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Evaluation of Manganese Minerals for Chemical Looping Combustion

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Supporting Information

ABSTRACT: The use of mineral materials as oxygen carriers for Chemical Looping Combustion (CLC) is an attractive option due to their low cost. This paper reports an experimental study of four manganese minerals as potential oxygen carriers focusing on the behavior in CLC as well as in Chemical Looping with Oxygen Uncoupling (CLOU). Experiments were carried out in a thermogravimetric apparatus (TGA) and a fluidized-bed reactor. Repeated tests with all the minerals showed no sufficient CLOU properties. Then, they can only be used in CLC applications involving the redox pairs Mn_3O_4/MnO and Fe_2O_3/Fe_3O_4 . Oxygen transport capacity and reactivity to the main fuel gases (H_{22} , CO, and CH₄) were determined in a TGA. Manganese minerals were also tested during 35–54 h in a fluidized-bed reactor to evaluate the evolution of reactivity to H_{22} , CO, and CH₄ as well as their attrition rate and mechanical strength. The reactivity of the materials decreased in the first 10 cycles with CH₄ and then became quite stable for the rest of the cycles performed with CH₄, CO, and H₂. In comparison to previously tested Fe-based minerals, lower reactivity with CH₄ was found for the manganese minerals. However, in terms of CO and H₂ combustion, their reactivity was adequately high. During the tests, agglomeration and defluidization were never found for any material. Mechanical crushing strength of the particles decreased with cycles, which led to the increase of attrition rate of some materials above acceptable levels. Nevertheless, materials with adequate crushing strength and low attrition were identified. Combining the reactivity and attrition resistance, the manganese minerals MnSA and MnGBHNE from South Africa and Gabon, respectively, can be suggested as promising materials for CLC with coal.

1. INTRODUCTION

The use of fossil fuels as a primary source for energy production cannot be completely substituted in the coming years, while other alternatives, such as renewable energies, become mature.¹ In the process of energy transformation from fossil fuels, the CO₂ emission is a crucial issue, due to its impact on the greenhouse effect. In order to mitigate CO₂ emissions, clean coal technologies have been developed during the past years. Among these technologies, Chemical Looping Combustion (CLC) emerged as a low-cost technology for CO₂ capture.² A CLC system is mainly composed of two reactors: air reactor and fuel reactor. An oxygen carrier, normally a metal oxide (MeO_x) , circulates between the two reactors to provide the oxygen source for fuel (C_nH_{2m}) combustion, while the mixing of air and fuel is prevented. The oxygen carrier is reduced to MeO_{x-1} in the fuel reactor, via reaction R1, and then reoxidized to the oxygen-enriched form in the air reactor, via reaction R2. Subsequently, the regenerated oxygen carrier is returned to the fuel reactor allowing the continuous combustion of fuels. Ideally, the gas products from fuel reactor contain only CO_2 and H_2O (reaction R1). Since the separation of CO₂ from H₂O can be accomplished via a simple steam condensation, the CLC process can be considered as a low-cost technology with intrinsic CO₂ separation.

Fuel reactor:
$$C_n H_{2m} + (2n+m) MeO_r$$

$$\rightarrow (2n+m) \operatorname{MeO}_{x-1} + n \operatorname{CO}_2 + m \operatorname{H}_2 \operatorname{O}$$
(R1)

Air reactor:
$$MeO_{x-1} + 1/2O_2 \rightarrow MeO_x$$
 (R2)

In the case of using solid fuels in CLC, e.g. coal, reaction R1 between oxygen carrier and fuels becomes extremely slow.³ Therefore, two approaches were proposed to improve the combustion: *in situ* Gasification Chemical Looping Combustion $(iG\text{-}CLC)^4$ and Chemical Looping with Oxygen Uncoupling (CLOU).⁵ In the *iG*-CLC process, the solid fuel is directly introduced to fuel reactor, where char gasification by H₂O and/ or CO₂ and the reaction of the gasification products with oxygen carrier occur simultaneously. Considering the volatiles from the pyrolysis of coal, the main gases to be burnt in *iG*-CLC are CH₄, CO, and H₂. In the CLOU process a different mechanism is involved. In this case, the oxygen carrier releases gaseous oxygen for fuel combustion.

One of the concerns both in the iG-CLC and CLOU approaches is the losses of oxygen carrier that may be associated with the drainage of the ashes produced during



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solid fuel combustion which should be compensated with a makeup stream of new oxygen carrier. In order to minimize the operating costs associated with ash drainage, low-cost oxygen carriers are desirable in CLC for solid fuel combustion. In this context, iron-based and manganese-based minerals were considered during the last years as promising materials for CLC.⁶ In the case of iron-based minerals, experimental tests in continuous pilot units allowed the identification of several oxygen carriers for the further scale-up of the *i*G-CLC process.^{7–11} Among them ilmenite was extensively investigated and was recently considered for the scale-up to 50 kW-1 MW for coal combustion.^{8,11,12}

Manganese minerals are also attractive as oxygen carriers for CLC, although they received less attention than Fe-based materials in terms of the number of papers published. Theoretically, Mn₂O₃ contained in manganese minerals can release gaseous oxygen in the transformation Mn₂O₃/Mn₃O₄.⁵ However, with the presence of Fe_2O_3 together with Mn_2O_3 , the cases become different. A recent thermodynamic calculation suggests that the bixbyte in the Fe-Mn-O system is responsible for the release of gaseous oxygen.¹³ However, this component cannot be regenerated by air at temperatures higher than 900 °C.¹³ Therefore, in practical conditions a low CLOU effect was observed for manganese minerals.¹⁴ Moreover, the CLOU effect decreased with the number of cycles which can be attributed to the presence of Mn₃O₄ instead of Mn₂O₃ in the manganese material after several cycles.¹⁵ These findings imply that Mn₂O₃ cannot be regenerated at practical temperatures, and, therefore, manganese minerals seemed to be more suitable for gas fuelled CLC or iG-CLC for solid fuels, based on the redox pair Mn₃O₄/MnO.

Some investigations were carried out using manganese minerals for CH_4 , H_2 , or syngas combustion in a thermogravimetric apparatus (TGA) or in a fluidized-bed reactor.^{15–19} Results showed a high reactivity of manganese materials to H_2 and syngas;^{17,18} however, poor mechanical stability, poor fluidizing properties, and even agglomeration were also encountered.¹⁸ A recent investigation with a Brazilian manganese mineral in the combustion of CH_4 showed high conversion of CH_4 and good fluidization.¹⁵ Nevertheless, the reactivity of this mineral with CO and H_2 , the main gases involved in char gasification, could not be determined.¹⁵

The Brazilian manganese mineral was further used in experiments in a 10 kW_{th} CLC unit burning coal.²⁰ High combustion efficiencies and CO_2 captures were obtained. These good results were attributed to an improvement in the rates of char gasification in the presence of manganese minerals up to five times when compared to similar experiments using ilmenite.^{14,15,21} Despite this, high attrition rate and low mechanical strength were observed for the Brazilian manganese mineral during the operation in the continuous unit.²⁰ The high attrition rate of the manganese minerals might be the most important problem to overcome for the long-term use in CLC. In this sense, more research is necessary to develop manganese minerals with high reactivity and low attrition rate for the applications to CLC.

Thus, the objective of the present study is to contribute to the development of manganese minerals as potential oxygen carriers for CLC. With this aim, different manganese minerals were tested in experiments both using TGA and batch fluidized-bed reactor in order to analyze their reactivity with the main gases presented in CLC. Mechanical properties, including attrition, crushing strength, and agglomeration behavior were also examined.

2. EXPERIMENTAL SECTION

2.1. Manganese Minerals. Four different manganese minerals were tested in this work for the possible use as oxygen carriers in CLC. In order to identify them throughout this paper, they were denoted as MnSA, MnGBHNE, MnGBMPB, and MnBR depending on their origin countries: South Africa (SA), Gabon (GB), and Brazil (BR), respectively. Furthermore, in order to identify the two minerals from Gabon throughout this work, the supplier names were abbreviated in the denominations, i.e. HNE and MPB. The MnSA and MnGBHNE manganese minerals, respectively from South Africa and Gabon, were both supplied by Hidro Nitro Española S.A. The second sample from Gabon, MnGBMPB, was a commercially distributed manganese mineral provided by Mario Pilato Blatt S.A. Finally, the MnBR from Brazil is a similar material to that used elsewhere,^{14,15} which was supplied by Mineração Buritirama. Once received, they were dried at room temperature for 72 h to remove the humidity and then crushed and sieved to particles in the range of $100-300 \ \mu$ m. According to previous works,^{3,7,9,10,12} this particle size interval was considered as a desirable size to get better fluidization and reaction behaviors in fluidized bed reactors. More indeed, according to a recent experimental investigation²² a particle within 100-300 μ m showed satisfactory fluidization behavior in loop-seal involved in the CLC system. In addition, particle size in this range does not affect the reactivity of materials. Prior to the tests, all the samples were calcined in air at 800 °C for 2 h to ensure the complete oxidation.

The chemical composition of the materials was determined by inductively coupled plasma atomic emission spectroscopy (ICP-AES) with a Jobin Ybon 2000 apparatus. Crystalline chemical species were identified by powder X-ray diffraction (XRD) patterns acquired in an X-ray diffractometer Bruker AXS D8ADVANCE using Ni-filtered Cu $K\alpha$ radiation equipped with a graphite monochromator. The force needed to fracture a particle was determined using a Shimpo FGN-5X apparatus. The crushing strength of each material was taken as the average value of at least 20 measurements. The Air Jet Attrition Index (AJI) was calculated as the percentage of fines after a 5 h test in a Ma. Tec. ATTRI-AS air jet attrition tester, following the ASTM D5757-00 standard.^{23,24} The porosity and pore size of the particles were measured by Hg intrusion in a Quantachrome PoreMaster 33. The specific surface area of the particles was determined by the Brunauer-Emmett-Teller (BET) method through adsorption/desorption of nitrogen at 77 K in a Micromeritics ASAP-2020 instrument. Particle size distribution (PSD) of the materials was determined with a LS 13 320 of Beckman Coulter equipment. Finally, the microstructure of the materials was observed in an ISI DS-130 type scanning electron microscope (SEM). The main chemical and physical properties of all the manganese minerals mentioned are presented in Tables 1 and 2.

In Table 1, XRD results showed that the main crystalline phases are Mn_2O_3 , Mn_3O_4 , Fe_2O_3 , and the inactive SiO_2 in the calcined particles, depending on the materials. However, the Mn_2O_3 cannot be regenerated by air after reduction. In this case, the active composition of the manganese minerals in redox cycles was determined in TGA tests following the procedure detailed in the Supporting Information. Mn_3O_4 and Fe_2O_3 were considered as the stable phases as it is shown in Table 2. The oxygen transport capacity, R_{OC} , was calculated through the following equation

$$R_{\rm OC} = \frac{m_{\rm o} - m_{\rm r}}{m_{\rm o}} \tag{1}$$

where m_0 and m_r are the registered masses of fully oxidized and reduced manganese mineral in TGA tests, respectively. The crushing strength of these calcined materials varied from 1.0 to 4.6 N (Table 2) which is adequate for the utilization in fluidization systems based on a previous study.²⁵

2.2. Experimental Setup and Procedure. 2.2.1. Thermogravimetric Apparatus (TGA). The reactivity of the manganese minerals was first analyzed in a CI Electronics type TGA at atmospheric Table 1. Composition of the Four Manganese Minerals Calcined at 800 °C for 2 h^a

	MnSA	MnGBHNE	MnGBMPB	MnBR		
XRD Main Phases						
	Mn_2O_3 , Mn_3O_4	Mn_2O_3	Mn_2O_3	Mn ₂ O ₃ , Mn ₃ O ₄		
	Fe ₂ O ₃					
		SiO ₂	SiO ₂	SiO ₂		
Element Distribution (wt %)						
Mn	39.8 (39.9)	46.6 (47.1)	53.0 (56.0)	44.6 (50.3)		
Fe	14.6 (13.9)	5.1 (5.7)	3.6 (3.5)	5.2 (4.0)		
Si	3.3	4.2	1.6	4.1		
Al	0.2	3.8	2.8	4.1		
0	36.2	38.4	37.5	38.9		

^{*a*}Composition was determined by ICP-AES, except for numbers in brackets, which were determined by TGA, described in the Supporting Information. Minor amounts of Mg, Ti, Ca, Na, and K were also detected. Oxygen was determined by difference.

pressure. The TGA was well-defined to explore the reactivity and the available oxygen stored in the materials. A detailed description of the TGA apparatus can be found elsewhere.²⁶ In all the experiments, around 50 mg of the particles was loaded in a platinum basket. Mn-based materials are potentially valid for CLC and CLOU applications; therefore, the reactivity with gaseous fuels and the oxygen uncoupling capability were tested, respectively.

To evaluate the reactivity with gaseous fuels, the manganese materials were exposed to alternating reduction and oxidation cycles at 950 °C. During the reductions, 5% H₂ + 40% H₂O, 15% CO + 20% CO₂, and 15% CH₄ + 20% H₂O, balanced by N₂ respectively, were used as reducing mixtures. H₂O mixed with CH₄ and CO₂ mixed with CO were used to avoid possible carbon formation and to slow Boudouard reaction, respectively. When steam was desired in the reacting mixture, the gases were bubbled through a saturator containing water at the saturation temperatures 60 and 75 °C to reach the steam concentrations 20 and 40%, respectively. For the oxidations, air was used after the purge to regenerate the previously reduced materials.

The concentration of 15% for CO and CH₄ was used to facilitate the reactivity comparison between different materials following the methodology proposed by Johansson et al.²⁵ However, the H₂ concentration was 5% in this work to limit the reduction of iron oxide to Fe₃O₄ by using a low H₂ concentration joined to a high H₂O concentration.²⁷ Later, the reactivity was normalized to a concentration of 15% following the same methodology. An improved methodology to compare reactivity of different materials was proposed by Abad et al.,²⁸ but it requires the knowledge of the reaction order.

To evaluate the oxygen uncoupling capability, temperatureprogrammed decomposition and oxidation (TPD and TPO) as well as cycles at 800 °C were performed to evaluate the oxygen release and uptake. The TPD tests were carried out in N_2 with increasing temperature from room level to 950 °C, at 20 °C/min, whereas the TPO was performed in air and decreasing the temperature from 950 $^{\circ}C$ to the room level at 7 $^{\circ}C/min$. The isothermal cycles were performed at 800 $^{\circ}C$ in alternating N_2 and air. During the cycles, 800 $^{\circ}C$ was selected for oxidation according to the thermodynamic equilibrium of the Mn_2O_3/Mn_3O_4 system to allow the necessary oxidation to $Mn_2O_3.^5$

2.2.2. Batch Fluidized-Bed Reactor. Experiments in a fluidized-bed reactor allow the investigation of the gas product distribution during reactions and the attrition behavior of the manganese minerals as well as the possible agglomeration phenomena.

Figure 1 presents the scheme of the batch fluidized-bed reactor. A detailed description can be found elsewhere.^{29–31} For each mineral,



Figure 1. Setup of the batch fluidized-bed reactor.

300 g of manganese mineral was loaded in the reactor. All the tests were performed at 950 °C. After the temperature was reached, the material was exposed to alternating reducing and oxidizing cycles. Gases for the reduction, purge, and oxidation were supplied and introduced into the reactor through a 4-way valve upstream the reactor, to simulate a CLC process. The flows of these gases were controlled separately by mass flow meters, except the steam flow. In the case of steam, a peristaltic pump supplied the necessary water flow, which was heated to produce steam by a resistance heater before entering the reactor. For comparison purposes, the reducing gas flows and concentrations were chosen to have the same oxygen consumption. The reducing mixtures used in the experiments were $25\% \text{ CH}_4 + 10\% \text{ H}_2\text{O}$, $50\% \text{ CO} + 20\% \text{ CO}_2$, and $50\% \text{ H}_2 + 20\% \text{ H}_2\text{O}$, respectively, balanced by N_2 . In all the cases, the gas velocity was 6-16times the minimum fluidization velocity. Oxidations of the reduced samples were carried out after a period of purge in nitrogen. Diluted air containing 10% O2 was used to avoid high temperature increase during oxidation. Depending on the materials used, the reduction and oxidation of each material took 3-4 and 10 min, respectively. Because of the potential use of manganese minerals for CLOU application, the bed was purged with N2 for 5 min after oxidation to observe the possible oxygen release. Additionally, N2 and air were respectively used during reactor heating and cooling to observe the possible oxygen release and uptake at different temperatures. Two identical hot filters were located downstream the reactor, which were used to sequentially

Table 2. Active Component Composition and Properties of Calcined and Used Manganese Minerals

	MnSA	MnGBHNE	MnGBMPB	MnBR
redox composition (wt %) a	65.6 Mn ₃ O ₄	67.5 Mn ₃ O ₄	80.7 Mn ₃ O ₄	71.8 Mn ₃ O ₄
	18.6 Fe ₂ O ₃	8.4 Fe ₂ O ₃	5.2 Fe ₂ O ₃	6.0 Fe ₂ O ₃
bulk density (kg/m³)	3510 (1860) ^b	2570 (2290)	2900 (1690)	2230 (1800)
BET surface area (m^2/g)	0.6 (0.3)	10.1 (0.5)	7.4 (0.5)	12.2 (0.5)
porosity (vol %)	12.3 (28.2)	35.7 (25.4)	30.2 (32.9)	47.3 (31.7)
crushing strength (N)	4.6 (2.0)	1.8 (1.4)	2.4 (0.5)	1.0 (0.4)
AJI (%)	5.5 (14.4)	16.8 (14.6)	16.6 (22.1)	26.6 (11.8)
oxygen transport capacity (wt %)	4.7	5.0	5.8	5.2

^aDetermined by TGA, see the Supporting Information. ^bFor the used materials.

retain the solids elutriated during the tests. The filter was composed of a steel tube, a ceramic fiber, and some glass wools to guarantee the complete retention of fines at the outlet of reactor. In order to obtain the attrition rates of the particles, the amount of fines in the filters was recorded regularly. The differential pressure drop between the bottom and top of the reactor was measured to detect possible agglomeration problems of the materials. All the pipes and the filters were heated to avoid steam condensation in the pipe lines. The gas product stream was sent to several gas analyzers to determine the gas concentrations. The concentrations of CH_4 , CO, CO_2 , and H_2O were measured by a nondispersive infrared (NDIR) analyzer, and the concentration of H_2 was measured by a thermal conductivity detector (TCD). Oxygen concentration was measured using a paramagnetic analyzer.

2.3. Data Evaluation. 2.3.1. TGA Data. The solid conversion for the reduction, X_{r} , and oxidation, X_0 , of the manganese minerals were calculated from the mass variations registered in TGA

Reduction:

$$X_{\rm r} = \frac{m_{\rm o} - m}{R_{\rm OC} m_{\rm o}} \tag{2}$$

Oxidation:

$$X_{\rm o} = \frac{m - m_{\rm r}}{R_{\rm OC}m_{\rm o}} \tag{3}$$

where m is the instantaneous mass of the sample.

Following the methodology provided by Johansson et al.,²⁵ a normalized rate index, (*rate index*)_{TGA}, was calculated from TGA experiments. This value can be used to evaluate the reactivity of the oxygen carriers with different gases, e.g. H₂, CO, CH₄, and O₂,²⁵

$$(rate index)_{TGA} = 100.60 \cdot R_{OC} \cdot \left(\frac{dX_i}{dt}\right)_{norm}$$
 (4)

where the constants 100 and 60 were used to get the unit %/min, and the normalized reactivity $(dX_i/dt)_{norm}$ is calculated as eq 5 following the first-order reaction assumption made by Johansson et al.²⁵

$$\left(\frac{\mathrm{d}X_i}{\mathrm{d}t}\right)_{\mathrm{norm}} = \frac{p_{\mathrm{ref}}}{p_{\mathrm{TGA}}} \cdot \left(\frac{\mathrm{d}X_i}{\mathrm{d}t}\right)_{\mathrm{exp}} \tag{5}$$

where p_{ref} is fixed to 0.15 atm for reduction, viz. when $X_i = X_{rr}$ and to 0.1 atm for oxidation, viz. when $X_i = X_0$, respectively.²⁵ p_{TGA} is the partial pressure of the gas used in the TGA experiments. For the calculation of $(rate index)_{TGA}$ during the oxidation process, the oxidation after the reduction with 5% H₂ + 40% H₂O was considered. In this way reduction of iron oxide to FeO was avoided, and the oxidation of Fe₃O₄ to Fe₂O₃ was guaranteed.

2.3.1. Fluidized-Bed Reactor Data. From the gas product distribution obtained in the fluidized-bed reactor, it is possible to calculate the molar rate of oxygen transferred from the manganese minerals to the fuels as a function of time, $r_{\rm O}(t)$, according to an oxygen balance

Reduction:

$$r_{\rm O}(t) = (\chi_{\rm CO} + 2\chi_{\rm CO_2} + \chi_{\rm H_2O})_{\rm out} \cdot F_{\rm out} - (\chi_{\rm CO} + 2\chi_{\rm CO_2} + \chi_{\rm H_2O})_{\rm in} \cdot F_{\rm in}$$
(6)

Oxidation:

$$r_{\rm O}(t) = (2\chi_{\rm O_2})_{\rm in} \cdot F_{\rm in} - (2\chi_{\rm O_2})_{\rm out} \cdot F_{\rm out}$$
(7)

where $F_{\rm in}$ and $F_{\rm out}$ are respectively the inlet and outlet molar flows of gases from the reactor, and χ_i is the molar fraction of the gas i (i = CO, CO_2 , H_2O , or O_2) at the reactor inlet, $(\chi_i)_{\rm in}$, or the reactor outlet, $(\chi_i)_{\rm out}$.

The evolution of the conversion of solids during reduction and oxidation in the fluidized-bed reactor was evaluated by calculating the oxygen conversion as a function of time, X_0 . The value of X_0 was calculated by the integration of $r_0(t)$ from the beginning of the reaction ($t = t_{0,r}$ for reduction or $t = t_{0,0}$ for oxidation) to a time t, see eq 8 and eq 9

Reduction:

$$X_{\rm o}(t) = 1 - \frac{M_{\rm O}}{R_{\rm OC}m_{\rm o}} \int_{t_{\rm 0,r}}^{t-t_{\rm 0,r}} r_{\rm O}(t) \cdot {\rm d}t$$
(8)

Oxidation:

$$X_{o}(t) = X_{o,redf} + \frac{M_{O}}{R_{OO}m_{o}} \int_{t_{0,o}}^{t-t_{0,o}} r_{O}(t) \cdot dt$$
(9)

where $M_{\rm O}$ is the molar weight of oxygen, m_0 is the mass of fully oxidized solids in the reactor, and $X_{\rm o,redf}$ is the oxygen conversion at the end of the reduction step.

The reaction rate in batch experiments can be obtained as the following equation

$$\left(-\frac{\mathrm{d}X_{\mathrm{o}}}{\mathrm{d}t}\right)_{\mathrm{exp}} = \frac{M_{\mathrm{O}}}{R_{\mathrm{OC}}m_{\mathrm{o}}} \cdot r_{\mathrm{O}}(t) \tag{10}$$

The rate index in the fluidized-bed tests, $(rate index)_{FB}$, was also calculated. This parameter was used to evaluate and compare the reactivity of different materials in fluidized-bed reactor. The $(rate index)_{FB}$ can be obtained through eq 11

$$(rate index)_{\rm FB} = 100.60 \cdot R_{\rm OC} \cdot \left(-\frac{dX_o}{dt}\right)_{\rm norm}$$
 (11)

where $(-dX_o/dt)_{norm}$ is the normalized reaction rate calculated through eq 12 assuming first-order of reactions in all the cases. To facilitate the comparison, experimental data at $X_0 = 0.8$ was used.

$$\left(-\frac{dX_{o}}{dt}\right)_{\text{norm}} = \left(-\frac{dX_{o}}{dt}\right)_{\text{exp}} \frac{p_{\text{ref}}}{p_{\text{m}}}$$
(12)

In eq 12 p_{ref} refers to a reference partial pressure (0.15 atm), which would approximately correspond to a full conversion of the gas, and p_m refers to the average partial pressure of the gaseous fuel in the reactor being calculated by eq 13 through an integration process. Here assumption of a first-order reaction was made²⁸

$$p_{\rm m} = \frac{(p_{\rm in} - p_{\rm out}) \cdot p_{\rm in}}{(p_{\rm out} - p_{\rm in}) \cdot \varepsilon_{\rm g} + (p_{\rm in} + \varepsilon_{\rm g} \cdot p_{\rm out}) \cdot (1 + \varepsilon_{\rm g}) \cdot \ln\left(\frac{p_{\rm in} + \varepsilon_{\rm g} \cdot p_{\rm out}}{(1 + \varepsilon_{\rm g}) \cdot p_{\rm out}}\right)}$$
(13)

where $p_{\rm in}$ and $p_{\rm out}$ represent the partial pressures of the gaseous fuel at the inlet and outlet of the reactor, respectively, and $\varepsilon_{\rm g}$ is the expansion factor, i.e. $\varepsilon_{\rm g} = 0$ for CO and H₂ and $\varepsilon_{\rm g} = 0.5$ for reduction with 25% CH₄.

3. RESULTS AND DISCUSSION

3.1. Determination of the Redox Active Compounds. First, the investigation of the CLOU effect as well as the redox active compounds of the four manganese minerals was carried out. Regarding the CLOU behavior, four types of tests were performed to detect the possible cyclic oxygen release and uptake for the Mn_2O_3/Mn_3O_4 redox system. In the TGA, temperature-programmed decomposition and oxidation tests (TPD-TPO) together with cycles in N_2 and air at a fixed temperature were performed. In the fluidized-bed reactor, heating in N_2 and cooling in air together with long-term purges between cycles were done.

During TPD tests in nitrogen in TGA, mass losses were observed for all the uncalcined materials. However, during the following TPO in air no mass gains were seen, suggesting that reoxidation of these materials cannot occur at any temperature in air atmosphere. Besides these tests, successive cycles at 800 °C in alternating air and N₂ atmospheres in TGA showed some mass losses during the first exposure in N₂, which corresponded to 0.3, 2.0, 1.3, and 0.8 wt % of the mass of MnSA, MnGBHNE, MnGBMPB, and MnBR, respectively. These weight losses could be related to the decomposition of Mn_2O_3 , because the calcined minerals may initially contain Mn_2O_3 phase according to the XRD results in Table 1. However, after the first cycle, the regeneration of Mn_2O_3 by air hardly proceeded, which led to a decrease in the mass losses observed in the following cycles in N_2 . The mass losses of the third cycle corresponded to 0.2, 0.1, 0.1, and 0.1 wt % for MnSA, MnGBHNE, MnGBMPB, and MnBR, respectively. The oxygen release rate in this cycle is in the range 10^{-6} kgO/(s·kgOC), which corresponds to 1-3% of the decomposition rates calculated for other CLOU oxygen carriers such as CuO/CuAl₂O₄³² and CuO/MgAl₂O₄.³³ This comparison suggests that the rate of oxygen release from the manganese minerals tested in this work can be considered negligible.

The importance of the CLOU effect was also checked during the tests in the fluidized-bed reactor. The first heating of the sample introduced in the fluidized-bed-reactor was done using N₂, in order to observe the oxygen release. The maximum oxygen concentration obtained with the materials here tested was 2.8%. However, in the subsequent long-term purges in nitrogen between reduction and oxidation of the samples, oxygen release was never found. Moreover, the fluidized-bed was always cooled in air, and under these conditions no oxygen uptake was noticed in any case. Therefore, the following irreversible reaction, reaction R3, can be proposed for the transformation between Mn₂O₃ and Mn₃O₄ in N₂ and air cycles for all the manganese minerals.

$$6Mn_2O_3 \leftrightarrow 4Mn_3O_4 + O_2(g) \tag{R3}$$

The results from TGA and the fluidized-bed tests suggested that the CLOU effect in the manganese minerals tested in this work is not significant, because of the impossibility of Mn_3O_4 regeneration to Mn_2O_3 under practical operating conditions. Therefore, Mn_3O_4/MnO as well as Fe_2O_3/Fe_3O_4 systems were considered as the redox active pairs throughout this work. The oxygen transport capacity R_{OC} for these manganese minerals was determined by TGA tests considering the reductions from Mn_3O_4 to MnO and from Fe_2O_3 to Fe_3O_4 as detailed in the Supporting Information. As seen in Table 2, the values of R_{OC} fall into the range of 4.7–5.8 wt %, which are high enough for CLC applications.²⁸

3.2. Reactivity of the Calcined Materials in TGA. Figure 2 shows the solid conversion as a function of time for the different manganese minerals. At 950 °C rapid conversion of the manganese minerals was observed. Using CH_4 , CO, and H_2 , complete conversion of the MnSA, MnGBHNE, MnGBMPB, and MnBR samples were observed in less than 90 s. In the case of oxidation, rapid regeneration of the materials was also observed with reaction time less than 40 s (not shown in Figure 2).

The $(rate index)_{TGA}$ was calculated in order to compare the reactivity of these manganese minerals to that corresponding to other low-cost materials. Table 3 summarized the $(rate index)_{TGA}$ for the calcined and used manganese minerals in this work. Results corresponding to a previously tested Mn ore,¹⁷ Fe–Mn based materials,¹⁹ and some Fe-based materials³⁴ were also included in Table 3. Compared to the results corresponding to the Mn ore,¹⁷ the reactivity of the calcined MnSA is slightly lower in the reaction with CH₄ and H₂. However, higher or similar reactivity to that of the Mn ore was observed for the combustion of CH₄ and H₂ for the calcined MnGBHNE, MnGBMPB, and MnBR samples.¹⁷ In comparison



Figure 2. Solid conversion as a function of time at 950 °C for the calcined manganese minerals during the reactions with CH₄ (—, black), CO (—, red), H₂ (—, blue) and for the used materials with CH₄ (— —, black), CO (—·—, red), H₂ (…, blue).

Table 3. Rate Index (%/min) from TGA Tests at 950 °C, for the Calcined and Used Manganese Minerals, a Mn Ore, Two Fe–Mn Based Materials, and Some Fe-Based Materials

	CH_4	СО	H_2	O ₂
MnSA	$5.0^{a} (0.8)^{b}$	5.1 (5.3)	14.2 (11.8)	11.3 (8.1)
MnGBHNE	9.2 (1.3)	6.4 (1.4)	19.2 (9.0)	8.6 (9.7)
MnGBMPB	9.3 (2.2)	9.0 (2.4)	26.4 (14.8)	8.4 (9.4)
MnBR	12.7 (1.7)	8.2 (3.0)	20.5 (12.5)	8.5 (9.9)
Mn ore ¹⁷	7.2 [°]		19.8 [°]	
SINAI-A ¹⁹	3.6		5.4	
GUIZHOU ¹⁹	0.9		1.3	
ilmenite ³⁴	5.0	2.5	7.9	9.7
bauxite waste ³⁴	3.4	3.9	10.5	4.1
hematite ³⁴	3.3	3.4	12.4	5.5
^{<i>a</i>} Calcined mater	ial. ^b Used mat	terials. ^c Avera	ages between 9	00 and 1000

to the recently studied Fe–Mn based materials, SINAI-A and GUIZHOU,¹⁹ available results show higher reactivity of our manganese minerals than the Fe–Mn based materials in terms of CH₄ and H₂ combustion. It is important to highlight that the calcined manganese minerals exhibited higher reactivity to CO, H₂, and CH₄ than ilmenite, a bauxite waste, and a Fe-ore (hematite).³⁴ The reactivity to oxygen was also higher, except for the case of ilmenite. All the Fe-based materials have been successfully tested in continuous CLC units obtaining high carbon captures and combustion efficiencies.^{7,9,10,35} This fact could encourage the use of the manganese minerals here tested as oxygen carriers in the *i*G-CLC process, as they could even improve the conversion of the H₂ and CO generated during char gasification and lower the value of oxygen demand in the continuous CLC operation.

3.3. Reactivity in the Fluidized-Bed Reactor. *3.3.1. Gas Concentration Distribution.* The experiments in the fluidized-bed reactor allowed knowing the distribution of gas products during fuel combustion. As an example, Figure 3 depicted stable cycles for the combustion at 950 °C of CH_4 , CO, and H_2 with MnSA. The wet-basis concentrations of all the gases in Figure 3



Figure 3. Wet-basis gas concentration and solid conversion in the combustion of 25% CH₄, 50% CO, and 50% H₂ using the MnSA manganese mineral at 950 °C; red = reduction, oxi = oxidation.



Figure 4. Rate index, (*rate index*)_{FB}, for MnSA (\blacksquare , black), MnGBHNE (\blacktriangledown , red), MnGBMPB (\blacktriangle , green) and MnBR (\bigcirc , blue), as a function of cycles during the combustion of CH₄, CO, and H₂ in the fluidized-bed reactor; rate index at 950 °C of activated ilmenite (----) estimated from Cuadrat et al.²⁹ and for bauxite waste (----) from Mendiara et al.³⁰

were corrected following a deconvolution method,³⁶ considering the residence time distribution in the system.³⁷ In the combustion of CH₄, the main reaction products were CO₂ and H₂O, although low amounts of CO and H₂ were also detected. During the reduction period methane was not completely converted to CO₂ (only around 50%). The solid conversion $X_O(t)$ during reduction decreased to a value of approximately 0.6. After the purge with N₂, the MnSA material was regenerated using 10% O₂. During the first 4 min after feeding oxygen, all the oxygen supplied was consumed. At the end of oxidation, the initial oxygen concentration was recovered, and full regeneration of the manganese mineral was reached.

Shorter reducing times were used in the reaction of MnSA with CO and H_2 although higher degrees of reduction were observed, which was related to the reactivity and concentration of the fuels. Thus, the lower reactivity and lower concentration of CH₄ resulted in lower degree of solid conversion than using CO or H_2 as fuel. Obviously, the more reduction the longer period was required for reoxidation.

In the case of the other manganese minerals, i.e. MnGBHNE, MnGBMPB, and MnBR, similar behaviors to that shown in Figure 3 were observed for CH_4 , CO, and H_2 combustion. The unconverted CH_4 , CO, and H_2 that could be observed in the experiments might be attributable to two factors, i.e. the bubbling effect and the amount of manganese mineral introduced into the reactor. If more bed material was fed, complete conversion of the gases would be reached. However, incomplete gas conversion was desired in these experiments in order to calculate the wet-basis average concentrations of fuels via eq 13.

In total, 35 h of operation for MnSA and MnBR and 54 h for MnGBHNE and MnGBMPB were registered corresponding to 40–58 cycles in the batch reactor. It should be mentioned that in all the cycles, satisfactory fluidization was kept, and no agglomeration was found, even when a high degree of solid conversion was reached.

Article

3.3.2. Evolution of (rate index)_{FB} with Cycles. Stable reactivity of oxygen carriers is desirable for large-scale CLC plants; thereby it is necessary to examine the stability of these materials during cycles. The evolution of reactivity of the four manganese minerals as a function of cycle number is shown in Figure 4 where the (*rate index*)_{FB} calculated according to eq 11 was included for evaluation purpose.

In the first tests with methane shown in Figure 4, it can be observed that the values of $(rate index)_{FB}$ decreased until the 10th cycle and then remained stable. MnGBMPB and MnBR manganese minerals showed similar reactivity with CH4, which was slightly higher than the other samples. The stable values of (rate index)_{FB} were ~0.4-0.6%/min for all the manganese minerals during CH4 combustion. These values were lower than the value of $(rate index)_{TGA}$ determined in TGA and shown in Table 3, although all rate indexes were normalized to the same gas concentration, i.e. 15 vol %. This fact can be attributed to the bubbling effects in the fluidized-bed where some reacting gases inside the bubbles caused by fluidization have to diffuse to the emulsion, slowing their reactions with oxygen carrier particles.³⁸ After 20 cycles with CH₄, the materials were subjected to cycles with CO and H_2 . The (rate $index)_{FB}$ for all the materials was quite stable during the 20 cycles with CO and H₂, which indicated that after the cycles with CH₄ the manganese minerals had reached stable reactivity. During CO combustion, the highest (rate index)_{FB} was observed for MnSA (around 1%/min), which was 1.5-2.5 times that of the other samples. This finding is in accordance with the reactivity observed in TGA in the tests with used materials (Table 3). The highest rate index for gaseous fuel combustion was obtained in the experiments with H₂



Figure 5. Rate index (*rate index*)_{FB} for various fuels as a function of solids conversion using MnSA (—, black), MnGBHNE (—, red), MnGBMPB (—, green), and MnBR (—, blue) as bed materials in the fluidized-bed reactor.

combustion. The stable values of $(rate index)_{FB}$ during the cycles with H₂ were around 1.5–2%/min.

Subsequently, the stable values of $(rate index)_{FB}$ were used to compare with the available data in the literature, see the horizontal lines in Figure 4. In a previous work, ilmenite was tested in the same batch reactor and gaseous fuels as this work.²⁹ In this case, a gain in the reactivity was observed with the number of redox cycles. Stable values of $(rate index)_{FB}$ after the ilmenite activation step were here considered. The temperature used in these ilmenite tests was 900 °C, which was different than that used in the present work, 950 °C. Nevertheless, the improvement of the reaction rate due to the temperature increase can be calculated. According to Abad et al.,³⁹ a 50% increment of reduction rate for activated ilmenite was attained by changing the temperature from 900 to 950 °C. Thus, in order to compare the reactivity with manganese minerals this 50% increment was used in the present work for the same activated ilmenite. The $(rate index)_{FB}$ calculated in the present work suggests that the reactivity of the manganese minerals was similar to or even higher than that found for ilmenite with CH₄. In addition, the reactivity of manganese minerals with CO and H₂ was higher than that of ilmenite. Figure 4 also compares the results from manganese minerals to those obtained in the same batch reactor by Mendiara et al.³⁰ with a bauxite waste which was later successfully used in a continuous unit.⁹ In the case of CH₄ combustion, the rate index at 950 °C for bauxite waste was 1.1%/min, which is higher than that corresponding to the manganese minerals in this work. For H₂ and CO combustion, the rate index values for bauxite waste were not calculated due to the complete combustion of these gases.³⁰

As a conclusion, it can be said that the reactivity of the four manganese minerals decreased during the first 10 cycles using CH₄. Afterward, quite stable reactivity can be achieved. The stable rate index for the manganese minerals during CH₄ combustion was lower than that obtained under similar conditions with ilmenite^{29,39} and a bauxite waste.³⁰ For the combustion of CO and H₂, higher reactivity of the manganese minerals compared to ilmenite was found. Among the manganese minerals, the reactivity with CO and H₂ of MnSA, MnGBHNE, and MnGBMPB was higher or similar to that of MnBR, which could be taken as reference, as similar samples to this used here have been tested before for CLC applications,¹⁵ including continuous tests in a CLC unit with coal.²⁰ These facts seem to indicate that the manganese minerals in this work possess enough reactivity to be used in CLC. The high reactivity observed for MnSA during CO

combustion is especially interesting, which can be considered as an advantage for further tests of this material in continuous CLC units.

3.3.3. Evolution of (rate index)_{FB} with the Extent of *Reduction.* Figure 5 shows the $(rate index)_{FB}$ versus the solid conversion for all the manganese minerals during the cycles where reactivity remained stable. Here the evolution of the reactivity at different extents of reduction can be observed. The (rate index)_{FB} decreased slightly at high degrees of reduction in the combustion of CH₄ and CO. This suggested the decrease of reactivity during reduction in the fluidized-bed reactor. Nevertheless, the decrease of reactivity was not relevant for the combustion of H_{22} since the rate index did not change with the value of solid conversion in Figure 5. The stable rate index during H₂ combustion could be attributable to the stable and high reactivity of the manganese minerals with H₂. The decrease of reactivity can also be illustrated by the TGA results shown in Figure 2 where the materials used in the fluidized-bed exhibited a decreasing slope of the X_r -t curve at high extent of reduction. However, when high circulation rate of oxygen carrier was required to transfer heat between air and fuel reactors, short residence time of oxygen carrier in fuel reactor would be expected. In this condition, low extent of reduction of the oxygen carrier always encounter, thus the decrease of reactivity of these manganese minerals would not be of relevance.

3.4. Reactivity of the Used Materials in TGA. The used materials from the fluidized-bed reactor were tested with CH4, CO, and H₂ in TGA at 950 °C to see the possible changes in reactivity. Figure 2 and Table 3 gathered the solid conversion evolution with time and $(rate index)_{TGA}$ for the used materials. It was remarkable that in Figure 2 slower conversion of all the manganese materials in the reaction with CO, H₂, and CH₄ was obtained after being used in the fluidized-bed reactor. This led to lower (*rate index*)_{TGA} for the used materials (Table 3) which could be attributable to the decrease in the BET surface areas observed in Table 2. Despite this, most of the manganese minerals had higher reactivity than ilmenite in terms of CO and H₂ combustion.³⁴ The used materials exhibited full solid conversion in any case, which indicated that the oxygen transport capacities remained the same as that of the calcined material. The reactivity data of the used materials also allowed estimating the minimum solids inventory in fuel and air reactors to reach 100% combustion of CH₄, CO, and H₂ in the fuel reactor (see Table 4).^{39,40} The solids inventory in the fuel reactor needed for CH₄ combustion varied from 222 kg/MW_{th} using MnGBMPB to 629 kg/MW_{th} using MnSA. These values

Table 4. Minimum Solids Inventory (kg solids per MW_{th} of fuel) Required in Fuel and Air Reactors To Reach 99.9% Combustion Efficiency of CH₄, CO, and H₂ at 950 °C

	CH ₄	СО	H_2
MnSA	$629^{a} (54)^{b}$	63 (34)	34 (49)
MnGBHNE	380 (71)	236 (39)	44 (41)
MnGBMPB	222 (44)	139 (32)	27 (43)
MnBR	289 (55)	112 (38)	32 (40)
ilmenite ³⁹	167 (42)	105 (30)	42 (35)
^{<i>a</i>} For fuel reactor. ^{<i>l</i>}	'For air reactor.		

are larger than the value corresponding to ilmenite (167 kg/ MW_{th}). However, during CO combustion lower solids inventory than for ilmenite was required for the MnSA material (63 kg/MW_{th}) due to the high reactivity of MnSA with CO. In the case of H₂, similar solids inventory to that corresponding to ilmenite were required for all the materials (27–44 kg/MW_{th}). On the other hand, the inventory required for the oxidation process is remarkably lower than that for the reduction step, and it is in the range 30–71 kg/MW_{th} depending on the fuel and material used.

The low inventory for CO and H_2 combustion using the MnSA can be accounted as an advantage for the use of this material in *i*G-CLC with coal as CO and H_2 are the main gases from char gasification. Therefore, the MnSA mineral can be considered as a promising oxygen carrier for further testing in *i*G-CLC with coal.

4. PHYSICAL CHARACTERIZATION

4.1. Porosity Distribution. The distribution of pores in both calcined and used samples is shown in Figure 6. Before the



Figure 6. Pore size distributions of calcined (continuous lines) and the used (dash-dot lines) manganese minerals.

tests in the fluidized-bed reactor the pores were composed of macropores (d > 50 nm) and some mesopores (2 < d < 50 nm), except for MnSA which showed only macropores. After reaction in the fluidized-bed reactor, mesopores were completely developed to macropores, resulting in the disappearance of pores smaller than 50 nm. For calcined and used MnGBHNE, MnGBMPB, and MnBR a bimodal distribution could be observed regarding the macroscale with d < 10000 nm and d > 10000 nm. A decrease in porosity of the

used MnGBHNE and MnBR was observed at d < 10000 nm in Figure 6, and the values are included in Table 2. Results in Figure 6 can be also linked to the changes observed by SEM analysis shown in Figure 7. The big holes in Figure 7 are related to the peak at d > 10000 nm in the used materials and cannot be identified as porosity. For the MnSA mineral, bimodal distribution of pore size and holes inside the particle were not observed both for calcined and used samples. Nevertheless, the amount of pores smaller than 10000 nm was raised in the used MnSA, which led to the increase of porosity in Table 2. Combining the shifts of pores from meso to macro and the holes appeared in Figure 7, the decrease of bulk density of the used particles in Table 2 was reasonable.

It was noticed that in Figures 6 and 7 pores inside the particle were developed after the use in fluidized-bed reactor. Therefore, Figure 8 compared the particle size distributions of the calcined and the used manganese minerals. In accordance with the findings in a previous work,¹⁵ significant expansion of particles was observed for the used materials, except for the MnGBHNE. The change in the particle size distribution found for MnSA might be caused by the increase in porosity of this material, see Table 2. In the cases of MnGBMPB and MnBR, the big holes appearing in the used particles in Figure 7 might be the reason for the increase of particle size. However, for MnGBHNE smaller holes were found in the used particles, which resulted in less expansion.

4.2. Fluidization Behavior: Agglomeration and Attrition Rate. For all the manganese minerals, neither defluidization nor agglomeration of particles was observed during the tests in the fluidized-bed reactor, even when particles were highly reduced. This fact was confirmed by SEM analysis of the used particles in Figure 7 where agglomeration was not observed. Attrition generated by collision and abrasiveness produced fine particles. In this work, the attrition rate was calculated considering the elutriated particles smaller than 40 μ m as unrecoverable fines. Figure 9 (a) plotted the attrition rates for the manganese minerals as a function of cycles. It was observed that the MnSA maintained low attrition rate for all the cycles. In contrast, the attrition rate of MnBR notably increased with the number of cycles up to 0.73%/h in the end of tests. For MnGBHNE and MnGBMPB the attrition rates were lower than for MnBR after 40 cycles, although slightly higher than that found for MnSA. Notably, the MnSA and MnGBHNE minerals possessed similar attrition rates to ilmenite²⁹ which was considered for scale-up to 50 kW-1MW continuous CLC units.^{8,11,12} The attrition rate can be related to the crushing strength of the particles. Samples were extracted at different operation cycles and subjected to determining the crushing strength. Figure 9 (b) plotted the crushing strength of the four manganese minerals as a function of the number of cycles. Generally, the crushing strength of each mineral decreased with cycles, attributable to the development of pores and the appearance of holes in the particles, as shown in Figures 6 and 7. In more detail, the MnSA initially possessed a high crushing strength of 4.6 N that was decreased to 2.0 N after 40 cycles (Table 2). In the case of MnGBHNE, the crushing strength was reduced slowly during operation from 1.8 to 1.4 N after 58 cycles. In both cases, the crushing strength is similar to that for activated ilmenite after a similar number of cycles,²⁹ which was higher than the minimum value to allow operation in a CLC system, 1 N. Thus, low attrition rates for MnSA and MnGBHNE were observed in Figure 9 (a). On the contrary, the crushing strength of MnGBMPB and MnBR was lower than



Figure 7. Morphology analyses of the calcined (a-d) and used materials (e-l) in terms of the cross sections, for the manganese minerals: MnSA (a, e, i), MnGBHNE (b, f, j), MnGBMPB (c, g, k), and MnBR (d, h, l).



Figure 8. Particle size distributions of the calcined (continuous lines) and the used (dash-dot lines) manganese minerals.

1N after 40 cycles, which may cause the increase in the attrition rate observed for both MnGBMPB and MnBR in Figure 9 (a).

Normally, attrition can affect the lifetime of the materials used for CLC. A high attrition rate can increase the amount of fine particles produced, thus decreasing the lifetime. The attrition rate of MnSA and MnGBHNE was moderate, which corresponded to an average lifetime of 1600 and 1000 h, respectively. For MnGBMPB and MnBR the corresponding lifetime would be 360 and 150 h, respectively, making them less attractive for long-term use. In the case of MnGBMPB, this is in accordance with the value of AJI calculated for the used sample. Following the results from Gayán et al.,²³ the oxygen carriers with AJI values higher than 15-20% after their use in a CLC unit should improve their attrition performance before being selected for scale-up. However, despite the high attrition rate observed and the low crushing strength after use in the fluidized-bed reactor, the MnBR sample presented a value for AJI lower than 15%. According to the particle size distributions shown in Figure 10, this can be due to the breaking behavior of MnBR particles under the conditions used in the attrition tester. Figure 10 compared the distribution corresponding to the MnBR sample after use in the fluidized-bed reactor and that corresponding to the used particles after testing following the



Figure 9. Attrition rates (a) and crushing strength (b) with cycles for the manganese minerals MnSA ($-\blacksquare$, black), MnGBHNE ($-\Psi$, red), MnGBMPB ($-\blacktriangle$, green), MnBR ($-\Phi$, blue), the horizontal dash line with the value 1 N represents the minimum desirable crushing strength of particles.



Figure 10. Particle size distribution of the MnBR sample after use in the fluidized-bed batch reactor (dash-dot line) and after the AJI test (dash lines).

ASTM D5757-00 standard.²⁴ As it is observed, the original distribution turns into a bimodal distribution, which supports the idea of breakage of the particles during the tests. The average size of the smallest fraction of those broken particles is around 40 μ m. However, these particles remained in the bed after the tests in attrition tester. The particles considered to calculate the AJI value are only those recovered in the filter at the outlet of the system for the AJI determination, and they are less than 20 μ m in size. However, these broken particles do escape from the fluidized-bed batch reactor during the experiments with gaseous fuels and were retained in the filters at the outlet of fluidized-bed reactor and therefore considered in the calculation of the attrition rate. This would explain the apparent discrepancy between the attrition rate and the AJI value.

5. CONCLUSION

Four manganese minerals, i.e. MnSA, MnGBHNE, MnGBMPB, and MnBR, were studied in a TGA apparatus and a fluidized-bed batch reactor with gaseous fuels, CH_4 , CO, and H_2 . The performance of these materials in chemical looping combustion (CLC) and chemical looping with oxygen uncoupling (CLOU) was evaluated. The four studied minerals exhibited no cyclic release and uptake of gaseous oxygen due to the unfeasibility of the regeneration of Mn_2O_3 from Mn_3O_4 , so these materials were not suitable for CLOU. During the redox cycles with gaseous fuels in a fluidized-bed reactor, absence of agglomeration was found, even at high degrees of solids conversion.

Decreases of reactivity were seen for all the manganese minerals for the first 10 cycles in the fluidized-bed reactor. Stable reactivity of the materials was achieved after 10 cycles with CH_4 , which was maintained to the end of the tests involved in this work. The reactivity with CH_4 of these manganese minerals is lower than previously tested Fe-based materials. However, high reactivity and low solids inventory were found during the combustion of CO and H_2 with these manganese minerals. Remarkably, extremely low solids inventory was required for the use of MnSA during the combustion of CO and H_2 .

After the tests with the gaseous fuels, some changes were found in the materials. Pores distributed inside the particles were developed, which resulted in the transition of mesopores to macropores in the after-used particles. Therefore, expansion of particles and decrease of crushing strength were observed for the manganese materials during the tests. The MnBR sample showed a sharp decrease of crushing strength and produced a high amount of fines, which makes this material unsuitable after 40 cycles of operation. This trend, but a lower degree, was observed for MnGBMPB. Low attrition rates of particles and adequate crushing strength were found for the MnSA and MnGBHNE manganese minerals during the operation period. When both reactivity and attrition rate were evaluated, MnSA and MnGBHNE are more suggestible for further tests in CLC.

ASSOCIATED CONTENT

S Supporting Information

The Supporting Information is available free of charge on the ACS Publications website at DOI: 10.1021/acs.energy-fuels.5b01293.

Determination of the active composition and the oxygen transport capacity of manganese minerals from TGA experiments (PDF)

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Notes

The authors declare no competing financial interest.

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NOMENCLATURE

d = pore size of solid particles, nm

 $(dX_i/dt)_{exp}$ = reaction rate of solids calculated from experimental data for reduction (*i* = r) or oxidization (*i* = o), s⁻¹

 $(dX_i/dt)_{norm}$ = normalized reactivity of solids for reduction (*i* = r) or oxidization (*i* = o), s⁻¹

 F_i = molar flow rate of stream at inlet (*i* = in) or outlet (*i* = out), mol·s⁻¹

m = instantaneous mass of the manganese mineral material, kg

 $M_{\rm O}$ = molecular weight of oxygen, 16·10⁻³kg/mol

 m_i = mass of fully oxidized (*i* = o) or reduced (*i* = r) manganese minerals, kg

 p_i = partial pressure of gaseous fuels inlet (i = in) or outlet (i = out) the fluidized-bed reactor, atm

 $p_{\rm m}$ = average pressure of gaseous fuels in the fluidized-bed reactor, atm

 $p_{\rm ref}$ = a reference partial pressure, atm

 p_{TGA} = partial pressure of gas used in TGA, atm

 $(rate index)_{FB}$ = rate index determined in fluidized-bed tests, %/min

 $(rate index)_{TGA}$ = rate index determined in TGA tests, %/min

- $R_{\rm OC}$ = oxygen transport capacity for the manganese mineral $r_{\rm O}(t)$ = molar rate of atomic oxygen transference from the manganese minerals to the fuels, $molO \cdot s^{-1}$
- t = reaction time, s

 $t_{0,i}$ = time at the beginning of oxidation (i = 0) or reduction (i = r) process, s

V = volume of pores in the manganese mineral particles, cm^3/g

 X_i = solids conversion referred to the reduction (*i* = r) or oxidization (i = 0) process

 $X_{o,redf}$ = solids conversion at the end of reduction step

Greek symbols

- expansion factor of gas mixture
- $\mathcal{E}_{g}(\chi_{i})_{in}$ molar fraction of the gas i ($i = CO_1 CO_2, H_2O_1$ or O_2) inlet the reactor
- $(\chi_i)_{out}$ molar fraction of the gas *i* (*i* = CO, CO₂, H₂O, or O₂) outlet the reactor

Acronvms

- AJI Air Jet Attrition Index
- BET Brunauer-Emmett-Teller
- CLC chemical looping combustion
- CLOU chemical looping with oxygen uncoupling
- ICP-AES inductively coupled plasma atomic emission spectroscopy
- *i*G-CLC in situ gasification chemical looping combustion
- NDIR nondispersive infrared analyzer
- OC oxygen carrier
- PSD particle size distribution
- SEM scanning electron microscope
- TCD thermal conductivity detector
- TGA thermogravimetric analyzer
- TPD temperature-programmed decomposition
- TPO temperature-programmed oxidation
- XRD X-ray diffraction

REFERENCES

(1) IEA. World Energy Outlook 2013; OECD/IEA: Paris, France, 2013.

(2) Kerr, H. R. Capture and Separation Technology Gaps and Priority Research Needs. In Carbon Dioxide Capture for Storage in Deep Geologic Formations; Elsevier Science: Amsterdam, 2005.

- (3) Mendiara, T.; García-Labiano, F.; Gayán, P.; Abad, A.; de Diego, L. F.; Adánez, J. Fuel 2013, 106, 814-826.
- (4) Dennis, J. S.; Scott, S. A.; Hayhurst, A. N. J. Energy Inst. 2006, 79, 187-190.
- (5) Mattisson, T.; Lyngfelt, A.; Leion, H. Int. J. Greenhouse Gas Control 2009, 3, 11-19.
- (6) Adánez, J.; Abad, A.; García-Labiano, F.; Gayán, P.; de Diego, L. F. Prog. Energy Combust. Sci. 2012, 38, 215-282.
- (7) Cuadrat, A.; Abad, A.; García-Labiano, F.; Gayán, P.; de Diego, L. F.; Adánez, J. Int. J. Greenhouse Gas Control 2011, 5, 1630-1642.
- (8) Markström, P.; Linderholm, C.; Lyngfelt, A. Int. J. Greenhouse Gas Control 2013, 15, 150-162.
- (9) Mendiara, T.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Abad, A.; Adánez, J. Int. J. Greenhouse Gas Control 2013, 17, 170-182.
- (10) Mendiara, T.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Abad, A.; Adánez, J. Fuel 2014, 126, 239-249.
- (11) Ströhle, J.; Orth, M.; Epple, B. Appl. Energy 2014, 113, 1490-1495.

- (12) Abad, A.; Pérez-Vega, R.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Adánez, J. Appl. Energy 2015, DOI: 10.1016/j.apenergy.2015.03.094.
- (13) Larring, Y.; Braley, C.; Pishahang, M.; Andreassen, K. A.; Bredesen, R. Energy Fuels 2015, 29, 3438-3445.
- (14) Frohn, P.; Arjmand, M.; Azimi, G.; Leion, H.; Mattisson, T.; Lyngfelt, A. AIChE J. 2013, 59, 4346-4354.
- (15) Arjmand, M.; Leion, H.; Mattisson, T.; Lyngfelt, A. Appl. Energy 2014, 113, 1883-1894.
- (16) Sundqvist, S.; Arjmand, M.; Mattisson, T.; Leion, H.; Rydén, M.; Lyngfelt, A. In Proceedings of the 11th International Conference on

Fluidized Bed Technology, Beijing, 2014. (17) Fossdal, A.; Bakken, E.; Øye, B. A.; Schøning, C.; Kaus, I.;

- Mokkelbost, T.; Larring, Y. Int. J. Greenhouse Gas Control 2011, 5, 483-488.
- (18) Leion, H.; Mattisson, T.; Lyngfelt, A. Energy Fuels 2009, 23, 2307-2315.
- (19) Larring, Y.; Pishahang, M.; Sunding, M. F.; Tsakalakis, K. Fuel 2015, 159, 169-178.
- (20) Linderholm, C.; Lyngfelt, A.; Cuadrat, A.; Jerndal, E. Fuel 2012, 102, 808-822.
- (21) Arjmand, M.; Leion, H.; Lyngfelt, A.; Mattisson, T. Int. J. Greenhouse Gas Control 2012, 8, 56-60.

(22) Yazdanpanah, M.; Forret, A.; Gauthier, T. Appl. Energy 2015, DOI: 10.1016/j.apenergy.2015.07.015.

(23) Gaván, P.; García-Labiano, F.; Cabello, A.; de Diego, L. F.; Abad, A.; Adánez, J. 22nd Fluidized Bed Conversion Conference, Turku, Finland, 2015.

(24) ASTM D5757-00: Standard Test Method for Determination of Attrition and Abrasion of Powdered Catalysts by Air Jets, ASTM: Philadelphia, United States, 2000.

- (25) Johansson, M.; Mattisson, T.; Lyngfelt, A. Ind. Eng. Chem. Res. 2004, 43, 6978-6987.
- (26) Adánez, J.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Abad, A.; Palacios, J. M. Energy Fuels 2004, 18, 371-377.
- (27) Abad, A.; García-Labiano, F.; de Diego, L. F.; Gayán, P.; Adánez, J. Energy Fuels 2007, 21, 1843-1853.
- (28) Abad, A.; Adánez, J.; García-Labiano, F.; de Diego, L. F.; Gayán, P.; Celaya, J. Chem. Eng. Sci. 2007, 62, 533-549.
- (29) Cuadrat, A.; Abad, A.; Adánez, J.; de Diego, L. F.; García-Labiano, F.; Gayán, P. Fuel Process. Technol. 2012, 94, 101-112.
- (30) Mendiara, T.; Abad, A.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Adánez, J. Energy Fuels 2012, 26, 1420-1431.
- (31) Adánez-Rubio, I.; Gayán, P.; Abad, A.; de Diego, L. F.; García-Labiano, F.; Adánez, J. Energy Fuels 2012, 26, 3069-3081.
- (32) Mei, D.; Abad, A.; Zhao, H.; Adánez, J. Fuel Process. Technol. 2015, 133, 210-219.
- (33) Adánez-Rubio, I.; Gayán, P.; Abad, A.; García-Labiano, F.; de Diego, L. F.; Adánez, J. Chem. Eng. J. 2014, 256, 69-84.
- (34) Mendiara, T.; Pérez, R.; Abad, A.; de Diego, L. F.; García-Labiano, F.; Gayán, P.; Adánez, J. Ind. Eng. Chem. Res. 2012, 51, 16216-16229.
- (35) Ortiz, M.; Gayán, P.; de Diego, L. F.; García-Labiano, F.; Abad,
- A.; Pans, M. A.; Adánez, J. J. Power Sources 2011, 196, 4370-4381.
- (36) Abad, A.; Mattisson, T.; Lyngfelt, A.; Johansson, M. Fuel 2007, 86, 1021-1035.
- (37) Levenspiel, O. Chemical reaction engineering, 2nd ed.; Wiley: New York, 1972.

(38) Abad, A.; Adánez, J.; García-Labiano, F.; de Diego, L. F.; Gayán, P. Combust. Flame 2010, 157, 602-615.

- (39) Abad, A.; Adánez, J.; Cuadrat, A.; García-Labiano, F.; Gayán, P.; de Diego, L. F. Chem. Eng. Sci. 2011, 66, 689-702.
- (40) Adánez, J.; Cuadrat, A.; Abad, A.; Gayán, P.; de Diego, L. F.; García-Labiano, F. Energy Fuels 2010, 24, 1402-1413.