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Intermittent waste heat recovery via ORC in coffee torrefaction

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Abstract

Coffee torrefaction is carried out by means of hot air at average temperature of 200–240°C and with intermittent cycles where a lot of heat is discharged from the stack. CHP systems have been investigated to provide heat to the process. However, much of the heat released in the process is from the afterburner that heats up the flue gas to higher temperatures to remove volatile organic compounds and other pollutants. In this paper, the techno-economic feasibility of utilising waste heat from a rotating drum coffee roasting with partial hot gas recycling is assessed. A cost analysis is adopted to compare the profitability of two systems configurations integrated into the process. The case study of a major coffee torrefaction firm with 500 kg/hr production capacity in the Italian energy framework is taken. The CHP options under investigation are: (i) regenerative topping micro gas turbine (MGT) coupled to the existing modulating gas burner to generate hot air for the roasting process; (ii) intermittent waste heat recovery from the hot flue gas through an organic Rankine cycle (ORC) coupled to a thermal storage buffer. The results show that the profitability of these investments is highly influenced by the natural gas/electricity cost ratio, by the coffee torrefaction production capacity and intermittency level of discharged heat. In this case study, MGT seems to be more profitable than waste heat recovery via ORC due to the intermittency of the heat source and the relatively high electricity/heat cost ratio.

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1. Introduction

Coffee roasting is a highly energy intensive process, requiring 11.2×10^{12} kJ/fuel energy annually, with 75% of the energy being wasted as heat through the stack [1,2]. Around 50% of industrial processes such as coffee roasting, sintering, dairy production and pharmaceuticals use batch processes to improve the quality and consistency of the products [3–6]. The drawback to batch processes is the substantial amount of waste heat emitted intermittently, preventing conventional methods from being used for recovery. Heat recovery of batch processes across industries has been investigated in the literature to reduce energy use [7–9], and heat storage systems have proven to be successful in recovering wasted heat [10–15].

The operation of batch gas-fired coffee roasters equipped with afterburners is well known and widely described in literature [16–19]. The methods include a heating air process, generally by means of a modulating gas burner, followed by passing the hot air through the coffee beans passively with a batch process that lasts around 10 minutes. The hot air usually travels from the roasting chamber through an afterburner to remove the VOCs and carbon

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monoxide and is finally discharged at high temperature into the atmosphere. This single-pass roasting method is often substituted to a semi-closed loop where part of the roaster gases are recycled and mixed to the hot air flow in order to roast the beans with the desired time-temperature curve. A big challenge for roasting is to rapidly heat the air before introducing it into the batch. To achieve this rapid heating, the roasters use a very energy intensive and quite low efficiency process. The afterburner is also quite energy intensive and releases the final gas at a very high temperature. In this paper, energy and material balances of a rotating drum coffee roasting with partial hot gas recycling and costs assessment methodologies are adopted to compare the profitability of two systems integrated into the process. The case study of a major coffee processing firm with 500 kg/hr production capacity and the Italian electricity/NG cost scenario are assumed to carry out the thermo-economic assessment. The CHP options under investigation are: (i) regenerative topping MGT coupled to the existing modulating gas burner to generate hot air for the roasting process; (ii) fluctuating waste heat recovery from the hot flue gas through an ORC with a thermal storage buffer. The sensitivity of investment profitability to the main techno-economic process parameters and in particular to the daily roasting operating hours and electricity/NG cost ratio is presented.

2. The coffee roasting process

The coffee roasting process is a chemical process that turns green coffee beans into beans that can be ground, brewed and consumed with a complex aroma and flavor. For a batch coffee roaster, the roasting happens in cycles as the green coffee beans enter the roasting drum and are heated to the desired temperature of 200–250 °C before being transferred to a cooling chamber [20,21]. A cycle lasts 10 to 20 minutes depending on the desired degree of roasting. Hot air is generated in a combustion chamber, usually fueled by NG, and passes through the roasting drum, heating the beans to the desired temperature. The first stage of coffee roasting involves the evaporation of water from the bean when it reaches temperatures between 160 °C and 190 °C [18,22–24]. After water removal, the bean undergoes a set of chemical reactions in pyrolysis, which give the coffee its final flavor and aroma at a temperature of 200–250 °C. During pyrolysis, there is also a release of volatile organic compounds, carbon dioxide, and particulate matter from the roasted chaff that are added to the flue gas in the roasting drum. When too many volatiles are released from the bean at temperature higher than 200 °C, the aroma of the bean can decrease. For this reason, the temperature gradient during the process must be constant. This is done by means of a modulating boiler that can control the output of heat into the roasting drum. Some of the flue gas is returned to the combustion chamber at 180–230 °C through a heat exchanger to utilize some of the waste heat and increase the efficiency of the burner. For the rest of the flue gases leaving the roasting drum, the VOCs, PM and CO₂ must be removed before being emitted to the environment. The flue gas goes through a cyclone to remove particulate and then an afterburner to burn pollutants that are released from the beans during the roasting phase and is then finally released through a chimney into the atmosphere at temperatures of 300–350 °C.

Due to the nature of the batch coffee roasting process, waste heat is released intermittently during each 10-min cycle. This intermittency prevents a constant flow of energy to recover and reutilize in systems such as an ORC. This can be done using a thermal storage system, either sensible or latent heat thermal storage. Sensible heat storage systems are most common and use a medium such as pressurized water. As the flue gas passes through the heat exchanger, its temperature drops, causing a potential issue of fouling. The VOCs and other compounds in the flue gas could condense on the heat exchanger, lowering the efficiency of the heat transfer.

The production site being investigated is a large coffee roasting firm that has roasting capacity of 500kg/hr. The firm operates 6 hours per day and 1,560 hours per year. There are two torrefaction units that roast 250 kg of beans per cycle with each cycle lasting 10 minutes. The modulating boiler and afterburner are fueled by NG. Each cycle requires 1.5 MJ of natural gas fuel per kg of beans, or 375 MJ per cycle. With the firm operating at 6 hours per day and 5 days a week, it requires 7020 GJ of natural gas and 800 MWh of electricity per year. The modulating boiler operates in the range 175–940 kW. The temperature of the exhaust gas from the stack is in the range 300–350 °C and the flow rate is 4500 Nm³/hr. This hot air flow rate is assumed constant through the system. A flow chart of the proposed torrefaction process is shown in Fig 1.

3. The scenarios under investigation

The roasting process is carried out at a relatively low temperature, hence it is possible to improve the overall efficiency recovering thermal energy from a topping MGT in cogenerative configuration or recovering the thermal energy content of the flue gas exiting the afterburner section via an ORC, as described in the following.

3.1. Case Study 1: Regenerative CHP-MGT

A possible solution where a topping MGT is installed before the fresh air inlet is reported in Fig. 2. Therefore the MGT exhaust gas is mixed with the recirculated stream and driven to the furnace, where the temperature is adjusted by means of the existing burner before entering the roasting drum. Assuming that the roasting process is carried out at average temperature of 200°C and a typical exhaust gas temperature of a regenerative gas turbine is 270°C the heat recovered can be calculated from:

$$Q = \dot{m}(h_{in} - h_{drum}) \quad (1)$$

where Q is the heat recovered; \dot{m} is the gas turbine exhaust gas flow rate; h_{in} is the enthalpy of the flue gas at the gas turbine outlet temperature; h_{drum} is the enthalpy of the flue gas at the drum temperature

Assuming the data of a commercial 200 kWe MGT, with exhaust gas flow rate of 1.3 kg/s and outlet temperature 270°C, and assuming the flue gas composition of Table 1, the heat recovered Q is 58.32 MJ in a 10 min cycle.

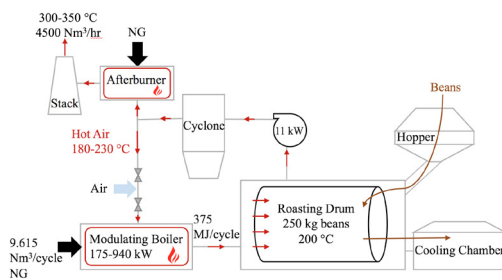


Fig. 1. Flow chart of the coffee torrefaction process.

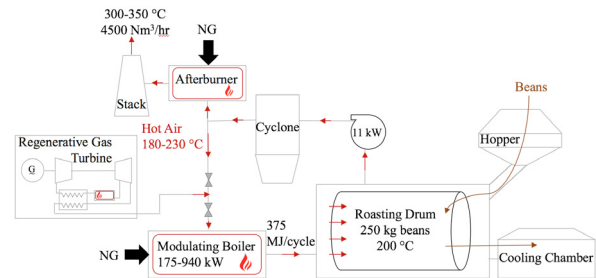


Fig. 2. Schematic of CHP-MGT in Case Study 1: Electricity from the MGT is fed to the grid or on site consumed and cogenerated heat is used for torrefaction process.

Assuming a heat demand of 1.5 MJ/kg of green beans and a processed mass of 250 kg, the heat required per cycle is 375 MJ. The MGT gas flow rate of about 3700 Nm³/h is coherent with the flow rate discharged into atmosphere in a commercial roasting process, which means a minor perturbation to the original process.

Case Study 2: Waste-heat recovery via ORC

Another viable option for the recovery of waste heat from the coffee torrefaction process is via an ORC system. The torrefaction process generally lasts 10 minutes after which the ‘torrefied’ coffee is extracted and the torrefaction air with coffee peel is discharged to a post burner. In the post burner, the coffee peel is further burnt to avoid emissions of coffee peels and the eventual flue gas exits the after-burner at around 350 °C and at a flow rate that varies linearly between 2,500 and 4,500 Nm³/h in 10 minutes. This variability requires a thermal energy storage system (TES) in the form of a pressurized-water loop (or other technologies such as hot-oil loop), to recover the discharged heat from the flue gas; the flue gas is then discharged to the environment at a lower temperature of 120 °C. This pressurized-water loop acts as the heat source to the (subcritical and recuperative) ORC system, ensuring a constant availability of heat to the ORC system. The schematic of the coffee torrefaction process with the ORC power system is shown in Fig. 3. A temporal energy balance around the heat recovery loop is described in Eqn (2), where T_{hr} , m_{hr} and $c_{p,hr}$ are the temperature, mass and specific heat capacity of the pressurized-water in the heat recovery loop. $\dot{Q}_{hr}(t)$ is the

(fluctuating) rate of heat transfer from the flue gas to the heat recovery loop; the profile is provided in Fig. 3. \dot{Q}_{ORC} is the (constant) heat transfer rate from the pressurized-water stream to the ORC working fluid in the ORC evaporator. The thermal storage system is designed (by suitably modifying m_{hr} and \dot{m}_{hr}) such that ~ 288 kW of heat is constantly supplied to the ORC system during the coffee torrefaction process. In all the heat exchangers, the minimum pinch temperature difference is assumed to be 10°C .

$$m_{\text{hr}} c_{p,\text{hr}} \frac{dT_{\text{hr}}}{dt} = \dot{Q}_{\text{ORC}} + \dot{Q}_{\text{hr}}(t), \text{ where } \dot{Q}_{\text{ORC}} = \dot{m}_{\text{hr}} c_{p,\text{hr}} (T_{\text{hr,sublv}} - T_{\text{hr,return}}). \quad (2)$$

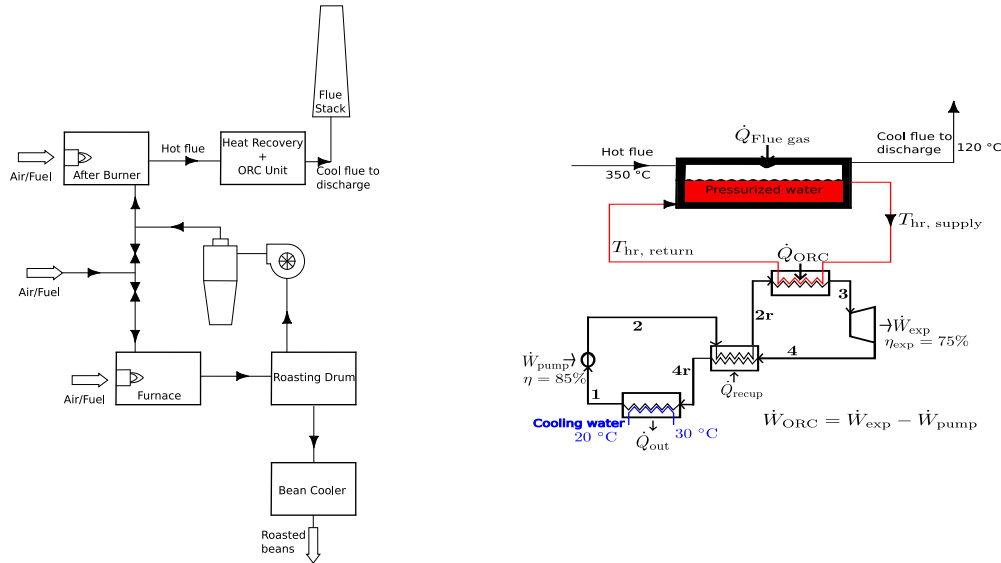


Fig. 3. Case Study 2: Waste heat recovery from flue gas via the organic Rankine cycle (ORC) system: LEFT – Schematic of coffee torrefaction process showing heat recovery and ORC power generating unit; RIGHT – Schematic of pressurized-water heat recovery unit and recuperative ORC engine.

A few of the common ORC working fluids were also considered here and are listed in Table 1, where the results of maximizing the net power output (\dot{W}_{ORC}) from the ORC system (at $T_{\text{hr,sublv}} = 120^\circ\text{C}$, $T_{\text{hr,return}} = 90^\circ\text{C}$) are presented. The ORC systems delivers in excess of 26 kWe. The inclusion of a recuperator in the ORC system is also seen to improve its efficiency as up to 98 kW of heat is ‘internally recovered’ within the ORC system.

Table 1. The maximum net power output and cycle design conditions of the ORC system.

| | $T_{\text{hr,sublv}} = 120^\circ\text{C}$ | | | | |
|-------------------------------|---|---------|--------|--------|---------|
| Working fluid | Butane | Pentane | R227ea | R245fa | R1234ze |
| \dot{W}_{ORC} (kW) | 26.2 | 25.9 | 26.6 | 26.3 | 26.3 |
| η_{ORC} (%) | 9.07 | 8.99 | 9.21 | 9.10 | 9.13 |
| \dot{Q}_{ORC} (kW) | 288 | 288 | 288 | 288 | 288 |
| \dot{Q}_{out} (kW) | 262 | 263 | 262 | 262 | 262 |
| \dot{Q}_{recup} (kW) | 44.8 | 43.1 | 58.3 | 42.8 | 31.9 |

Table 2. Main energy balance results and techno-economic data for the proposed case studies referred to a torrefaction cycle time of 10 min.

| | Case 1 | Case 2 |
|--|---------|---------|
| Plant size kW _e | 200 | 26 |
| NG saved Nm ³ /cycle | 1.65 | - |
| Electricity generated kWh/cycle | 33.33 | 4.3 |
| NG consumption Nm ³ /cycle | 11.31 | - |
| Saving Eur/cycle | 5.65 | 0.65 |
| total Cost Eur/cycle (fuel +O&M) | 4.75 | 0.06 |
| Balance Eur/cycle | 0.9 | 0.59 |
| Investment (Eur) | 200,000 | 120,000 |
| Pay back cycles | 222,222 | 203,390 |
| Pay back time (years) (6 hr/day operation) | 23.7 | 21.7 |

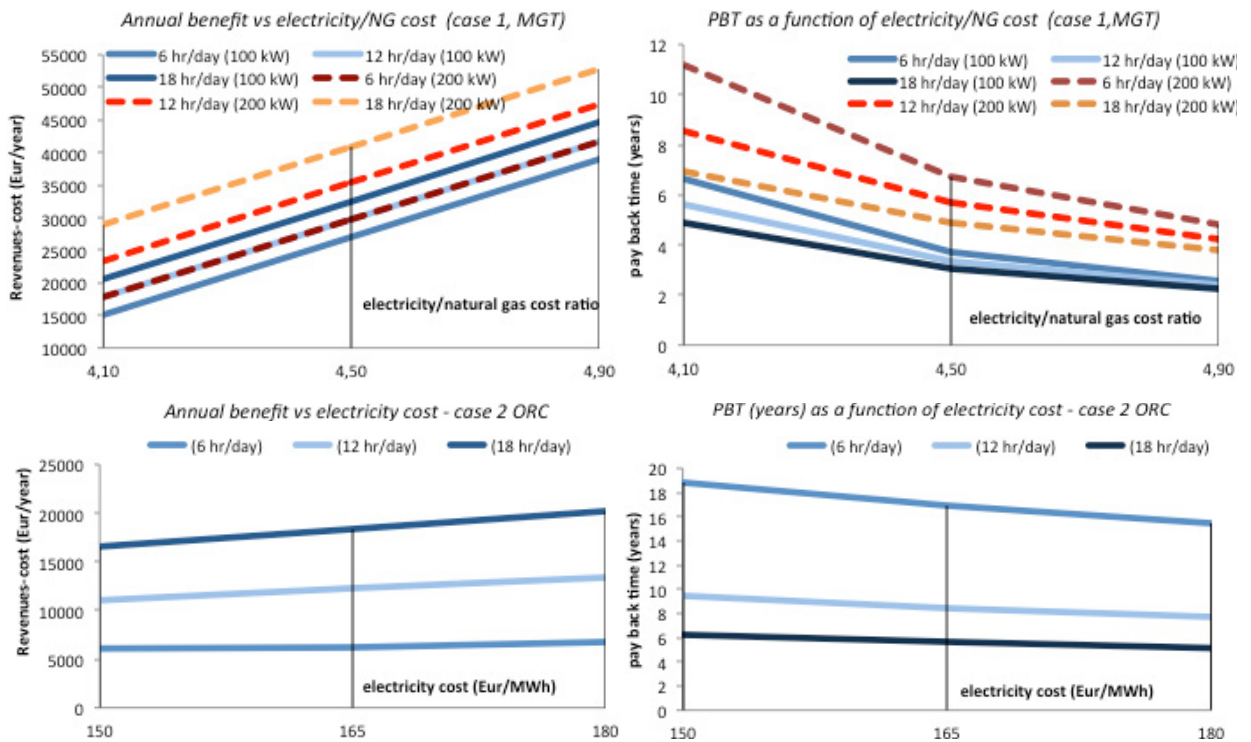


Fig. 4. Top: Profitability analysis results for Case Study 1 (as a function of electricity/NG cost and CHP size (100 and 200 kW respectively); the CHP operates baseload at 8,000-4,000 hours/year (100-200 kW size) and the heat demand is varied (1000, 2000, 3000 hr/year). Bottom: profitability analysis results for Case Study 2 (as a function of electricity cost and process working hours (hr/day) and ORC size (26 kW_e)

4. Profitability and sensitivity analysis

The annual electricity demand of the firm results of about 800 MWh with a total cost of supply of 150 kEur/year and a correspondent avoided cost for on site generated electricity of 150 Eur/MWh. The NG supply cost is 0.385 Eur/Nm³ with a LHV of 10.46 kWh/Nm³. O&M costs for all cases are assumed of 12 Eur/MWh of electricity generated. On the basis of the energy performance and investment/operational costs, a preliminary cost/benefit analysis is presented. Investment costs are assumed from manufacturers of MGT [25] and small scale ORC [26]. Table 2 reports the electricity output in each 10-min cycle and the correspondent NG consumption and savings for the three case studies. On the basis of fuel costs, O&M costs and investment costs, the economic balances are reported in Table 2, assuming the torrefaction capacity rate of the firm under investigation (36 cycles per day and

9,360 cycles per year). As can be seen, at the relatively low production capacity of the proposed case study, and with the assumed electricity and NG costs, the investment profitability is quite low. The results assume that the CHP plant operates only during the coffee torrefaction process to avoid discharged heat from cogeneration. The investment profitability increases when operating the CHP plant baseload and in net metering to the grid, in order to produce, on a yearly basis, the same amount of electricity consumed by the firm. With the assumed hypothesis, the highest profitability for case study 1 is achieved when the CHP electricity output on a yearly basis equals the on site electricity demand of the firm. In order to evaluate the influence of the electricity/natural gas cost ratio and of the torrefaction production capacity (and consequently heat demand for the process), a sensitivity of investment profitability of the case 1 (MGT in CHP mode) is proposed in Figure 4. The two sizes of 100 and 200 kW are compared (with operating hours of 8000 and 4000 h/year respectively) at different heat demand intensity (which correspond to different production capacity intensity rates, and different cycles per year). As can be seen, at high electricity/natural gas cost ratio and high production capacity intensity, the profitability of the investment increases, and the small scale MGT appears more profitable since it can better match the electricity demand of the firm operating for 8,000 hr/year baseload instead of 4,000 hr/year of the 200 kW size CHP. In the same figure, the sensitivity of investment profitability for case 2 (ORC) is reported, as a function of the torrefaction production capacity (respectively 6.12 and 18 working hours/day) and avoided cost of electricity, and for the 26 kWe ORC size. As can be seen, when increasing the operating hours of the MGT (case 1), the investment profitability increases and exceeds the one of the ORC, which is highly influenced by the production capacity.

4. Conclusions

In this paper, energy balances of a rotating drum coffee roasting with partial hot gas recycling and costs assessment methodologies are adopted to compare the profitability of three cogeneration systems integrated into the torrefaction process. The case study of a major coffee processing firm with 500 kg/hr production capacity is assumed to carry out the thermo-economic assessment. The CHP options under investigation are: (i) regenerative topping MGT coupled to the existing modulating gas burner to generate hot air for the roasting process and electricity for on site consumption; (ii) fluctuating waste heat recovery from the hot flue gas via a TES and organic Rankine cycle (ORC); (iii) not regenerative topping MGT with direct recovery of turbine outlet air for the roasting process by means of afterburner to modulate heat demand of roasting. MGTs are the typical choice for implementing CHP systems in these types of roasters, therefore ORCs are investigated to determine their feasibility of utilising more waste heat from the afterburner. The techno-economic results report a PBT for ORC ranging between 5 and 6 years in case of high production capacity (18 hr/day) and 15 to 18 years if the production capacity is low (6 hr/day). On the contrary, the benchmarking investment in a regenerative MGT operated baseload and in net metering to cover annual on site electricity demand presents a PBT in the range of 2 to 6 years (according to the electricity/NG cost ratio) and is more independent from the torrefaction production capacity. It can be concluded that, with the assumed hypotheses, the on site CHP via MGT (case 1 or 3) is more profitable than the heat recovery via ORC (case 2) whatever the production capacity level and the electricity/NG cost, and the relative profitability is strongly influenced by the electricity/gas cost ratio and heat demand intensity.

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Nomenclature

Symbols

| | |
|----------------|--|
| VOCs | volatile organic compounds |
| NG | natural gas |
| CHP | combined heat and power |
| MGT | micro gas turbine |
| ORC | organic Rankine cycle |
| PM | particulate matter |
| TES | Thermal Energy Storage |
| \dot{Q} | thermal power recovered from cycle (kJ/s) |
| \dot{m} | gas turbine mass flow rate (kg/s) |
| h_{drum} | enthalpy of flue gas in drum (kJ/kg) |
| h_{in} | enthalpy of flue gas at MGT outlet (kJ/kg) |
| R | heat capacity of flue gas (J/kg K) |
| ρ | density of flue gas (kg/Nm ³) |
| T_{hr} | T pressurised water in the TES (°C) |
| m_{hr} | mass of water in the TES (kg) |
| \dot{m}_{hr} | mass flow rate of water in TES (kg/s) |

| | |
|-----------------|--|
| $c_{p,hr}$ | specific heat capacity of pressurised water in TES (J/kg°C) |
| \dot{Q}_{hr} | intermittent rate of heat transfer from flue gas to TES (kJ/s) |
| \dot{Q}_{ORC} | constant heat transfer rate from pressurised water to ORC (kJ/s) |
| $T_{hr,supply}$ | supply temperature of water in TES (°C) |
| $T_{hr,return}$ | return temperature of water in TES (°C) |
| \dot{W}_{ORC} | net power output from ORC system (kW) |
| η_{ORC} | efficiency of ORC working fluid (%) |
| P_{evan} | ORC working fluid evaporator limits (bar) |
| P_{cond} | ORC working fluid condenser limits (bar) |
| LHV | lower heating value |