Economic Potential of Gold in Batouri (Eastern Cameroon)

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Abstract

The aim of this work is the valorization of the economic potential of gold in the Batouri region. The study is undertaken on five sites of which two alluvials (Djengo and Mongonam localities) made up of flat and river gold, two eluvials (Kambele and Bote) containing gold of veins in quarries, and one semi mechanized exploitation (METALICON) working on the two previous types. Laboratory works consist of traditional melting, determination of the various grades of gold through densimetry and spectrometry analysis and refining using the Miller Chlorination method. The main results from these analyses are: i) recovery concentration is low, (about 0.5 g/t) for the traditional mining and higher with the semi mechanization (1.5-2 g/t). Densimetry and spectrometry analyses show that gold of semi mechanized sites has an average grade of about 24 carats, 22 carats and 20 for alluvial and eluvial gold respectively. ii) For 26 kg of gold refined, a weight of 16.681 kg is obtained at a cost of 4 051 946 (four million fifty one thousand and nine hundred forty six) CFA. Spectrometry analyses reveal the presence of silver and copper impurities, elements that can still be valorized through the presence of a gold refining unit. Hence, the absence of a gold refining unit in our country leads to poor transformation of its ores and loss of devices.

Keywords: traditional mining exploitation, traditional melting, refining, gold grade, Batouri

1. Introduction

The diversity of the geological formations (plutonic, volcanic, metamorphic and sedimentary) in Cameroon give it a rich subsoil full of important and interesting minerals which, if well exploited within founding projects, will inescapably lead to an emerging country by 2035 (Strategy Document for Growth and Employment ECSD, 2009). In this respects, we distinguish base metals, precious stones and precious metals notably gold.

Nowaday, the exploitation of these minerals is not mechanized, but manually done. As for the gold ores, there are two (2) types: Artisanal mining in the strict sense and semi-mechanized artisanal mining.

Mining is one of the major domains on which Cameroon directs its emergence. Indeed, the sector whose research covered up to 40% of the country (Ntep Ngwet, 2001), contributed less than 1% to the Gross Domestic Product (GDP) until August 2014 (Cadre d’Appui et de Promotion de l’Artisanat Minier, CAPAM, 2015). But now with the legal provisions of August 1st 2014, and the joint order Ministry of Finance and Ministry of Mines and Technology Development (MINFI / MINMIDT) of June 1st 2015, which empowers the CAPAM to collect the ad valorem mineral tax and corporation tax in companies engaged in less mechanized artisanal mining, there is an amelioration of economic growth (CAPAM, 2015).

Extricate metallurgy is the art of extracting metals from the earth and purifying them (Larousse, 2007). In the processing chain of gold, refining or purification process of the impurities is the last step. In Cameroon, as in countries of the sub region, there is no refining unit. The gold mined in Cameroon is refined in Dubai. Transaction costs and transport, and the by-products from refining represent a loss for the state. In addition, technical operations and treatments used in panning of gold are archaic, require much physical effort, and cause much loss. Thus it is necessary to investigate the operations and the various transformations involved in the exploitation of gold in the Batouri locality.

The objective of this work is to enhance the economic potential of gold. This involves the identification of sites of artisanal mining of gold in Batouri, the description of the existing artisanal gold-mining techniques; the treatment techniques used in determining the titres of gold in various exploitation sites, the identification of shortcomings and propose eventual solutions.
2. Field Descriptions and Geological Setting

The Batouri area is located in the Eastern region of Cameroon (04° 21' to 04° 29' North and 14° 17' and 14° 29' East), within a humid tropical climatic zone. The geomorphology of the study area is a savanna plain with low altitudes (400–800 m). It covers an area of approximately 408 km². It comprises the localities of Djengo, Mongonam, Kambele, Bote and Metalicon (fig. 1), where samples were collected. The study area is located within the so-called « Eastern golden district » which is the most mineralized area in Cameroon. The metallogenic importance of the area is related to the geology of the whole region. It belong to the polycyclic granitic basement of Adamaoua (Toteu et al., 2004; Van Schmus et al., 2008), an intrusive syntectonic complex of N130 orientation. On the petrographic point of view, it comprises alkali granite, calc-alkali granite and charnockites; with minor quartz-syenite, syenite, quartz-diorite and gabbros. In these rocks, they are veins and lenticular bodies of quartz containing primary gold. The secondary gold (eluvial and alluvial) of Batouri is probably their alteration product and erosion.

3. Material and Methods

The data collection phase was done through field campaign and laboratory works. The field campaign comprised the choice of study sites, the identification of various sites and the presentation of list of questions to stakeholders. Samples used were collected after the washing step. Laboratory work involved traditional melting of gold and the determination of different grades of gold by densimetry using by the principle of Pushed of Archimedes and X-ray fluorescence spectrometry using SKY Ray instrument, EDX pocket III model.

The concentration criterion is given by equation (1):

$$C = \frac{6_L - 6_p}{6_L - 6_p}$$

Where $6_L$ represents the density of the heavy solid phase, $6_L$ the density of the light solid phase, and $6_p$ the density of the liquid used. Water is usually the liquid in which the particles are washed.

In the laboratory, artisanal gold melting and determination of different grades of gold by densimetry and X-ray-fluorescence spectrometry were done on 22 samples.

The different losses were calculated using the following formulas:
- Loss after acid attack denoted \( P_a \):
  \[ P_{a} = P_1 - P_2 \]  
  (2)

- Loss rate denoted \( T_a \):
  \[ T_{a} = \frac{P_{a} \times 100}{P_1} \]  
  (3)

- Loss on fusion denoted \( P_f \):
  \[ P_{f} = P_2 - P_3 \]  
  (4)

- Loss rate denoted \( T_f \):
  \[ T_{f} = \frac{P_{f} \times 100}{P_1} \]  
  (5)

- Absolute loss denoted \( P_T \):
  \[ P_{T} = P_1 - P_3 \]  
  (6)

- Loss rate denoted \( T_G \):
  \[ T_{G} = \frac{P_{T} \times 100}{P_1} \]  
  (7)

The determination of the gold titre by densimetry is illustrated by the following formulas:

\[
d_{Au} = \frac{\rho_{Au}}{\rho_e} = \frac{\rho_{Au} \cdot V_e}{\rho_e \cdot V_e} = \frac{m_{Au}}{m_e} \]

(7.1)

\( m_{Au} = \rho_{Au} \cdot V_e \) Where \( V_e \) corresponds to the volume of water displaced, and also equal to the volume of gold bullion. 

\( V_e = V_{Au} = V_t \) and \( m_{Au} \neq m_t \)

\[
d_{t} = \frac{\rho_t \cdot V_e}{\rho_e \cdot V_e} = \frac{m_l}{m_e} \]

(7.2)

\[
\frac{d_{Au}}{dt} = \frac{m_{Au}}{m_e} \cdot \frac{m_e}{m_l} = \frac{m_{Au}}{m_l} \]

(7.3)

\[
\frac{d_{Au}}{dt} \times 100 = \frac{m_{Au}}{m_l} \times 100 = \%Au \]

(7.4)

24 carats \[ 100 \]

X carats \[ \% Au \]

Where:
- \( X = \) titre of gold bullion obtained  
- \( d_{Au} = \) relative density of pure gold  
- \( m_{Au} = \) mass of pure gold  
- \( m_t = \) mass of gold bullion obtained after weighing  
- \( \rho_{Au} = \) density of pure gold  
- \( \rho_e = \) density of water

X-ray fluorescence spectrometry permitted the determination of the gold titre using the equation below:

\[
CARAT = \frac{\%Au \times 24}{100} \]

(9)

Where \( \%Au = \) Percent gold

Concerning refining, the technique involves chlorination of raw gold by the Miller process where pure gold is obtained at 99.5% before undergoing a second treatment, that of the Wohlwill’s electrolysis where pure gold is obtained at 99.99%.

4. Results

The use of these various methods enabled us to obtain the results presented in the following lines.

Table 1 summarizes the sites visited and the various gold grades obtained. The five sites employ about 350 craftsmen, with Djengo and Bote presenting the highest number (about 100 each). The depth of sites varies from 5 to 12 m and the concentration 0.24 to 2 g/t. The METALICON site is thickest and presents the highest concentration.
Table 1. Site characteristics and corresponding gold grades

<table>
<thead>
<tr>
<th>Exploitation sites</th>
<th>Type of exploitation</th>
<th>Geographical coordinates</th>
<th>COMMENTS</th>
</tr>
</thead>
</table>
| DJENGO             | Artisanal            | N04°27'23.2'' E014°24'37.9'' 617m | - Number of craftsmen: about 100  
                      |                      |                          | - Operation "gold flats"  
                      |                      |                          | - Depth: 5m  
                      |                      |                          | - Particle size: <5 mm  
                      |                      |                          | - Concentration: about 0.32 g/t |
| MONGONAM           | Artisanal            | N04°26'44.3'' E014°24'47.9'' 628m | - Number of craftsmen: 50.  
                      |                      |                          | - Operation: "water gold"  
                      |                      |                          | - Particle size <5 mm  
                      |                      |                          | - Concentration: about 0.36 g/t |
| KAMBELE            | Artisanal            | N04°28'42.3'' E014°25'47.9'' 628m | - Number of craftsmen: 70  
                      |                      |                          | - Operation "gold quarry"  
                      |                      |                          | - Depth: 10 m  
                      |                      |                          | - Particle size <5 mm  
                      |                      |                          | - Concentration: about 0.28 g/t |
| BOTE               | Artisanal            | N04°27'25.1'' E014°25'47.2'' 649m | - Number of craftsmen: 100  
                      |                      |                          | - Operation: "gold quarry"  
                      |                      |                          | - Depth: between 8 and 12 m  
                      |                      |                          | - Particle size <5 mm  
                      |                      |                          | - Concentration: about 0.24 g/t |
| METALICON          | semi mechanized      | N04°26'25.1'' E014°26'47.9'' 640m | - Cameroonian employees 30  
                      |                      |                          | - Excavation shovels: 03  
                      |                      |                          | - Washing units 02  
                      |                      |                          | - Depth: about 12 m  
                      |                      |                          | - Concentration: 1.5-2 g/t |

The results obtained on the different loss rates are presented in Table 2:

Table 2. Results obtained on the different loss rates

<table>
<thead>
<tr>
<th>Exploitation site</th>
<th>Gold type</th>
<th>Initial weight in laboratory $P_1$ (g)</th>
<th>Weight obtained after melting $P_2$ (g)</th>
<th>Absolute loss $P_3=P_1-P_2$ (g)</th>
<th>Loss rate (%)</th>
</tr>
</thead>
<tbody>
<tr>
<td>DJENGO</td>
<td>Alluvial &quot;water gold&quot;</td>
<td>210.56</td>
<td>195.01</td>
<td>15.55</td>
<td>7.38</td>
</tr>
<tr>
<td>MONGONAM</td>
<td>Alluvial &quot;flat gold&quot;</td>
<td>22.94</td>
<td>19.81</td>
<td>3.13</td>
<td>13.64</td>
</tr>
<tr>
<td>BOTE</td>
<td>Eluvial &quot;Quarry gold&quot;</td>
<td>500.15</td>
<td>40512</td>
<td>95.03</td>
<td>19.00</td>
</tr>
<tr>
<td>KAMBELE</td>
<td>Eluvial &quot;Quarry gold&quot;</td>
<td>53.94</td>
<td>44.23</td>
<td>9.71</td>
<td>18.01</td>
</tr>
<tr>
<td>METALICON</td>
<td>Semi-mechanized</td>
<td>218.20</td>
<td>209.47</td>
<td>8.73</td>
<td>4.02</td>
</tr>
</tbody>
</table>
The different loss rates in Batouri Region

The eluvial gold presents the highest loss rate on melting with values amounting 19%, on the contrary, alluvial gold which yields an average loss rate of 7%. It should be noticed that semi mechanization reduces loss rate to 4.02% (fig. 2).

Table 3 represents the results obtained by densimetry and table 4 those through X-ray Fluorescence Spectrometry.

In Batouri alluvial gold has a mean titre of 22 carats, 20 carats for eluvial gold, both artisanal and approximately 24 carats for the semi-mechanized gold (table 3). The impurities in gold are silver and copper that can still be valorised (table 4).

However, refining is not yet being practiced in Cameroon. Molten gold was refined in Dubai and the results are presented in the table 5.

Table 3. Gold titre of different exploitation sites

<table>
<thead>
<tr>
<th>Exploitation site</th>
<th>Gold type</th>
<th>Weight of boullion</th>
<th>% Au</th>
<th>V_e = V_Au (cm³)</th>
<th>Titre (carats)</th>
</tr>
</thead>
<tbody>
<tr>
<td>DJENGO</td>
<td>alluvial &quot;water gold&quot;</td>
<td>195.01</td>
<td>93.21</td>
<td>9.38</td>
<td>22.37</td>
</tr>
<tr>
<td>MONGONAM</td>
<td>alluvial &quot;flat gold&quot;</td>
<td>19.81</td>
<td>91.71</td>
<td>0.94</td>
<td>22.01</td>
</tr>
<tr>
<td>BOTE</td>
<td>eluvial &quot;Quarry gold&quot;</td>
<td>405.12</td>
<td>85.08</td>
<td>17.83</td>
<td>20.42</td>
</tr>
<tr>
<td>KAMBELE</td>
<td>Eluvial &quot;Quarry gold&quot;</td>
<td>44.23</td>
<td>83.75</td>
<td>1.92</td>
<td>20.10</td>
</tr>
<tr>
<td>METALICON</td>
<td>Semi-mechanized</td>
<td>209.47</td>
<td>96.17</td>
<td>10.42</td>
<td>23.08</td>
</tr>
</tbody>
</table>

Table 4. Gold analytical results in the various exploitation sites

<table>
<thead>
<tr>
<th>Exploitation site</th>
<th>Weight (g)</th>
<th>Cu (%)</th>
<th>Zn (%)</th>
<th>Au (%)</th>
<th>Ag (%)</th>
<th>Titre (carat)</th>
</tr>
</thead>
<tbody>
<tr>
<td>DJENGO « alluvial gold »</td>
<td>Test 1</td>
<td>195.01</td>
<td>0.41</td>
<td>0.00</td>
<td>93.49</td>
<td>6.10</td>
</tr>
<tr>
<td>Test 2</td>
<td>195.01</td>
<td>0.39</td>
<td>0.00</td>
<td>94.18</td>
<td>5.43</td>
<td>22.60</td>
</tr>
<tr>
<td>Average</td>
<td>195.01</td>
<td>0.40</td>
<td>0.00</td>
<td>93.84</td>
<td>5.76</td>
<td>22.52</td>
</tr>
<tr>
<td>MONGONAM « alluvial gold »</td>
<td>Test 1</td>
<td>19.81</td>
<td>0.00</td>
<td>0.00</td>
<td>92.06</td>
<td>7.94</td>
</tr>
<tr>
<td>Test 2</td>
<td>19.81</td>
<td>0.00</td>
<td>0.00</td>
<td>91.88</td>
<td>8.12</td>
<td>22.05</td>
</tr>
<tr>
<td>Average</td>
<td>19.81</td>
<td>0.00</td>
<td>0.00</td>
<td>91.97</td>
<td>8.28</td>
<td>22.07</td>
</tr>
<tr>
<td>BOTE « eluvial gold »</td>
<td>Test 1</td>
<td>405.12</td>
<td>4.10</td>
<td>0.00</td>
<td>85.06</td>
<td>10.84</td>
</tr>
<tr>
<td>Test 2</td>
<td>405.12</td>
<td>3.22</td>
<td>0.00</td>
<td>86.38</td>
<td>10.40</td>
<td>20.73</td>
</tr>
<tr>
<td>average</td>
<td>405.12</td>
<td>3.66</td>
<td>0.00</td>
<td>85.72</td>
<td>10.62</td>
<td>20.57</td>
</tr>
<tr>
<td>KAMBELE « eluvial gold »</td>
<td>Test 1</td>
<td>44.23</td>
<td>0.00</td>
<td>0.00</td>
<td>82.98</td>
<td>17.02</td>
</tr>
<tr>
<td>Test 2</td>
<td>44.23</td>
<td>0.00</td>
<td>0.00</td>
<td>84.94</td>
<td>15.06</td>
<td>20.39</td>
</tr>
<tr>
<td>Average</td>
<td>44.23</td>
<td>0.00</td>
<td>0.00</td>
<td>83.96</td>
<td>16.04</td>
<td>20.15</td>
</tr>
<tr>
<td>METALICON « Semi mecanised »</td>
<td>Test 1</td>
<td>209.47</td>
<td>0.00</td>
<td>0.00</td>
<td>96.22</td>
<td>3.78</td>
</tr>
<tr>
<td>Test 2</td>
<td>209.47</td>
<td>0.00</td>
<td>0.00</td>
<td>96.36</td>
<td>3.64</td>
<td>23.13</td>
</tr>
<tr>
<td>Average</td>
<td>209.47</td>
<td>0.00</td>
<td>0.00</td>
<td>96.29</td>
<td>3.71</td>
<td>23.11</td>
</tr>
</tbody>
</table>
Table 5. Results obtained for gold refining in Dubai (Capam, 2011)

<table>
<thead>
<tr>
<th>Amount of refined gold (kg)</th>
<th>Quantity obtained after refining (kg)</th>
<th>Unit Cost (FCFA)</th>
<th>Total cost (FCFA)</th>
<th>Titre of gold obtained in (%)</th>
<th>Total Loss (kg)</th>
</tr>
</thead>
<tbody>
<tr>
<td>26,000</td>
<td>16,681</td>
<td>155 844</td>
<td>4 051 946</td>
<td>99,5</td>
<td>9, 319</td>
</tr>
</tbody>
</table>

For 26 Kg of gold refined in Dubai, a weight of 16.681 Kg was obtained at a cost of 4,051,946 FCFA, averaging 10 kg loss that could contribute to an increase in the GDP (Gross Domestic product, table 5).

5. Discussion

The results from densimetric and X-ray fluorescent spectrometric methods are slightly different. This could probably be due to the fact that the densimetric method is quite complex and sometimes poses difficulties in reading the balance. The X-ray fluorescence spectrometer has the advantage that it gives not only the chemical composition of the analyzed metal highlighting the associated elements of the gold, but also, the results are more reliable.

The values of titres for alluvial gold were higher than those of the eluvial gold. Furthermore, the lower the carat, the higher the loss rate. This is due to the fact that the eluvial gold grains often contain small boulders difficult to separate, and the presence of poorly washed ferromagnesian particles constitute impurities which in the course of melting are dissolved in nitric acid, thus increasing their loss rate.

Washing techniques used in Batouri (pan and sluice) lead to considerable loss rates of over 50%.

Calculation of the concentration criterion C in the Batouri site

We have $\delta L = 15 - 19.3$; Take $\delta L = 16; \delta l = 2.65$ (about 3 for some gangue minerals)

$\delta p = 1$

Finally $C = 7.5$

This concentration criterion was compared to Tagart’s criterion table (Weiss, 1985, table 6).

Table 6. Various values of the concentration criterion C or Tagart’s criterion (Weiss, 1985)

<table>
<thead>
<tr>
<th>Criterion</th>
<th>Applicability</th>
</tr>
</thead>
<tbody>
<tr>
<td>C &gt; 2.5</td>
<td>Easy separation</td>
</tr>
<tr>
<td>1.75 &lt; C &lt; 1.50</td>
<td>Possible separation for the particles of dimension &gt; to 0.2 mm</td>
</tr>
<tr>
<td>1.50 &lt; C &lt; 1.25</td>
<td>Possible separation for the particles of dimension &gt; to 1.65 mm</td>
</tr>
<tr>
<td>C &lt; 1.25</td>
<td>Almost impossible separation</td>
</tr>
</tbody>
</table>

The criterion C is significantly higher than 2.5. The Batouri gold is that of easy separation (Weiss, 1985). At first the concentration methods used on different sites (sluice and panning) which are physical processes based on gravity, will give satisfactory results.

During our investigations, we found that the gold washing methods result in large losses (more than 50%) while the actual recovery rate would be around 70%.

Most of the artisanal sluices used in Batouri contain neither riffles nor strips. These traps can undoubtedly improve metal recovery yield. These processes based on proven physical laws are used on an industrial scale, an equally important aspect of the slope of the sluice. In industrial practice, the slope is on average 5°. In the sites visited, this angle is between 13 and 15°. According to International Labour Organization (ILO, 2009), for maximum gold recovery, the construction of a washing ramp provides: a manifold equipped with a grating at the bottom and a flow channel of 30 to 50 cm width and about one to two metres long. This channel must be fitted with a wedged under carpet webs. The sluice must be disposed at an inclined angle plane between 4° and 6° to claim a recovery rate of about 80%.

Also, aluminum bowls or non-grooved calabash serving pan is the last stage of recovery of concentrates from Long Tom. But for a good recovery, Keita (2001) proposed to use dark grooved frames in sahelian region. He believes that the grooves can retain the gold nuggets and the colour for easy identification at night.

Hence, to reduce the strain at work and increase efficiency, this paper recommends to use the gold wash table designed and tested by the Precasem team described on figure 2:
Figure 2. Gold-washing table

<table>
<thead>
<tr>
<th>N°</th>
<th>Item</th>
<th>Long Tom Traditional</th>
<th>SLUICE (Long Tom enhanced)</th>
</tr>
</thead>
<tbody>
<tr>
<td>01</td>
<td>Number of washed gravel gold pans</td>
<td>52</td>
<td>52</td>
</tr>
<tr>
<td>02</td>
<td>gravel issue washed</td>
<td>0.402 m³</td>
<td>0.402 m³</td>
</tr>
<tr>
<td>03</td>
<td>Quantity</td>
<td>1 tonne</td>
<td>1 tonne</td>
</tr>
<tr>
<td>04</td>
<td>working time</td>
<td>05 hours</td>
<td>01 hour</td>
</tr>
<tr>
<td>05</td>
<td>Number of pump-motor</td>
<td>02</td>
<td>01</td>
</tr>
<tr>
<td>06</td>
<td>Quantity of gasoline used</td>
<td>04 Litres</td>
<td>02 Litres</td>
</tr>
<tr>
<td>07</td>
<td>Expenditure physical energy</td>
<td>Very high</td>
<td>Low</td>
</tr>
<tr>
<td>08</td>
<td>“GOLD” PRODUCTION</td>
<td>0.301 grams</td>
<td>0.750 grams</td>
</tr>
</tbody>
</table>

6. Conclusion

The present work had as main objective the enhancement of the economic potential of gold. For this, the study was conducted in five sites, two alluvial, two eluvial, and a semi-mechanized, combined with laboratory work consisting of artisanal gold melting and determination of different titres. It follows: at the extraction, a low recovery rate for craft sites 0.5 g / t in contrast to the semi mechanization 1.5-2 g / t. In the laboratory, the different loss rates were evaluated. Also, different titres of gold calculated by densimetric measurement and X-ray fluorescent spectrometric analysis show that semi mechanized sites have an average capacity approaching 24 carats, that of alluvial gold around 22carats and the eluvial gold around 20. At refining, for 26 kg of gold which were refined, a weight of 16.681 kg was obtained at a cost of 4,051,946 (four million fifty one thousand nine hundred and forty-six) FCFA. However, due to the lack of a gold refining unit, the state has a shortfall in processing these ores. Also, the organization of artisans in CIG (Common Initiative Group)
and their enlightening on modern recovery techniques will increase their production rate.

Acknowledgement
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References


Laws, decrees and orders related to this work

Order No. 064 / PM of 25 July 2003 Relative to the creation, organization and functioning of Small Scale Mining Support and Promotion Framework Unit (CAPAM).

Law No. 001 of 16 April 2001, bearing mining code.

Decree No. 2002/648 / PM of 26 March 2002 laying down the procedures for implementing Law No. 001 of April 16 on the Mining Code.

Law N° 2010/011 of 29 July 2010 amending and supplementing that of No. 001 of 16 April 2001 relating to the Mining Code.


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