	963083	580195				
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795	PQ 1689	963046	580158	834	PQ 1728	963085	580197		PQ 1791 (Amdmt)	981729	597005
796	PQ 1690	963047	580159	835	PQ 1729	963086	580198		· · · · · · · · · · · · · · · · · · ·		

797	PQ 1691	963048	580160	836	PQ 1730	963087	580199				
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861	PQ 1793	981731	591446	872	PQ 1804	981742	591457	883	PQ 1815	981753	591468
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	PQ 1794 (Amdmt)	981732	597005		PQ 1805 (Amdmt)	981743	597005		PQ 1816 (Amdmt)	981754	597005
863	PQ 1795	981733	591448	874	PQ 1806	981744	591459	885	PQ 1817	981755	591470
	PQ 1795 (Amdmt)	981733	597005		PQ 1806 (Amdmt)	981744	597005		PQ 1817 (Amdmt)	981755	597005
864	PQ 1796	981734	591449	875	PQ 1807	981745	591460	886	PQ 1818	981756	591471
	PQ 1796 (Amdmt)	981734	597005		PQ 1807 (Amdmt)	981745	597005		PQ 1818 (Amdmt)	981756	597005
865	PQ 1797	981735	591450	876	PQ 1808	981746	591461	887	PQ 1819	981757	591472
	PQ 1797 (Amdmt)	981735	597005		PQ 1808 (Amdmt)	981746	597005		PQ 1819 (Amdmt)	981757	597005
866	PQ 1798	981736	591451	877	PQ 1809	981747	591462	888	PQ 1820	981758	591473
	PQ 1798 (Amdmt)	981736	597005		PQ 1809 (Amdmt)	981747	597005		PQ 1820 (Amdmt)	981758	597005
867	PQ 1799	981737	591452	878	PQ 1810	981748	591463	889	PQ 1821	981759	591474
	PQ 1799 (Amdmt)	981737	597005	·	PQ 1810 (Amdmt)	981748	597005		PQ 1821 (Amdmt)	981759	597005
868	PQ 1800	981738	591453	879	PQ 1811	981749	591464	890	PQ 1822	981760	591475
	PQ 1800 (Amdmt)	981738	597005		PQ 1811 (Amdmt)	981749	597005		PQ 1822 (Amdmt)	981760	597005
869	PQ 1801	981739	591454	880	PQ 1812	981750	591465	891	PQ 1823	981761	591476
	PQ 1801 (Amdmt)	981739	597005	·	PQ 1812 (Amdmt)	981750	597005		PQ 1823 (Amdmt)	981761	597005
870	PQ 1802	981740	591455	881	PQ 1813	981751	591466	892	PQ 1824	981762	591477
	PQ 1802 (Amdmt)	981740	597005		PQ 1813 (Amdmt)	981751	597005		PQ 1824 (Amdmt)	981762	597005
893	PQ 1825	981763	591478	909	PQ 816	982297	591685	945	PQ 852	982333	591721
	PQ 1825 (Amdmt)	981763	597005	910	PQ 817	982298	591686	946	PQ 853	982334	591722
		T		911	PQ 818	982299	591687	947	PQ 854	982335	591723
894	PQ 1826	981764	591479	912	PQ 819	982300	591688	948	PQ 855	982336	591724

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895	PQ 1827	981765	591480	915	PQ 822	982303	591691	951	PQ 858	982339	591727
	PQ 1827 (Amdmt)	981765	597005	916	PQ 823	982304	591692	952	PQ 859	982340	591728
				917	PQ 824	982305	591693	953	PQ 860	982341	591729
896	PQ 1828	981766	591481	918	PQ 825	982306	591694	954	PQ 861	982342	591730
	PQ 1828 (Amdmt)	981766	597005	919	PQ 826	982307	591695	955	PQ 862	982343	591731
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897	PQ 1829	981767	591482	921	PQ 828	982309	591697	957	PQ 864	982345	591733
	PQ 1829 (Amdmt)	981767	597005	922	PQ 829	982310	591698	958	PQ 865	982346	591734
				923	PQ 830	982311	591699	959	PQ 866	982347	591735
898	PQ 1830	981768	591483	924	PQ 831	982312	591700	960	PQ 867	982348	591736
	PQ 1830 (Amdmt)	981768	597005	925	PQ 832	982313	591701	961	PQ 868	982349	591737
				926	PQ 833	982314	591702	962	PQ 869	982350	591738
899	PQ 1831	981769	591484	927	PQ 834	982315	591703	963	PQ 870	982351	591739
	PQ 1831 (Amdmt)	981769	597005	928	PQ 835	982316	591704	964	PQ 871	982352	591740
				929	PQ 836	982317	591705	965	PQ 872	982353	591741
900	PQ 1832	981770	591485	930	PQ 837	982318	591706	966	PQ 873	982354	591742
	PQ 1832 (Amdmt)	981770	597005	931	PQ 838	982319	591707	967	PQ 874	982355	591743
				932	PQ 839	982320	591708	968	PQ 875	982356	591744
901	PQ 1833	981771	591486	933	PQ 840	982321	591709	969	PQ 876	982357	591745
	PQ 1833 (Amdmt)	981771	597005	934	PQ 841	982322	591710	970	PQ 877	982358	591746
				935	PQ 842	982323	591711	971	PQ 878	982359	591747
902	PQ 1834	981772	591487	936	PQ 843	982324	591712	972	PQ 879	982360	591748
	PQ 1834 (Amdmt)	981772	597005	937	PQ 844	982325	591713	973	PQ 880	982361	591749
				938	PQ 845	982326	591714	974	PQ 881	982362	591750
903	PQ 810	982291	591679	939	PQ 846	982327	591715	975	PQ 882	982363	591751
904	PQ 811	982292	591680	940	PQ 847	982328	591716	976	PQ 883	982364	591752
905	PQ 812	982293	591681	941	PQ 848	982329	591717	977	PQ 884	982365	591753
906	PQ 813	982294	591682	942	PQ 849	982330	591718	978	PQ 885	982366	591754
907	PQ 814	982295	591683	943	PQ 850	982331	591719	979	PQ 886	982367	591755

908	PQ 815	982296	591684	944	PQ 851	982332	591720	980	PQ 887	982368	591756
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983	PQ 890	982371	591759	1026	PQ 932	983437	592595	1065	PQ 1788	990007	598737
984	PQ 891	982372	591760	1027	PQ 933	983438	592596				
985	PQ 892	982373	591761	1028	PQ 934	983439	592597	1066	PQ 935	993806	600222
986	PQ 893	982374	591762					1067	PQ 936	993807	600223
987	PQ 894	982375	591763	1029	PQ 989	983440	592598	1068	PQ 937	993808	600224
988	PQ 895	982376	591764	1030	PQ 990	983441	592599	1069	PQ 938	993809	600225
989	PQ 896	982377	591765	1031	PQ 991	983442	592600	1070	PQ 939	993810	600226
990	PQ 897	982378	591766	1032	PQ 992	983443	592601	1071	PQ 940	993811	600227
991	PQ 898	982379	591767	1033	PQ 993	983444	592602	1072	PQ 941	993812	600228
992	PQ 899	982380	591768	1034	PQ 994	983445	592603	1073	PQ 942	993813	600229
993	PQ 900	982381	591769	1035	PQ 995	983446	592604	1074	PQ 943	993814	600230
994	PQ 901	982382	591770	1036	PQ 996	983447	592605	1075	PQ 944	993815	600231
995	PQ 902	982383	591771	1037	PQ 997	983448	592606	1076	PQ 945	993816	600232
996	PQ 903	982384	591772	1038	PQ 998	983449	592607	1077	PQ 946	993817	600233
997	PQ 904	982385	591773	1039	PQ 999	983450	592608	1078	PQ 947	993818	600234
998	PQ 905	982386	591774	1040	PQ 1000	983451	592609	1079	PQ 948	993819	600235
999	PQ 906	982387	591775					1080	PQ 949	993820	600236
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1001	PQ 908	982389	591777	1042	PQ 790	983502	592830	1082	PQ 951	993822	600238
1002	PQ 909	982390	591778	1043	PQ 791	983503	592831	1083	PQ 952	993823	600239
1003	PQ 910	982391	591779	1044	PQ 792	983504	592832	1084	PQ 953	993824	600240
1004	PQ 911	982392	591780	1045	PQ 793	983505	592833	1085	PQ 954	993825	600241
1005	PQ 912	982393	591781	1046	PQ 794	983506	592834	1086	PQ 955	993826	600242
1006	PQ 913	982394	591782	1047	PQ 795	983507	592835	1087	PQ 956	993827	600243
1007	PQ 914	982395	591783	1048	PQ 796	983508	592836				
1008	PQ 915	982396	591784	1049	PQ 797	983509	592837	1088	PQ 1001	993828	600244
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1145	PQ 1122	993885	600301	1189	PQ 1751	993929	600164	1233	PQ 771	994396	600681
1146	PQ 1123	993886	600302	1190	PQ 1752	993930	600165	1234	PQ 772	994397	600682
1147	PQ 1124	993887	600303	1191	PQ 1753	993931	600166	1235	PQ 773	994398	600683
1148	PQ 1125	993888	600304	1192	PQ 1754	993932	600167	1236	PQ 774	994399	600684
1237	PQ 775	994400	600685	1281	PQ 988	994444	600745	1324	PQ 1093	994487	600788
1238	PQ 776	994401	600686					1325	PQ 1094	994488	600789
1239	PQ 777	994402	600687	1282	PQ 1029	994445	600746	1326	PQ 1095	994489	600790
1240	PQ 778	994403	600688	1283	PQ 1030	994446	600747	1327	PQ 1096	994490	600791
1241	PQ 779	994404	600689	1284	PQ 1031	994447	600748	1328	PQ 1097	994491	600792
1242	PQ 780	994405	600690	1285	PQ 1032	994448	600749	1329	PQ 1098	994492	600793
1243	PQ 781	994406	600691	1286	PQ 1033	994449	600750	1330	PQ 1099	994493	600794
1244	PQ 782	994407	600692	1287	PQ 1034	994450	600751	1331	PQ 1100	994494	600795
1245	PQ 783	994408	600693	1288	PQ 1035	994451	600752	1332	PQ 1101	994495	600796
1246	PQ 784	994409	600694	1289	PQ 1036	994452	600753	1333	PQ 1102	994496	600797
1247	PQ 785	994410	600695	1290	PQ 1037	994453	600754	1334	PQ 1103	994497	600798
1248	PQ 786	994411	600696	1291	PQ 1038	994454	600755	1335	PQ 1104	994498	600799
1249	PQ 787	994412	600697	1292	PQ 1039	994455	600756	1336	PQ 1105	994499	600800
				1293	PQ 1040	994456	600757	1337	PQ 1106	994500	600801
1250	PQ 957	994413	600714	1294	PQ 1041	994457	600758	1338	PQ 1107	994501	600802
1251	PQ 958	994414	600715	1295	PQ 1042	994458	600759	1339	PQ 1108	994502	600803

	Claim	BLM	Elko		Claim	BLM	Elko		Claim	BLM	
		Serial	County			Serial	County			Serial	
	Names	(NMC #)	Doc#		Names	(NMC #)	Doc #		Names	(NMC #)	
1252	PQ 959	994415	600716	1296	PQ 1043	994459	600760	1340	PQ 1109	994503	
1253	PQ 960	994416	600717	1297	PQ 1044	994460	600761	1341	PQ 1110	994504	
1254	PQ 961	994417	600718	1298	PQ 1045	994461	600762	1342	PQ 1111	994505	
1255	PQ 962	994418	600719	1299	PQ 1046	994462	600763	1343	PQ 1112	994506	
1256	PQ 963	994419	600720	1300	PQ 1047	994463	600764	1344	PQ 1113	994507	
1257	PQ 964	994420	600721	1301	PQ 1048	994464	600765	1345	PQ 1114	994508	
1258	PQ 965	994421	600722	1302	PQ 1049	994465	600766				
1259	PQ 966	994422	600723	1303	PQ 1050	994466	600767	1346	PQ 1126	994509	
1260	PQ 967	994423	600724	1304	PQ 1051	994467	600768	1347	PQ 1127	994510	
1261	PQ 968	994424	600725	1305	PQ 1052	994468	600769	1348	PQ 1128	994511	
1262	PQ 969	994425	600726	1306	PQ 1053	994469	600770	1349	PQ 1129	994512	
1263	PQ 970	994426	600727	1307	PQ 1054	994470	600771	1350	PQ 1130	994513	
1264	PQ 971	994427	600728	1308	PQ 1055	994471	600772	1351	PQ 1131	994514	
1265	PQ 972	994428	600729	1309	PQ 1056	994472	600773	1352	PQ 1132	994515	
1266	PQ 973	994429	600730	1310	PQ 1057	994473	600774	1353	PQ 1133	994516	
1267	PQ 974	994430	600731	1311	PQ 1058	994474	600775	1354	PQ 1134	994517	
1268	PQ 975	994431	600732	1312	PQ 1059	994475	600776	1355	PQ 1135	994518	
1269	PQ 976	994432	600733	1313	PQ 1060	994476	600777	1356	PQ 1136	994519	
1270	PQ 977	994433	600734					1357	PQ 1137	994520	
1271	PQ 978	994434	600735	1314	PQ 1083	994477	600778	1358	PQ 1138	994521	
1272	PQ 979	994435	600736	1315	PQ 1084	994478	600779	1359	PQ 1139	994522	
1273	PQ 980	994436	600737	1316	PQ 1085	994479	600780	1360	PQ 1140	994523	
1274	PQ 981	994437	600738	1317	PQ 1086	994480	600781	1361	PQ 1141	994524	
1275	PQ 982	994438	600739	1318	PQ 1087	994481	600782			End	
1276	PQ 983	994439	600740	1319	PQ 1088	994482	600783				
1277	PQ 984	994440	600741	1320	PQ 1089	994483	600784				
1278	PQ 985	994441	600742	1321	PQ 1090	994484	600785				
1279	PQ 986	994442	600743	1322	PQ 1091	994485	600786				

PQ 987

PQ 1092

Elko

County

Doc #

AuEx Ventures Inc. NI 43-101 Technical Report on Resources, "West Pequop Gold Exploration Project, Nevada, USA", effective date of March 23, 2010.

Dated this Thursday, July 15, 2010.

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QP Signature

Allan V. Moran Principal Geologist SRK Consulting (U.S.) Inc.