

# THE INTERNATIONAL FOOD OPERATIONS & PROCESSING SIMULATION WORKSHOP

*SEPTEMBER 21-23 2015*  
BERGEGGI, ITALY



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**THE INTERNATIONAL FOOD OPERATIONS & PROCESSING  
SIMULATION WORKSHOP, FOODOPS 2015**  
*SEPTEMBER 21-23 2015, BERGEGGI, ITALY*

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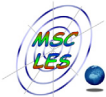
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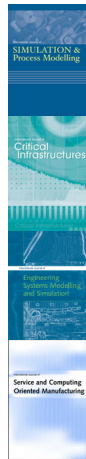


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## CHAIRS' MESSAGE

On behalf of the Organization Committee, we would like to welcome and thank all the people who have contributed to the First edition of “The international Food Operations & Processing Simulation Workshop” (FoodOPS 2015). In this first edition FoodOPS workshop is intended to extend a track we managed in last years for the I3M conference on “Modelling and Simulation of Food Processing and Operations” bringing new ideas, solutions, approaches and plenty more. The aim of this initiative is to promote the particular nature of the food sector, being able, at the same time, to integrate new perspectives and raise new challenges.

FoodOPS Workshop offers the possibility to present results and critical issues concerning processes design and operations management in food industry using Modeling and Simulation as methodological approach. Nevertheless, different topics concerning processes and operations management in food industry has been well-accepted. We received high-quality, original scientific contributions and we will also consider them for publication in a Special Issue on the ISI indexed Journal “International Journal of Food Engineering”. To this end, we would like to thank you the Editor in Chief of this prestigious Journal, Prof. Dong Chen, who agreed to support this workshop on this particular field also by contacting the Associate editor Ruben Mercade Prieto, who agreed to be here as “General Co-Chair” of the workshop.

Based on the peculiarities of the Food modelling and simulation sector, the scientific committee structured the workshop on 4 tracks dealing with:

1. Simulation of Food Supply Chain
2. Modelling and Simulation of food processing
3. Modelling and Simulation of food properties
4. Sustainability of Food processes

As regard to these topics, FoodOPS 2015 is an opportunity to meet and discuss and create new opportunities for cooperation and also joint research projects. FoodOPS 2015 is co-located with the I3M 2015 multi-conference, that includes 8 additional international conferences, in the wonderful location of Bergeggi, Italian Riviera. We would like to wish to all the FoodOPS and I3M attendees a fruitful conference and a wonderful stay in Bergeggi.



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The FoodOPS 2015 International Program Committee (IPC) has selected the papers for the Conference among many submissions; therefore, based on this effort, a very successful event is expected. The FoodOPS 2015 IPC would like to thank all the authors as well as the reviewers for their invaluable work.

A special thank goes to all the organizations, institutions and societies that have supported and technically sponsored the event.

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# CFD ANALYSIS OF COFFEE PACKAGING IN CAPSULES USING GAS FLUSHING MODIFIED ATMOSPHERE PACKAGING

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## ABSTRACT

The aim of this work is to analyze, by means of CFD, the gas flow in a packaging machine used to fill polymeric capsules with coffee. The final goal is to optimize the geometric shape of some of the mechanical components in the machine's sealing station in order to reduce inert gas consumption (N<sub>2</sub> in this case) and, at the same time, achieve an O<sub>2</sub> residual which is constantly equal or below 1% by mass in the center of the capsule.

The fluid domain has been obtained starting from the 3D CAD model of the sealing station of the packaging machine. The CAD software SolidWorks (version 2014) has been used to design the system, while Ansys CFX 14.5 software has been used for the CFD analysis.

The CFD model has been validated by comparing the simulation results with those obtained by experimental tests. The modified solution allows reducing the O<sub>2</sub> residual from almost 3% to less than 1%.

Keywords: CFD, Food Packaging, Modified Atmosphere Packaging, Gas Flushing

## 1. INTRODUCTION

Modified Atmosphere Packaging (MAP) is a packaging technique which implies the addition (or removal) of gases from storage rooms, transportation containers or packages in order to manipulate the level of gases such as oxygen, carbon dioxide, nitrogen, ethylene etc., and achieve an atmospheric composition different to that of normal air around the food (Floros 1990). This technique is used generally to enhance the shelf-life of food products or to preserve their organoleptic features.

Nowadays MAP is an important part of food industry and it is used especially in the processing of fresh products such as fruits and vegetables, fish and fresh meat, or in the processing of products which have to maintain unaltered for a long period of time their aromatic features such as coffee, tea or spices.

MAP is nowadays frequently used in the packaging of coffee in capsules. Today, in fact, the demand of coffee packaged in capsules is constantly increasing, enhancing the attention in product quality.

To fulfill the coffee quality requirements, producers started to ask for coffee packaging lines, which are able to minimize the O<sub>2</sub> residual inside the capsules. A low oxygen level in capsules help limiting the oxidation process of the final product, thus enhancing its shelf-life and reducing the organoleptic alteration of coffee. Some systems could be used to achieve this goal, such as: vacuum filling, vacuum filling with inert gas injection and many others. Among these techniques the most used is the gas flushing Modified Atmosphere Packaging. According to this method the capsule is fluxed with an inert gas before the filling phase until the sealing phase, to remove air from the inner side of the capsule itself and consequently reducing the O<sub>2</sub> residual in the closed package. The inert gases usually used for these purposes are N<sub>2</sub>, CO<sub>2</sub>, Argon or mixtures of them.

To make modified atmosphere packaging effective, three conditions must be satisfied:

1. the packaging phase must be preceded by a correct coffee degassing process, in order to minimize the quantity of air blocked in the coffee itself;
2. the packaging phase needs a very effective monitoring to be sure that air is completely removed by the capsules and Nitrogen is instead inside them;
3. the capsule and closing material must have a low permeability, in order to limit the permeation of the inert gas through the package wall (aluminium or multilayer polymeric films are usually used in this case).

The third point attends to technological aspects of the container manufacturers, while the first one attends to the food equipment manufacturer, whilst the second one is on the responsibility of food packaging manufacturer.

Considering its importance, MAP is now studied in a more methodical way, also by means of advanced techniques as CFD (Computational Fluid Dynamics). This technique was in particular adopted in the design of postharvest facilities or cooling facilities for the storage of fruits and vegetables (Delele et al., 2011 and Ferrua et al., 2008); recently it has also been used to investigate mass and heat transfer of bio-substrates in modified atmosphere packaging (De Bonis et al., 2013) or to

evaluate the losses of CO<sub>2</sub> through a PET bottle containing a carbonated beverage (Carrieri et al., 2011). Nevertheless, CFD has never been used to evaluate the flow of gas in packaging machines and as a tool to optimize the packaging process.

Aim of this work is then to analyze, by means of CFD, the gas flushing of N<sub>2</sub> on a coffee in capsules packaging machine in order to optimize the geometric shape of some of the mechanical components in the sealing station of the machine itself. The new configuration should allow reducing inert gas consumption and, at the same time, achieving an O<sub>2</sub> residual in the center of the container which is constantly equal or below 1% by mass.

## 2. MATERIALS AND METHODS

Based on what stated in the introduction about the influence of the coffee degassing phase, both in the simulation and in the subsequent experimental phase the capsules were left empty in order to avoid the influence of the coffee degassing process on the results (since its accuracy could not be monitored).

### 2.1. Equipment

The coffee packaging machine is made by a conveyor belt which moves in a discontinuous way (one step every 0.4s). On all the elements composing the conveyor belt are present four holes and in every hole is housed a little basket which contains a capsule. The conveyor belt advances and stops at every station present on the packaging line. In particular, four station are present:

1. a filling station, where the capsules are filled with coffee;
2. an aspiration station, where coffee powder is aspirated from the edge of every capsule in order to facilitate the following sealing operation;
3. a station where a micro-perforated film is placed on the "free surface" of coffee;
4. a sealing station, where the capsules are closed with a polymeric or aluminium lid.

The second station is not always activated, depending on coffee granulometry.

From immediately before the filling station, to the sealing station, there is a tunnel where it is fluxed nitrogen gas in order to saturate the environment and remove air that may cause an excessive O<sub>2</sub> residual inside the closed package.

This packaging lines can produce more than 340 capsules per minute. The overall dimensions of the packaging machine are 4690 x 555 x 1340 mm.

Figure 1 shows a block diagram which resumes the functioning of the previously described packaging line.

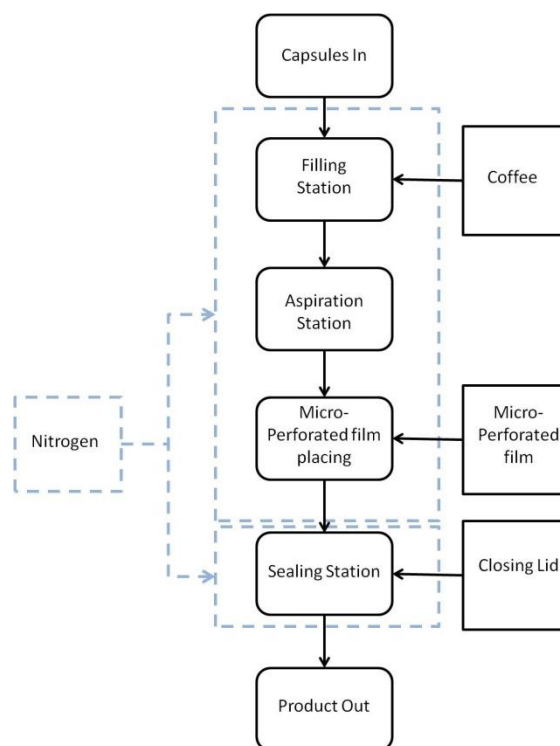


Figure 1: block diagram of the coffee packaging line considered in this study

As previously explained, the CFD analysis has been focused only on the sealing station of the packaging machine. The reason is that this is the area where the N<sub>2</sub> flow rate is higher. In Figure 2 is shown a 3D simplified geometry of the capsules' sealing station.

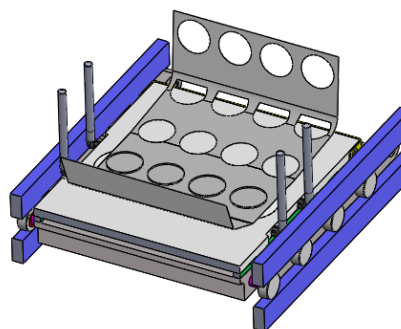


Figure 2: capsules' sealing station

The main components present in this station are: a stainless steel plate, within which a groove for the distribution of gaseous N<sub>2</sub> is formed by milling, four diffuser rings, which address N<sub>2</sub> inside the capsules, four small pipes (internal diameter equal to 8 mm) which direct N<sub>2</sub> flow inside the groove realized in the plate, the closing lid and, finally, four electric heaters welding the lid on the capsules' edge. Figure 3 shows a section of the stainless steel plate.



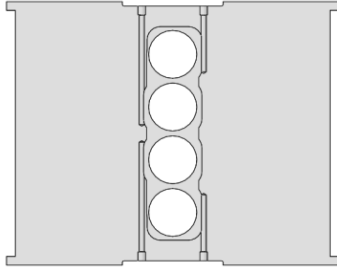


Figure 3: plate section

The one just outlined is the original configuration of the sealing station, which is currently installed on the packaging line. In addition to this, a modified station has been analyzed.

After the modeling phase, the simulations will be validated with an experimental phase. Both in the simulations and in the experimental phase. The considered capsule is the model "Blue", made by Lavazza, which has a truncated-cone shape, with a maximum diameter of 40 mm, a minimum diameter of 32 mm and an overall height of about 26 mm. It is important to specify the capsule's model because from it depends the kind of diffuser ring that has to be mounted.

After the validation of the CFD model some simulation with other two kind of capsule that the packaging line can process have been performed. The involved capsules model are "A Modo Mio", made by Lavazza and a Nespresso-Like capsule. The first one has a maximum diameter of 40 mm, a minimum diameter of 32 mm and an overall height of 17 mm, while the second one has a maximum diameter of 30 mm, a minimum diameter of 15.7 mm and an overall height of 28 mm.

Based on the obtained results some changes in the shape of the plate will be made and tested.

## 2.2. CFD Modelling

To set up the physics and solve the calculation, Ansys CFX version 14.5 has been used, because this software allows to simulate multi-component fluids. In fact, a 3D multi-component simulation has been set up, considering a gas mixture of Argon, N<sub>2</sub> and O<sub>2</sub>.

In a mixture each component has its own equation for conservation of mass. After Reynolds-Averaging it can be expressed as:

$$\frac{\delta \tilde{\rho}_i}{\delta t} + \frac{\delta (\tilde{\rho}_i \tilde{U}_j)}{\delta x_j} = - \frac{\delta}{\delta x_j} [\rho_i (\tilde{U}_{ij} - \tilde{U}_j) - \overline{\rho_i'' U_j''}] + S_i \quad (1)$$

where:

$\tilde{\rho}_i$  is the mass-average density of fluid component i in the mixture

$\tilde{U}_j = \sum \frac{(\tilde{\rho}_i \tilde{U}_{ij})}{\tilde{\rho}}$  is the mass-average velocity field,

$\tilde{U}_{ij}$  is the mass-average velocity of fluid component i,

$\rho_i (\tilde{U}_{ij} - \tilde{U}_j)$  is the relative mass flux,

$S_i$  is the source term for component i which includes the effects of chemical reactions.

The mixture, instead, is generally treated as an "Ideal Mixtures", meaning that its properties are calculated directly by means of a mass averaging of the component materials properties. Thus, for instance, the mixture density  $\rho$  can be calculated from the mass fraction  $Y_i$  and the thermodynamic density of each component  $\rho_i$  by means of equation 2.

$$\frac{1}{\rho} = \sum_{i=A,B,\dots}^{Nc} \frac{Y_i}{\rho_i} \quad (2)$$

In the mixture, N<sub>2</sub> has been considered as constraint component, meaning that its mass fraction is calculated to ensure that all the component mass fraction sum to unity in every instant. This means that the mass fraction of nitrogen is set equal to the total mass fraction of the other components in the Ideal Mixture subtracted from unity.

The full buoyancy model has been implemented in the domain, to consider motion due to density differences and a heat exchange model has been considered, to take account of the heat generated by the electrical resistances used to fix the closing lid over the capsules. In particular, the "Total Energy" model has been adopted, which is usually recommended by Ansys CFX solver theory guide and Ansys CFX solver modeling guide when dealing with flow of compressible substances.

Furthermore, the Shear Stress Transport turbulence model has been adopted. The SST model, proposed by Menter (1994), is an eddy-viscosity model, which is a combination of a k- $\omega$  model and k- $\epsilon$  model. The first is used in the inner boundary, while the second in the outer region and outside of the boundary layer (Bottani et al, 2008). The SST model has been used in order to overcome the problems of both the methods. These features make the SST model more accurate and reliable for a wider class of flows than the standard k- $\omega$  and k- $\epsilon$  models.

### 2.2.1. Mesh setting for the fluid domain

The fluid domain has been obtained starting from the 3D CAD model of the sealing station by means of the CAD software SolidWorks (version 2014), while the discretization of the fluid domain has been performed using the software Icem CFD version 14.5.

The mesh was initially set by creating a uniform subdivision, and then thickened in the critical areas of the fluid volume. In particular, a finer mesh was used near the capsules, in the groove realized inside the stainless steel plate and near the diffuser rings which address the N<sub>2</sub> flow inside the containers.

At the end of the operation, an unstructured grid with a total of 4,700,000 nodes and 27,600,000 tetrahedrons was obtained.

Figure 4 shows a section of the obtained volume mesh.

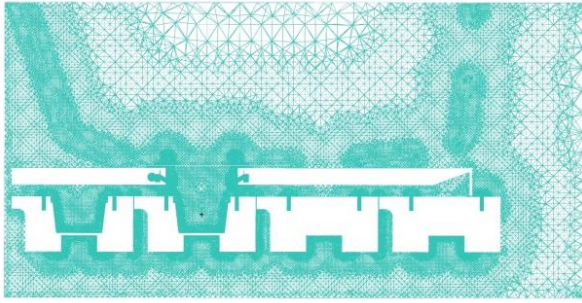


Figure 4: section of the fluid domain volume mesh

### 2.2.2. Simulation setting

A 3D multi-component simulation has been set up considering a gas mixture of Argon, N<sub>2</sub> and O<sub>2</sub>. These three substances were considered as ideal gases, while the mixture has been set as "Ideal Mixture". N<sub>2</sub> has been used as constrain component, meaning that its mass fraction is set equal to the total mass fraction of the other components in the Ideal Mixture subtracted from unity. Since Argon Ideal Gas is not present in Ansys CFX material library it has been created using the settings showed in Table 1.

Table 1: Argon ideal gas settings

Argon Ideal Gas Settings	
Thermodynamic State	Gas
Equation of State	Ideal Gas
Molar Mass	39.948 $\frac{g}{mol}$
Specific Heat Capacity	520 $\frac{J}{kg K}$
Specific Heat Type	Constant Pressure
Reference Temperature	25 °C
Reference Pressure	1 atm
Dynamic Viscosity	2.217e-05 Pa·s
Thermal Conductivity	0.016 $\frac{W}{m K}$

At the beginning of the simulation the fluid domain has been considered, in order to simulate air presence, as composed by 21% of O<sub>2</sub>, 1% of Argon and 78% of N<sub>2</sub>, while on the inlets only nitrogen was considered.

The boundary and initial condition for the original case are shown in Table 2.

Table 2: boundary and initial conditions for original case

Original Case Boundary Conditions	
Inlet_1; Inlet_2; Inlet_3; Inlet_4	Volumetric Flow-Rate = 20 l/min T = 25°C
Opening	Composition: 1% Ar, 78% N <sub>2</sub> , 21% O <sub>2</sub> T = 25°C p = 1 bar
Resistenze	Free Slip Wall T = 90°C
Wall	Free Slip Wall Adiabatic
Original Case Initial Conditions	
Domain Composition	1% Ar, 78% N <sub>2</sub> , 21% O <sub>2</sub>
Velocity	0 m/s
T	25°C
p	1 bar

The simulations have been performed both in steady state and both in transient mode because the flow of N<sub>2</sub> is not time dependent but the capsules stay in the sealing station only for a fixed period of time (0.4s). In particular, in transient mode, 4s have been simulated with a time step of 0.02 s. The results have been checked at 0.4s.

### 2.3. Experimental Method

To validate the CFD model some experimental tests have been performed on the packaging machine object of this study. In particular, the machine with the new plate has been tested.

As previously stated, the capsules considered during experimental phase were the "Blue" model and they were left empty, so the coffee dosing system has been excluded from the machine configuration. Furthermore also the tunnel has been excluded, in order to evaluate only the effects of gas flushing in the sealing station.

The sensor used in the tests was Dansensor CheckPoint, especially created to evaluate O<sub>2</sub> or CO<sub>2</sub> concentration in the headspace of containers processed in modified atmosphere. This sensor has a needle which must be inserted in the closed package. From the needle a small quantity of the internal gas mixture is aspirated and then the probe evaluates the O<sub>2</sub> concentration in the mixture.

Before the test, few packaging cycles have been performed, in order to reach steady state condition in the flux of N<sub>2</sub>.

The respected experimental procedure is the following:

1. execution of the packaging cycle;
2. taking of the capsules coming out from the packaging line;
3. placing on the capsules closing lid of a small piece of adhesive rubber, so to avoid the flow

of external air in the container during the perforation;

4. introduction of the sensor's needle;
5. test execution;
6. writing of the value of O<sub>2</sub> residual showed by sensor's monitor.

Before tests' execution is important to check the correct calibration of the sensor, measuring various time the O<sub>2</sub> concentration of the air around the sealing station and verifying that the showed value is always close to 21%.

### 3. RESULTS AND DISCUSSION

#### 3.1. CFD simulation of the original case

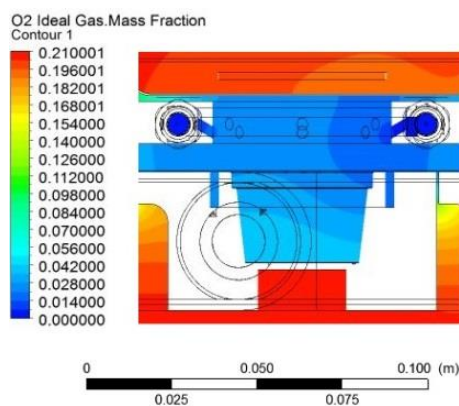
With the "probe" tool in post process section of the used CFD code, that allows to measure the value of a given parameter in a specified point of the fluid domain, the O<sub>2</sub> residual in the geometrical center of each capsule.

The detected values are shown in Table 3.

**Table 3: O<sub>2</sub> residual in capsules' center for current configuration**

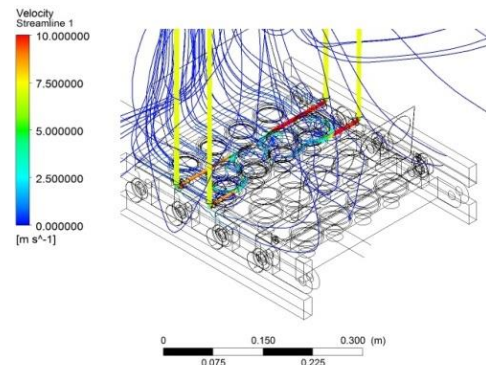
O <sub>2</sub> Residual in Capsules' Center		
Capsule	N <sub>2</sub> Concentration	O <sub>2</sub> Residual
Capsule 1	90.66 %	1.96 %
Capsule 2	83.51 %	3.46 %
Capsule 3	83.59 %	3.45 %
Capsule 4	90.88 %	1.92 %
Mean O <sub>2</sub> Residual		2.70 %

As can be noticed from the previous Table, in each capsule O<sub>2</sub> residual is higher than the target value of 1% and the mean O<sub>2</sub> residual is equal to 2.70%. It is interesting to note also that the two capsules located in the center position of the conveyor belt (indicated with Capsule 2 and Capsule 3) reach a lower N<sub>2</sub> concentration and consequently show a higher O<sub>2</sub> residual. Figure 5 shows the O<sub>2</sub> concentration inside capsule 2.



**Figure 5: O<sub>2</sub> mass fraction inside one of central capsules**

Streamlines have been analyzed too. In fact they represent the preferred trajectories of gas particles therefore their visualization could be useful to identify any N<sub>2</sub> leakage points. Figure 6 provides an illustration of the fluid streamlines within the domain.

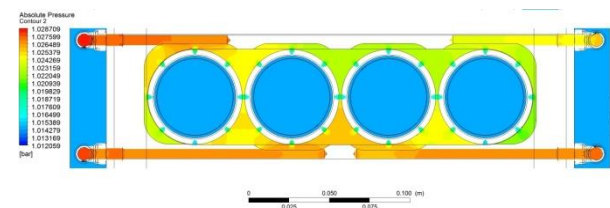


**Figure 6: streamlines within the domain**

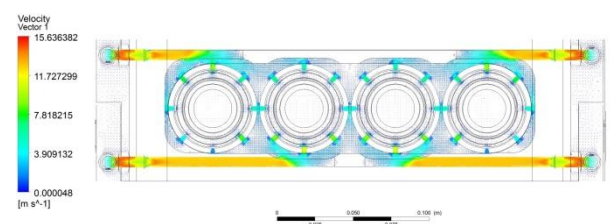
It was not found any particular point of leakage, with the exception of the connecting zone between the sealing station and the tunnel, where there is the N<sub>2</sub> flushing with previously mentioned. This zone was however not considered like a real leakage point since the gas flowing out from here is not lost but goes inside the tunnel and thus contributes anyhow to the effectiveness of the packaging process.

Other analyzed results (respectively shown in Figure 7 and Figure 8) are the absolute pressure and the velocity distribution inside the groove for N<sub>2</sub> distribution, in order to observe the conditions which lead to the O<sub>2</sub> residuals indicated in Table 1.

In particular it has been noticed that the pressure distribution inside the groove is not perfectly homogeneous and that low velocities especially near the holes which address N<sub>2</sub> in the capsules. This may be the cause of the higher O<sub>2</sub> residual detected in the central capsules.



**Figure 7: absolute pressure inside groove for N<sub>2</sub> distribution (original case)**



**Figure 8: velocity vectors inside groove for N<sub>2</sub> distribution (original case)**

### 3.2. Experimental phase validation

The CFD model has been validated with an experimental phase, which has been performed on the described packaging machine.

The followed experimental procedure is the one described in paragraph 2.3. Table 4 shows tests' results.

Table 4: experimental validation results

Experimental Tests Results				
Test no.	Capsule	O <sub>2</sub> Residual [%]	N <sub>2</sub> Flow-Rate [l/min]	Pressure [bar]
1	1	2.1	80	2.5
	2	3.4		
	3	3.6		
	4	1.9		
2	1	1.8	80	2.5
	2	4.0		
	3	3.7		
	4	2.0		
3	1	2.1	80	2.5
	2	3.5		
	3	3.7		
	4	2.0		

The experimental validation phase showed good agreement between CFD model and the real case.

### 3.3. Design of a new N<sub>2</sub> distribution system

Considering the results obtained with the first CFD simulation and during the model validation phase a new N<sub>2</sub> distribution system has been developed. In particular, the main variation is in the shape of the groove realized within the plate. A section of the new plate is shown in Figure 9.

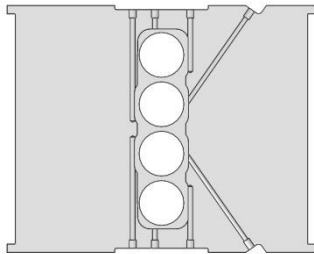


Figure 9: new plate section

As can be noticed from the previous picture the introduced changes have resulted in an increase in the number of N<sub>2</sub> inlet points (from 4 to 8).

After obtaining a new fluid domain a CFD simulation of the modified case has been set up. For the mesh generation the settings described in paragraph 2.2.1 have been adopted, while boundary and initial conditions are the same of the ones shown in Table 1, except for the Inlets. In fact, as previously explained, with the new plate eight Inlets are present thus on each one of them a Volumetric Flow-Rate of 10 l/min, and a Temperature of 25°C have been fixed, in order to keep the total flow rate constant.

### 3.4. CFD simulation of the new case

The results obtained for O<sub>2</sub> residual with the new plate are shown in Table 5.

Table 5: O<sub>2</sub> residual in capsules' center for new configuration

O <sub>2</sub> Residual in Capsules' Center				
Capsule	N <sub>2</sub> Concentration	O <sub>2</sub> Residual	Increase in N <sub>2</sub> Conc.	Decrease in O <sub>2</sub> Residual
Capsule 1	96.44 %	0.75 %	6.4 %	- 61.7 %
Capsule 2	97.34 %	0.56 %	16.6 %	- 83.8 %
Capsule 3	97.42 %	0.55 %	16.5 %	- 84.1 %
Capsule 4	96.25 %	0.78 %	5.6 %	- 59.4 %
Mean O <sub>2</sub> Residual		0.66 %		
Mean Increase in N <sub>2</sub> Conc.			11.28 %	
Mean Decrease in O <sub>2</sub> Residual				- 72.25 %

As can be noticed from previous Table, the introduced changes allow to reach the target goal of an O<sub>2</sub> residual lower than 1% by mass. In particular the new configuration shows a mean decrease in O<sub>2</sub> residual of - 72.25 %, allowing to reach a mean O<sub>2</sub> residual of 0.66% (max residual 0.78 %). Furthermore, with this configuration the most disadvantaged capsules are no longer the ones in central position but are Capsule 1 and Capsule 4.

Also in this case, absolute pressure and velocity distribution inside the groove realized within the plate have been analyzed (respectively Figure 10 and Figure 11).

In particular it has been noticed a more homogeneous pressure distribution, with values closer to atmospheric pressure and higher velocity values in correspondence of some of the injection holes (maximum velocity of about 16 m/s).

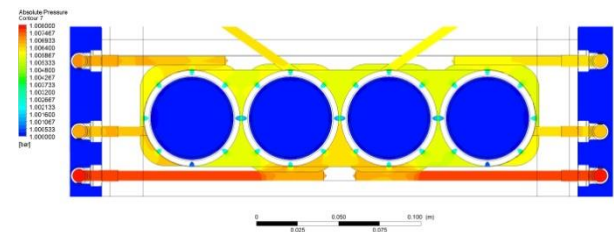


Figure 10: absolute pressure distribution inside groove for N<sub>2</sub> distribution (modified case)

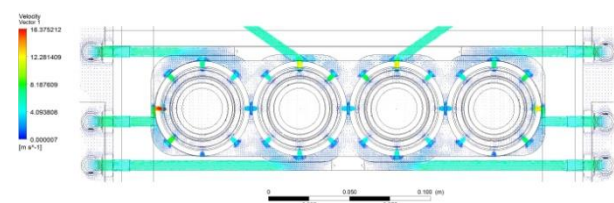


Figure 11: velocity vectors inside groove for N<sub>2</sub> distribution (modified case)

Clearly, maintaining the total flow-rate constant but increasing the number of inlets, N<sub>2</sub> velocity inside the distribution pipes is lower than in the original case,

and also this fact can be noticed comparing Figure 9 and Figure 11.

### 3.5. Experimental phase results

As previously stated, experimental tests have been executed also on the new configuration of the packaging machine, in order to validate the CFD model.

The followed experimental procedure has been described in paragraph 2.3. Table 6 shows tests' results.

**Table 6: experimental phase results**

Experimental Tests Results				
Test no.	Capsule	O <sub>2</sub> Residual [%]	N <sub>2</sub> Flow-Rate [l/min]	Pressure [bar]
1	1	0.5	80	1.8
	2	0.4		
	3	0.4		
	4	0.4		
2	1	1.80	80	2.1
	2	0.7		
	3	0.7		
	4	1.0		
3	1	0.6	80	2.5
	2	0.6		
	3	0.6		
	4	0.7		

The experimental tests, especially test no. 3, show a good correspondence between CFD modeling and experimental phase.

### 3.6. Other CFD simulations

The CFD simulations with the other type of capsules have been performed only in steady state mode.

The O<sub>2</sub> residual results are presented in Table 7 for the model "A Modo Mio" and in Table 8 for the Nespresso-Like capsule.

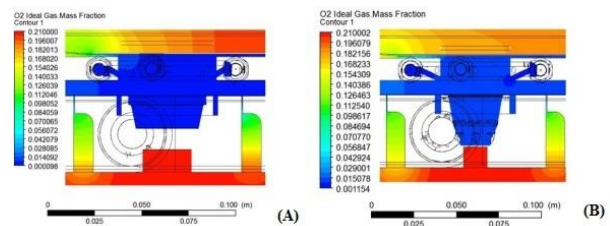
**Table 7: O<sub>2</sub> residual in "A Modo Mio" capsules' center**

O <sub>2</sub> Residual in Capsules' Center		
Capsule	N <sub>2</sub> Concentration	O <sub>2</sub> Residual
Capsule 1	96.55 %	0.73 %
Capsule 2	97.63 %	0.49 %
Capsule 3	96.94 %	0.64 %
Capsule 4	96.65 %	0.71 %
Mean O <sub>2</sub> Residual		0.643 %

**Table 8: O<sub>2</sub> residual in capsules' center for Nespresso-Like capsules**

O <sub>2</sub> Residual in Capsules' Center		
Capsule	N <sub>2</sub> Concentration	O <sub>2</sub> Residual
Capsule 1	94.15 %	1.22 %
Capsule 2	95.08 %	1.03 %
Capsule 3	95.27 %	0.98 %
Capsule 4	95.06 %	1.04 %
Mean O <sub>2</sub> Residual		1.068 %

In Figure 12 is compared the O<sub>2</sub> mass fraction inside the container for this two kind of capsules.



**Figure 12: O<sub>2</sub> mass fraction inside one of central capsules for the model "A Modo Mio" (A) and the Nespresso-Like capsule (B)**

## 4. CONCLUSIONS

Modified atmosphere packaging of coffee in capsule is increasingly used because producers need to preserve as long as possible final product quality. In particular it is increasingly felt the need to obtain a O<sub>2</sub> residual in the head space of containers equal to or less than 1% by mass. Thus, the aim of this work was to optimize, by means of CFD analysis, the geometrical shape of some components in the sealing station of a coffee packaging machine, in order to achieve the desired O<sub>2</sub> residual with an equal or lower inert gas (N<sub>2</sub> in this case) consumption. In particular, it has been modified a groove realized within a stainless steel plate which composes the sealing station of the machine and that is used for N<sub>2</sub> distribution. To avoid the need to consider the effectiveness of the coffee degassing process capsules were considered empty. The model of the capsule used in the simulation is "Blue", produced by Lavazza.

The original configuration of the packaging machine was first simulated, using the plate currently installed, showing how this design could not allow to reach an O<sub>2</sub> residual lower than 1% for each capsule. In particular a mean residual of 2.70 % was detected for the simulation.

The CFD model has been validated with an experimental phase, executed on the packaging machine with the old plate installed on it. The tests showed a good correspondence between simulations and experimental data.

Subsequently, it has been simulated the new configuration of the packaging line, with the modified

plate installed. This analysis showed how the new plate can allow to reach the prefixed goal: in fact, a mean O<sub>2</sub> residual of 0.66 % has been detected for the simulation, which means that the introduced changes have led to a mean decrease in unwanted gas residuals of - 72.25 %. Thus it is confirmed that the new sealing station is more effective in reaching a low level of O<sub>2</sub> residual than the previous one.

Other two simulations have been performed in order to analyze if the introduced changes were effective also with other capsules models which can be processed by the machine object of this study. The analyzed capsules in this case were the model "A Modo Mio", made by Lavazza, and a Nespresso-Like capsule. In the first case a mean O<sub>2</sub> residual of 0.643% has been obtained, while in the second case the mean O<sub>2</sub> residual was equal to 1.068%.

These new simulations also have highlighted a correlation between the capsule shape and the level of O<sub>2</sub> residual that can be reached. In particular it has been noticed that higher is the capsule height, harder is to reach a low level of O<sub>2</sub> residual and that smaller is the capsule diameter, harder is to reach a low residual.

Future researches could be addressed to improve the CFD simulation process and make the analysis more accurate, adding a moving mesh, in order to take account of the motion of the conveyor belt.

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# DYNAMIC FREIGHT FLOW MODELLING FOR RISK EVALUATION IN FOOD SUPPLY

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## ABSTRACT

The research presented in the paper defines a new dynamic model of food supply. The model includes 51 food categories including four different temperature ranges. It differentiates between the different actors in a supply chain: food producers, food retailers, wholesalers, logistics service providers, and the consumer. It works on an aggregate level of 402 regions within Germany as well as the most important trading nations. In the model, inventories for every food category, every group of actors, and every region are recorded in a single data cube. This data cube is recalculated incrementally every day, considering the production, relocation of food products, and consumption. Gravity models calibrated with data of the Federal Transport Plan generate the aggregate commodity flows between the regions. A detailed sectorial input output model for food is calculated for the year 2012 based on data from public authorities, food-related associations, and professional data providers.

Keywords: risk evaluation, simulation, German food supply, freight transport modelling

## 1. INTRODUCTION

This paper introduces a new concept of modelling dynamic commodity flows in supply networks. Where supply relations exist, commodities flow between locations from production via processing and distribution to the points of sale. These commodity flows are modelled for all supply relations of all companies of one specific sector on a national level. The dynamic aspect of the commodity flows is realized by modelling changeable relations between the different locations as well as by modelling the chronological sequence of production and transportation.

The modelling of dynamic commodity flows is necessary for the risk assessment of supply networks because in supply networks processing depends on the availability of preliminary products. If a supplying location fails or a connection gets disrupted, the affected actors in the supply network have to react in order to keep their production running. For example, commodity flows have to be rerouted, new connections have to be established, or safety stocks have to be used. The better a supply network can react to disruptions the more agile it is, lowering the risk of a supply shortfall.

The measurement of risk and agility is of growing importance because increasingly widespread supply networks are prone to specific kinds of risks. In addition, principles like lean, where amongst others safety stocks are reduced, are intensifying this need.

The concept of modelling dynamic commodity flows presented in this paper has the ability to meet this claim by mapping the agile supply network and its evasive reactions to disruptions on a detailed level. In the following sections, the concept is explained with a test case and first results are shown. The food supply sector in Germany is chosen as the test case. It is one of the most important sectors because it is the foundation for the life of the population. Since food supply shortfalls could endanger the lives of many people, risks have to be identified correctly and possible reactions have to be planned accordingly.

## 2. METHODOLOGY

The model consists of four tiers representing the supply chain: agricultural production, processing, warehousing, and consumption. It covers the German land area divided into 402 NUTS3-regions. In these 402 regions the most important stakeholders are aggregated. These are food producers, food retailers and wholesalers as well as logistics service providers. For the tiers agricultural production and processing, the relevant supplying regions are considered as well. The foods are categorised into 51 groups including four different temperature ranges, which represent the majority of the German food supply network from agricultural raw products like crops or raw milk to consumable foods like bread and pastries or cheese.

Since production, processing, warehousing, and consumption of food products are not taking place in the very same regions, the food products have to be transported between the regions while passing all four tiers. Transport within the tiers processing and warehousing is possible too, in case some foods are processed more than one time or the supply chain of a retailer has a hierarchic warehouse structure. The decision which region is supplying which region will be modelled with a gravity model for each food category.

In the model, inventories for every food category, every group of actors, and every region are recorded in a single data cube. This data cube is recalculated incrementally every day, considering the production, relocation of food products, and consumption (see Figures 1 and 2).

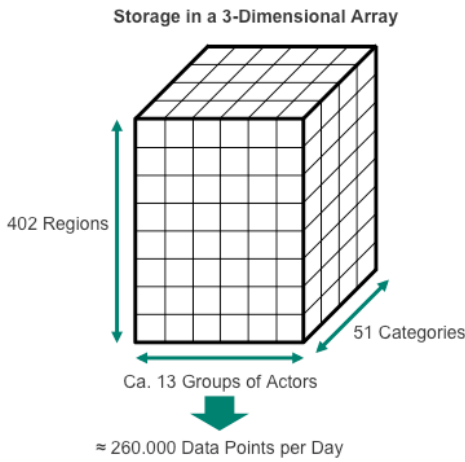


Figure 1: Dimensions of the Data Cube

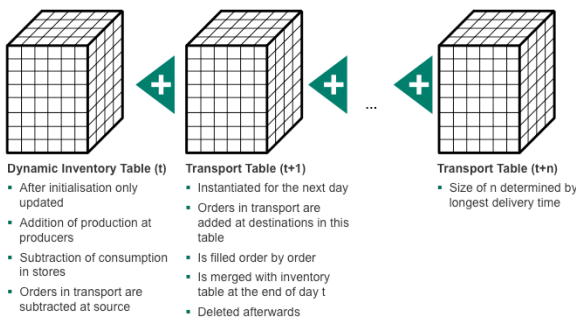


Figure 2: Interactions of Data Cubes for Inventories and Transports

The required data is collected for the year 2012 from the Federal Statistical Office, the Federal Ministry of Food and Agriculture, the Federal Ministry of Transport and Digital Infrastructure, food-related associations, and professional data providers. The gathered data and modelled processes are verified by interviews with experts from the production, logistics, and retailing sector. A model with more recent data will be possible, but it is important to consider only data from the same time period to get a consistent overall picture. Lots of data could be collected for the years 2013 and 2014, but especially national accounts are updating slowly and are still only available for 2012.

### 3. FINDINGS

So far, the data sources have been consolidated and aggregated to match 51 food categories. For each category the amount of metric tons produced, imported, and exported has been determined on a national level. In addition, the usage of the resulting amounts for consumption, other food products and other productions

has been identified. The result is a detailed input-output-table for the German food sector.

In a second step, the consumption and production has been split from the national level to the 402 regions. While the consumption has been split up relative to the distribution of population, production has been split up by data from the Federal Ministry of Food and Agriculture as well as employment data.

Using the detailed regional production data and the results of the gravity model as well as an optimal interaction distribution the two average distances from the Federal Transport Plan for agricultural products and processed products could be detailed to all 51 food categories.

Combining all this with the data concerning the dates of harvest, processing and consumption, the level of stocks over time are calculated allowing statements regarding times of minimal stock and necessary minimum stock levels.

### 4. RESEARCH LIMITATIONS

The amount of data used in the model is a problem for different reasons. The gathering of data is complex and time consuming. Other issues are restrictions due to excessive time necessary for computing such big data, making model generation and visualisation with software tools difficult.

### 5. PRACTICAL IMPLICATIONS

The dynamic simulation of supply networks of single companies should be interesting for all supply chain managers. The dynamic simulation of a whole sector offers new insights to companies as a kind of market analysis. The main application of such dynamic sector model will be the national emergency preparedness. In this case, the public authorities will use the model to identify weaknesses in the food supply and take measures to improve the security and reliability of the food supply in Germany.

This is done using different scenarios, which can be integrated into the normal scenario of 2012. This enables predictions about stock levels in regions affected by disruptions. In addition, different measures taken by public authorities could be tested for their effects on stock levels and freight transport demand.

### 6. ORIGINALITY

A model is proposed, which uses system dynamics and operations research methods to show commodity flows in multi-level supply networks dynamically developing over time. It improves market and risk analyses and makes the assessment of agile networks possible.



# FINITE ELEMENT ANALYSIS INVESTIGATION OF RESPONSE OF PUMPKIN TISSUE UNDER UNIAXIAL LOADING

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## ABSTRACT

Finite element analysis (FEA) models of uniaxial loading of pumpkin peel and flesh tissues were developed and validated using experimental results. The tensile model was developed for both linear elastic and plastic material models, the compression model was developed only with the plastic material model. The outcomes of force versus time curves obtained from FEA models followed similar pattern to the experimental curves however the curve resulted with linear elastic material properties had a higher difference with the experimental curves. The values of predicted forces were determined and compared with the experimental curve. An error indicator was introduced and computed for each case and compared. Additionally Root Mean Square Error (RMSE) values were also calculated for each model and compared. The results of modelling were used to develop material model for peel and flesh tissues in FEA modelling of mechanical peeling of tough skinned vegetables.

Keywords: mechanical properties, tensile, compression, Finite Element Model, peel and flesh tissues, Root Mean Square Error, error indicator.

## 1. INTRODUCTION

Reducing the volume of loss and waste in post harvesting and processing stages of agricultural products can increase food availability globally (Hodges, Buzby et al. 2011). Applying Finite Element methods and software in modelling industrial operations is a new trend among researchers, particularly in the food and beverage industry. Researchers focus on determining and optimising the best possible conditions and highest quality of products considering conditional variables, mainly application of these methods in analysing stress versus strain for food materials under external loading, as it is usually difficult to use common experimental methods (Li, Li et al. 2013). Additionally, computer based models are capable of predicting the outputs of operations even before manufacturing the equipment (Schaldach, Berger et al. 2000). Considering the advantages of these methods and software, there are some limitations which need to be well thought-out. Material modelling, boundary conditions, dimensions and geometric aspects are some of these crucial parameters. Additionally, regarding the complexity of industrial operations, it is necessary to simplify the models in order to reduce the computation time and error. The response of agricultural crops and food

materials differs in terms of damage on different sections and magnitude of load; specifically under large deformations the rate of loss can be high due to the nature of tissue. This paper details the research carried out in development of FE models of the mechanical loading process of pumpkin peel and flesh samples utilizing Finite Element Method (FEM). Tensile and compression tests are two common methods of evaluating mechanical response of food materials under loading. In order to select an appropriate material model, results of compression and tensile tests (Shirmohammadi, Yarlagadda et al. 2011, Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013, Shirmohammadi and Yarlagadda 2014) were used to develop FE model. The model development was designed to present accurately the mechanical response of peel and flesh samples. The model has been constructed to create a numerically stable and efficient representation of tough skinned vegetable response under large deformation. The model was developed using the LS DYNA v971 (2001) program which is utilized in large deformation static and dynamic behaviours of materials (Hallquist 2006).

### 1.1. Response of material

Response of structure under loading is an important aspect in modelling and simulation of any engineering system. This response can be classified as linear or non-linear, which is highly related to the type of structure and operations that have been modelled. Linear response is defined as a direct relationship between stress and strain values (Kilcast 2004), which is normally limited to a very low rate of load for a short period of time related to the softness of material. In a real world operation the response to the loading process is usually non-linear. In the case of agricultural and food materials, the linear response of materials is limited only to the small deformation condition, while the raise in the deformation value leads to non-linear behaviours of materials (Lu and Puri 1991, Lu and Puri 1992, Lu and Chen 1998). Although an elastic portion is usually considered as response of material to simplified mechanical loading tests, even this portion is restricted to the deformation condition, value of yield stress and the limit of elasticity. In this study the material behaviour is considered to be nonlinear and the failure criterion was used to develop FE models. The bio-yield and rupture points of samples obtained from

experimental results were considered as effective parameters in modelling procedure.

## 1.2. Material Model Evaluation Using Experimental Results

Agricultural and food material behaviours under external loading can be complex as these materials are naturally soft and there are various parameters affecting their response. For example, the level of moisture content, variety and ripening stage are some of these influential parameters. FE modelling of their response in mechanical operations is a challenge. The important aspect is selecting an appropriate material model which exhibits similar properties of materials in both elastic and plastic regions. LS DYNA package is used in this project to develop the model; in terms of material modelling the main concerns were the accuracy of the FE model behaviour under loading in comparison with the experimental results. Stress versus strain and force versus deformation were the parameters to compare the results of modelling with the empirical outcomes. In order to evaluate the material response to the external loading, two compression and tensile models were developed and the results of simulations were validated using the experimental tests results. A piecewise model with the capability of inputting elastic and plastic section details has been chosen to create the material model. The bio-yield stress was used as a point in which the failure starts in the tissue structure, and the details of plastic section of stress strain curve were entered as the guide for plastic changes in material. Piecewise-linear plasticity formulation (MAT\_24) is one of the available material types in the LS DYNA package which classifies material response to elastic and plastic sections. In the elastic zone, material's modulus of elasticity (E) is the main parameter and the limit of elasticity is defined by Yield stress. The other option available is to define 8 points after yielding as effective plastic stress and effective plastic strain values, to define the actual stress versus strain curve in plastic region. After yielding the tangent modulus of material is required for the material model. However when the stress versus strain curves is input as the effective plastic stress and strain data, defining the values of tangent modulus were not required (2001).

Failure phenomenon was another essential parameter to be defined and considered. This failure breaks the cell walls and creates elastic and inelastic deformations in agricultural crops such as apple, which have been defined as bio yield point by Mohsenin (Mohsenin 1986). For the harder materials such as kernels this phenomenon creates cracks that can be visible or invisible on the surface of grain, in meat however it has been defined as the tearing and separating that happens in the connected parts of tissues (Mohsenin 1986). For agricultural materials the failure is directly related to the cell wall resistance to the applied load. Plastic deformation basically is known as a state of failure in material structure (Fischer-Cripps 2000) which can happen due to various changes on structure of tissue

such as cleavage, slip and bruise (Holt and Schoorl 1983). There are different failure criteria depending on the type of material and external loading. In this study the Yielding stress (Von Mises Criterion) has been considered as the main factor of causing failure in the material. In this failure condition mainly deviatoric stress creates the changes rather than the hydrostatic stress (Fischer-Cripps 2000). Based on experimental results the Von Mises failure criterion is more accurate than the Tresca failure criterion (Leckie and Dal Bello 2009).

## 2. MATERIAL MODEL

Experimental results (Shirmohammadi 2013, Shirmohammadi, Yarlagaadda et al. 2013) were used to develop and validate the FE model using both elastic and plastic properties of tissue. The Piecewise\_Linear Plasticity model uses the Cowper Symonds model with the following formulation for scaling the strain rate (2001):

$$\begin{aligned} \sigma_y(\dot{\epsilon}_{eff}^p, \epsilon_{eff}^p) \\ = \sigma_y^s(\epsilon_{eff}^p) + SINGY \cdot \left(\frac{\dot{\epsilon}_{eff}^p}{C}\right)^{1/p} \end{aligned} \quad \text{Equation 1}$$

Equation 1 is used in dynamic case where the *SINGY* > 0, however for static problems when the *SINGY* = 0 the model will apply the following solution:

$$\sigma_y(\dot{\epsilon}_{eff}^p, \epsilon_{eff}^p) = \sigma_y^s(\epsilon_{eff}^p) \left[ 1 + \left(\frac{\dot{\epsilon}_{eff}^p}{C}\right)^{1/p} \right] \quad \text{Equation 2}$$

In Equation 2, the *SINGY* is the Yield stress,  $\epsilon_{eff}^p$  and  $\dot{\epsilon}_{eff}^p$  are effective strain and strain rates, and  $\sigma_y^s(\epsilon_{eff}^p)$  is the static stress. Nonlinearity and large strains happening in small stress condition in mechanical loadings of food materials (Mohsenin 1986), and the possibility of applying both elastic and plastic behaviours of tissue with failure criteria at the bio-yield point was the reason of selecting material MAT\_24 for FE model.

### 2.1. Model development

Compression and tensile models were developed based on the size and dimensions of the samples in experimental tests (Maryam Shirmohammadi, 2013). The following assumptions were considered:

The compression loading process happens with the constant rate of 20 mm/min.

Tensile samples were dog bone-shaped with a narrow section in the middle. The length of samples was 40 mm with a width of 10 mm.

Shell element was selected for the tensile model and the thickness of samples was 3 mm according the experimental dimensions of samples.

The material is assumed to be homogenous and moisture content assumed to be constant.

According to the unit consistency in LS DYNA the system of units: tonnes, mm, s, N, Mpa and N.m (2013) was selected for all the models.

The flesh and peel samples had different heights according to the experimental specimen dimensions.

Material properties were obtained from the empirical results.

Both flesh and peel models were developed using solid element and triangular mesh.

A cylindrical model of flesh and peel samples with diameter of 40 mm were developed for compression test. The height of peel and flesh samples were 5 mm and 34.44 mm respectively. The values of stress versus strain curve were input for the material model. The elastic modulus was considered to be the slope of stress-strain curve. Poisson's ratio value for flesh samples also was obtained from experimental tests (Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013). Poisson's ratio value for flesh considered 0.43, for peel samples however the Poisson's ratio was considered as the value determined for unpeeled samples (0.33) (Shirmohammadi and Yarlagadda 2014). A set of nodes was defined on the bottom surface of the model and fixed support was applied at this area limiting the translational and rotational movement in X, Y and Z directions. The compressive movement also modelled as a displacement-time as applied on a set of nodes on the top side of samples which moved in a negative direction of Z axis. The termination was applied as the time that the compressive loading experiment has been stopped. A Prescribed\_Motion\_Set was applied for a set of nodes that was defined on top surface of the samples with the displacement in Z direction and the load curve obtained from experimental tests. The termination time for all models was considered to be the termination time of experimental tests with the time step of 0.1 s.

Dog bone-shaped samples with length of 40 mm, width of 10 mm and thickness of 3 mm were modelled (Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013). The tests were performed on flesh and peel samples of Jap pumpkin previously as a part of PhD study (Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013) under deformation rate of 20 mm/min using an INSTRON universal testing machine. A fully integrated element type was applied for the FE model, which is commonly used for plasticity problems in LS DYNA software (Hallquist 2006). A displacement was applied on a set of nodes defined on one side of the model; a fixed boundary condition was applied on this set. Prescribed\_Motion\_Set was applied on a set of nodes defined on the free side of the model. The motion assumed to be a displacement in X direction and the detail of elongation versus time was obtained from experiments. Stress versus strain curve has been calculated using the results of the force-extension curve obtained from experimental tests from literature (Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013).

True stress and true strain values were used in the modelling (Pruitt and Chakravartula 2012):

$$\sigma_t = \sigma(1 + e) \quad \text{Equation 3}$$

$$\varepsilon_t = \ln(1 + e) \quad \text{Equation 4}$$

In Equation 3 and Equation 4,  $\sigma_t$ ,  $\varepsilon_t$ ,  $\sigma$  and  $e$  are true stress, true strain, engineering stress and engineering strain respectively. Termination time was obtained from experimental testing and the time step was 0.1s. The model was saved as a Key file and sent to LS DYNA solver, the post-processing step was completed afterward using the results of simulation.

### 3. RESULTS AND DISCUSSION

Finite element models with a non-linear elasto-plastic material type for pumpkin tissues were developed assuming that material is isotropic and homogenous. Models were created and simulated based on the following performance criteria: the FE models of compression and tensile loadings are numerically compatible with the experimental and constitutive results, and the material model selected for each model should accurately represent the mechanical behaviours of tough skinned vegetables under loading. The performance criteria were employed to facilitate developing an accurate FE model for each part. Results of FE models were compared with the results of experimental compression and tensile tests (Shirmohammadi 2013, Shirmohammadi, Yarlagadda et al. 2013), an error indicator values were defined based on the following description:

$$e_{force} = \frac{|F_{exp} - F_{FEA}|}{|F_{exp}|} \times 100 \quad \text{Equation 5}$$

In Equation 5,  $e_{force}$ ,  $F_{exp}$  and  $F_{FEA}$  are error indicators for the force predicted, experimental force value and FEA predicted force value. For each FEA set of results, the individual differences between experimental values and values predicted by model, an error percentage was determined.

#### 3.1. Development and Validation of Finite Element Model of Tensile Test

The tensile model was developed for two different material behaviours including linear elastic and plastic material response. As existing FE modelling studies in literature mainly used linear elastic material model for development and validation of their models (Wu and Pitts 1999, Hernandez and Belles 2005, Lewis, Yoxall et al. 2008, Sadrnia, Rajabipour et al. 2008, Li, Li et al. 2013), it was decided to create and validate a tensile model in order to be able to compare the results with available literature. Afterward, models of tensile and compression tests were developed using plastic material model.

The displacement versus time curve was applied up to the elastic limit for the loading and as is mentioned before, the stress versus strain relationship assumed to be linear with the slope equal to elastic modulus value. The density, elastic modulus and Poisson's ratio of

tissues were determined from experimental study (Table 1) (Shirmohammadi and Yarlagadda 2014).

Table 1: Mechanical properties of samples input for the tensile model.

	Density(ton/mm <sup>3</sup> )	Elastic Modulus (Mpa)	Bio-Yield Stress (Mpa)	Poisson's Ratio
Peel	$0.903 \times 10^{-9}$	25.02	1.5	0.33
Flesh	$0.934 \times 10^{-9}$	7.63	0.58	0.434

The models were solved and the outcomes of force versus time were determined as shown in Fig. 1. The maximum stress happened in the narrow middle section of the samples, which is the section where rupturing happened in experimental tensile tests as well. This was expected as the cross sectional area is smaller than the sides of sample, and as a result the value of stress will be higher on the middle section in comparison with the sides. Maximum tensile load reached 22.8 N, which was close to the experimental value 20.21 N for peel samples. The results of force in flesh samples were 7.4 N while the experimental value was 7.8 N. The difference between errors for the predicted value from the FE model and the actual experimental values were calculated as error percentage values and the percentage of error versus time curves was determined for both samples as is shown in Fig. 1(b).

Table 2: RMSE and maximum load predicted by FE model using linear elastic material properties.

Linear Elastic Material		RMSE	Maximum Force (N)
FEA	Peel	1.06	22.8
	Flesh	0.39	7.87
EXP	Peel	-	20.21
	Flesh	-	7.47

The Root Mean Square Error (RMSE) values also were determined to compare the accuracy of the FE models, the RMSE values for peel and flesh samples were 1.03 Mpa and 0.36 Mpa respectively. These RMSE values show the difference between the FE predicted value with the actual values observed in experimental tests. The ratio of individual differences over the experimental values for force was also determined as error indicator (see Equation 1) for sets of data obtained from FE and experimental results.

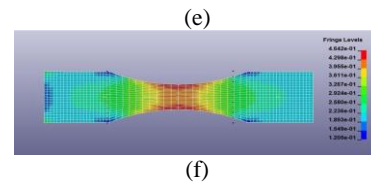
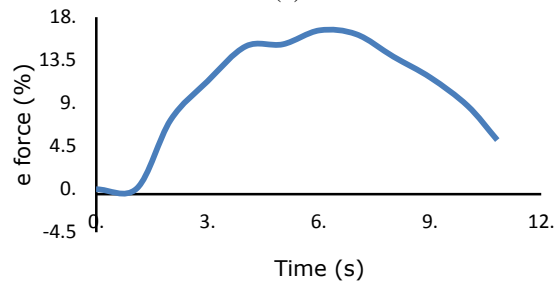
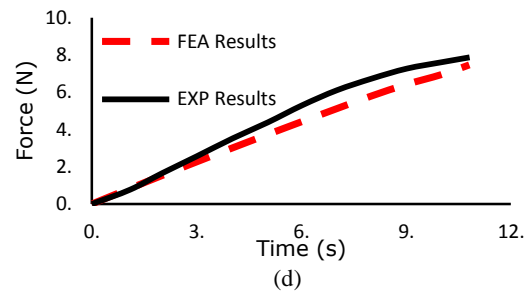
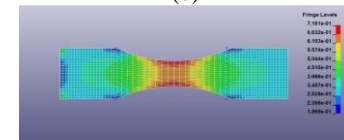
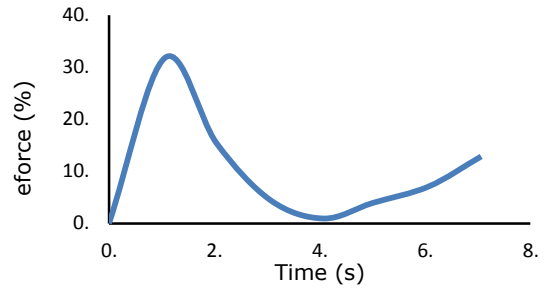
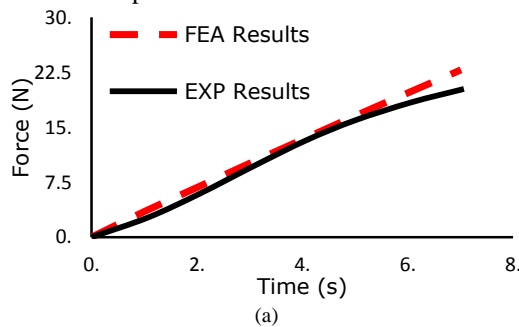


Fig. 1: Tensile force versus time for peel samples (a), error percentage between predicted and experimental values (b) and (c) the stress distribution resulted from FE model, (d), (e), (f) results of flesh model..

Values of error shown in Fig. 1(b) showed that the maximum difference between predicted and experimental value happened where the experimental curve bends, while the FE results is a linear line regarding the material type selected. Maximum error value was 31% for peel samples. The results of stress for flesh samples with elastic material properties showed the maximum error near the force peak value which was 16% (Fig. 1(b)).

The comparison between FE and experimental work shows a better agreement for smaller deformation; similarly it is reported by Dintwa et al. (Dintwa, Jancsó et al. 2011) for tomato under compressive loading. Where the results of the FE model showed a very close result under smaller deformation (less than 0.2 mm), the error values for flesh samples were much

lower than the error percentage for peel samples, according to the shape of experimental curve. For flesh samples, under smaller deformation the difference between predicted values was lower while for peel samples, smaller errors appeared under larger deformation (see Table 2 and Fig. 1). In comparison between peel and flesh results also there was a better correlation between experimental and FE results for flesh samples; the percentage of error and RMSE were lower. This indicates the development of permanent plastic deformations are starting to occur and the stress versus strain relationship is changing to a nonlinear. As is mentioned before regarding the curved shape of peel results, the first section of peel results was less linear than the flesh samples, which were flatter than peel samples. The modulus of elasticity, density, Poisson's ratio, bio yield stress, and the true stress versus strain curve were input as material properties (see Table 1) for material properties to develop a PieceWise\_Linear\_Plasticity model. This material model is a failure-based material type which requires the details of yielding point and the effective plastic stress versus strain curves. The curves in Fig. 2 feature a force versus time curve resulted from FE modelling; the differences between experimental values and predicted values with FE in two different cases have also been presented. Root Mean Squared Error also was calculated for the obtained results and shown in Table 2 and Table 4. The RMSE was also 0.44 Mpa which is lower than the values determined for FEA with the experimental input model (1.36 Mpa). For the first two seconds of the loading, the FEA model results followed a close pattern with the experimental curve Fig. 2(a) and the maximum predicted values of force was higher than the experimental results. The results of FE models for flesh samples have been presented in Fig. 2. The maximum force predicted by the FE model was 8.53 N and the RMSE was 0.18 Mpa and error indicator factor for peel samples showed the maximum error of 18.5%.

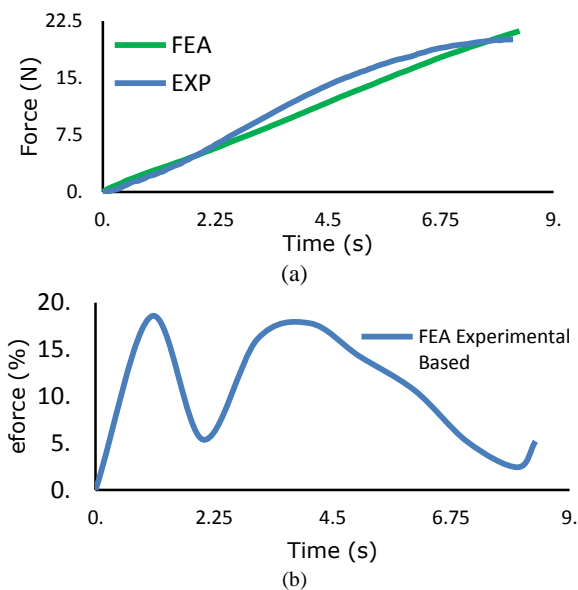


Fig. 2: Results of FE model with experimental input properties, and percentage of error for peel samples under tensile loading.

The comparison between peel results showed that both predicted maximum load values for peel samples were higher than the experimental values. From the error point of view however, maximum error for FEA model of flesh sample was over 35%, which happened in under 2 seconds of test (see Fig. 3(a)).

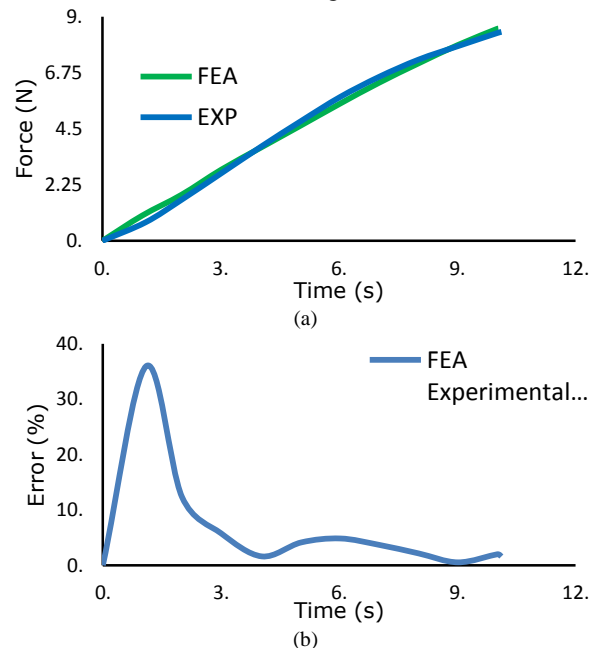


Fig. 3: Results of FE model with experimental and constitutive input properties, and the error percentage for flesh samples under tensile loading.

### 3.2. Development and Validation of Finite Element Model of Compression Test

The FE model of compressive loading consisted of two parts, including the compression test of flesh and peel samples. Model geometry was developed based on experimental sample size (Shirmohammadi 2013). Poisson's ratio value for flesh samples was calculated using the results of experiments from literature, for the peel samples Poisson's ratio assumed to be 0.33 equal to the value calculated for unpeeled samples under uniaxial compression (Shirmohammadi 2013, Shirmohammadi, Yarlagaadda et al. 2013). The results of experimental tests were used to develop stress versus strain curves for both samples (Fig. 4). Bio-yield point was defined for the samples as a point where the compressive force value decreases or stays unchanged with the increase in deformation value and elastic modulus was defined as the slope of curve in that limit (see Table 3) for material properties of tissue).

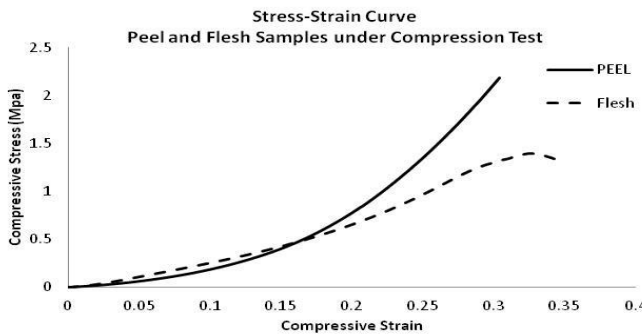


Fig. 4: Stress versus strain for peel and flesh samples obtained from experimental study under compressive loading.

Force versus time results have been presented in Fig. 5, the first curve (a) in Fig. 5 shows the results of FE modelling test for peel samples. The input data was obtained from experimental results and included the density, elastic modulus, Bio-Yield stress and the effective plastic stress versus strain curve. Regarding the failure criterion, the effective stress and effective strain were required in order to consider the material behaviours after yielding happens where failure in materials is assumed to occur.

Table 3: Material properties were defined for peel and flesh samples under compressive loading obtained from experimental study.

Material	Density (ton/mm <sup>3</sup> )	Elastic Modulus (Mpa)	Bio-Yield Stress (Mpa)	Poisson's ratio
Peel	0.903×10 <sup>-9</sup>	2.59	2.18	0.33
Flesh	0.934×10 <sup>-9</sup>	4.19	1.39	0.434

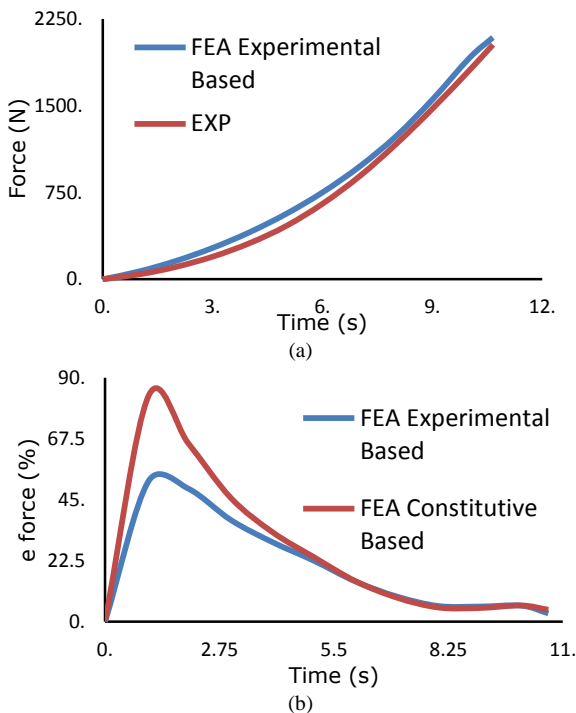


Fig. 5: Force versus time curve resulted from FE model, and comparison with experimental results for peel samples.

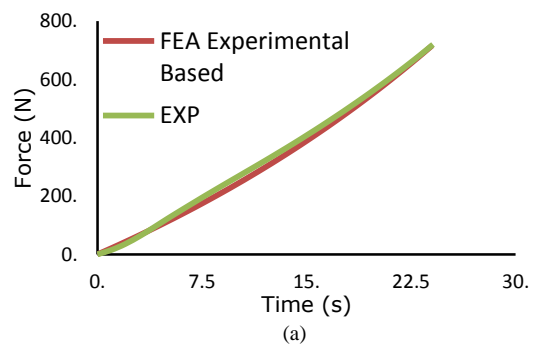
Fig. 5 (b) represents error indicator for predicted values for force in comparison with experimental values for FEA. As it has been shown, the errors were higher for

the first 2 seconds of test while the error values dropped gradually after a peak and reached minimum after 8 seconds. Despite the difference between percentages of error during the model running time, FE curve results had a similar pattern in comparison with experimental curve. The Root Mean Squared Error (RMSE) value for experimental based FEA was 16.77 Mpa. The values for peel samples were 76.8 Mpa for FEA.

Table 4: Comparison between FEA and experimental results and RMSE values for tensile ad compression tests.

Tensile	RMSE		Maximum Tensile Load (N)	
	Peel	Flesh	Peel	Flesh
FEA	1.36	0.18	21.19	8.53
Experimental	-	-	20.14	8.39
Compression	RMSE		Maximum Compressive Load (N)	
	Peel	Flesh	Peel	Flesh
FEA	76.98	16.77	2088.81	1073.68
Experimental	-	-	2026.658	1176.19

As is presented in Fig. 6, the error calculated for experimental based FE model is higher for the first 20 seconds of compression loading test, while the error percentages are lower after 20s. The error indicator had a peak for both peel and flesh samples in the first 2 seconds of the test with 42%, while the error value in peel results was much higher than the flesh. This high rate of error was due to the curve shape of force versus time diagram (see Fig. 5(b)). The FEA results also showed a stress distribution from the top to the bottom where the force is applied on top of the cylindrical samples (Fig. 6 (d)); the same pattern was observed for peel samples. This phenomenon was similar to what happened in experiments; damage was observed as crushed layer of tissue on the side of samples where load was applied was higher and the deformation was clear.



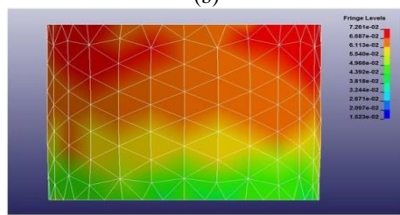
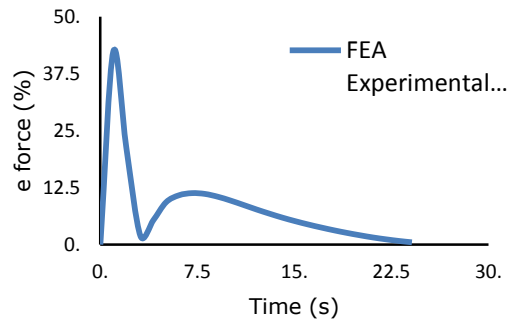


Fig. 6: Force versus time curve resulted from FE model and comparison with experimental and constitutive results for flesh sample.

(a) Results of FEA with experimental input, (b) the error percentages for FE model with experimental and constitutive based input, (c) the FEA stress distribution results.

The RMS Error for compression model was lower for flesh samples in both experimental and constitutive input cases (Table 4) Maximum predicted force values were higher than the experimental values for peel samples for both FE models, while the flesh results were both lower than the experimental values.

#### 4. CONCLUSION

The predicted values for force in the tensile model with linear elastic material properties were 22.8 N and 7.4 N where the experimental data was 20.21 N and 7.8 N for peel and flesh respectively. In general, the outcomes of FE model followed a linear line as expected regarding the type of material was chosen. For the tensile model with piecewise material model where the plastic deformation of tissue was considered, maximum load was predicted 21.19 N and 8.35 N while the experimental values were 20.14 N and 8.39 N for peel and flesh respectively. As mentioned in the previous sections, the force-deformation curve for food and agricultural materials are usually different from other engineering materials. The curve resulted from FE model was similar to the experimental curve in terms of pattern on force versus time curve. Obviously, the Picewise\_linear Plasticity model illustrated better outcomes however the models were developed for one deformation rate according to the experimental condition while further studies on rate of deformation can provide a clearer details of tissue behaviour.

In compression model the maximum compressive load was determined by FE model 2088 N and 1073 N for peel and flesh respectively while the experimental values were 2026 N and 1176 N for peel and flesh respectively. The maximum stress in both tensile and compression models occur on the expected region were

the highest deformation and rupturing on the tissue surface.

This study was one of the few attempts on modelling actual behaviours using FEA approach, however it is recommended to apply this method for other types of tissues where mechanical properties of material plays an important role in reducing volume of unwanted deformation on tissue during post harvesting and processing operations.

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# BREAKAGE MATRIX COMPARISON OF GRANULATED FOOD PRODUCTS FOR PREDICTION OF ATTRITION DURING LEAN-PHASE PNEUMATIC CONVEYING

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## EXTENDED ABSTRACT

### 1. INTRODUCTION

Pneumatic conveying is widely used in the production and handling processes of many food products. These food products can take the form of powders or flake type bulk materials, including items such as tea leaves, granulated sugar, flour, and flavorings. The main advantages of using pneumatic conveying to transport materials throughout a process include the maintenance of product hygiene, and the potential for flexibility in pipework routing in comparison to mechanical conveying. However, the fulfilment of the many benefits of pneumatic conveying can only be realised if the system is correctly configured and operated optimally. In instances where this is not the case, a common repercussion, is bulk material attrition due to excessive impact forces exerted on the particles as they pass through the system, and encounter changes in direction at bends. The final result can often be the generation of excess quantities of degraded particle size fractions – often in the form of fines or dust. The presence of these smaller particle sizes can have significant implications for the reliability of subsequent process operations or customer perception.

#### 1.1. Motivation

The research presented in this document will provide a comparative breakage analysis of two types of food products, one crystalline in structure, and the other non-crystalline. This information may then be used to predict with a greater degree of confidence, the magnitude of anticipated attrition when pneumatically conveying particulate materials.

#### 1.2. Lean Phase Pneumatic Conveying

There are three main modes of pneumatic conveying:

1. Dense-phase.
2. Lean-phase.
3. Pulse/dune flow.

Mode 1 describes conveying conditions when the entire cross-section of the pipe is occupied by particles at relatively low velocity (approximately 2-8 m/s). Mode 2

describes conveying conditions when the particles are fully suspended in the air flow under high velocity conditions (greater than approximately 12 m/s). Mode 3 describes transient conveying conditions between Modes 1 and 2, and describes the flow of ‘dunes’ of material progressing through the pipeline.

This study will consider lean phase pneumatic conveying effects specifically, as the combination of high velocity and low particle concentration yields the most likely conditions to induce an increased level of particle attrition.

#### 1.3. Repercussions of excess fines content

The term ‘fines’ in this research shall refer to the  $d_{10}$  content of the virgin cumulative size distribution. Should the material within this size range make up too large a proportion of the overall mass, adverse material handling and performance issues can occur. These types of issues are extensive, and include:

- Unreliable discharge from hoppers,
- Dust generation, leading to explosion or health hazards,
- Segregation of mixtures,
- Excess dust in commercial products leading to customer complaints,
- Error in volumetric dosing operations due to variations in bulk density.

The occurrence of such issues typically results in additional time and financial costs to the producers of the product. These costs are consumed in amending the configuration or operating conditions of the existing process, or in reformulating the nature of the material handled.

#### 1.4. Advantages of predictive analysis

This research seeks to provide an improved tool that can be applied to identify risk within processes. The approach will use predictive analysis of the product through empirical testing. Particle impact conditions can be carefully controlled within a centrifugal accelerator attrition tester, to provide comparative data

for qualitative analysis. Then, based on a comparison with products previously handled, insight into the breakage behavior may be attained and its subsequent attrition behavior estimated.

A selection of advantages that predictive analysis would yield include:

- Reduced requirement for post-installation troubleshooting,
- Enhanced understanding of the operating parameters that cause an unacceptable degree of attrition,
- Heightened stakeholder awareness of the nature of the material, with regard to particle attrition.

## 2. LITERATURE

There has been significant research conducted into lean phase pneumatic conveying, with respect to the erosion of pipeline bends. A number of models exist which seek to predict the level of erosion through modelling and empirical testing (Burnett 1996; Deng 2005; Macchini 2013; Sato 1995). As an issue that is closely related to erosion, there is presently a lack of research focusing solely on modelling of attrition of bulk material particles during lean phase pneumatic conveying.

Kalman (2000) conducted an extensive programme of pneumatic conveying tests, exploring the effect of a wide number of parameters on the attrition of various bulk materials. The equipment used in this study has the potential to cause particle attrition, and this has been addressed in the manuscript. The output of the research is limited to general relationships and recommendations in limiting the degree of particle attrition caused by a pneumatic conveying system.

Bridle (2000) performed an in-depth study of particle attrition through the testing and comparison of results between two testing facilities: a single bend test apparatus, and an industrial scale apparatus. The drawbacks of this study were that only three bulk materials were considered: granulated sugar, basmati rice, and malted barley. In addition to this, the testing programme did not consider the attrition behavior of narrow size fractions, nor could the test be performed in a conventional laboratory due to floor space requirements.

Chapelle P (2004) demonstrated a method for taking measurements from a bench top tester and scaling them up to predict the particle attrition behavior in an industrial-scale pneumatic conveying system. While this work lays the foundation for obtaining a scale-up procedure from a bench top scale, the range of variables for this form of testing were not fully explored, omitting the effect of solids loading ratio and impact angle. Additionally, a 5 gram sample was used, making the data exceedingly sensitive to variation in sample size distribution.

van Laarhoven, (2012) developed a new attrition tester of a scale suitable for placement on a laboratory desk. This tester contained the test sample within an

oscillating box, where the particles would impact against the walls, causing attrition. Impact velocities estimated to be in the order of 1-5 m/s were attained. The significantly larger impact velocities observed in pneumatic conveying systems (up to 40 m/s) were simulated through increasing the number of impacts in the tester. The study was also limited to two granular materials, and tested a wide sample size distribution.

## 3. METHOD

Section 3 will describe the methods used to acquire the results in this course of research.

### 3.1. Materials

Two materials were selected for this course of research: golden breadcrumbs, and cooking salt. The former has a non-crystalline structure, and the latter exhibits a crystalline structure. Additionally, each material has a significantly different size distribution, with respect to the mean and range of particle sizes.

### 3.2. Statistically Representative Sampling

Due to the generally highly variable nature of bulk materials, it is essential that the testing be conducted using statistically representative samples obtained from the master batch. Should this prerequisite be neglected, subsequent tests could be conducted using samples with significant variation in size distribution or mixture composition. Therefore, statistical subdivision was conducted through the use of two apparatuses: a spinning riffler and riffle boxes.

#### 3.2.1. Spinning Riffler

This piece of equipment is used for statistically subdividing samples of up to 40 litres in volume and is depicted in Figure 1.



Figure 1: Spinning Riffler

A mass-flow hopper is positioned above a vibratory feeder, which in turn dispenses the bulk material master sample into eight sub-samples on a rotating carousel. The amplitude of the vibratory feeder can be varied to

control the bulk material dispensary rate. This method produces eight statistically representative samples from a master sample.

### 3.2.2. Riffle Boxes

Riffle boxes are used for statistical subdivision of sample sizes up to approximately 500 millilitres. Figure 2 depicts the riffle boxes used for this programme of research.

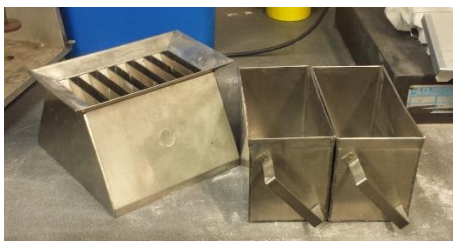


Figure 2: Riffle Boxes

Material is poured into the top of the separating section, which contains a series of slots, alternating between two receiving vessels underneath. This method produces two statistically representative samples from a master sample.

### 3.3. Particle Size Measurement

In order to determine the virgin size distribution of the sample and the post-test size distribution, a stack of mechanical sieves was used. These were cleaned in an ultrasonic bath prior to use in order to remove any contaminants.

The sieve size progression chosen for this course of research followed a  $\sqrt{2}$  progression increase in aperture size. Due to the interest in the smaller size fractions generated, as detailed in the introduction, the sieve progression will begin at the smallest size of 45  $\mu\text{m}$ . This will enable closer inspection of the material residing within small size fractions generated as a result of the degradation testing.

The bulk material sample was loaded into the top of the sieve stack, after which the stack was placed on a vibrating plate, depicted in Figure 3. The subsequent sieving was performed for 15 minutes.

Once sieving was complete, a set of electronic scales was used to determine the mass retained on each sieve to the nearest 0.01 g.

### 3.4. Attrition Test

The attrition testing was performed on a centrifugal accelerator style attrition tester, depicted in Figure 3.



Figure 3: Centrifugal Attrition Tester

The sample material is inserted into the upper mass-flow hopper of the apparatus. It is then fed into the tester below via an increasing-capacity screw. The rate of screw rotation was kept constant throughout the course of the testing, and consistency in material delivery rate was stable. The function of the screw feeder is to ensure that the volumetric flow rate of material into the attrition test is controlled – a trait that is not closely controlled in many forms of attrition testing.

The material fed from the screw feeder then enters the center of a rotating spindle, and accelerated outwards through a series of radial tubes. The accelerated particles then impact with a target array at a known angle of impact.

The post-test sample is collected at the base of the impact chamber, and taken for size analysis.

#### 3.4.1. Sample Preparation

To obtain the required sample size for each test, the master sample was riffled down to a suitable size for separation via mechanical sieving. In the knowledge that each material batch fed into the sieving unit is statistically representative of the master sample, the size fraction retained on each sieve may also be treated as statistically representative samples. Therefore, samples of known size range with upper and lower bounds may be prepared for testing in the centrifugal accelerator attrition tester.

## 4. RESULTS

The results for this course of research consist of the initial virgin size distribution, the resultant size distributions when a statistically representative sample is degraded under known conditions, and the resultant size distribution when a known size fraction of material is degraded under known conditions.

It should be noted that the value of '0  $\mu\text{m}$ ' for the sieve size represents the material retained in the pan at the base of the sieve stack.

### 4.1. Virgin Size Distribution

The initial analysis undertaken was to determine the virgin size distribution of the 16 kg master sample for

golden breadcrumbs and salt, presented in Figure 4 and Figure 5 respectively.

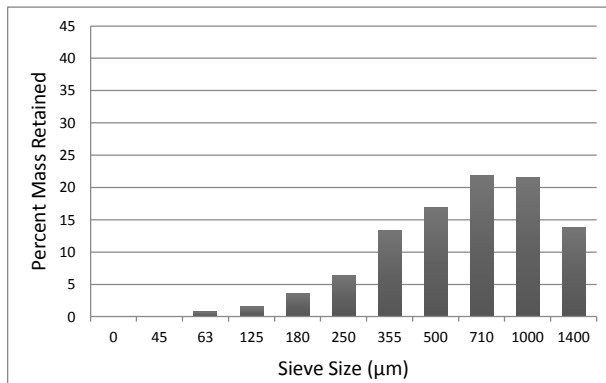


Figure 4: Virgin Size Distribution of Golden Breadcrumbs

No particles used for testing were retained on a 2000 µm sieve – the next sieve in the  $\sqrt{2}$  progression. Though inspection of the size distribution of golden breadcrumbs, the three size fractions representing the largest proportions of material by mass were chosen to undergo narrow size fraction degradation tests. These size fractions consist of the material retained on 500, 710 and 1000 µm sieves. These fractions represent approximately 58% of the total sample distribution by mass.

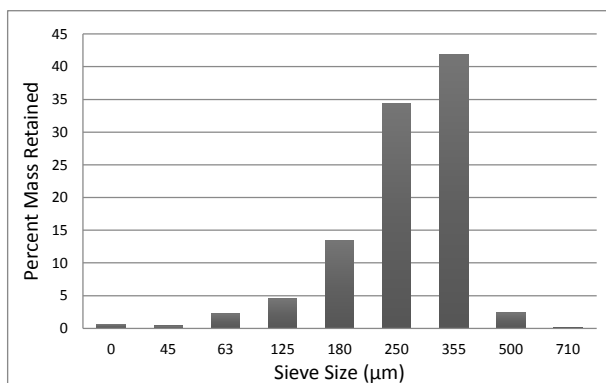


Figure 5: Virgin Size Distribution of Salt

No particles used for testing were retained on a 1000 µm sieve – the next sieve in the  $\sqrt{2}$  progression. Through inspection of the size distribution of salt, two size fractions, 250 and 355 µm, were chosen for narrow size fraction degradation testing. These fractions represent approximately 75% of the total sample distribution by mass.

#### 4.2. Size Distribution Post-Breakage

This section presents the results of the degradation tests performed with the full virgin size distribution as the input material. Three velocities were tested: 15, 25 and 35 m/s. The results for the golden breadcrumbs are presented in Figure 6, and the results for the salt are presented in Figure 7.

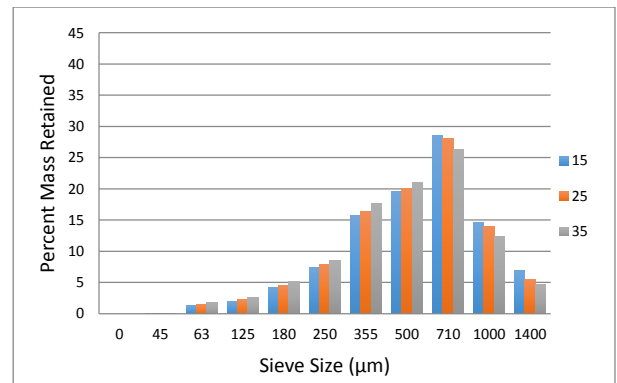


Figure 6: Results of Full Size Distribution Tests under Three Impact Velocities (measured in m/s) for Golden Breadcrumbs

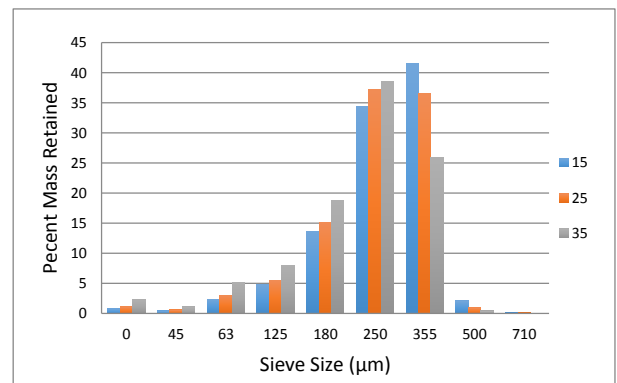


Figure 7: Results of full size distribution tests under Three impact velocities (measured in m/s) for salt

#### 4.3. Narrow-Fraction Size Distribution Post-Breakage

This section presents the results of degradation tests of narrow size fractions across three impact velocities: 15, 25, and 35 m/s. Figures 8, 9 and 10 present the results for golden breadcrumbs, and Figures 11 and 12 present the results for salt, all at impact velocities of 15, 25 and 35 m/s.

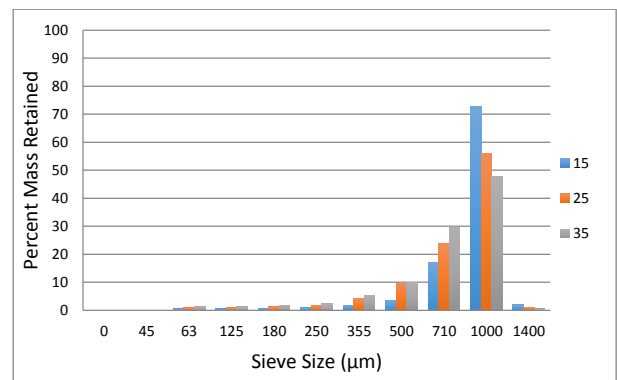


Figure 8: Results of 1000-1400 µm size fraction tests under three impact velocities (measured in m/s) for golden breadcrumbs

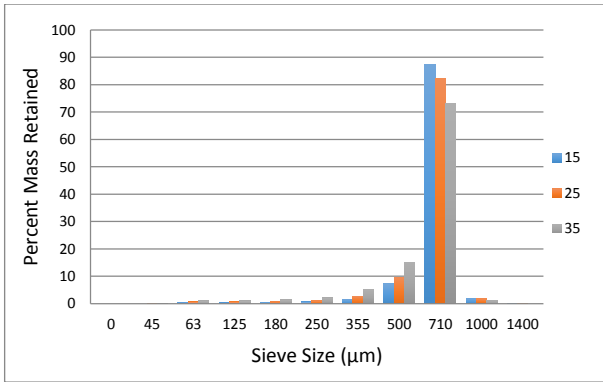


Figure 9: Results of 710-1000 µm Size Fraction Tests under Three Impact Velocities (measured in m/s) for Golden Breadcrumbs

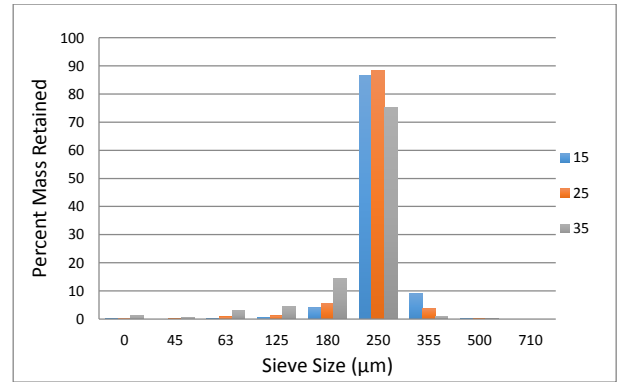


Figure 12: Results of 250-355 µm Size Fraction Tests under Three Impact Velocities (measured in m/s) for Salt

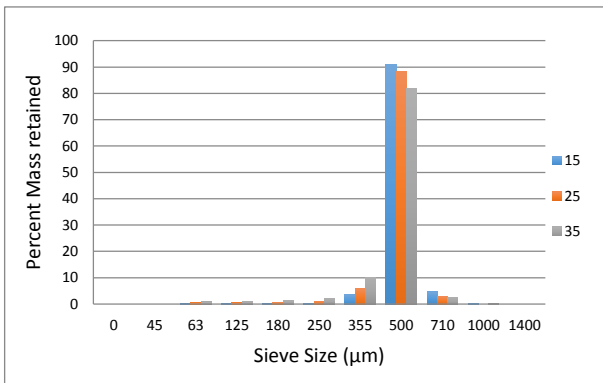


Figure 10: Results of 500-710 µm Size Fraction Tests under Three Impact Velocities (measured in m/s) for Golden Breadcrumbs

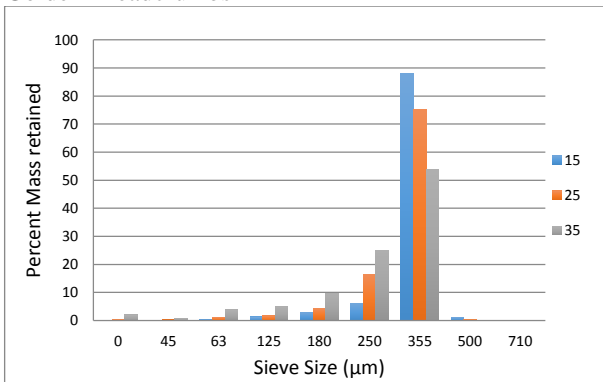


Figure 11: Results of 355-500 µm Size Fraction Tests under Three Impact Velocities (measured in m/s) for Salt

## 5. DISCUSSION

This section will present the analysis of the results, looking at each of the materials and their representative size fractions in turn.

### 5.1. Virgin Size Distributions

There are considerable differences observable between the virgin size distributions of the subject materials in this research. Firstly, the golden breadcrumbs exhibit a wider size distribution, where considerable sample mass is clearly spread across seven size fractions from 180 to 1400 µm. Additionally, smaller size fractions are represented with smaller mass percentages. In comparison, the salt exhibited a much narrower overall size distribution, with 75% of the sample by mass residing in two size fractions – on the 250 and 355 µm sieves.

### 5.2. Full Size Distribution Tests

Observations of the full size distribution tests display behavior typical of attrition tests for most common materials. In the larger size fractions, more mass is retained for low impact velocities in comparison to the high impact velocities. This indicates that more of the larger particles in a sample break under high impact energies. Additionally, it is evident that less mass is retained in the smaller size fractions at low impact velocities in comparison to the higher impact velocities. As expected, an increased number of smaller particles are generated at higher impact velocities.

Figure 13 shows the percentage gain or loss in material for golden breadcrumbs in each size fraction, based on the virgin size distribution for each of the three impact velocities considered.

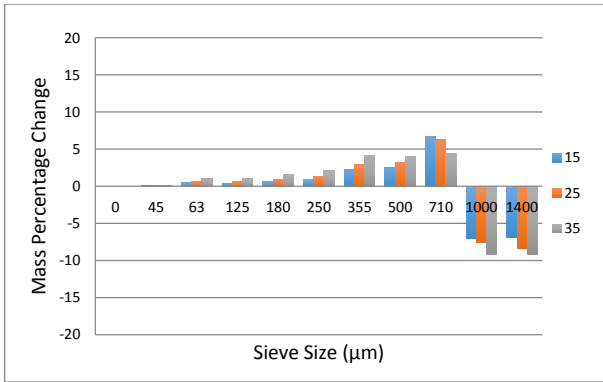


Figure 13: Mass Percentage Change of Golden Breadcrumbs under Impact Velocities (measured in m/s)

The only size fraction that does not follow the expected trend with regards to impact velocity, is the 710  $\mu\text{m}$  sieve. Rather than gaining more material as the impact velocity increases, the reverse is true. This may indicate that this particular size fraction loses more material through particle breakage, than it gains from particle breakage of larger size fractions, namely the 1000 and 1400  $\mu\text{m}$  fractions. By increasing the impact velocity, and hence the impact energy, this behavior is emphasized. Figure 14 presents the results of the same analysis for salt.

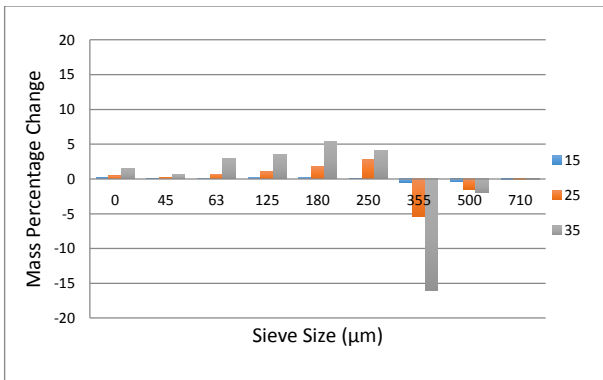


Figure 14: Mass Percentage Change of Salt under Impact Velocities (measured in m/s)

It is evident from the minimal migration of material across the size fractions at 15 m/s, that the impact energy is not sufficient to cause a significant amount of degradation. However, it can be concluded that at 25 and 35 m/s, the impact energy is sufficient to generate smaller particles from the larger size fractions. In particular, large changes are observed at 35 m/s, where a significant proportion of particles undergo size changes.

### 5.3. Narrow Size Fraction Tests

General observations of the narrow size fraction attrition tests clearly show that larger particles are more susceptible to breakage in comparison to smaller particles for the materials studied. Additionally, the narrow size fractions display similar behavior to that

observed in the full size distribution tests, where more particles are broken at higher impact velocities.

Figures 15, 16 and 17 show the percentage change in the size distribution for narrow size fractions of 1000-1400, 710-1000, and 500-710  $\mu\text{m}$  for golden breadcrumbs respectively.

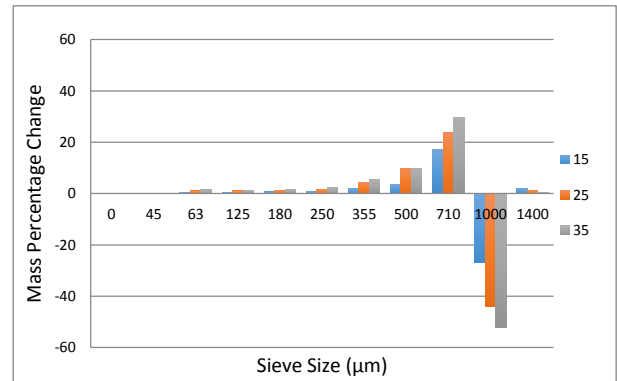


Figure 15: Mass Percentage Change of Golden Breadcrumbs under Impact Velocities (measured in m/s) for the Size Fraction 1000-1400  $\mu\text{m}$

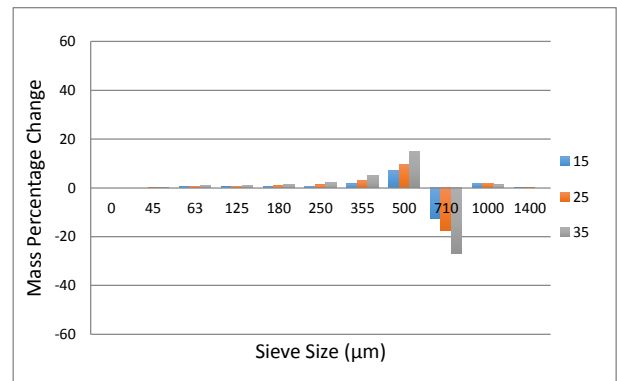


Figure 16: Mass Percentage Change of Golden Breadcrumbs under Impact Velocities (measured in m/s) for the Size Fraction 710-1000  $\mu\text{m}$

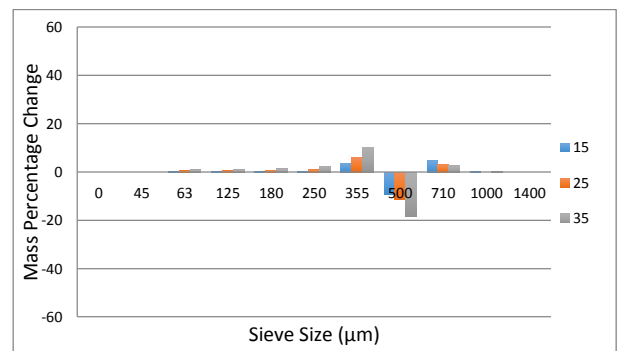


Figure 17: Mass Percentage Change of Golden Breadcrumbs under Impact Velocities (measured in m/s) for the Size Fraction 500-710  $\mu\text{m}$

This analysis shows some mass retained on sieves larger than the size fraction under scrutiny. This is due to the irregular shape of the particles, which have diameters dependent on the orientation of the particle with respect to the sieve mesh.

In comparing the analysis of three adjacent size fractions for golden breadcrumbs, particle sizes greater than 1000  $\mu\text{m}$  in diameter break to a considerably greater degree than either of those retained on 500 or 710  $\mu\text{m}$  sieves. This suggests that once particles reach 1000  $\mu\text{m}$  in diameter, the particle material does not have enough internal strength to withstand the momentum transfer of the impact. However, with the reduced diameter of the 710 and 500  $\mu\text{m}$  particles, the mass is sufficiently small to maintain the structure of more particles post-impact.

Figures 18 and 19 show the mass percentage change analysis of the 355-500 and 250-355  $\mu\text{m}$  size fractions for salt respectively.

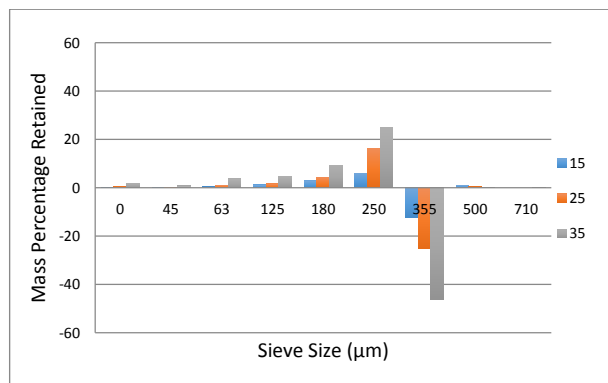


Figure 18: Mass Percentage Change of Salt under Impact Velocities (measured in m/s) for the Size Fraction 355-500  $\mu\text{m}$

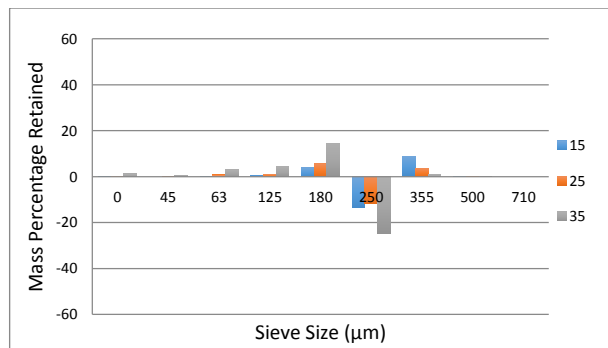


Figure 19: Mass Percentage Change of Salt under Impact Velocities (measured in m/s) for the Size Fraction 250-355  $\mu\text{m}$

Even though the size fractions considered for salt are smaller than those considered for golden breadcrumbs, the magnitude of breakage occurring is comparable under some impact velocities. For example, the amount of material lost for salt in the 355  $\mu\text{m}$  size fraction tested at 35 m/s, is of the same order of magnitude as that lost for golden breadcrumbs in the 1000  $\mu\text{m}$  size fraction at 35 m/s. While there is little relationship between the two materials with respect to particle size, this demonstrates the influence of the material with regard to the particle strength.

#### 5.4. Particle Breakage Relationships

The full size distribution of a sample material post-breakage should be composed of breakage of each size fraction if a logical progression is followed. Therefore, the breakage of the narrow size fractions will be used to construct a predictive size distribution for the full distribution of material at each of the breakage velocities. The simulated size distributions will be calculated from three components:

1. The percent mass of the virgin size distribution,
2. The amount of material lost in each of the size fractions tested for their breakage characteristics,
3. The amount of material contributed to each size fraction from the breakage tests address in Component 2.

This approach will make the following assumptions/limitations:

- Material collected on sieves greater than the diameter of size fractions of interest in the narrow size fraction post-breakage measurements will be omitted and treated as an error in the experimental method,
- Material in the virgin size distribution greater in diameter than that considered in narrow size fraction testing will be omitted from the results displayed, however will still be accounted for in overall sample size to calculate percentage masses of subsequent size fractions,
- Material in size fractions smaller than those tested under the narrow size fraction tests will be treated as stationary (undergo no breakage in the simulation),
- Only a limited number of size fractions will be considered in the testing regime, as addressed in Section 4.3.

The generalized equation used to calculate the simulated percentage mass retained on the sieve for a given size fraction is given in Equation 1.

$$m_{frac,sim} = m_{frac,vir} + \sum_{Frac} \left( \frac{p_{frac}}{100} \times m_{tot,sim} \right) \quad (1)$$

Where:

- $m_{frac,sim}$  is the simulated mass retained in a simulated breakage,
- $m_{frac,vir}$  is the mass retained for a given fraction in the material virgin size distribution,
- $m_{tot,sim}$  is the total sample mass of material input into the simulated breakage and is composed of the full size distribution,
- $p_{frac}$  is the percentage change in mass of each fraction as a function of the sample mass.

Figures 20, 21 and 22 compare the experimental results (denoted by FSD) and the simulated results (denoted by SIM) for golden breadcrumbs at impact velocities of 15, 25, and 35 m/s respectively.

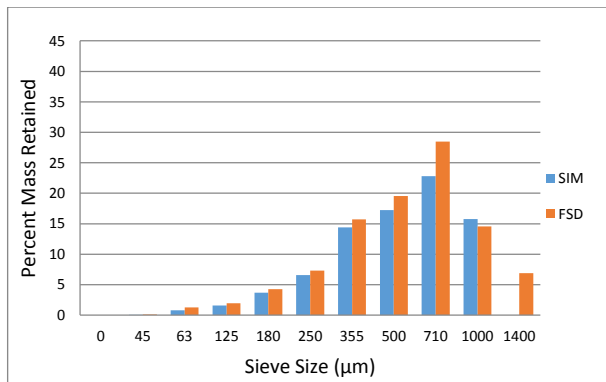


Figure 20: Simulated and Experimental Post-Test Golden Breadcrumb Size Distributions at 15 m/s Impact Velocity

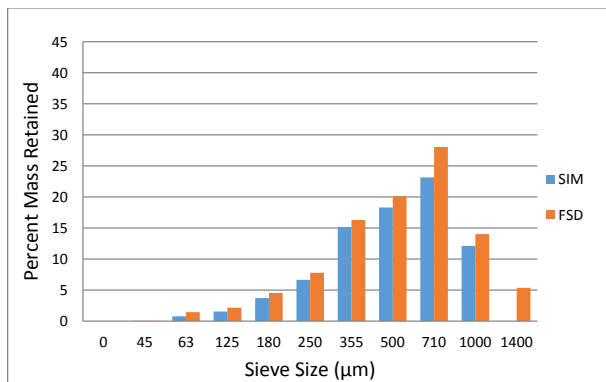


Figure 21: Simulated and Experimental Post-Test Golden Breadcrumb Size Distributions at 25 m/s Impact Velocity

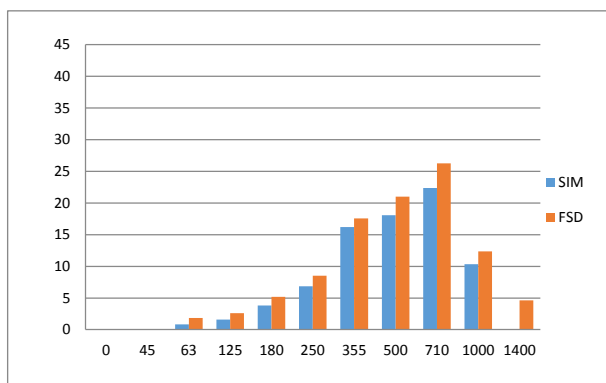


Figure 22: Simulated and Experimental Post-Test Golden Breadcrumb Size Distributions at 35 m/s Impact Velocity

Considering the assumptions made in the simulation method, the predictions made fit reasonably well with the full size distribution breakage results. The most notable differences that can be observed lie primarily in the larger size fractions, namely 710 and 1000 µm. This

observation is expected, as the 1400 µm size fraction was not considered in the narrow size fraction breakage analysis. It would be logically sound to state that this size fraction would contribute to the 710 and 1000 µm fractions, increasing agreement between the simulated and experimental values.

With regard to the smaller size fractions, the simulation underestimates the magnitude consistently. This may be explained by the simulation failing to account for the movement of all size fractions below those for which results are presented. It is expected that all fractions will break to some degree, thus contributing to the stated error.

By taking into consideration both of the aforementioned statements, the accuracy of the simulated predictions could be increased.

Figures 23, 24 and 25 compare the experimental results (denoted by FSD) and the simulated results (denoted by SIM) for salt, at impact velocities of 15, 25, and 35 m/s respectively.

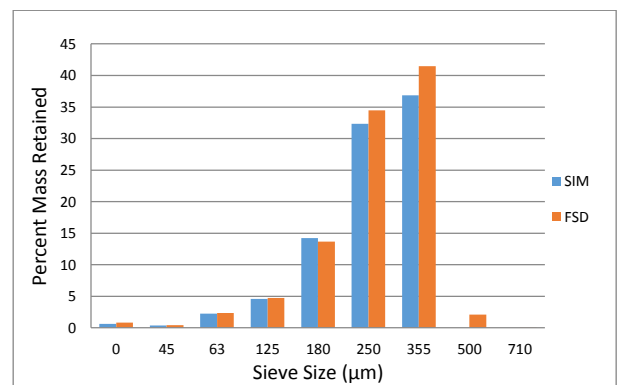


Figure 23: Simulated and Experimental Post-Test Salt Size Distributions at 15 m/s Impact Velocity

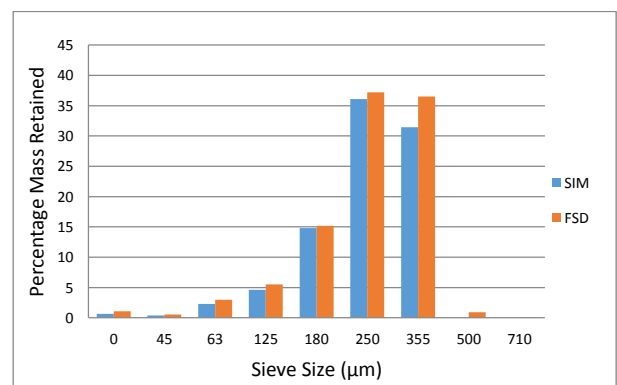


Figure 24: Simulated and Experimental Post-Test Salt Size Distributions at 25 m/s Impact Velocity



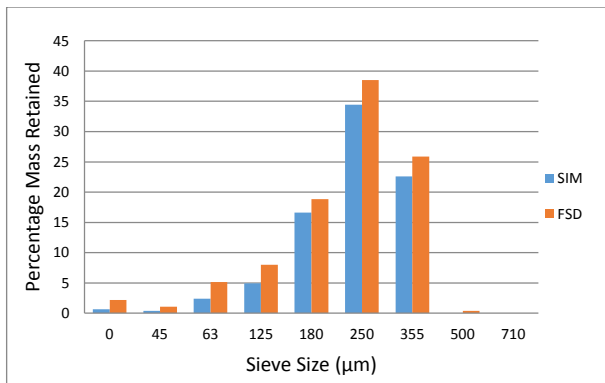


Figure 25: Simulated and Experimental Post-Test Salt Size Distributions at 35 m/s Impact Velocity

As with the golden breadcrumbs, a good agreement between the simulated and experimental results for salt is achieved. Almost all size fractions are under-predicted once again, and as with the golden breadcrumbs, the logical explanations for this behavior lie with the omission of larger size fractions than what underwent testing, in addition to the migration of smaller size fractions.

## 6. CONCLUSIONS

Research into the breakage characteristics of two granulated food products, one crystalline in structure and the other non-crystalline, has been performed. The virgin size distributions were determined, and then the resulting size distributions were tested under three impact velocities: 15, 25 and 35 m/s. Subsequently, three size fractions for golden breadcrumbs and two size fractions for salt were tested under the same conditions as the full size distributions. The results of the latter tests were used to simulate the conditions of the full size distribution breakage characteristics.

The simulated results found good agreement across each of the two materials and the three impact velocities considered. The discrepancies between the experimental and simulated results have been attributed to a combination of omitting the larger size fractions not considered in the testing, and neglecting the migration of the smaller size fractions down the material size distribution. Both discrepancies may be addressed with further testing.

## 7. FUTURE DIRECTION

The future direction of this work falls primarily into two categories: enhancing the predictive capabilities of this method of analysis, and to enhance understanding of how this analysis may be used in the design and/or optimization process.

Enhancing the predictive capabilities of this method includes looking more holistically at the size distribution and testing the breakage characteristics of every available size fraction. From this, increasing the accuracy of the final size distribution prediction should be achievable.

The application of this analysis would involve obtaining a greater understanding of the process in question,

subsequently followed by identifying the undesirable material qualities that result from excess attrition. Determining the ratio of size fractions that exhibit these undesirable qualities will then enable quantification of the process conditions to be avoided.

## ACKNOWLEDGEMENTS

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# OPTIMIZATION OF TRANSPORTATION DECISIONS UNDER EXCLUSIONARY SIDE CONSTRAINTS IN FOOD SUPPLY CHAIN

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## ABSTRACT

The main objective of this paper is to develop an effective model of planning food supplies, when exclusionary constraints are present. In order to achieve this objective, we performed two stages of research, each of which had its own partial goals. The first part of the research aims for identification of the types of the exclusionary constraints that exist in the real world. The objective of the second stage is to develop mathematical models of transportation problems that include the exclusionary constraints existing in reality.

Keywords: optimization of transportation, exclusionary constraints, food supply chain

## 1. INTRODUCTION

A supply chain is a network of organizations working together in different processes and activities in order to bring products and services to the market, with the purpose of satisfying customers' demands (Ahumada and Villalobos 2009). In the case of a food supply chain, as any other supply chain, it is an integrated process where raw materials are acquired, converted into products and then delivered to the consumer; speaking colloquially, it flows from the plant to the plate. The chain is linked with the feed forward flow of materials and the feedback flow of information (Jiang 2009). Tijssens et al. (2001) conclude that the main fact that differentiates food supply chains from other supply chains is a continuous change in quality from the time the raw materials leave the grower to the time the food reaches the consumer.

The topic of the food supply chain is very popular among scientists. This is evidenced by a wide range of research and publications. They concern general problems in food supply chains (Vorst 2000), but there are also investigations which tackle specific problems, such as safety and security issues in the fresh good supply chain (Bruzzone, Longo, Massei, Nicoletti and Agresta 2014), optimization of the fresh food supply chain (Rong, Akkerman and Grunow 2011), evaluating the effectiveness of different policies in managing frozen goods supply chain, (Bruzzone, Massei and Poggi 2007), sustainability and efficiency of logistics in the integrated food district (Bottani, Rizzi and Vignali

2014), traceability (De Cindio, Longo, Mirabelli and Pizzuti 2011a) and track & trace system (De Cindio, Longo, Mirabelli and Pizzuti 2011b) for a food supply chain.

According to Mardsen, Banks and Bristow (2000), creation, operation and evolution of food supply chains are key dimension in the new patterns of rural development now emerging.

The food supply chain is also an interesting subject for institutions which create some rules and best practices for business. In 2010 the European Commission (EC) introduced The High Level Forum for a Better Functioning Food Supply Chain, which ensures the follow up of the recommendations of the High Level Group on the Competitiveness of the Agro-Food Industry (ec.europa.eu 2015). According to EC, the food supply chain consists of three important issues – agriculture, the food processing industry and the distribution system. Those three sectors together account for more than 5% of the European added value and 7% of employment. It is worth to emphasize that the food supply chain has direct consequences for all European citizens, because food represents 16% of the European household expenditures (Commission of the European Communities 2009).

A characteristic thing of the food supply chain is a large variety of entities: farmers, food processors, traders, wholesalers, distributors, logistics companies and retailers. Among these, there are also very large companies and small and medium-sized enterprises which are simultaneously competitors, suppliers or customers. In this industry, companies increasingly cooperate in order to optimize the supply chain. This is reflected most often as: joint commercialization agreements, tying and bundling, joint purchasing agreements (buying alliances) and the increasing use of private labels (Commission of the European Communities 2009).

A lot of challenges are related with food supply chains. One of them is price pressure which forces supply chain leaders into constant efforts to decrease product prices, even when the competitive strategy is primarily focused on such characteristics as quality or delivery time. Globalization and internationalization of enterprises have also contributed to the fact that a large number of

European and American companies produce or commission production in the Asian market (Anholcer and Kawa 2012). According to the McKinsey consultancy, by 2020, production of approximately 80% of goods will take place in a country different from the one where it will be consumed. It will cause huge changes in the movement and consumption of goods (Balou 2007).

What is closely connected with the flow of goods among the entities of the food supply network is transport, which is considered to be one of the most important elements of a logistic system and requires careful planning and control. Lack of planning, ineffective decision making, and poor visibility in transportation can cause companies to overpay, miss delivery targets, and loose valuable business. Transportation costs can be a significant part of a company's overall logistics spending (Murray 2014). According to various estimates, transportation constitutes one-third of the logistics costs (Tseng et al. 2005). Moreover, transportation is the largest end-use contributor toward global warming in many developed countries. Transportation has a significant impact on the food sector because it often involves long-distance shipments, most frequently by road (Wakeland, Cholette and Venkat 2012). Pirog et al. (2001) notice that nearly half of all fruit sold in the US is imported, and that produce grown in North America travels an average of 2.000 km from the source to the point of sale. All the costs are transferred to the customer who needs to pay more for the goods. That is why cutting transportation costs is the major target for companies.

There is a number of transportation strategies that can be applied by management to help improve performance (Murray 2014). One of them is logistics outsourcing. Companies commonly use outside resources, particularly in transport. Logistics service providers often handle loads that come from many different customers. These are products having different physical and chemical properties. They also show different susceptibility to transport, which causes some problems. That is why in many cases it is necessary to impose some exclusionary constraints on the transportation process. In particular, it can be the case when some types of goods cannot be transported by the same mean of transport (like alive animals and frozen fruits and vegetables). Another situation where such constraints have to be imposed is when some suppliers do not want their products to be delivered to the same customer (e.g. because of some reasons concerning the marketing strategy of the company). However, as we can see below, it is not easy to find effective methods of dealing with such types of constraints.

For these reasons the main objective of this paper is to develop an effective model of planning food supplies, when the exclusionary constraints are present.

## 2. EXCLUSIONARY CONSTRAINTS

For the purpose of this paper we have performed two stages of research. In the first stage empirical research

was conducted with the use of secondary and primary sources. On the basis of the secondary sources (scientific papers, industry materials, market reports), we identified exclusionary constraints that were subjected to primary research. Next, we conducted the primary empirical research based on qualitative research in the form of FGI (Focus Group Interview). The main objective of this research was to diagnose the types of exclusions emerging in transport. In July 2015 we conducted FGIs in three cities in Poland using questionnaires with managers from transport and distribution companies. The sample size was three groups, each of which consisted of 5-7 persons.

In order to ensure comparability, the studies in the three cities proceeded along the same scenario. A moderator supervised the procedure, and was also responsible for making all participants contribute to the discussion as equally as possible and, if necessary, for stopping those who excessively tried to impose their views on the other group members. The interviews took place in professional focus rooms, making the experiment comfortable. They took approximately one and a half hours, the procedure was recorded (audio and video). On the basis of those focus group interviews, the following exclusionary constraints that occur in shipment were determined:

- Sensitivity to duration of transport.
- Sensitivity to carriage temperature.
- Sensitivity to air humidity.
- Sensitivity to effects of light.
- Sensitivity to absorption of odors.
- Chemical composition of the product.
- Perishable goods.
- Dangerous goods.
- Animals.
- Competitive products.
- Exclusive transport agreement.

The majority of the above agree with the exclusionary constraints presented in the literature.

In the second stage, based on the determined exclusionary constraints existing in reality, we developed mathematical models of the transportation problems.

## 3. TRANSPORTATION PROBLEM MODEL

The transportation (and, more generally, network) problems with linear side constraints were considered e.g. in (Glover et al. 1978; Klingman and Russel 1975; Thompson and Setbi 1986). It seems that the transportation problem with exclusionary constraints (TPESC), which are in turn nonlinear, appeared for the first time in (Cao 1992). The author considered an ordinary transportation problem with additional condition that for each supplier there are some pairs of destinations that cannot be served at the same time, i.e.:

$$x_{ij} \cdot x_{ik} = 0 \quad (1)$$

for certain triples  $(i, j, k)$ , where  $i$  is number of a supplier and  $i \neq j$  are numbers of destinations. The author provided a branch and bound method to solve this problem. In (Cao and Uebe 1995) the same problem was solved with tabu search. In (Sun 2002), Sun proposed two branch and bound methods to solve the 0-1 mixed integer formulation of the discussed problem. Goossens and Spieksma in (Goossens and Spieksma 2009) studied the computational complexity of the TPESC, proving that it is NP-hard and becomes solvable in pseudo-polynomial time when the number of supply nodes is fixed. They also studied the TPESC with identical exclusionary sets. Recently, Vancroonenburga et al. (2014) studied the so-called red-blue transportation problem, being a special case of the TPESC, where suppliers are assigned to one of two disjoint sets, and each customer is supposed to obtain the deliveries only from the suppliers from one of these sets.

Of course all the above problems have very specific form, as the transportation network in each case has the form of bipartite graph. This is usually not the case in the real-life problems. Another issue is that the authors rather arbitrarily assumed the type of possible exclusions, at least they did not refer to any real-life data.

Also other types of networks were studied. Darmann et al. in (2011) consider the Minimum Spanning Tree Problem, the Maximum Matching Problem and the Shortest Path Problem with binary disjunctive constraints. Pferschy and Schauer in (2013) studied, among others, the Maximum Flow Problem with additional exclusionary constraints, where the exclusions were not restricted to the arcs starting in one vertex. As one can easily see, the last mentioned problems, although involving various types of possible networks, do not focus on the optimization of the transportation process. In some sense, this was the goal of Zhang et al. (2011), who considered the Minimum Spanning Tree Problem with exclusionary constraints and of Öncan et al. (2013), who considered the Minimum Cost Perfect Matching Problem with exclusionary constraints. In both cases, however, the transportation network has so simple structure, that it is far away of the reality.

To summarize, no one so far has considered the minimization of the transportation costs in the networks (especially in food supply chain), which structure is close to the real-life networks and exclusionary constraints are present. Only some attempts were made in the last years to investigate simple models. In consequence no one has studied the methods of solution of problems similar to the real-life problems. From this point of view, the model presented below has pioneering nature.

Let us start with the structure of a typical logistic food supply chain. If we consider a single product, such a chain has the structure of layered network, as presented on Fig.1. Between the layers all the connections are

possible, however, in practice, only few of them are in use (it follows from the properties of the solutions of network problems and consistent with the common sense).

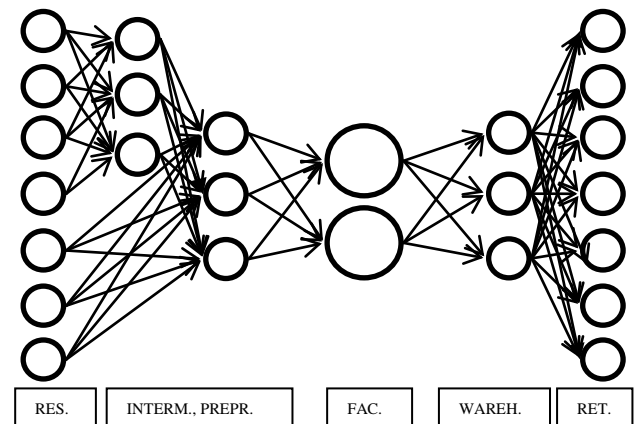


Figure 1: Scheme of food supply chain

What is important, there are many suppliers of the resources, then there are some intermediaries and factories pre-processing the resources and very few (sometimes even one) factories producing the final product. Then the chain starts to be wider again – there are usually more warehouses/logistic centers than factories and much more retailers/ final customers. If we divide the chain into two parts, one can easily see that the part before the main factories can be a little bit complicated (some resources are delivered to the first layer of factories, some of them to the second layer, sometimes the same resource may be necessary in various layers), which sometimes may even cause the network loose the layer structure. What is more important for us – the chains of various products are usually disjoint in this stage. According to the performed research, the problems with exclusions more likely appear in the second part of the chain. Even if the exclusionary constraints rarely touch the products leaving one factory, they start to be important on the level of warehouses/logistic centers, where various products are combined. It is a direct consequence of the fact that the logistic chains of many products intersect at this stage. In the remainder of this section we will thus focus on the second part of the chain.

As we already observed, it can be assumed that the transportation network on this stage consists of the layers and the transportation is possible only between two neighboring ones. We number the layers with consecutive integers and denote the number of a layer with  $l, l = 1, 2, \dots, L$  (where  $L$  denotes the total number of layers). The number of agents in layer  $l$  (factories, centers, warehouses, shops etc.) will be denoted by  $n(l)$  and the index of chosen agent by  $i(l)$ . The total number of goods to be transported is  $G$ , and the index of a good will be denoted by  $g$ . The amount of good  $g$  accessible at any node of the network in the considered period will be denoted by  $a(l, i(l), g)$  – this numbers are positive for all the layers except the last one, where the negative

values correspond with the demand (all those numbers are parameters in the presented model, but can be treated as variables if one considers a dynamic version, see the section “Conclusions and future work”). We assume that the links between nodes are not capacitated (e.g., it is always possible to send more trucks in order to transport all goods). The capacities of the nodes are denoted by  $u(l, i(l))$  (they are defined only for the inner nodes, i.e. the nodes in all the layers except the first and last ones). The volume of a unit of good  $g$  is  $v(g)$ . The unit costs of transferring good through respective node (e.g. the production costs, storage costs etc.) will be denoted by  $d(l, i(l), g)$ . The unit cost of transportation of good  $g$  between agent  $i(l)$  from layer  $l$  and agent  $i(l+1)$  from layer  $l+1$  will be denoted by  $c(l, i(l), i(l+1), g)$ . The variables will correspond with the flows of goods between the nodes/agents and to the amounts of good transferred through the respective nodes. In particular, the amount of good  $g$  transported between agent  $i(l)$  from layer  $l$  and agent  $i(l+1)$  in layer  $l+1$  will be denoted by  $x(l, i(l), i(l+1), g)$ . The total amount of good  $g$  transferred through node  $i(l)$  in layer  $l$  will be denoted by  $y(l, i(l), g)$ .

Since we are interested in the minimization of the cost, the objective function (to be minimized) has the form:

$$f(x) = \sum_{g=1}^G \sum_{l=1}^{L-1} \sum_{i(l)=1}^{n(l)} \sum_{i(l+1)=1}^{n(l+1)} c(l, i(l), i(l+1), g) x(l, i(l), i(l+1), g) + \sum_{g=1}^G \sum_{l=1}^L \sum_{i(l)=1}^{n(l)} d(l, i(l), g) y(l, i(l), g) \quad (2)$$

The capacity constraints have the form:

$$\sum_{g=1}^G v(g) y(l, i(l), g) \leq u(l, i(l)), \quad (3)$$

$$l = 2, \dots, L-1, i(l) = 1, \dots, n(l)$$

$$x(l, i(l), i(l+1), g) \geq 0, \quad (4)$$

$$l = 1, \dots, L, i(l) = 1, \dots, n(l),$$

$$i(l+1) = 1, \dots, n(l+1), g = 1, \dots, G$$

The mass balance constraints are:

$$\sum_{i(2)=1}^{n(2)} x(1, i(1), i(2), g) \leq a(1, i(1), g), \quad (5)$$

$$i(1) = 1, \dots, n(1), g = 1, \dots, G$$

$$\sum_{i(l-1)=1}^{n(l-1)} x(l-1, i(l-1), i(l), g) + a(l, i(l), g) = y(l, i(l), g), \quad (6)$$

$$l = 2, \dots, L-1, i(l) = 1, \dots, n(l), g = 1, \dots, G$$

$$\sum_{i(l+1)=1}^{n(l+1)} x(l+1, i(l), i(l+1), g) \leq y(l, i(l), g), \quad (7)$$

$$l = 2, \dots, L-1, i(l) = 1, \dots, n(l), g = 1, \dots, G$$

$$\sum_{i(L-1)=1}^{n(L-1)} x(L-1, i(L-1), i(L), g) \geq -a(L, i(L), g), \quad (8)$$

$$i(L) = 1, \dots, n(L), g = 1, \dots, G$$

Finally, we need to include the exclusionary constraints. As mentioned before, sometimes two goods cannot be transported together because of their properties. Assume that good  $g_1$  and  $g_2$  cannot be included in the same transport. Then the constraints could take the form:

$$x(l, i(l), i(l+1), g_1) x(l, i(l), i(l+1), g_2) = 0, \quad (9)$$

$$l = 1, \dots, L-1, i(l) = 1, \dots, n(l), i(l+1) = 1, \dots, n(l+1)$$

This is a nonlinear constraint. However, we can preserve the linearity of the model by introducing sufficiently large number  $M$  (for example equal to the maximum capacity of a node in the supply chain) and binary variables  $\alpha(g_1, g_2, l, i(l), i(l+1))$  (equal to 1 if the good  $g_1$  can be transported and  $g_2$  cannot, and 0 otherwise). Then for each  $l, i(l)$  and  $i(l+1)$ , the constraint (9) can be written in an equivalent form (note that we need to use two constraints here instead of one):

$$x(l, i(l), i(l+1), g_1) \leq \alpha(g_1, g_2, l, i(l), i(l+1)) M \quad (10)$$

$$x(l, i(l), i(l+1), g_2) \leq (1 - \alpha(g_1, g_2, l, i(l), i(l+1))) M \quad (11)$$

Another kind of exclusion can take place if two goods cannot be transferred simultaneously through chosen nodes. In such case the constraint, for defined  $l$  and  $i(l)$ , would be as follows:

$$y(l, i(l), g_1) y(l, i(l), g_2) = 0 \quad (12)$$

Again we can use a similar transformation as in the case of constraints (9), by using respectively defined variables  $\beta(g_1, g_2, l, i(l))$ :

$$y(l, l(i), g_1) \leq \beta(g_1, g_2, l, i(l)) M \quad (13)$$

$$y(l, l(i), g_2) \leq (1 - \beta(g_1, g_2, l, i(l))) M \quad (14)$$

#### 4. CONSLUSIONS AND FUTURE WORK

The exclusionary constraints are defined for goods – e.g. if some pair of good cannot be transported together, the constraint looks same for all the combinations of nodes. This together with the possibility of involving the binary variables suggests that the problem can be solved by a branch-and-bound method. It is well-known, however, that such methods work in a polynomial time, which is not acceptable when taking

into account the number of the binary variables even in small instances of the problem. Thus usage of some heuristics or metaheuristics could be useful. The design of various solutions methods for the problem described above will be our main interest in the nearest future.

What is worth mentioning, the above model is quite simple. At least four modifications could be introduced and will be analyzed in the future. The first one is introduction of time to the model, i.e. changing the static model described in this article into a dynamic one. The second modification is the generalization to the supply chains where transportation is possible also between two non-neighboring layers (or, in other words, the network is no more a layered network). Third possible modification is considering the fixed charge costs. The fourth one – considering the model where the costs are nonlinear (in particular piecewise linear). Also in the case of these four modifications the development of effective solution methods will be one of our interests.

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# EFFECT OF FOAMING AGENT, FOAM STABILIZER AND WHIPPING TIME ON DRYING PROCESS OF TOMATO PASTE UNDER DIFFERENT DRYING EQUIPMENT

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## ABSTRACT

Laboratory experiments were conducted to determine the effect of foaming agent (egg white, EW), foam stabilizer (carboxyl methyl cellulose, CMC) and whipping time on drying rate and quality of tomato (*Lycopersicon esculentus*) paste dried under air oven, microwave oven and mechanical dryer. A 4<sup>3</sup> factorial experiment in Randomized Complete Block Design (RCBD) was used to study the effect four levels each of foaming agent (5, 10, 15 and 20% EW), foam stabilizer (0.15, 0.30, 0.45 and 0.60% CMC) and whipping time (3, 5, 7 and 9 min) on drying rate and quality of foam-mat dried tomato powder in the three drying equipment. Each trial was performed in triplicates making a total number of 576 samples that were individually tested and measured. 25 g sample of the paste were dried to a moisture content of 7.60 % (wb) for 8 h in mechanical dryer and oven dryer at temperatures of 55 °C and 50 °C respectively and 10 min in microwave oven at 540W. Data obtained from the experiments were statistically analyzed using the analysis of variance (ANNOVA) while the Duncan's Multiple Range Test was used to compare the means. Results showed that drying rate increased with increase in foaming agent, foam stabilizer and whipping with minimum values of 9.21 g/h obtained in mechanical dryer, 9.31 g/h in air oven, and 8.05 g/h in microwave oven. Increase in foaming agent, foam stabilizer and whipping time did not cause any adverse effect on vitamin C, ash, protein, fat, carbohydrate, crude fibre contents of the samples. Samples reached a stable moisture content of 7.60 % (wb) in less than 8 h in mechanical dryer and air and less than 10 min in microwave oven. The results of

the study showed that egg white, CMC and whipping time influenced the drying rate and quality of foam-mat dried tomato powder.

Keywords: foaming agent, foam-mat drying, foam stabilizer, drying rate, equipment, tomato

## 1. INTRODUCTION

Tomato (*Lycopersicon esculentus*) is the second most important vegetable in many regions of the world with the production in Nigeria being more than doubled in the last ten years. In 2001 alone, the production was 879,000 metric tonnes (FAO 2002) and these were largely consumed in the fresh state. Large quantities are processed into soups, juices, sauces, puree, paste and canned products. Transportation of tomatoes over long distance, hauls in lorries or trucks or in basket subject the tomatoes to vibration and bruising, hence cause loss of nutrient due to enzymatic actions.

Daily intake of tomatoes provides the body some nutrients such as carotene, vitamin, lycopene which lower the risk of cancer and cardiovascular diseases as recommended by the American Cancer Society (Block, Patterson, and Subar 1992). Tomatoes also have antioxidant components that are medically useful in the area of cataracts, bone metabolism and asthma. About 90 % of the developing nations' (such as Nigeria) food supply are wasted owing to postharvest deterioration each year. Tomato production is seasonal and it is a highly perishable commodity in its natural state after harvest. Thus processing it to a form that can be easily preserved is crucial to minimizing wastage and spoilage during



the production season and ensuring that maximum nutritional values of the fruit are retained. Literature and past researches (Sachin, Janghan, and Mujumdar 2011) have shown that drying is one of the best methods of preserving fruits, vegetable and foods.

Various methods of drying fruits, vegetables and food exist; such methods include open sun drying, controlled solar drying, hot-air drying, freeze drying, vacuum drying and more. Idah, Obajemihi, Adeboye, and Olaniyan (2014) studied the assessment of osmotic dehydration-assisted drying of tomato at different drying temperatures under a hot-air dryer. They reported that the osmotic dehydration pretreatments had great influence on the drying kinetics and quality of tomato. The drying characteristic of banana chips and physicochemical changes during drying process were investigated by Itang (2010a, 2010b). Result of the study revealed that a drying temperature and microwave energy of 55 °C and 2.5 W/g respectively were the most suitable drying parameters while the biggest physicochemical changes were found at the primary drying stage. Olaniyan and Omoleiyomi (2013) investigated the drying process of Okra (*Abelmoschus caillei*) under different conditions using a hot-air dryer. Result showed that an optimum drying rate could be achieved by subjecting okra to osmotic dehydration pretreatment prior to drying.

According to Jangan, Law, and Mujumdar (2011), the four major considerations in drying food, vegetables and fruits are speed of operation, energy efficiency, cost of operation and quality of dried products. It is essential that any food drying process should carefully establish these parameters as objective functions. This present study was conducted to investigate some of these parameters in drying tomato paste to tomato powder under different drying equipment: air oven, microwave oven and mechanical dryer. Therefore, the objectives of the study were: (i) to develop mathematical models to estimate the effects of foaming agent (egg white, EW), foam stabilizer (carboxyl methyl cellulose, CMC) and whipping time on drying rate and quality of tomato paste dried under air oven, microwave oven and mechanical dryer; and (ii) to optimize the drying process for maximum drying rate and quality of dried product. It is expected that this will elucidate

the problems of drying and enable the establishment of process parameters for optimal drying of tomato paste to tomato powder.

## 2. MATERIALS AND METHODS

### 2.1. Experimental Equipment

The following equipment and apparatus were used for the study: (i) a mechanical dryer designed and fabricated by Omodara, Olaniyan, and Oyewole (2011) as shown in Figure 1; (ii) a microwave oven (Model: HTMO- 2890EG); (iii) an air oven (Model: MINO/50-10G039, Genlab Engineers); (iv) an electronic weighing balance; (v) a stainless steel knife; (vi) desiccators; and (vii) a blender (Model: HR-2815).

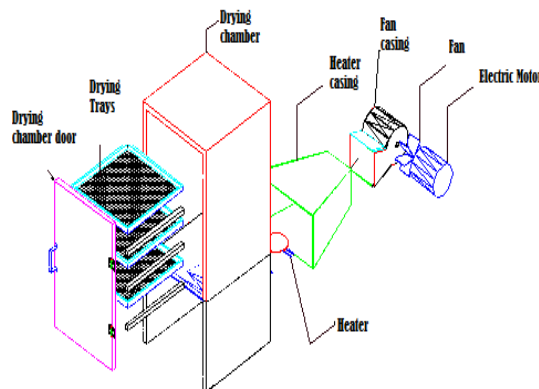


Figure 1: Exploded View of the Mechanical Dryer showing the Component Parts

### 2.2. Experimental Design

A 4<sup>3</sup> factorial experiment in a randomized complete block design was used in this study. The factors taken into consideration included 3 levels each of foaming agent, foam stabilizer, and whipping time. The range of foaming agent considered was 5, 10, 15 and 20 % EW; foam stabilizer was 0.15, 0.30, 0.45 and 0.60 % CMC while the whipping time was 3, 5, 7, and 9 min. Every trial was carried out in triplicate in the three drying equipment (mechanical dryer, air oven and microwave oven), making total of 576 samples that were individually tested and measured.

### 2.3. Experimental Procedure

The experiment was carried out in the Department of Chemical Engineering Laboratory, University of Ilorin. The average room temperature was 30 °C throughout the period of experimentation. Enough quantities of fresh and ripe tomatoes were bought from a local market in Ilorin metropolis. The tomatoes were graded to ensure uniformity of samples to be used for experimentation.

#### 2.3.1. Sample Preparation

The graded tomatoes were washed in clean water, sliced and deseeded with the use of the stainless steel knife on a stainless steel tray and then blended to form tomato concentrate. 100 ml of the concentrate was measured and foamed with 5, 10, 15 and 20 % EW and 0.15, 0.30, 0.45 and 0.60 % CMC. The mixture was then foamed for 3, 5, 7 and 9 min whipping time with the aid of the blender to form a ready-for-drying foamed mixture.

#### 2.3.2. Drying Procedure

The air oven and mechanical dryer were set at temperatures 50 °C and 55 °C respectively while the microwave oven was set at power of 540W. 25g of foamed samples prepared as described earlier were spread evenly on the labelled laboratory dishes at a foam thickness of between 2-3 mm and carefully placed inside each of the drying equipment for drying as shown in Figures 2, 3 and 4 below. The weights of the samples were taken every one hour and the drying operation continued until there was no more changes in the weight of the samples. The experiment was performed in triplicate in all the three drying equipment.



Figure 2: Samples in the Mechanical Dryer during Drying



Figure 3: Sample in the Air Oven during Drying



Figure 4: Samples in the Microwave Oven during Drying

### 2.4. Output Parameters

#### 2.4.1. Drying Kinetics

In this study, drying rate is the amount of moisture removed from product per unit time during a drying operation. It was determined by the equation used by Olaniyani and Omoleiyomi (2013) which is expressed below as:

$$R = \left( \frac{dM}{dt} \right) = \frac{m_i - m_f}{t} \quad (1)$$

where; R is the drying rate in g/h;  $dM$  is change in mass of okra in g;  $dt$  change in time in h; t is the total time of drying in h;  $m_i$  and  $m_f$  are the initial and final mass of okra samples respectively in g.

#### 2.4.2. Quality of Dried Product

Proximate analysis of the dried samples were carried out to determine their crude protein, crude fat,

vitamin C, ash content, carbohydrate, crude fibre and moisture contents using the standard procedures (AOAC 1990). These together with colour evaluation were used as quality determining criteria in this study.

## 2.5. Experimental Analysis

Using the SPSS (16.0) computer software package, model equations were developed by employing essential regression analysis. The model equations were used to predict the drying rate and quality of dried product based on foaming agent, foam stabilizer and whipping time. Furthermore, the drying process was optimized for maximum drying rate and maximum post-drying quality considered in the study.

## 3. RESULTS AND DISCUSSION

### 3.1. Development of Model Equations

The model was based on a three-factor essential regression model and the factors were foaming agent, foam stabilizer and whipping time. From the regression analysis, the following regression models (Equations 2 – 9) were obtained for drying rate, vitamin C content, ash content, fat content, crude fibre content, carbohydrate content, protein content and moisture content respectively.

#### 3.1.1. Drying Process in Mechanical Dryer

The model developed for the drying process of tomato paste in the mechanical dryer is described by the equations below:

$$Y_{ddr} = 10.96 - 0.232\alpha - 5.619\beta + 0.01177\alpha^2 + 1.508\alpha\beta$$

$$; R^2 = 0.89 \quad (2)$$

$$Y_{dvc} = -0.453\gamma + 1.575\alpha\beta - 0.00232\alpha^2$$

$$; R^2 = 0.91 \quad (3)$$

$$Y_{das} = -0.04316\alpha\beta + 0.000861\alpha\gamma + 0.08476\beta\gamma - 2$$

$$; R^2 = 0.85 \quad (4)$$

$$Y_{dfa} = 1.637 - 0.113\gamma - 0.000976\alpha^2 - 0.06560\alpha\beta + 0.00651\alpha\gamma + 0$$

$$; R^2 = 0.81 \quad (5)$$

$$Y_{def} = 3.323 - 0.00547\gamma - 0.127\alpha\beta + 0.995\beta^2 = 0.00431\alpha^2\beta$$

$$; R^2 = 0.85 \quad (6)$$

$$Y_{dca} = 4.170 + 0.000591\alpha^2\gamma - 0.001\alpha^2\gamma + 0.000297\alpha\beta\gamma,$$

$$R^2 = 0.90 \quad (7)$$

$$Y_{dpr} = 0.547 - 0.03686\alpha - 0.08069\gamma + 0.00284\alpha\beta^2 + 0.02757\alpha\beta + 0$$

$$R^2 = 0.91 \quad (8)$$

$$Y_{dmc} = 87.83 - 0.00381\alpha^2 + 0.252\alpha\beta - 4.071\beta^2 + 0.000539\alpha^2\gamma - 0.0$$

$$, R^2 = 0.88 \quad (9)$$

where  $Y_{ddr}$ ,  $Y_{dvc}$ ,  $Y_{das}$ ,  $Y_{dfa}$ ,  $Y_{def}$ ,  $Y_{dca}$ ,  $Y_{dpr}$  and  $Y_{dmc}$  are drying rate, vitamin C content, ash content, fat content, crude fibre content, carbohydrate content, protein content and moisture content respectively and  $\alpha$ ,  $\beta$  and  $\gamma$  are foaming agent, foam stabilizer and whipping time respectively.

#### 3.1.2. Drying Process in Air Oven

The model developed for the drying process of tomato paste in the air oven is described by the equations below:

$$Y_{adr} = 7.693 + 24.22\beta - 1.159\alpha\beta + 0.07100\alpha\gamma - 23.53\beta^2$$

$$- 2.352\beta \quad 43.36 - 8.527\beta$$

$$; R = 0.80 \quad (10)$$

$$Y_{avc} = 39.17 + 1.112\alpha\beta + 0.03877\alpha\gamma - 14.64\beta^2 - 0.05078\alpha^2\beta$$

$$+ 1.124 \quad 1.252$$

$$; R = 0.90 \quad (11)$$

$$Y_{aas} = 1.235 + 0.0778 \alpha - 0.00204 \alpha \gamma - 0.00260 \alpha^2 \beta + 0.0$$

$$; R^2 = 0.89 \quad (12)$$

$$Y_{afa} = 1.524 - 0.106 \gamma + 0.01515 \gamma^2 - 0.0001 \alpha^2 \gamma + 0.146$$

$$; R^2 = 0.83 \quad (13)$$

$$Y_{acf} = 3.953 - 0.05952 \alpha - 0.07786 \gamma - 2.237 \beta^2 + 0.308$$

$$; R^2 = 0.89 \quad (14)$$

$$Y_{aca} = 4.302 + 1.593 \beta - 0.08763 \gamma - 0.353 \alpha \beta + 0.01306$$

$$; R^2 = 0.91 \quad (15)$$

$$Y_{apr} = 1.843 + 0.04380 \alpha + 3.393 \beta - 0.288 \alpha \beta - 4.824 \beta^2$$

$$; R^2 = 0.85 \quad (16)$$

$$Y_{amc} = 88.46 - 12.20 \beta + 0.852 \alpha \beta - 0.02793 \alpha \gamma + 10.68 \beta^2$$

$$; R^2 = 0.81 \quad (17)$$

where  $Y_{adr}$ ,  $Y_{avc}$ ,  $Y_{aas}$ ,  $Y_{afa}$ ,  $Y_{acf}$ ,  $Y_{aca}$ ,  $Y_{apr}$  and  $Y_{amc}$  are drying rate, vitamin C content, ash content, fat content, crude fibre content, carbohydrate content, protein content and moisture content respectively and  $\alpha$ ,  $\beta$  and  $\gamma$  are foaming agent, foam stabilizer and whipping time respectively.

### 3.1.3. Drying Process in Microwave Oven

The model developed for the drying process of tomato paste in the microwave oven is described by the equations below:

$$Y_{mdr} = 4.383 + 0.559 \alpha + 2.424 \alpha^2 - 0.01861 \alpha^2 - 0.155 \alpha$$

$$; R^2 = 0.85 \quad (18)$$

$$Y_{mvc} = 41.33 - 0.315 \alpha + 9.552 \beta + 0.122 \alpha \gamma - 3.292 \beta \gamma - 0.00849 \alpha \gamma$$

$$; R^2 = 0.90 \quad (19)$$

$$Y_{mas} = 1.248 + 0.01769 \alpha - 0.392 \beta - 0.000995 \alpha^2 + 0.718 \beta^2 + 0.0003$$

$$; R^2 = 0.91 \quad (20)$$

$$Y_{mfa} = 1.336 - 0.07280 \alpha + 0.155 \gamma + 0.00308 \alpha^2 + 0.109 \alpha \beta - 0.196 \beta$$

$$; R^2 = 0.88 \quad (21)$$

$$Y_{mcf} = 3.715 - 4.013 \beta + 0.00132 \alpha^2 - 0.01627 \alpha \beta + 3.715 \beta^2 + 0.768$$

$$; R^2 = 0.85 \quad (22)$$

$$Y_{mca} = 4.257 - 0.00335 \gamma^2 + 0.00220 \alpha^2 \beta - 0.113 \alpha \beta^2 + 0.00869 \beta \gamma^2$$

$$; R = 0.90 \quad (23)$$

$$Y_{mpr} = 2.206 + 0.000121 \alpha^2 + 0.00200 \gamma^2 + 0.00346 \alpha \beta \gamma - 0.000164 \alpha$$

$$; R^2 = 0.91 \quad (24)$$

$$Y_{mmc} = 86.51 + 8.292 \beta - 0.00201 \alpha^2 + 0.02645 \alpha \gamma - 9.302 \beta^2 - 1.364$$

$$R^2 = 0.88 \quad (25)$$

where  $Y_{mdr}$ ,  $Y_{mvc}$ ,  $Y_{mas}$ ,  $Y_{mfa}$ ,  $Y_{mcf}$ ,  $Y_{mca}$ ,  $Y_{mpr}$  and  $Y_{mmc}$  are drying rate, vitamin C content, ash content, fat content, crude fibre content, carbohydrate content, protein content and moisture content respectively and  $\alpha$ ,  $\beta$  and  $\gamma$  are foaming agent, foam stabilizer and whipping time respectively.

### 3.2. Model Validation

The measured (observed) outputs of drying rate and post-drying quality attributes of tomato paste were compared with the predicted values from the developed models. The two sets of data were plotted on the same graph to see how favourably compared the two of the curves are and the plots are shown in Figures 5, 6 and 7. From the figure, it is obvious that little or no difference could be noticed in the trend of the paired curves generated. This shows that all the predicted values compared favourably with the observed values. Therefore, the models reliably predicted the drying characteristics of tomato paste very well in terms of these output parameters.

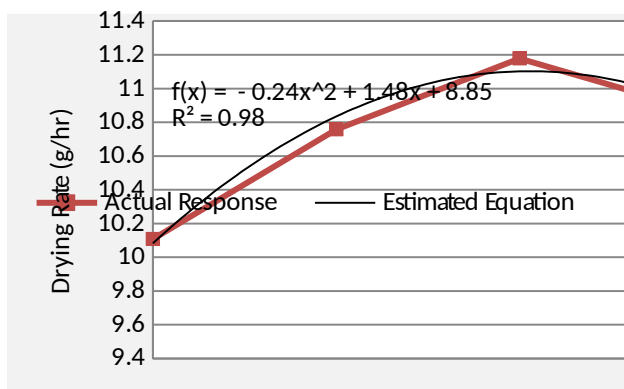


Figure 5: Effect of Foaming Agent on Drying Rate using Mechanical Dryer

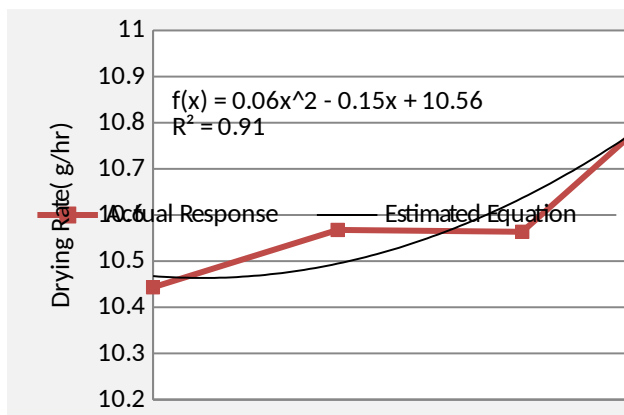


Figure 6: Effect of Foaming Agent on Drying Rate using Air Oven

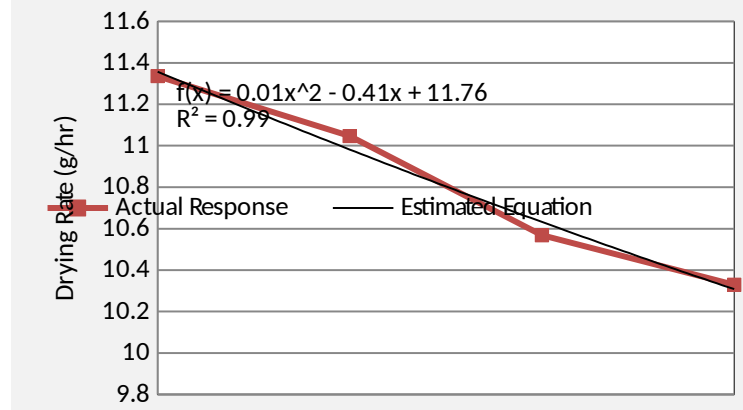


Figure 7: Effect of Foaming Agent on Drying Rate using Microwave Oven

### 3.3. Process Optimization

The optimum values of the process variables (foaming agent, foam stabilizer and whipping time) and optimized value of process outputs (drying rate and post-drying qualities) are presented in Table 2. The table shows that in order to maximize drying rate of tomato paste in the mechanical dryer, foaming agent, foam stabilizer and whipping time of 14.0 %EW, 0.48 %CMC and 9 min respectively must be combined to give the optimum drying rate of 11.36 g/h. For ash content, a combination 20 %EW foaming agent, 0.6 %CMC foam stabilizer and whipping time of 9 min resulted to the optimum value of 1.3 %.

An optimum value of 43.4 % was obtained for vitamin C at a foaming agent of 8.78 %EW, foam stabilizer of 0.6 %CMC and whipping time of 9 min. The crude fat content of tomato paste has 1.39 % as its optimum value and this corresponds to a combination of 5 %EW foaming agent, 0.15 %CMC foam stabilizer and 3 min whipping time. A foaming agent of 5 %EW, foam stabilizer of 0.15 %CMC and whipping time of 3 min should be selected to give the optimum crude fibre of 3.25 %.

Carbohydrate requires a foaming agent of 20 %EW, foam stabilizer of 0.6 %CMC and whipping time of 3 min to give the optimum value of 4.3 %. An optimum value of 2.5 % was obtained for protein at a foaming agent of 20 %EW, foam stabilizer of 0.15 %CMC and whipping time of 3 min. Ash content of dried tomato powder has 1.18 % as its minimum value and this

corresponds to a combination of 20 %EW foaming agent, 0.3 %CMC foam stabilizer and 3 min whipping time.

Table 2: Optimized Values of Process Parameters and

Process Outputs

Dryin g Equip ment	Output Para meter s	Fo a m St ab ili ze r pin ( % E W )	Fo a m St ab ili ze r pin ( % C M C )	Whi p pin g Ti me (m in)	Optim ized Value	Natur e of Soluti on
Mech anical Dryer	Dryin g Rate (g/h)	1 4. 0	0. 48 0.	9	11.36	Maxi mized Mini mized
	Vitam in C (%)	5 8. 7	15 0. 6	3	10.16	Maxi mized Mini mized
	Ash (%)	2 0 2	0. 6 0.	6	40.9	Maxi mized Mini mized
	Fat (%)	0 5 1	15 0. 3	3	1.39	Maxi mized
	Crude Fibre (%)	3. 2 5	0. 6 15	6.7	1.25	Mini mized Maxi mized
	Protei n (%)	4. 7 9	0. 47 9	9	3.0	Mini mized Maxi mized
	Carbo hydrat e (%)	2 0 2	0. 6 0.	3	4.3	Mini mized Mini mized
	Protei n (%)	0 9. 2	15 0. 15	3	2.5	Maxi mized Mini mized
		9	5.8 7	2.2		Mini mized

Dryin g Equip ment	Output Para meter s	Fo a m St ab ili ze r pin ( % E W )	Fo a m St ab ili ze r pin ( % C M C )	Whi p pin g Ti me (m in)	Optim ized Value	Natur e of Soluti on
Air Oven	Moist ure Conte nt (%)	2 0 2	0. 15 0.	9	8.8	Maxi mized Mini mized
	Dryin g Rate (g/h)	2 0 5	15 0. 6	6.7	11.57	Maxi mized Mini mized
	Vitam in C (%)	5. 6 5	0. 3 5	3	42.7	Maxi mized Mini mized
	Ash (%)	1 2 0	0. 6 0.	3	1.3	Maxi mized Mini mized
	Fat (%)	0 5 2	4 6 0.	3	1.39	Maxi mized Mini mized
	Crude Fibre (%)	0 5 1	4 2 0.	3	3.5	Maxi mized Mini mized
	Carbo hydrat e (%)	7 2 0	5 0 15	9	3.0	Maxi mized Mini mized
	Protei n (%)	2 0 2	0. 15 0.	3	2.4	Maxi mized Mini mized
	Moist ure Conte nt (%)	0 2 5	3 0. 4	9	2.2	Maxi mized Mini mized
	Dryin g Rate (g/h)	2 5 1	0. 15 0.	6	12.05	Maxi mized Mini mized
Vitam in C	3 2 0	6 0. 15	3	43.5	Maxi mized	

Drying Equipment	Output Parameters (%)	Foaming Agent (%)	Foam Stabilizer (%)	Whipping Time (min)	Optimized Value	Nature of Solution
		5	6	6.5	41	Minimized
Ash (%)	1	1	15	9	1.3	Maximized
	2	0	6	3	1.2	Minimized
Fat (%)	5	15	5.4	1.4		Maximized
	4	0				Minimized
	5	15	3	1.2		Maximized
Crude Fibre (%)	5	16	9	3.5		Maximized
	7	0	9	2.9		Minimized
Carbohydrate (%)	2	0	3	4.3		Maximized
	1	5	0			Minimized
	5	6	3	3.95		Maximized
Protein (%)	2	0	9	2.3		Maximized
	5	15	3	2.2		Minimized
Moisture Content (%)	1	2	0			Maximized
	9	6	6.4	8.8		Minimized
	5	15	3.2	8.7		Minimized

EW= Egg white, CMC= Carboxyl Methyl Cellulose, WT= Whipping Time

### 3.4. Practical Application

Preservation of tomato and quality retention could be best achieved if it is processed to a powdery form immediately after harvest. The process involves drying tomato paste to a powdery form with a shelf-stable moisture content. Since tomato is a food product, pretreatment methods, drying conditions and

other process parameters should be carefully controlled in order not to destroy or denature the sensory characteristics such as flavor, aroma and other quality attributes. The result of this research provides useful information for process monitoring, control and observation during the drying of tomato. The model can be used for optimizing the efficiency of commercial food drying process and the performance of commercial food dryers. An important aspect of the research is the microwave drying process which is a new technology for dehydrating fruits, vegetables and food without damaging their nutritive and sensory values. Therefore, a food drying process based on the result of this research would ensure a high process efficiency and appreciable retention of product quality.

### 4. CONCLUSION

The empirical models developed to relate the drying rate and post-drying qualities of tomato paste to foaming agent, foam stabilizer and whipping time were found to have described the drying characteristics of tomato very well and sufficiently predicted output parameters. Therefore these models can be applied for industrial drying of tomato under different process pretreatments and different drying conditions. Further analysis of the models by optimization showed that an optimum drying rate of 11.36 g/h could be achieved using a mechanical dryer if tomato paste is pretreated with foaming agent, foam stabilizer and whipping time of 14.0 %EW, 0.48 %CMC and 9 min respectively. The vitamins C content of the tomato powder could be maximized to an optimum value of 43.4 % if foaming agent, foam stabilizer and whipping time of 8.78 %EW, 0.6 %CMC and 9 min were used to pretreat tomato paste and then dried in the mechanical dryer. In the same development, tomato powder of water an optimum ash content of 1.3 % could be obtained with a foaming agent of 20 %EW, foam stabilizer of 0.6 %CMC and whipping time of 9 min. It could be inferred from this study that the drying process of tomato paste could be optimized under different pretreatments and drying conditions. Further studies should be carried out on the factors that affect the storability of tomato powder for long-time storage and preservation.

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## BIOGRAPHY OF THE AUTHORS

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been working on techniques, processes and equipment for processing agricultural and bioresources materials to food, fibre and industrial raw materials. Dr. Olaniyan's principal area of research is on Bioproduct Processing and Food Process Engineering, where he has carried out a number of projects and published a number of papers in local and international journals. He joined the service of the University of Ilorin in 1998 as an Assistant Lecturer in the Department of Agricultural and Biosystems Engineering and rose to the position of a Senior Lecturer in 2009. Currently, he is an Associate Professor at the Department of Agricultural and Bio-resources Engineering, Federal University Oye-Ekiti, Nigeria. Dr. Olaniyan has bagged several awards including the Award for the Best Paper (2007) in the Journal of Food Science and Technology, Mysore, India; Chinese Government Sponsorship (2008) for International Training Programme in Protected Agriculture at International Exchange Centre, Yangling, China; Netherlands Fellowship Programme (2009) for International Training Programme in Milk Processing at Practical Training Centre, Onkerk, the Netherlands; and Postdoctoral Fellowship (2011) of the Academy of Sciences of Developing Countries.

**2. Dr. Musliu Olushola Sunmonu** graduated with B.Eng, M.Eng and PhD in Agricultural Engineering from Federal University of Technology, Mina, Nigeria in 2000, 2006 and 2014 respectively. Since 2008, he has been working on postharvest processing and storage of agricultural products. Dr. Sunmonu's principal area of research is on postharvest storage and packaging, where he has carried out a number of projects and published a number of papers in local and international journals. He joined the service of the University of Ilorin in 2008 as an Assistant Lecturer in the Department of Agricultural and Biosystems Engineering and rose to the position of Lecturer I in 2013. Dr. Sunmonu has received some awards including the Netherlands Fellowship Programme (2010) for International Training in Milk Processing at Practical Training Centre, Onkerk, the Netherlands.



# LIFE CYCLE ASSESSMENT OF A NEW FEED PRODUCTION OBTAINED BY WASTED FLOUR FOOD COLLECTED FROM THE DISTRIBUTION AND RETAIL PHASES

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## ABSTRACT

Every year in Emilia Romagna more than 10000 tons of food are wasted during the retail and distribution phase. Landfill is nowadays the most adopted end of life, but working on a specific sorting system it would be possible to separate food from packaging and recover both of them in the most preferable way. In this case, Food Waste could be valorized by means of different technologies, such as anaerobic digestion, composting, and animal feed production. In this study the environmental performance of two Food Waste valorization scenarios using the Life Cycle Assessment (LCA) methodology has been analyzed. For both scenarios the Floor Food Products Fraction, like bread, pasta and biscuits, was valorized to produce animal feed. The environmental impacts of these new scenarios were compared with the impacts caused by the traditional feed production. The new scenarios lead to benefit for all the considered impact categories.

Keywords: Food waste, LCA, Waste valorization, Environmental impacts.

## 1. INTRODUCTION

Vermeulen et al., (2012) estimated that food production is responsible for up to 29% of the anthropogenic greenhouse gases emissions. One of the major problems related to food production is the waste generated throughout the Food Value Chain. In 2011, the Food and Agriculture Organization of the United Nations (FAO) determined that one third of the global production of food for human consumption is lost or passed out as waste, corresponding to approximately 1.3 billion tons each year (Gustavsson et al., 2011; FAO 2013a; FAO 2013b). Three main definitions of Food Waste (FW) can be found in literature. Firstly, the Food and Agriculture Organization (FAO) defines FW as wholesome edible material intended for human consumption, arising at any point in the food supply chain that is instead discarded, lost, degraded or consumed by pests (FAO 1981). Stuart (2009) adds to the FAO's definition, by stating that FW should also include edible material that is intentionally fed to animals or is a by-product of food processing diverted away from the human food chain. Finally, Smil (2004) suggests that FW covers the definitions above, but adds over-nutrition, the gap between the energy value of

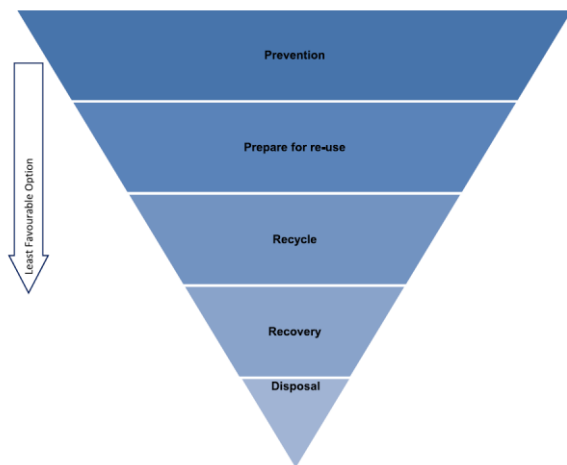
consumed food per capita and the energy value of food needed per capita.

Throughout the Food Value Chain, food is wasted in different stages, namely during production, food processing, distribution, and use. Several causes of this wastage exist (European Commission 2015), and they can be mainly attributed to:

- *Production*: overproduction, product and packaging damage from farmers and food processing;
- *Retail*: inefficient stock management, marketing strategies that lead to overbuying (e.g. 2 for 1, buy 1 get one for free), and aesthetic issues;
- *Catering*: the meal sizes and the difficulty to anticipate the number of clients;
- *Households*: lack of awareness towards food wastage, lack of shopping planning, misconception regarding the "best before" and "use by" date labels, and absence of knowledge/innovation to reuse and cook with the leftovers; and
- In general throughout the entire food value chain due to inappropriate storage and packaging methods.

The main reasons for food waste generated during distribution are examined in the literature, including inappropriate packaging, poor handling and transportation, failures on forecasting and storage (Kantor et al., 1997).

The management of FW should follow certain policies based on the 3R's concept, i.e., reduce, reuse, and recycle (Memon 2010; Sakai et al., 2011). In the European context, the Waste Framework Directive 2008/98/EC proposes the following waste management hierarchy: prevention, preparing for re-use, recycling, other recovery (e.g. energy recovery), and disposal (European Commission 2008). As illustrated in Figure 1, the best option is "prevention", and at the bottom of the inverted pyramid, the least favorable option is "disposal".



**Figure 1: The waste hierarchy. (Source: European parliament council 2008)**

Several studies evaluating the environmental performance of different FW management options can be found in literature. Lundie and Peters (2005) compared several FW disposal options, and showed that codisposal at landfill made a greater contribution than aerobic composting or central composting. Also Mendes et al., (2003) confirmed that the landfilling of biodegradable waste is generally the worst strategy from an environmental point of view. In addition, Güereca et al., (2006) reported that a composting based system had less impact on global warming than an incineration based system in the biowaste treatment of Barcelona, Spain. According to Khoo et al., (2009), using the FW for aerobic composting system is more environmentally sustainable than incinerators, but less compared to the anaerobic digestion process, due mainly to CO<sub>2</sub> and NH<sub>3</sub> emissions. Also in a study made in USA, anaerobic digestion had the best environmental results (Levis and Barlaz 2011). In the same study the authors showed that composting and landfill disposal with energy recovery presented lower environmental impacts than landfill disposal. Cherubini et al., (2009) showed that landfill systems is the worst alternative; the study also show that a sorting plant coupled with electricity and biogas production is the best option for waste management. Also from the study on municipal solid waste disposal by Eriksson et al. (2005) and Hong et al. (2006), landfill had more impact on global warming than incineration.

A few studies evaluated the use of FW to produce animal feed. According to Kim and Kim (2010) and Lee et al., (2007) using FW for feed production presented lower environmental impact than landfill disposal. Vandermeersch et al., (2014) showed that FW can be valorized through animal feed production, reducing the environmental impact mainly due to the avoided products from traditional supply chain.

The objective of this study was to evaluate which FW valorization scenarios could bring more environmental benefit. The environmental performance was evaluated by means of LCA method.

The remainder of the paper is organized as follows. LCA methodology has been explained considering all the phases according to ISO 14044. Environmental impact assessment has been performed considering the current scenario (disposal in landfill), comparing it with the new ones and evaluating the production impacts of existing feeds in comparison with those partly originated by FW. A sensitivity analysis has been then performed showing how changes could influence the results. Finally, conclusion section is reported in order to resume the main results of this work.

## 2. LIFE CYCLE ASSESSMENT METHODOLOGY AND DATA

The LCA methodology was applied according to the principles and requirements provided standards. LCA consists of four main steps of analysis: goal and scope definition; inventory analysis; impact assessment; and interpretation (ISO 14044, 2006). The SimaPro 8.0.5 LCA software was used in this assessment.

### 2.1. Goal and scope definition

The purpose of this work is to evaluate and compare the environmental impacts of 2 FW valorization scenarios by identifying the inputs and outputs for both option and their impacts.

#### 2.1.1. Functional unit

The functional unit provides a reference unit for which the inventory data are normalized (ISO 14040, 2006). The concept of the functional unit is key in LCA, as it facilitates the comparison of alternative products and services (ISO 14044, 2006). The functional unit of this study is 1 kg of animal feed product with 10% of FFPP. In scenario 1 we had compared the production of 1 kg corn feed with a new feed where 10% of corn was replaced with the same quantity of FFPP (Table 1). Instead in scenario 2 we compared the production of 1 kg animal feed (Table 1) with a new animal feed production where 10% of each ingredients was replaced with the same quantity of FFPP.

**Table 1: Recipe for 1 kg of animal feed in 2 scenarios**

Recipe for 1kg of animal feed					
Input scenario 1	Traditional animal feed	Animal feed with FFPP	Input scenario 2	Traditional animal feed	Animal feed with FFPP
Corn grains	100%	90%	Wheat grains	91.0%	81.9%
FFPP	0%	10%	Soybeans	5.0%	4.5%
			Tallow	4.0%	3.6%
			FFPP	0.0%	10.0%

We have chosen to replace only the 10%, in weight, with FFPP in the animal feed according to Prandini et al., (2007). In this study they have demonstrated that by replacing 10% of corn with the same amount of FFPP, no differences were shown for weight losses at different phases of seasoning and for the chemical analysis of seasoned hams.

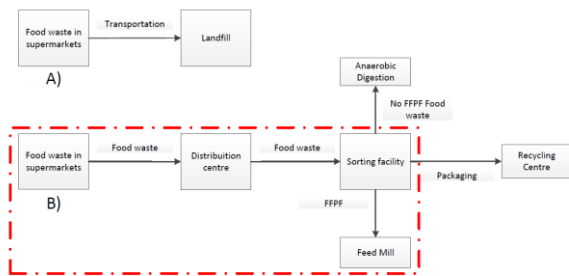
## 2.2. System boundaries and assumptions

In order to quantify the impact of the product analyzed, system boundaries need to be determined. The case study involves the food product wasted during the retail and distribution field, located in Emilia Romagna (northern Italy). The annual generation of FW was estimated at 14,600 tons in 2013. In this study were analyzed only the FFPF, corresponding to 19% of FW (Lipinski et al., 2013).

The generation of FW happens in several supermarkets spread out through Emilia Romagna. This FW is composed by food products that were no sold to consumers, i.e., products exceeded the expiration date stated on the package, deteriorated, or that no longer seemed to be edible (Lipinski et al., 2013, HLPE 2014; BCFN 2012). Nowadays this products are disposed of in landfills (Figure 2A and Table 1).

**Table 2: Summary of disposal n landfill**

Current Scenario		
Input	Unit	Quantity
Transport, freight, lorry 3.5-7.5 metric ton, EURO4	km	80
Landfill of biodegradable waste EU 27	ton	1



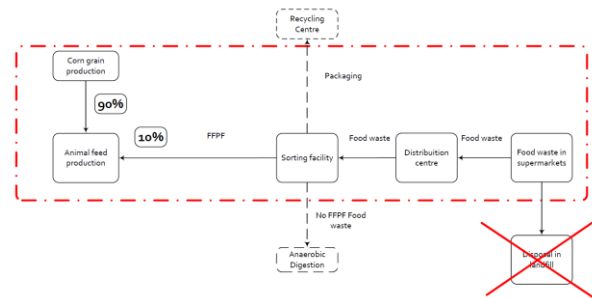
**Figure 2: Graphical scheme of the A) disposal in landfill and B) new valorization scenarios**

Instead in the new proposed scenarios the FW will be transported to a Distribution center and every 4 days. It will be transported from the distribution center to a centralized retort center, where a mechanical sorting will be performed, separating the organic fraction from packaging. The FFPF will be transported to a Feed mill where will be used to produce animal Feed. The other fraction of FW will be recovered in the most preferable rate such as through Anaerobic digestion (Vandermeersch et al., 2014). The sorted packaging materials will be then sent to specific recycling centres. The system boundaries can be visualized in Figure 2B, and included only the FFPF fraction.

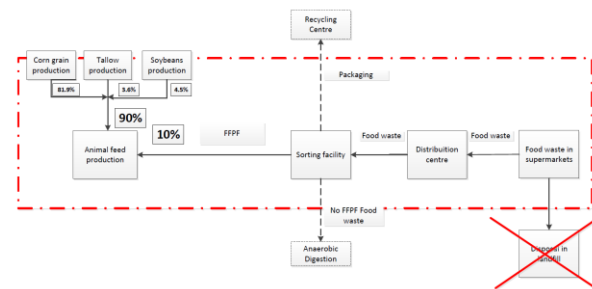
## 2.3. Inventory analysis and data collection

The life cycle inventory analysis quantifies the resources use, energy use, and environmental releases associated with the system being evaluated (ISO 14040 2006). Ecoinvent Database 3.1 is used for secondary data by considering data related to the Italian situation (Ecoinvent 2015).

In this paragraph we have defined the incoming and avoided phases regarding the new scenarios (Figure 3 and Figure 4). The considered incoming stages are: 1) Generation of FW; 2) Transportation of FW to distribution centre; 3) Storage in distribution centre; 4) Transportation of FW to Sorting facility; 5) Sorting of FW; 6) Transportation of FFPF to feed mill; 7) Feed production. While the avoided phases are: 8) Transportation of FW to the landfill, 9) Disposal product in landfill and 10) The produce of corn grain (Scenario 1) wheat grains, soybeans and tallow (Scenario 2), and 11) the relative transport.



**Figure 3: Graphical scheme of the scenario 1**



**Figure 4: Graphical scheme of the scenario 2**

Within the system boundary, the following assumptions and limitations are applied:

- The data from the transports was obtained based on the average distance traveled;
- Mass based allocation was used in order to account the share between FFPF, organic mass and packaging into the FW
- The average Italian electricity mix is used in this study.
- In this study, data from Italy were used when available. In case of unavailable data, relevant background data from Europe or Switzerland were used.
- Regarding the cut-off criteria applied, less than 1% of the energy and mass flows of the inputs and outputs was excluded from this analysis.

## 2.4. Methods of impact assessment

The data collected in the inventory analysis are the basis for the impact assessment phase whose aim is to evaluate the potential environmental impact of the system (ISO 14040 2006) caused by effluent emissions,

releases into the environment and resources consumption. The impact analysis was carried out using the ReCiPe method. The hierarchic perspective has been selected in this work, as it is considered to be the most balanced of the three proposed by the method (Egalitarian, Individualist and Hierarchist). Impact values were calculated at midpoint level for 18 impact categories, i.e. (i) climate change, (ii) ozone depletion, (iii) terrestrial acidification, (iv) freshwater eutrophication, (v) marine eutrophication, (vi) human toxicity, (vii) photochemical oxidant formation, (viii) particulate matter formation, (ix) terrestrial ecotoxicity, (x) freshwater ecotoxicity, (xi) marine ecotoxicity, (xii) Ionising radiation, (xiii) agricultural land occupation, (xiv) urban land occupation, (xv) natural land occupation, (xvi) water depletion, (xvii) metal depletion and (xviii) fossil depletion.

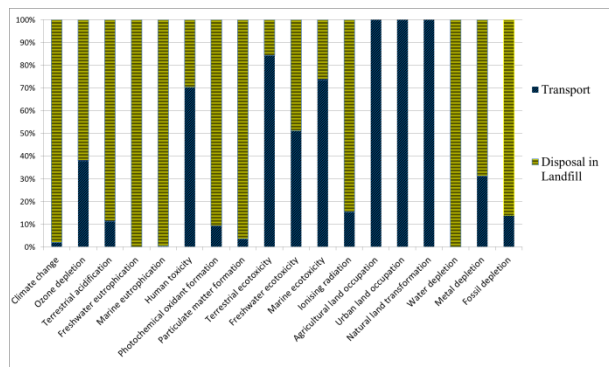
### 3. LIFE CYCLE IMPACT ASSESSMENT

#### 3.1. Current scenario: disposal in landfill

The results for the current scenario are reported in Table 2, while Figure 5 shows the relative contribution of each processing input.

**Table 3: Characterization results for the overall impact of the disposal in landfill scenario.**

Impact category	Unit	Total	Transport	Disposal in Landfill
Climate change	kg CO <sub>2</sub> eq	1.45E+06	2.89E+04	1.42E+06
Ozone depletion	kg CFC-11 eq	1.33E-02	5.05E-03	8.22E-03
Terrestrial acidification	kg SO <sub>2</sub> eq	9.71E+02	1.11E+02	8.60E+02
Freshwater eutrophication	kg P eq	1.27E+03	5.91E-01	1.27E+03
Marine eutrophication	kg N eq	1.18E+03	4.75E+00	1.18E+03
Human toxicity	kg 1,4-DB eq	6.00E+03	4.22E+03	1.78E+03
Photochemical oxidant formation	kg NMVOC	1.50E+03	1.38E+02	1.36E+03
Particulate matter formation	kg PM10 eq	1.36E+03	5.03E+01	1.31E+03
Terrestrial ecotoxicity	kg 1,4-DB eq	9.45E+00	7.97E+00	1.48E+00
Freshwater ecotoxicity	kg 1,4-DB eq	3.32E+01	1.70E+01	1.62E+01
Marine ecotoxicity	kg 1,4-DB eq	9.14E+01	6.73E+01	2.41E+01
Ionising radiation	kBq U235 eq	1.18E+04	1.82E+03	9.95E+03
Agricultural land occupation	m <sup>2a</sup>	4.32E+02	4.32E+02	0.00E+00
Urban land occupation	m <sup>2a</sup>	9.71E+02	9.71E+02	0.00E+00
Natural land transformation	m <sup>2</sup>	1.01E+01	1.01E+01	0.00E+00
Water depletion	m <sup>3</sup>	5.85E+02	-1.42E+01	6.00E+02
Metal depletion	kg Fe eq	5.34E+03	1.67E+03	3.67E+03
Fossil depletion	kg oil eq	7.21E+04	1.01E+04	6.20E+04



**Figure 5: Relative contribution of different phases to the overall impacts for disposal in landfill**

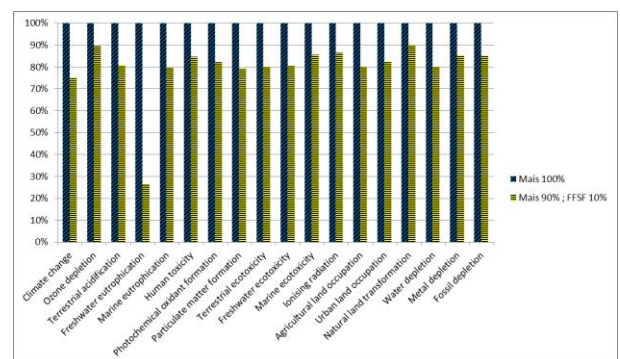
As can be seen from the total environmental impact divided according to stages (Table 1), disposal in landfill is the most important contributor to most impact categories; while, the transportation of FFPF to the landfill contributes mostly in 6 categories: Human toxicity, Terrestrial ecotoxicity, Marine ecotoxicity, Agricultural land occupation, Urban land occupation and Natural land transformation.

#### 3.2. Scenario 1

In this case we compared the production of 1 kg corn feed with a new animal feed production where 10% of corn was replaced with the same quantity of FFPF. The results for 18 midpoint indicators can be seen in Table 4 and Figure 6.

**Table 4: Characterization results for the scenario 1.**

Impact category	Unit	Mais 100%	Mais 90% ; FFPF 10%	Avoided impact
Climate change	kg CO <sub>2</sub> eq	5.86E-01	4.39E-01	1.47E-01
Ozone depletion	kg CFC-11 eq	3.51E-08	3.13E-08	3.73E-09
Terrestrial acidification	kg SO <sub>2</sub> eq	7.06E-03	5.70E-03	1.36E-03
Freshwater eutrophication	kg P eq	8.44E-05	2.24E-05	6.20E-05
Marine eutrophication	kg N eq	9.36E-03	7.45E-03	1.91E-03
Human toxicity	kg 1,4-DB eq	6.69E-02	5.66E-02	1.03E-02
Photochemical oxidant formation	kg NMVOC	1.87E-03	1.54E-03	3.35E-04
Particulate matter formation	kg PM10 eq	1.50E-03	1.19E-03	3.12E-04
Terrestrial ecotoxicity	kg 1,4-DB eq	3.68E-03	2.95E-03	7.31E-04
Freshwater ecotoxicity	kg 1,4-DB eq	2.97E-03	2.39E-03	5.80E-04
Marine ecotoxicity	kg 1,4-DB eq	8.53E-04	7.30E-04	1.23E-04
Ionising radiation	kBq U235 eq	1.36E-02	1.18E-02	1.85E-03
Agricultural land occupation	m <sup>2a</sup>	1.35E+00	1.08E+00	2.70E-01
Urban land occupation	m <sup>2a</sup>	2.45E-02	2.02E-02	4.31E-03
Natural land transformation	m <sup>2</sup>	6.53E-05	5.87E-05	6.60E-06
Water depletion	m <sup>3</sup>	1.71E-02	1.37E-02	3.41E-03
Metal depletion	kg Fe eq	2.69E-02	2.30E-02	3.98E-03
Fossil depletion	kg oil eq	9.89E-02	8.42E-02	1.47E-02



**Figure 6: Results of the LCA for the Scenario 1**

The new scenario showed better results in each impact categories analyzed; in almost all the categories the impact were resulted less than 10-20%, while for the Freshwater eutrophication the impact was resulted less than 73 %, mainly due to the avoided disposal in landfill of FFPF. For the other categories, the main reason for these results are the same; by valorizing the FFPF with feed production, avoids partly the production

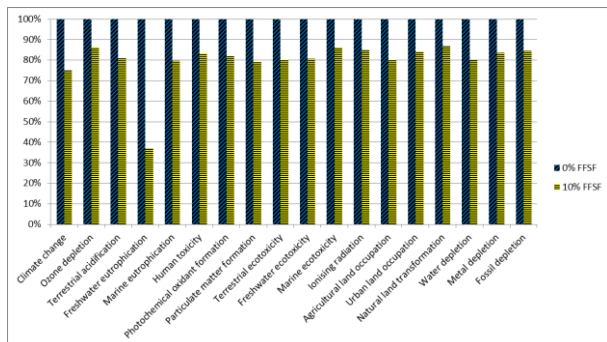
of animal feed from corn, avoiding the environmental impact that are directly related to agriculture activities, such as land occupation. The Relative contribution of each processing input for the new scenario can be seen in Appendix A.

### 3.3. Scenario 2

In scenario 2 we compared the production of 1 kg animal feed (Table 1) with a new animal feed where 10% of each ingredients was replaced with the same quantity of FFPF. The results for 18 midpoint indicators can be seen in Table 4 and Figure 7.

**Table 5: Characterization results for the scenario 2**

Impact category	Unit	0% FFPF	10% FFPF	Benefit
Climate change	kg CO <sub>2</sub> eq	6.04E-01	4.54E-01	1.51E-01
Ozone depletion	kg CFC-11 eq	5.30E-08	4.57E-08	7.32E-09
Terrestrial acidification	kg SO <sub>2</sub> eq	6.32E-03	5.11E-03	1.21E-03
Freshwater eutrophication	kg P eq	1.05E-04	3.87E-05	6.61E-05
Marine eutrophication	kg N eq	1.11E-02	8.88E-03	2.27E-03
Human toxicity	kg 1,4-DB eq	9.31E-02	7.75E-02	1.55E-02
Photochemical oxidant formation	kg NMVOC	2.23E-03	1.83E-03	4.07E-04
Particulate matter formation	kg PM10 eq	1.48E-03	1.17E-03	3.07E-04
Terrestrial ecotoxicity	kg 1,4-DB eq	4.96E-03	3.97E-03	9.87E-04
Freshwater ecotoxicity	kg 1,4-DB eq	2.38E-03	1.91E-03	4.63E-04
Marine ecotoxicity	kg 1,4-DB eq	7.86E-04	6.77E-04	1.10E-04
Ionising radiation	kBq U235 eq	1.81E-02	1.54E-02	2.75E-03
Agricultural land occupation	m <sup>2a</sup>	1.40E+00	1.12E+00	2.79E-01
Urban land occupation	m <sup>2a</sup>	1.40E-02	1.18E-02	2.22E-03
Natural land transformation	m <sup>2</sup>	9.47E-05	8.22E-05	1.25E-05
Water depletion	m <sup>3</sup>	3.26E-02	2.61E-02	6.51E-03
Metal depletion	kg Fe eq	4.01E-02	3.35E-02	6.62E-03
Fossil depletion	kg oil eq	1.11E-01	9.35E-02	1.70E-02

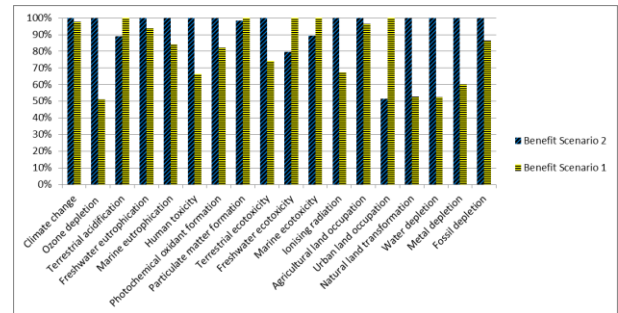


**Figure 7: Results of the LCA for the Scenario 2**

Also this scenario showed better results in each impact categories analyzed; in almost all the categories the impact were resulted less than 14-25%, while for the Freshwater eutrophication the impact was resulted less than 63%. The reasons are the same as already discussed in the previous sections. The Relative contribution of each processing input for the new scenario can be seen in Appendix B.

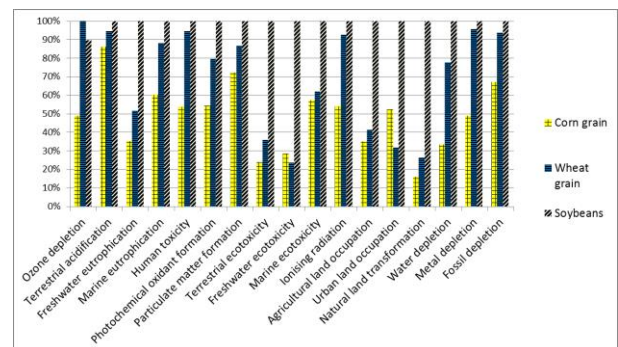
### 3.4. Comparison between Scenario 1 and scenario 2

In this paragraph we compared the environmental benefits (avoided impacts thanks to using FFPF in the animal feed) between scenario 1 and 2 (Figure 8).



**Figure 8: Comparison results between 2 scenarios**

Scenario 2 showed better results in 13 categories (Climate change, Ozone depletion, Freshwater eutrophication, Marine eutrophication, Human toxicity, Photochemical oxidant formation, Particulate matter formation, Terrestrial ecotoxicity, Ionising radiation and Natural land transformation), while scenario 1 had better results in 5 categories (Terrestrial acidification, Freshwater ecotoxicity, Marine ecotoxicity, Agricultural land occupation and Urban land occupation). Summarizing scenario 2 can bring more environmental gains than scenario 1 because the FFPF is valorized for producing Animal feed containing soybeans and wheat grain that they have most environmental impactful than producing corn grain (Figure 9).



**Figure 9: Total impact of vegetable productions**

**Table 6: Comparison results between 2 scenarios**

Impact category	Unit	Benefit Scenario 2	Benefit Scenario 1
Climate change	kg CO <sub>2</sub> eq	1.51E-01	1.47E-01
Ozone depletion	kg CFC-11 eq	7.32E-09	3.73E-09
Terrestrial acidification	kg SO <sub>2</sub> eq	1.21E-03	1.36E-03
Freshwater eutrophication	kg P eq	6.61E-05	6.20E-05
Marine eutrophication	kg N eq	2.27E-03	1.91E-03
Human toxicity	kg 1,4-DB eq	1.55E-02	1.03E-02
Photochemical oxidant formation	kg NMVOC	4.07E-04	3.35E-04
Particulate matter formation	kg PM10 eq	3.07E-04	3.12E-04
Terrestrial ecotoxicity	kg 1,4-DB eq	9.87E-04	7.31E-04
Freshwater ecotoxicity	kg 1,4-DB eq	4.63E-04	5.80E-04
Marine ecotoxicity	kg 1,4-DB eq	1.10E-04	1.23E-04
Ionising radiation	kBq U235 eq	2.75E-03	1.85E-03
Agricultural land occupation	m <sup>2a</sup>	2.79E-01	2.70E-01
Urban land occupation	m <sup>2a</sup>	2.22E-03	4.31E-03
Natural land transformation	m <sup>2</sup>	1.25E-05	6.60E-06
Water depletion	m <sup>3</sup>	6.51E-03	3.41E-03
Metal depletion	kg Fe eq	6.62E-03	3.98E-03
Fossil depletion	kg oil eq	1.70E-02	1.47E-02

Table 6 shows the comparison of the avoided environmental impact of the two scenarios; considering the whole use of the FFPF (2774 ton) to produce Animal feed (2774,000 ton).

#### 4. SENSITIVITY ANALYSIS

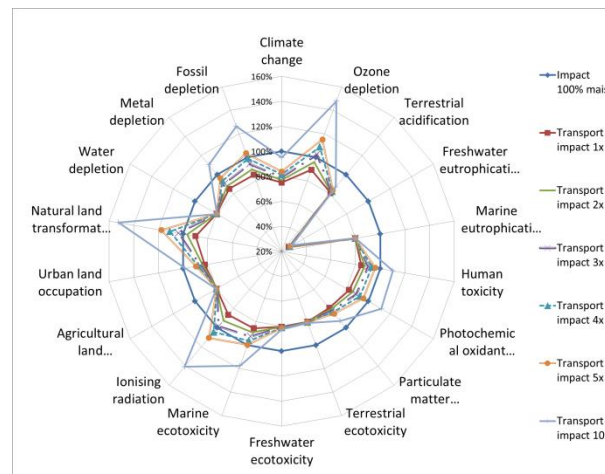
Sensitivity analysis is a systematic procedure for estimating the effects of the choices made regarding methods and data on the outcome of a study (ISO 14040, 2006). A sensitivity assessment was performed to evaluate the environmental performance when the transport phase (Transportation of FW to distribution center, transportation of FW to Sorting facility; and transportation of FFPS to feed mill) varies. We assume that no variations occur in the other processing phases when changing the transport phase. This sensitivity analysis considers different distances; in particular they were increased by 2-3-4-5-10 times to see the influence on the whole process.

**Table 7: Summary of transport phase**

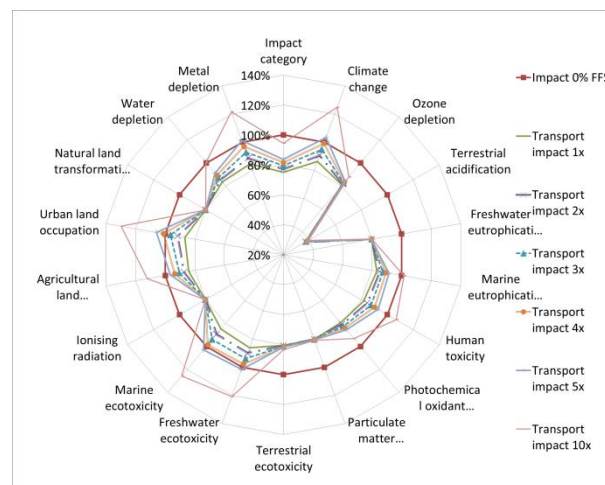
Transport Phase			
Phase	Means of transport	Unit	Quantity
Transportation of FW to distribution centre	Lorry 3.5-7.5 metric ton, EURO4	km	250
Transportation of FW to Sorting facility	Lorry 16-32 metric ton, EURO4	km	80
Transportation of FFSF to feed mill	Lorry 3.5-7.5 metric ton, EURO4	km	40

The results of the sensitivity analysis show that the variation in the transport phase (Table 7) has a limited influence on the environmental impacts (Figure 10 and Figure 11). In fact, increasing by 3 times the transport

phase in scenario 1, only in 2 categories (Ozone depletion and Ionising radiation ) the impact of the new scenario is higher the impact of the scenario where animal feed is produced only with corn. Instead in scenario 2 solely increasing by 4 times the transport phase, the new scenario showed a higher impact in one unique category (Natural land occupation) compared to the original scenario (Table 4).



**Figure 10: Graphical results of the sensitivity analysis for the variation in transport phase, scenario 1**



**Figure 11: Graphical results of the sensitivity analysis for the variation in transport phase, scenario 2**

#### 5. CONCLUSIONS

The purpose of this work is to evaluate and compare the environmental impacts of 2 FW valorization scenarios by identifying the inputs and outputs for both option and their impacts. Both scenarios ensure a reduction of the environmental impact in all categories considered as compared to the current scenario (landfill). The LCA study showed that scenario 2 can bring more environmental gains than scenario 1 (in particular in 13 categories such as Climate change, Ozone depletion, Freshwater eutrophication and Terrestrial ecotoxicity). The results showed that about 4E+6 kg CO<sub>2</sub>-eq. of greenhouse gases would be avoided valorizing the FFPF through both scenarios. This study proves clearly that the method

of manufacturing feed from FW for animals can be an environment friendlier method than landfilling.

Further studies could investigate the environmental performance of other FW treatment scenarios, for instance, separating the meat and fish fractions of FW for pet food production.

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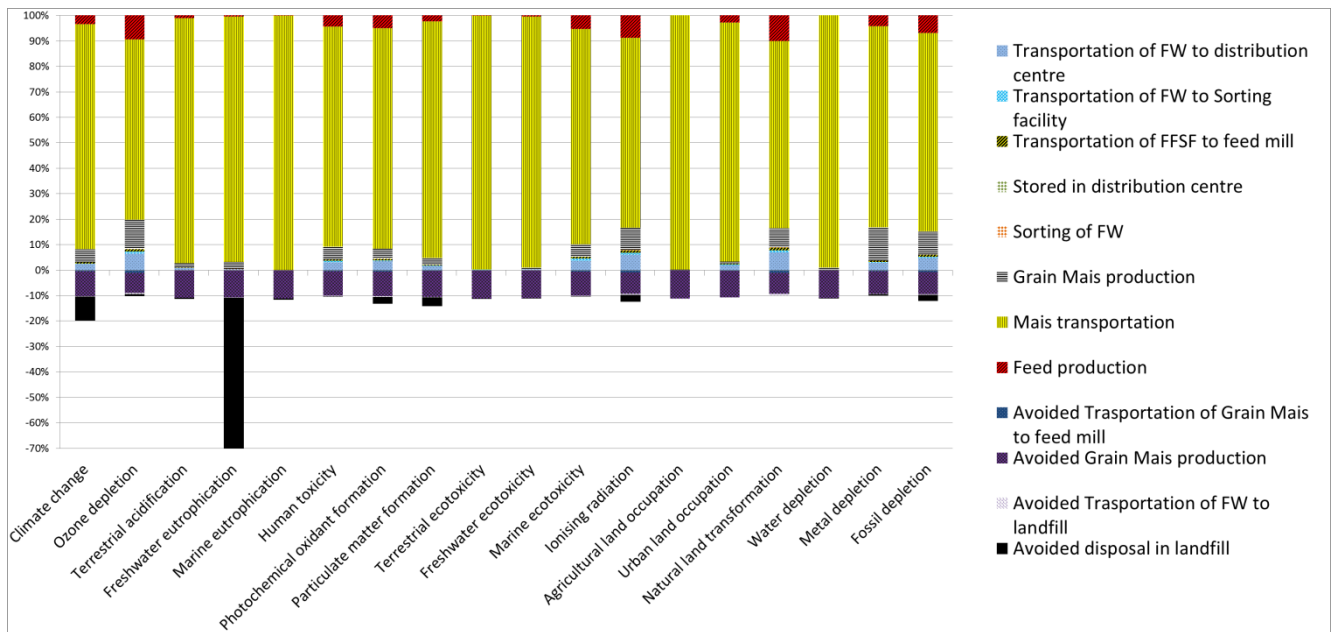
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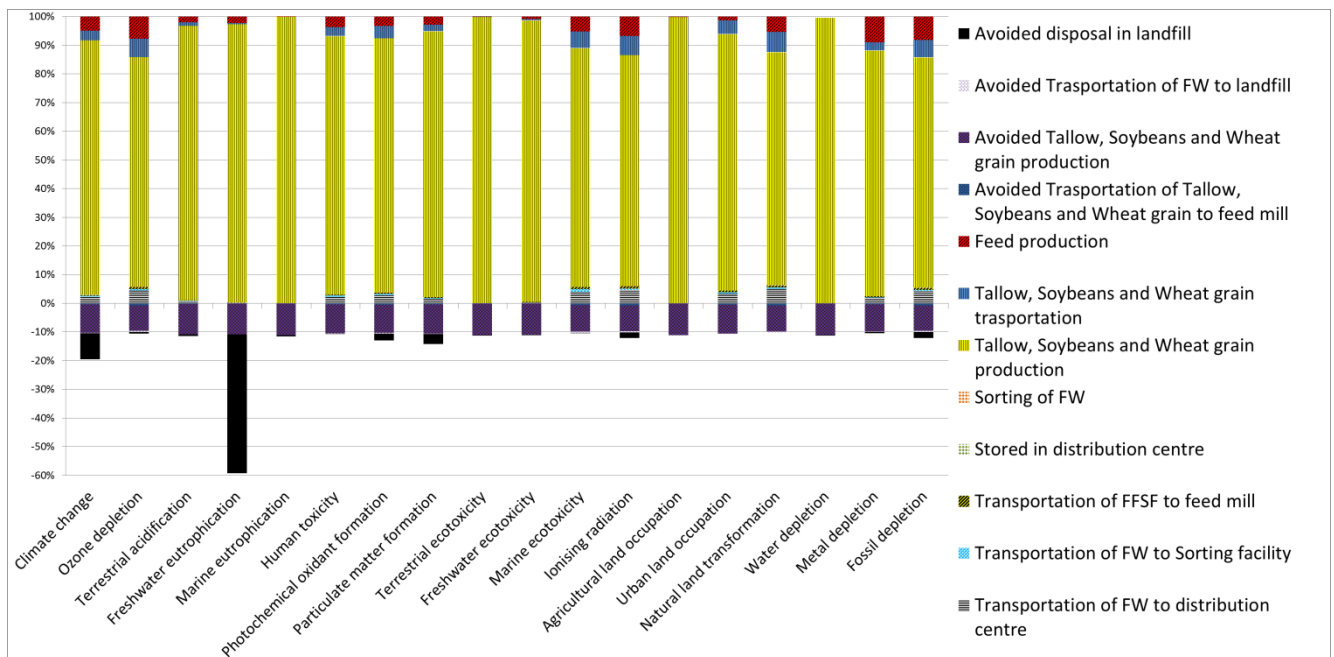
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## APPENDIX



Appendix A: Relative contribution of different phases to the overall impacts for scenario 1



Appendix B: Relative contribution of different phases to the overall impacts for scenario 2

# DEVELOPMENT OF A SMALL SCALE MACHINE FOR ROASTING SHEA KERNEL PRIOR TO SHEA BUTTER EXTRACTION

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## ABSTRACT

Shea kernel roasting prior to shea butter extraction is done to prepare the kernel in a good condition for the extraction of the fat (shea butter). Besides, roasting reduces the viscosity of oil to be extracted, ruptures the oil cells, adjusts the moisture content of the kernel to the optimum level for oil extraction and destroys or deactivates enzymes which inhibit the release of oil from the kernel. All these facilitate the extraction process. A small scale machine for roasting shea kernels prior to shea butter extraction was designed, constructed and tested. The machine consists of a hopper, top cover, roasting chamber, chamber support, an auger or screw conveyor, temperature-controlled heating device, a discharge outlet, a 2 hp electric motor and a speed-reduction gear box. The machine derives its power from the electric motor through a gear reduction box to the shaft of the screw conveyor. Shea kernel is fed through the hopper into the roasting chamber and then conveyed to the discharge end after being roasted. Heat is supplied to the kernel inside the roasting chamber from two 750 W ring-type heaters placed at both ends of the chamber. The roasting temperature is regulated by a thermostat connected to the electric heaters. The machine has a capacity of 20 kg/h and the production cost was USD100. Test results revealed weight losses of 12.1, 27.6, 31.8 and 32.8 % in the kernels at roasting temperatures of 50, 70, 90 and 110 °C respectively. The kernels were of uniform deep-brown colour with clear film of oil on the surface after roasting indicating that they experienced uniform roasting.

Keywords: development, roasting machine, roasting temperature, shea butter, shea kernel

## 1. INTRODUCTION

Oilseed materials with high protein content are usually roasted before pressing in order to coagulate the protein and free the oil for extraction. Shea kernel (SK) roasting prior to shea butter (SB) extraction is a common practice among SB processors in the

Nigerian rural communities because it facilitates oil extraction process. Besides, roasting reduces the viscosity of oil to be extracted, ruptures the oil cells, adjusts the moisture content of the kernel to the optimum level for oil extraction and destroys or deactivates enzymes which inhibit the release of oil from the kernel. SB is the fat content of SK which is used in food, soap, cosmetics, pharmaceutical, medical and engineering industries for the production of cooking oils, toilet soaps, pomade, drugs, ointments and metal cutting fluids respectively. The residual cake from which the oil is extracted – shea cake (SC) – can be an excellent ingredient for livestock feed production and a substitute for cocoa butter in the chocolate and confectionary industries. These products, in Nigeria, have high potential for export and as foreign exchange earners.

Olopade and Akinoso (2004) designed and fabricated 2 kg capacity roaster for roasting cassia sieberiana seeds. Visual observation while testing the roaster revealed that physical changes on the seeds were time dependent; 200 °C roasting temperature and 15 min roasting time gave the best samples of cassia sieberiana during the testing. Mahama, Mburidiba, Mensah, and Seidu (2004) developed a machine for roasting crushed SK in Northern Ghana. The roaster consists of an enclosed cylindrical drum with a lid welded to the opening to prevent the material from falling out during operation. Inside the drum are agitators which transfer the material into different positions in order to prevent burning, mixes the material in the drum for uniform roasting and sweeps the roasted materials properly in the drum in order to receive fresh one. However, the machine is very slow in operation and operates with much heat loss and too much heat to body contact with possibility of fire disaster during roasting (Mahama, Mburidiba, Mensah, and Seidu 2004).

Akinoso, Asiru, and Awoliyi (2004) developed a 300 kg/h manual cashew nut roasting machine which uses charcoal as a source of heat energy. Performance evaluation showed that the roaster could roast 15 kg of raw cashew nut having a moisture content of 12 %

wet basis for 3 minutes using 130 °C as a roasting temperature. Davids (2003) stated that roasting machines basically operate at temperatures between 240-275 °C for a period of time ranging from 3 to 30 minutes. Raemy and Lambelet (1982) explained that coffee roasting process is initially endothermic (absorbing heat) but becomes exothermic (giving off heat) at a temperature of about 175 °C. It implies that, at this temperature, the beans are heating themselves; thus, roaster design engineers should know that roaster's heat source should be adjustable and controllable. Basically, there are two common types of roasting machines which include drum roasters and hot-air roasters (Anonymous, 2012). Others that are less common include packed bed roasters, tangential roasters and centrifugal roasters (Anonymous, 2012). In hot air coffee roasters, heated air is forced through a perforated plate under the coffee beans with a force sufficient enough to lift the beans while tumbling and circulation of the beans within the fluidized bed aid the process of heat transfer to the beans (Anonymous, 2012).

In the traditional method of SB extraction, SK is roasted over fire for a certain period of time using some local devices. One of such devices is made of a perforated clay pot inside which the kernels are placed and then heated from the bottom using firewood. Another one consists of an empty drum with a perforated bottom while the kernels are fired underneath using firewood. This traditional method is crude, tedious, labour intensive, has no provision for ergonomic factors and also exposes the operators to excessive smoking and heat radiation. SB extracted from SK treated using this method has been found to be poor quality. A SB processor in the rural communities needs a SK roasting device that can easily be operated, used and maintained. Therefore, the objectives of this work are to design, construct and test a simple small scale machine for roasting SK prior to SB extraction in the rural communities. Such a machine would remove the hazards involved in the traditional roasting of SK and eventually improve the efficiency of SB extraction.

## 2. THE MACHINE

### 3.1. Description and Working Principles

The roaster consists of a hopper, top cover, roasting chamber, chamber support, an auger or screw conveyor, temperature-controlled heating device, a discharge outlet, a 2 hp electric motor and a speed-reduction gear box. The machine derives its power from the electric motor through a gear reduction box to the shaft of the screw conveyor. SK is fed through the hopper into the roasting chamber and then conveyed to the discharged end after being roasted. Heat is supplied to the kernel inside the roasting

chamber from two 750 W ring-type heaters placed at both ends of the chamber. The roasting temperature is regulated by a thermostat connected to the electric heaters. The isometric view of the machine is shown in Figure 1.

### 3.2. Design Considerations

Design considerations included machine capacity of 20 kg/h; roasting chamber adequately insulated to prevent heat loss during roasting operation; a speed-reduction mechanism to step down the motor to the desired shaft speed; temperature-control device to regulate the roasting temperature; a screw auger to serve the dual purposes of conveying and stirring the kernels in the roasting chamber thereby preventing burning of the kernels; roasting chamber designed to accommodate the required quantity of SK; roasting chamber designed to ensure maximum contact of individual kernel with the heat output in order to increase efficiency and reduce heating time; and a strong main frame to serve as structural support for the machine.

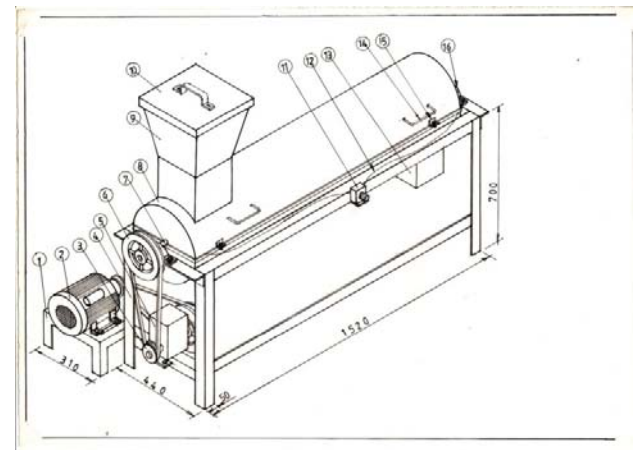


Figure 1: Isometric View of the Roasting Machine: 1-Electric motor stand, 2-Electric motor, 3-Speed reduction pulley, 4-Machine stand, 5-Speed reduction box, Machine pulley, 7-Conveyor shaft, 8-Roasting drum, 9-Feeding hopper, 10-Hopper lid, 11-Thermostat, 12-Thermocouple wire, 13-Kernel outlet, 14-Drum handle, 15-Drum flange, 16-Bearing housing

### 2.3. Design Theorems

#### 2.3.1 Transmission Shaft

The main shaft that makes the screw conveyor in the roasting chamber is subjected to bending moment and shear stress during rotation. The permissible diameter of the shaft, according to American Society of Mechanical Engineers (ASME) Code was determined as:

$$d^3 = \frac{16}{\delta_o \pi} \sqrt{M^2 + T^2} \quad (1)$$

where  $d$  is the permissible shaft diameter in m,  $M$  is the bending moment in Nm,  $T$  is the maximum torque in Nm,  $\delta_o$  is the yield stress in N/m<sup>2</sup> and  $\pi$  is a constant. With  $M = 13.13$  Nm,  $T = 60$  Nm, and  $\delta_o = 72 \times 10^6$  N/m<sup>2</sup>, hence  $d = 15.11$ mm. Therefore, a solid mild steel rod of diameter of 20 mm was used for the transmission shaft.

### 2.3.2. Auger Capacity

The theoretical capacity of the auger was calculated from the equation developed Onwualu, Akubuo, and Ahneku (2006) as:

$$Q = 60 \times \frac{\pi}{4} (D^2 - d^2) p N \phi \quad (2)$$

where  $Q$  is the theoretical capacity in m<sup>3</sup>/h,  $D$  is the screw diameter in m,  $d$  is the shaft diameter in m,  $p$  is the screw pitch in m,  $N$  is the shaft speed in rpm and  $\phi$  is the filling factor. Substituting  $D = 30$  cm,  $d = 20$  mm,  $p = 30$  cm,  $N = 30$  rpm and  $\phi = 0.8$ , hence,  $Q = 30.401$  m<sup>3</sup>/h.

### 2.3.3. Power Requirement

The power required to rotate the auger was determined using an adapted form of Onwualu, Akubuo, and Ahneku (2006)'s equation as:

$$P = \frac{QL\rho gF}{3600} \quad (3)$$

where  $P$  is the power required in W,  $L$  is the length of the conveyor in m,  $\rho$  is the density of shea kernel in kg/m<sup>3</sup>,  $g$  is the acceleration due to gravity in m/s<sup>2</sup> and  $F$  is the material factor. Substituting  $Q = 30.401$  m<sup>3</sup>/h,  $L = 150$  cm,  $\rho = 754.18$  kg/m<sup>3</sup>,  $g = 9.81$  m/s<sup>2</sup> and  $F = 0.4$  into Equation 2.3 and taking into consideration the speed-reduction ratio of 15:1, hence,  $P = 562.298$  W. To give allowance to power used in driving the pulley and other losses, the rated power was 570 W. The power of the electric motor to drive the system was estimated as:

$$P_m = \frac{P}{\eta} \quad (4)$$

where  $P_m$  is the power of electric motor in W and  $\eta$  is the efficiency of the motor in decimal. Given that  $\eta = 0.8$ , therefore,  $P_m = 712.5$  W or 0.96 hp. The machine can be driven by a 1 hp electric motor. However, a 2 hp single-phase electric motor was selected.

### 2.3.4. Operating Capacity

The operating capacity of the machine was calculated from the energy equations used by Akinoso, Asiru, and Awoliyi (2004) and Olopade and Akinoso (2004) as:

$$Q_o = mc\theta \quad (5)$$

and

$$t = \frac{Q_o}{P_h} \quad (6)$$

where  $Q_c$  is the quantity of heat required in kJ,  $m$  is the mass of shea kernel in kg,  $c$  is the specific heat capacity of shea kernel in kJ/kg °C,  $\theta$  is the change in temperature in °C,  $t$  is the roasting time in s and  $P_h$  is the rated power of the heaters in kW. With  $m = 7$  kg,  $c = 1.507$  kJ/kg °C,  $\theta = 120$  °C and  $P_h = 1.5$  kW substituted into Equation 2.4, hence,  $Q_c = 1265.88$  kJ and  $t = 14.07$  min. Therefore, the machine can roast 7 kg of SK in about 15 min.

## 3. MATERIALS SELECTION AND CONSTRUCTION

The feeding hopper was fabricated from a standard length of 1.5 mm thick metal sheet. Four pieces of dimension 300 x 450 mm were cut from the metal sheet to make the four sides of the hopper. A 50 mm length was cut from both sides of the width to a height 200 mm to form a base of dimension 200 x 200 mm for the hopper. The four pieces of metal sheet were welded together to form the required shape of the feeding hopper. The roasting chamber is essentially a double wall cylindrical container made from mild steel sheet of 1.5 mm and 1.2 mm for the outer and inner walls respectively. The outer cylinder was of dimension 1000 x 1500 mm while the inner one was 950 x 1450 mm and the clearance was filled with insulation materials. The metal sheet was cut by a treadle-operated guillotine while a rolling machine was used to fold the sheet into a cylindrical shape.

The main frame was fabricated using a 40 x 40 mm angle iron. The procedures involved measurement with steel tape, marking out with scribe and cutting using power hacksaw. After cutting each component to the required dimensions, they were welded together to serve as support and stand for the roasting chamber and prime mover thereby enhancing the stability of the whole machine. The screw conveyor (auger) was made from a mild steel shaft (Φ 20 mm) on which a 1.2 mm metal sheet was welded spirally to form a screw. The turning and facing of the shaft

were carried out on the lathe. The specification of construction materials is shown in Table 1. Fabrication was carried out at the Central Engineering Workshop, University of Ilorin. Figures 2 and 3 show the exploded and orthographic front views of the roasting machine respectively.

Table 1: Materials of Construction of the Shea Kernel Roasting Machine and their Specifications

Material of Construction	Specifications	Quantity
Metal sheet	1.5 mm thickness, standard size	1
Metal sheet	1.2 mm thickness, standard size	1
Mild steel rod	$\Phi$ 25 mm, $\frac{1}{2}$ standard size	1
Angle iron	40 X 40 mm, standard length	3
Electric heater	750 W	2
TDC*	0-150 °C	1
Electric wire	3 m length	1
Welding electrode	Gauge 12	18
Electric plug	15 A	2
Bolts and nuts	M12	10
Bolts and nuts	M13	10
V - belt	A 42, 12.5 X 1125 mm	1
V - belt	A 35, 12.5 X 925 mm	1
Insulating materials	25 kg	1

\*TDC: Temperature Control Device

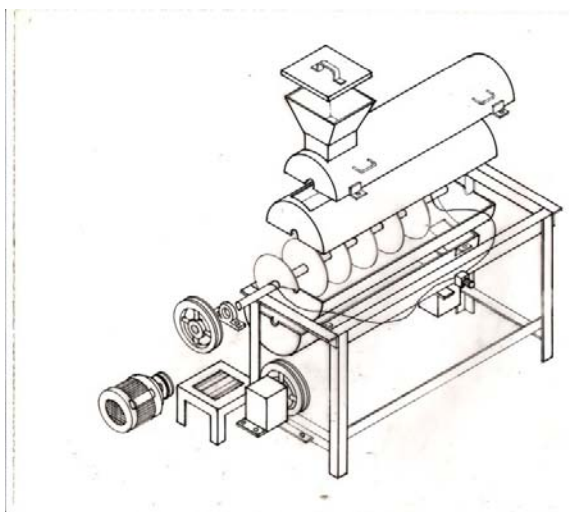


Figure 2: Exploded View of the Roasting Machine

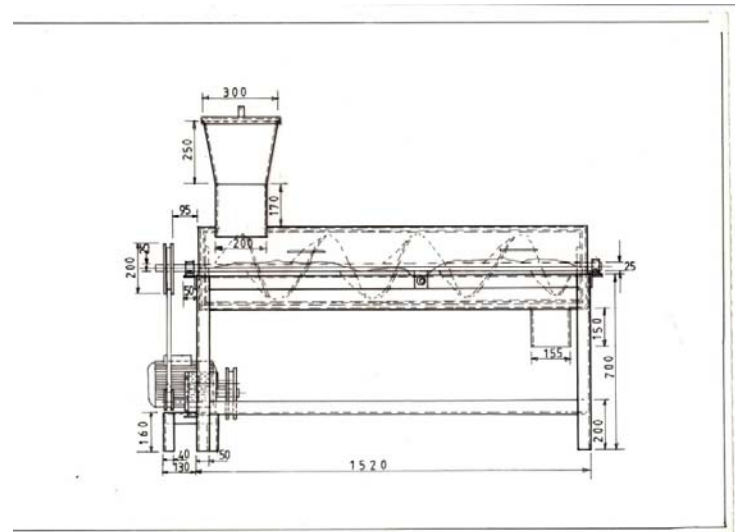


Figure 3: Front View of the Roasting Machine

#### 4. MATERIALS AND METHODS USED FOR PERFORMANCE EVALUATION

In order to test for the performance of the machine, 2 bags (100 kg) of shea nuts (SN) were obtained from a produce merchant in Ilorin. The nuts were cracked, winnowed, dried, packaged in moisture-proof bags and stored in the laboratory at room conditions. The moisture content of the kernels was determined according to Adeeko and Ajibola (1989). After the assembly of the machine, the heater was switched on and the roasting chamber was allowed to attain a desired temperature (50 °C) using the thermostat. The machine was then set into operation and a known quantity of SK was introduced through the hopper. The rotary motion of the auger (screw conveyor) provided stirring and tumbling of the kernels in the roasting chamber. In this way, burning was prevented while even roasting of the kernels was ensured. The kernels were discharged after 20 min of roasting and reweighed. Weight loss,  $W_{LS}$  in %, was then determined as follows:

$$W_{LS} = \frac{100(W_{BR} - W_{AR})}{W_{BR}} \% \quad (7)$$

where  $W_{BR}$  and  $W_{AR}$  are weight before roasting and weight after roasting respectively in g. The same procedure was repeated for 70, 90 and 110 °C with each test being replicated thrice. Samples of SK roasted by the machine were processed by two screw

press expellers and a model cage press for SB extraction. Oil yield, extraction efficiency and process loss were determined from standard formulae (Adeeko and Ajibola 1989; Olaniyan and Oje 2007) as follows:

$$O_Y = \frac{100W_{OE}}{W_{US}} \% \quad (8)$$

$$E_E = \frac{100W_{OE}}{xW_{US}} \% \quad (9)$$

$$P_L = \frac{100\{W_{US} - (W_{OE} + W_{RC})\}}{W_{US}} \% \quad (10)$$

where  $O_Y$ ,  $E_E$  and  $P_L$  are oil yield, extraction efficiency and process loss respectively in %;  $W_{OE}$ ,  $W_{RC}$  and  $W_{US}$  are weights of oil extracted, residual cake and shea kernel sample respectively in g and  $x$  is the oil content of shea kernel in decimal. Each trial was carried out in triplicates.

## 5. RESULTS AND DISCUSSION OF PERFORMANCE EVALUATION

The results of weight loss in kernel after roasting and that of the extraction experiments are shown in Tables 2 and 3 respectively. The average percentage weight loss increased from 12.1 % to 32.8 % when the roasting temperature was increased from 50 °C to 110 °C. From visual observation, it was revealed that the kernels were of uniform deep-brown colour with clear film of oil on the surface after roasting indicating that the kernels experienced uniform roasting. During roasting operation, the flavour and aroma peculiar with roasting of SK were perceived. As shown in Table 3, the roasted SK gave appreciable yield of SB and high extraction efficiency during extraction experiments with a model ram press and the screw expellers. Process losses were very small using the three oil extraction equipment. These results showed that the SK roaster performed efficiently and effectively with a high degree of satisfaction.

Table 2: Results of Investigation of Weight Loss of the Roasted Shea Kernels\*

Roasting Temperature (°C)	Weight Loss (%)
50	12.1
70	27.6
90	31.8
110	32.8

\* Each value is the mean of triplicates ± standard deviation

Table 3: Results of Investigation of Shea Butter Extraction from the Roasted Shea Kernels\*

Measured Parameters	20 kg/h Screw Press Expeller	250 kg/h Screw Press Expeller	Model Cage Press
Oil Yield (%)	28.6	40.56	35.1
Extraction Efficiency (%)	47.6	67.6	58.5
Process Loss (%)	18.9	5.0	2.8

\* Each value is the mean of triplicates ± standard deviation

## 6. CONCLUSION

A small scale machine for roasting SK prior to SB extraction was designed, constructed and tested. The construction materials were readily available locally and at affordable costs. With a production cost of USD100, the machine has an output capacity of 20 kg/h and was operated by a 2 hp single-phase electric motor. While testing the machine, the flavour and aroma peculiar with roasting of SK were perceived. Visual observation revealed that the kernels were of uniform deep-brown colour with clear film of oil on the surface after roasting indicating that the kernels experienced uniform roasting.

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#### NOMENCLATURE

$c$	Specific heat capacity	$\text{kJ/kg } ^\circ\text{C}$
$d$	Shaft diameter	m
$D$	Screw Diameter	m
$E_E$	Extraction efficiency	%
$F$	Filling factor	-
$g$	Acceleration due to gravity	$\text{m/s}^2$
$L$	Length of conveyor	m
$M$	Mass	kg
$N$	Rotational speed	rpm
$N_1$	Rated speed of electric motor	rpm
$N_2$	Speed of the shaft	rpm
$O_Y$	Oil yield	%
$P$	Power	W
$p$	Screw pitch	m
$P_m$	Power of electric motor	W
$P_h$	Rated power of the heater	W
$P_L$	Process loss	%
$Q$	Auger capacity	$\text{m}^3/\text{h}$
$Q_o$	Operating capacity	kJ
$T$	Torque	$\text{N/m}^2$
$t$	Roasting time	s
$W_{AR}$	Weight after roasting	kg
$W_{BR}$	Weight before roasting	kg
$W_{US}$	Weight of shea kernel sample	kg
$W_{OE}$	Weight of oil extracted	kg
$W_{LS}$	Weight loss	%
$W_{RC}$	Weight of residual cake	kg
$\delta_o$	Yield stress	$\text{N/m}^2$
$\rho$	Density	$\text{kg/m}^3$
$\theta$	Change in temperature	$^\circ\text{C}$
$\phi$	Filling factor	-
$\eta$	Efficiency	%

#### BIOGRAPHY OF THE MAIN AUTHOR

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# ANALYSIS, SIMULATION AND OPTIMIZATION OF THE MILKING PROCESS IN A COWSHED FOR THE PRODUCTION OF PARMIGIANO REGGIANO

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## ABSTRACT

The aim of this study is to optimize the current milking process of a cowshed, which provides milk for the production of Parmigiano Reggiano cheese. The ultimate goal of the analysis is to reduce the time required for milking operations, thus optimizing the whole management of the farm processes. The analysis described is based on a real case study, referring to a farm located near Parma (Italy). To optimize the milking process, a discrete-event simulation model was designed under Simul8<sup>TM</sup> professional platform. The model reproduces the main processes of the cowshed, and in particular, the movements of the animals inside the cowshed, during milking. Real data were collected to allow the reproduction of the milking process. Exploiting again the simulation model, two new configurations of the cowshed layout were designed and tested, and their performance, in terms of the total time required for milking, was compared to the current one, showing interesting savings.

Keywords: simulation; animal behavior; milking process; Parmigiano Reggiano production.

## 1. INTRODUCTION

In Europe, the dairy sector is one of the most important sectors within the food industry. The production of milk intended for the dairy industry is estimated around 159 million tons per year in 2013 (European Commission, 2015). The main cheese producers of Europe are Germany, France, the United Kingdom, Poland, the Netherlands and Italy, which together account for almost 70% of the EU production (European Commission, 2015). Parmigiano Reggiano is a typical Italian hard cheese, known among the most typical products of the Italian food industry (Bellesia et al., 2003) and as one of the oldest cheeses in Europe, dating back to the 13<sup>th</sup>-14<sup>th</sup> century (Zannoni, 2010).

The production of Parmigiano Reggiano is located mainly in the North of Italy, in a limited geographic area, to comply with the protected designation of origin of the product (Zannoni, 2010). A specific consortium (i.e., *Consorzio del Parmigiano Reggiano*) collects the small and medium sized cheese factories and

cooperatives that operates in the production of Parmigiano Reggiano. In 2013, the main figures of the Parmigiano Reggiano production sector were as follows: over 15.6 million tons of milk processed per year, which led to the production of more than 2.9 million wheels of cheese in 2013. The production of Parmigiano Reggiano cheese employs approx. 88% of the total production of cow's milk of Italy. The breeders operating in the Parmigiano Reggiano production chain are more than 3,100, with approx. 340 active dairy companies (Regione Emilia-Romagna, 2014). Approx. 33% of the production is exported outside Italy.

Parmigiano Reggiano is produced using raw, semi-skimmed milk, which can not in any way be subjected to chemical treatment nor been addicted with any substance. A specific set of rules, indeed, regulates the production of Parmigiano Reggiano cheese, to comply with the quality standards and protected designation of origin of the product (Consorzio del Parmigiano-Reggiano, 2011). Among others, this set of rules defines the feed for cows, the milk, the milking process and the manufacturing process of the cheese.

In this paper, we focus on the milking process for a cowshed that produces milk for the manufacturing of Parmigiano-Reggiano cheese. The farm analyzed is located in the area of Parma (Italy). The objective of the analysis was, overall, to improve the current milking process carried out at the farm, in terms of the total time, thus enhancing the productivity of the whole process. The analysis was supported by an *ad hoc* discrete-event simulation model, developed under Simul8<sup>TM</sup> professional platform. The remainder of the paper is organized as follows: section 2 provides an overview of the farm analyzed and describes the current (AS IS) milking process. Section 3 describes the data collection related to the AS IS scenario. Section 4 details the simulation model used to reproduce the milking process of the farm. Section 5 describes two alternative (TO BE) configurations that were designed to improve the AS IS milking process, as well as their analysis by means of the simulation tool. Section 6 summarizes the main findings of the study, the related implications and limitations, and indicates possible future research activities.



## 2. THE CONTEXT AND THE “AS IS” MILKING PROCESS

The farm analyzed owns 171 cows, and carries out two milking operations per day, one in the early morning and one in the afternoon. Milking processes are carried out in an appropriate room (milking room), where up to 16 cows can be handled simultaneously. Milking operations usually take approx. 3 hours, which is lower than the maximum allowed time of 4 hours, according to the requirements of Consorzio del Parmigiano-Reggiano, (2011). Currently, the milking process employs two people per work shift, resulting, overall, in 4 people involved in the milking activities. A scheme of the cowshed is proposed in Figure 1.

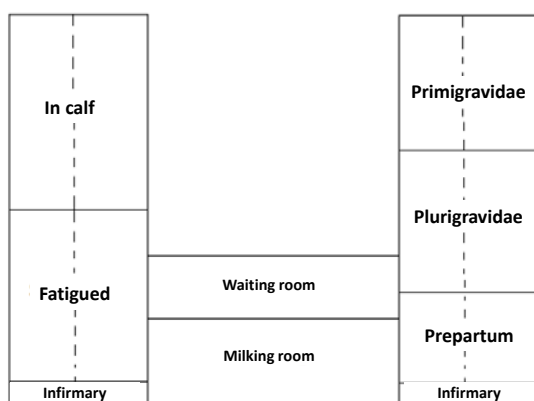


Figure 1: scheme of the cowshed and animals' groups

The milking process requires the cows to move from their positions in the cowshed (where they are located for most of the day) to the waiting room and then to the milking room. The waiting room is directly linked to the cowshed, so that the cows can easily reach it. One group of animals is moved at a time. Indeed, the cows can be classified in different groups, called 1-fatigued, 2-in calf, 3-primigravidae; 4-plurigravidae. The first group to be milked is that of fatigued cows, which consists of 48 animals. This group includes the oldest cows, which do not have a high production of milk and that require, under some circumstances, more time to complete the milking process. The second group to be milked is that of the in calf cows, which consists of 48 animals. Their milk production may not be so high, although the time required for milking is, in general, shorter. The third group is the smallest one (24 animals) and includes cows that, up to that moment, had only one pregnancy. The last group includes 51 animals, which, up to that moment, had more than one pregnancy. The last two groups of cows provide, in general, the highest quantity of milk, although the production may vary significantly during the year. Indeed, the daily production of the plant can slightly vary, reflecting the variation in the production of milk by cows across the year, due, in turn, to breed, season, or other management factors (e.g., nutrition or frequency of milking) (Jialina et al., 1998). In addition, the status of an animal can also change in time: for instance, a cow

belonging to group 3 (primigravidae) could become a member of group 4 (plurigravidae) after some time. Sometimes, cows can be moved to the farm infirmary, because of health problems. The milk produced by these animals should not be used for the production of Parmigiano Reggiano for a given time period, because of the possible presence of antibiotics.

The whole amount of milk produced by the milking process is moved to a tank, where it is stored at 18°C, and subsequently moved to the cheese manufacturing company.

## 3. DATA COLLECTION

On the basis of the analysis of the current milking process, the aim of the work done in this paper was to look for alternatives configurations of that process, at the same time keeping the layout of the cowshed unchanged. As mentioned, the analysis was supported by a discrete-event simulation model, whose construction required some preliminary data, primarily related to the time required to move the animals in the cowshed. Those data were collected through direct measurements. It was found that the time required to move the animals ranges from three minutes for the groups closer to the milking room (i.e., plurigravidae and fatigued) to four minutes for the groups more distant (primigravidae and in calf). This time was considered as adequate, in the light of the number of animals to move. The time required for milking operations was also recorded. The milking process has a defined structure, including:

- Preparation phase, including washing of the breast and of the immediate surrounding areas, and subsequent accurate drying, made with paper, which is then trashed;
- Attachment phase, where the employee checks the health status of the breast and sticks the teats of the milking to the cows' nipples. From this moment on, the data for the milk production, time and cow identity are recorded in the database of the farm;
- Detachment phase, which is the opposite of the attachment phase. This is an automatic step;
- Post-dipping phase, which is carried out with an appropriate detergent used on the cows' nipples.

The data related to the milking process were collected exploiting both the farm's information system and direct measurements. Specifically, the data related to 9 days (18 milking processes) were extracted from the farm's database. This set of records included the following pieces of information: (1) animal number, (2) group number; (3) milk production; (4) milking start time; (5) milking end time; (6) milking duration. During the period of measurement, the farm's database recorded 48 fatigued cows, 48 in calf, 24 primigravidae and 51 plurigravidae.

The analysis of the data recorded showed that, on average, a cow produced 10.24 l of milk per day in the analyzed period. At the same time, however, the production is quite variable (see Figure 2).

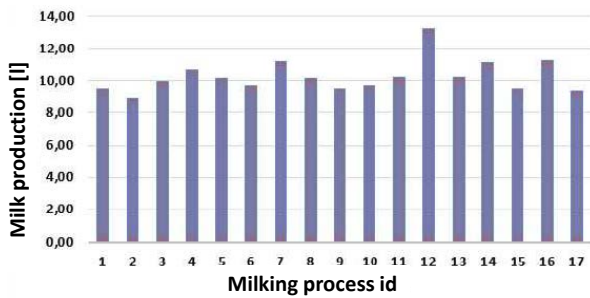


Figure 2: milk production [l]

With respect to the time required for milking, the data elaborated from the farm’s information system led to the results reported in Figure 3.

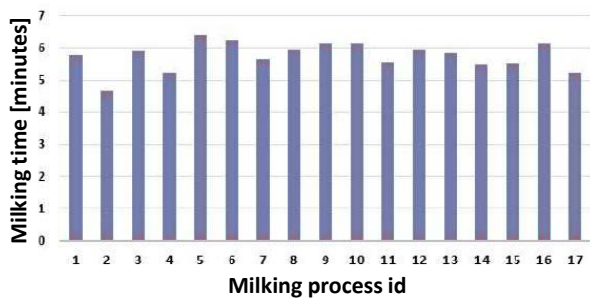


Figure 3: time required for milking.

Moreover, for 10 days (20 milking processes), direct measurements were made at the farm. During these measurements, the time required to complete the following activities on the animals was recorded:

- Cows’ movement from their position to the waiting room;
- Exit of the cows from the milking room and return in their positions.

Also, some measurements were carried out to assess the time required to the employees to complete some specific activities, such as:

- Preparation Phase
- Attachment phase
- Detachment phase.

The time required for milking was not measured because it could be directly derived from the data recorded in the farm’s database.

Finally, the detailed movements of the cows in the cowshed during milking were analyzed. As mentioned, the milking room is equipped with two sets of milking machines, each one able to handle 8 cows, resulting in 16 animals handles simultaneously. As an example, let us consider the first group of cows to be handled (i.e.,

group 1-fatigued); their initial movement inside the cowshed is described in Figure 4.

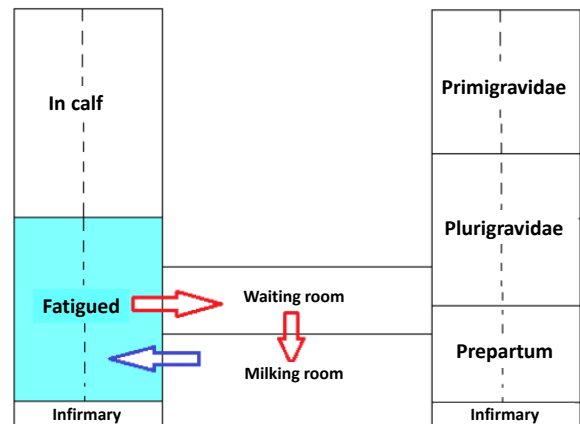


Figure 4: example of cows’ movements in the cowshed.

Once the milking process of group 1 starts, the cows should be moved back to their positions, while group 2 of cows should be moved to the milking room, as shown in Figure 5.

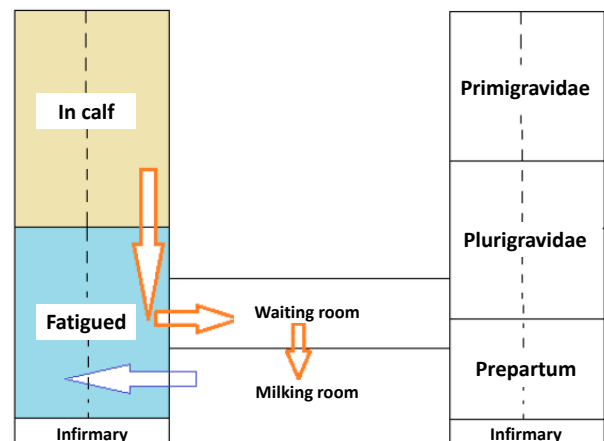


Figure 5: example of cows’ movements in the cowshed.

Similarly, the movements of group 3 and 4 of cows were analyzed.

#### 4. THE MODEL

The “AS IS” status of the milking process was reproduced in a simulation model, developed *ad hoc* exploiting the commercial software Simul8™ (Visual Thinking International Ltd). Simul8™ uses dynamic discrete simulation and is commonly exploited to simulate systems that involve processing of discrete entities at discrete times. Examples of those systems are production, manufacturing, logistic or service provision systems. As output, it generates statistics of performance parameters and metrics of the production system examined (Concannon et al., 2007).

The scheme of the model is proposed in Figure 6.

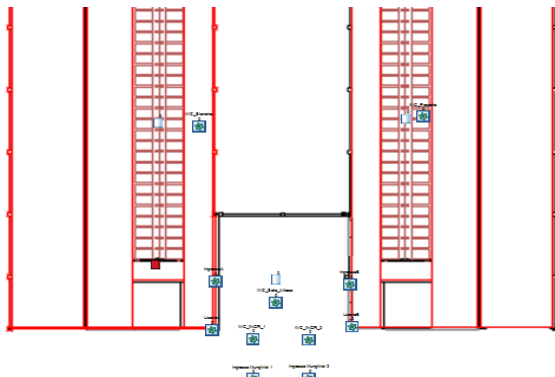


Figure 6: scheme of the Simul8™ model – AS IS scenario.

The model makes use of the input data described in section 3. Appropriate logics were embodied in the model to reproduce the behavior of the cows in the cowshed (cf. Figures 4 and 5), as well as the random order by which the cows move to the waiting room and milking room. Overall, the model includes 61 variables and 13 labels and takes the input data from 16 excel files.

To validate the model, we compared the results returned, in terms of the total time required for the milking process, with those measured during the data collection phase. In the real process, the total time required for milking varies from 2 hours 40 minutes to 3 hours. The results obtained through simulation reproduce the real scenario quite well, as can be seen from the frequency distribution in Figure 7.

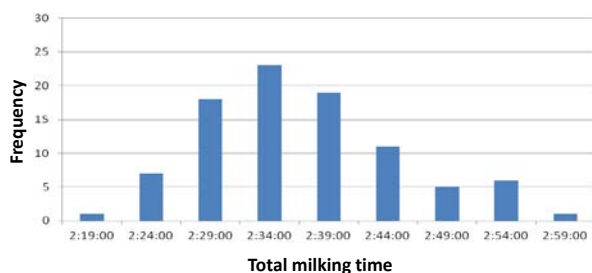


Figure 7: simulated results for the milking time - AS IS scenario.

The model was used to derive some further outcomes, such as the total amount of milk produced and the total production of milk per animal (cf. Figures 8 and 9). In can be seen from these figures that the model estimate the milk production per cow to range from approx. 11 to approx. 12 liters/cow, which is in line with the data

collected from the real process. Therefore, the model could be considered as validated.

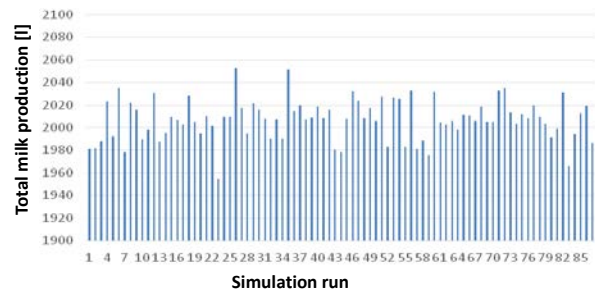


Figure 8: simulated results for the total milk production – AS IS scenario.

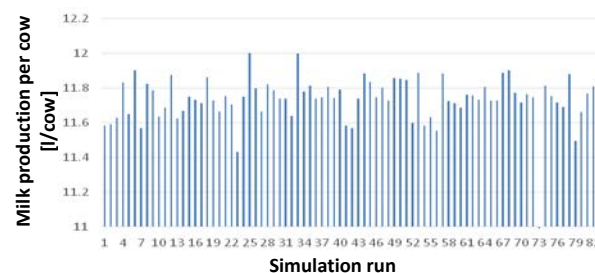


Figure 9: simulated results for the total milk production per cow – AS IS scenario.

## 5. TO BE CONFIGURATIONS

### 5.1. Definition of the TO BE scenarios

Once validated, the simulation model was used to examine two alternative configurations for the milking process, resulting in as many TO BE scenarios.

1. The first configuration examined considers a milking room with the same structure as that of the AS IS scenario. The equipment installed in the milking room, however, has a higher capacity, allowing to handle 24 cows (instead of 16) simultaneously. To simulate this TO BE configuration, some adjustments were made to the original simulation model, as well as to some of the input data. From a practical perspective, such a configuration would also require a slight reorganization of the space available in the milking room, to be implemented. Nonetheless, it is reasonable to expect that increasing the production capacity of the milking room will result in a decrease in the time required to complete the milking activities, resulting in a more efficient process.
2. The second TO BE configuration considered requires the complete reorganization of the cowshed layout, with the milking room and waiting room located in a different position compared to the AS IS scenario. A scheme of this TO BE configuration is proposed in Figure 10. The new positions of the milking room and waiting room would be useful, in particular, to

facilitate the movement of in calf and primigravidae cows, compared to the original structure of the cowshed. The second TO BE scenario would obviously modify the movements of the cows in the cowshed. Therefore, the model logics were slightly changed to reproduce the new cow movements. It is also hypothesized that 24 cows can be handled simultaneously in this configuration.

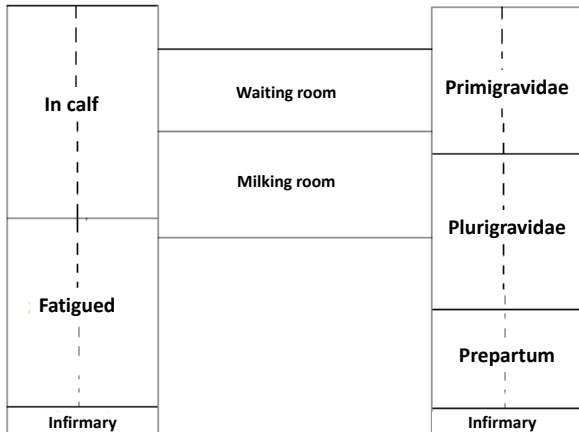


Figure 10: scheme of the second TO BE configuration.

### 5.2. Results for the first TO BE scenario

Simulating this TO BE configuration required some changes in the input data of the model (MS Excel™ files). For instance, the time required for preparation, washing, attachment and detachment were recalculated, considering the presence of 24 cows instead of 16 in the milking room. Conversely, data related to the behavior of the cows (e.g., the time for their movement or the waiting time) or those related to the milking process (e.g., the milking time and milk production) were not changed. With respect to the simulation model, simulating this TO BE scenario required adding 4 work centers in the milking process, to take into account the increased number of cows that can be handled simultaneously. The new model is proposed in Figure 11. Specific logics were defined for these new work centers. The relevant results obtained for the first TO BE configuration (with 300 simulation runs) are reported in Figure 12.

It can be seen from Figure 12 that, with this new configuration, the milking time decreases (overall) by approx. 20 minutes, equally shared among the different groups of cows (i.e., 5 minutes per group, on average). By saving 20 minutes in the milking process, employees can be allocated to different activities for a total of approx.  $20 \times 2 \times 2 = 80$  minutes per day.

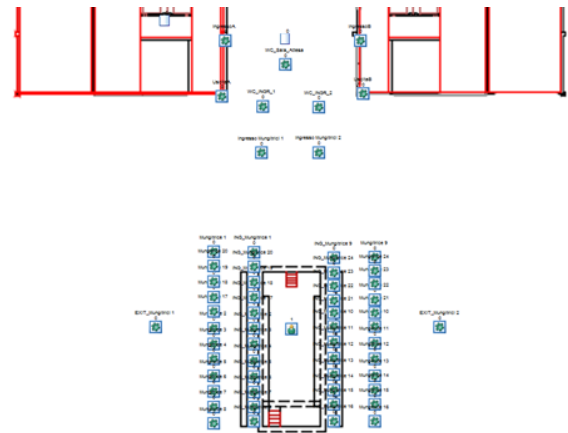


Figure 11: scheme of the Simu8™ model – TO BE scenario #1.

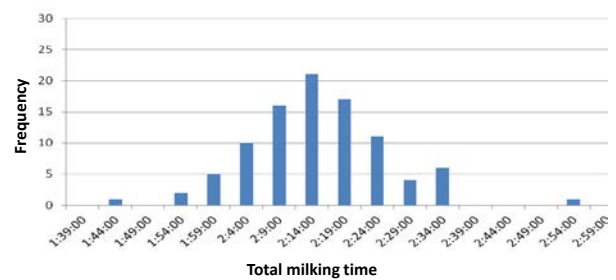


Figure 12: simulated results for the total milking time – TO BE scenario #1.

### 5.3. Results for the second TO BE scenario

Simulating the second TO BE configuration required some modifications to the input data, in terms of the time required to the cows to enter the milking room and the waiting room, as well as to go back to their locations in the cowshed. These data were recalculated based on the modified distances to be covered in the new configuration. Conversely, the simulation model was not changed in its structure.

The relevant results obtained for the second TO BE configuration (with 300 simulation runs) are reported in Figure 13.

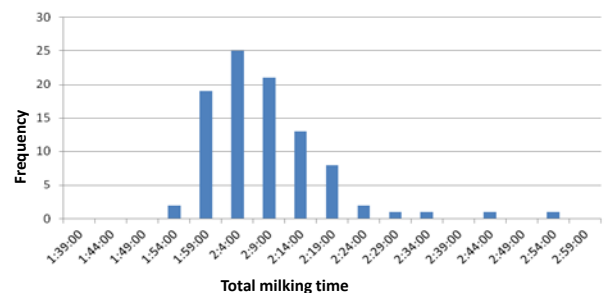


Figure 13: simulated results for the total milking time – TO BE scenario #2.

As can be seen from Figure 13, with this new configuration the time required for milking reduces to approx. 2 hours (between 2 hours 5 minutes to 2 hours 10 minutes), resulting in a significant saving compared to the original scenario. Only some simulation runs

returned a milking time close to 3 hours. Approx. 20,000 liters of milk are produced at every milking process, with an average production of 11.7 liters of milk per cow. As per the previous scenario, by saving approx. 30 minutes per milking process, employees could be allocated to different activities for a total of  $30 \times 2 \times 2 = 120$  minutes per day.

## 6. CONCLUSIONS

The aim of this study was to optimize the current milking process of a cowshed, which provides milk for the production of Parmigiano Reggiano cheese. The ultimate goal of the analysis was to reduce the time required for milking operations, thus optimizing the whole management of the farm processes.

To reach this aim, an *ad hoc* simulation model was developed to reproduce the milking process at the cowshed. The discrete-event simulation model was designed under Simul8<sup>TM</sup> professional platform. The main processes of the cowshed, and in particular, the behavior of the animals (e.g., their movements, waiting time and so on), were reproduced in the model, grounding on real data collected at the cowshed.

In line with the fact that the animals' behavior can significantly vary in time, the data collected were found to be characterized by high variance. Nonetheless, the simulation model allowed to estimate the time required for milking with a good approximation. Then, exploiting again the simulation model, two alternative configurations of the cowshed layout were tested and the related performance, in terms of the total time required for milking, was compared to the current one.

Both the new configurations of the milking process show interesting savings, in terms of the total time required for milking, compared to the current scenario. The first solution, in particular, allows saving approx. twenty minutes, compared to the AS IS scenario, allowing milking operations to be completed in approx. two hour and a quarter. As for the second configuration, the time saving is even greater, due to the greater proximity of all four groups of cows to the waiting room. On average, the groups most isolated, today, employ four minutes to reach the milking room. The final time saving is approx. half an hour compared to the initial situation.

From the practical perspective, both solutions could be considered for implementation by the company analyzed, although the first one is easier to be implemented, because it does not require modifications to the cowshed layout.

From the scientific point of view, this paper shows that simulation can be used also to reproduce very stochastic situations, such as the behavior of animals, with a good approximation, thus providing useful findings to the cowshed owners.

Among the limitations of the present study, it should be mentioned that the two TO BE scenarios examined does not cover all possible alternative configurations for the milking process examined. Other TO BE scenarios could be designed considering, for instance,

modifications to the equipment used for milking. Therefore, future research could be directed toward the development of additional TO BE scenarios. Another point is that, as recalled earlier in the paper, the production of milk can significantly vary during the year (Jialina et al., 1998). In this work, the data collection phase was carried out at the beginning of the summer, reflecting the period where the milk production is lower. Repeating the analysis in a different period of the year could be useful to confirm the validity of the model, as well as to assess the sensitivity of the results as a function of the milk production.

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# OPERATING STRATEGIES TO MINIMISE THE METHANOL CONTENT IN DISTILLATES OBTAINED IN ALEMBICS

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## ABSTRACT

The aromatic composition of fruit distillates not only depends on the raw material, but also on the distillation system and its operating conditions. One of the several compounds recovered during distillation is methanol, which is toxic for human consumption. In this paper, a multi-objective optimal control problem is formulated to minimise methanol's recovery and maximise ethanol's recovery in a batch distillation system, by adapting the heating trajectory. The dynamic optimisation problem is solved using orthogonal collocation on finite elements. Compared with a standard operation policy, the optimal heating trajectory reduced the methanol concentration in the distillate by 50%, without significantly affecting the ethanol recovery.

Keywords: aroma distillation, methanol, multi-objective optimisation, spirits.

## 1. INTRODUCTION

Distilled alcoholic beverages are produced worldwide from a wide variety of fruits and cereals, for example, Whisky (UK, Ireland), Cachaça (Brazil), Tequila (Mexico), Cognac/Brandy (France, Spain) and Pisco (Peru, Chile), to name a few. In the case of Pisco, the fermented juice is made from Muscat grapes containing 94-98% v/v of a mixture of water and ethanol, where the rest is comprised of hundreds of volatile compounds (Thorngate 1998; Voilley & Lubbers 1998). These minority volatile compounds are called congeners and essentially define the quality of the fruit distillates. The preparation of many of these distillates goes through several stages, and distillation is the most relevant in the final composition of the spirit. For distillation, Charentais copper alembics are the most frequently used in small scale production facilities. This system is operated in batch mode, recovering three cuts, head, heart and tail, where the heart cut gives rise to the final product. Although the operation of this alembic is

relatively simple, its operation is exposed to many disturbances that generate variability in the composition of the final product.

Nowadays, consumers are more demanding, asking for new products that are safe and aromatically distinctive. Therefore, the distillate should be rich in positive aromas as well as free from off-flavours and toxic compounds. Moreover, these products should keep their quality from year to year.

In this study, we focus on minimising the methanol content of the distillates, which is recovered in the three cuts. Methanol is slowly metabolized in the human body, producing formaldehyde and formic acid which are extremely toxic in high concentrations. Excessive intake of methanol generates various ailments such as fatigue, thirst, headache, stomach pains, nausea, vomiting, sensitivity to light and noise, lack of concentration and attention, tremors, excessive sweating and hypertension (Swift & Davidson 1998). Hence, in many countries, methanol content in alcoholic beverages is regulated. In Chile, methanol concentration must be less than 1.5 g/L of absolute alcohol (L.a.a).

Mathematical models have been developed for exploring new operating strategies in food and chemical processes, in order to obtain reproducible products with a predefined composition. Several phenomenological models have been presented in the literature for wine batch distillation, for example, in plate rectification columns (Osorio et al., 2004), packed rectification columns (Carvallo et al., 2011) and alembics (Scanavini et al 2010; Scanavini et al., 2012; Sacher et al., 2013). In addition, several numerical techniques have been applied to design dynamic optimization strategies for batch distillation. These techniques approach the design

as an optimal control problem, where the goal is to optimise a specific performance index (minimum time, maximum distillate, maximum quality or maximum profit) by searching for the best control path. For example, for a plate column batch distillation system, Osorio et al. (2005) solved the optimal control problem finding the trajectory of the cooling flow rate in the partial condenser. The objectives were to maximise the terpene recovery and minimise the fatty acid recovery in a Muscat wine distillate. The process model was very complex, with many differential and implicit algebraic equations; therefore, the formulation solution of a full optimisation problem is extremely difficult. To simplify the numerical solution of the optimisation problem, the cooling flow rate was established as a predefined smooth function of time, derived from previous experiences. The optimisation resulted in the parameters of the predefined function that maximised a multi-objective cost function. The optimal strategy obtained was validated in lab scale experiments. De Lucca et al. (2013) explored various operating policies, with a simulator of a packed bed column, to minimise the methanol content in the distillate. They used the same predefined function with adjustable parameters of Osorio et al. (2005) to find an optimal cooling flow rate trajectory. In these studies, the predefined function restricted the cooling flow rates to a sub-optimal solution space.

In this paper, a full optimal control problem is formulated and solved for a Charentais alembic distillation. The aim is to minimise the methanol recovery and maximise the ethanol recovery in the distillate in a fixed operation time, by manipulating the heating power trajectory. A ternary version of a previous alembic model (Sacher et al., 2013) was used in the optimisation.

## 2. MATERIALS AND METHODS

### 2.1. Distillation system

The distillation system is an automatic controlled Charentais copper alembic (Figure 1) which has the following features and elements:

- 4.8 L capacity
- Total condenser
- Three Pt100 sensors to measure the temperature in the boiler, partial condenser and surroundings
- Electrical resistance inside the kettle (1200 W)

- A Programmable Logic Controller (PLC) receives data from the temperature sensors and sends the control signal to the electrical resistance in the boiler.

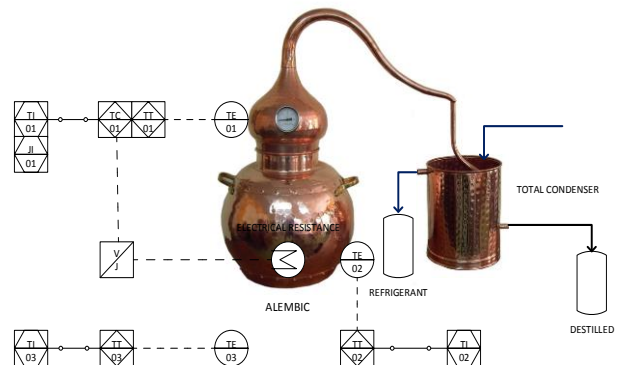


Figure 1: P&ID of the distillation system.

### 2.2. Experimental conditions

The Muscat wine was approximated as a synthetic ternary mixture of 13% v/v of ethanol and 1.5 g/L.a.a. of methanol (Osorio et al. 2004). The alembic was initially loaded with 1.8 L, and the ambient temperature was constant at 24 °C.

### 2.3. Modelling

To simplify, the liquid-vapour (L/V) equilibrium was considered to be a non-ideal "quasi binary" mixture of ethanol/water plus a congener (Sacher et al., 2013). The model includes mass and energy balances for the two equilibrium stages, the boiler and the partial condenser. The congeners' L/V equilibrium depends on ethanol concentration only, due to the assumption of "quasi binary" mixture. Activity coefficients were calculated using the UNIFAC group contribution method.

The mathematical model used in this paper is a simplified version of Sacher et al. (2013), which only considers a mixture of water, methanol and ethanol.

*Mass (total, ethanol, methanol) and energy balances in the boiler,*

$$\frac{d(M_B)}{dt} = L - V_B \quad (1)$$

$$\frac{d(M_B \cdot x_B^e)}{dt} = L \cdot x_L^e - V_B \cdot y_B^e \quad (2)$$

$$\frac{d(M_B \cdot x_B^m)}{dt} = L \cdot x_L^m - V_B \cdot y_B^m \quad (3)$$

$$\frac{d(M_B \cdot u_B)}{dt} = L \cdot h_L - V_B \cdot H_B + \dot{Q}_B \quad (4)$$



Mass (total, ethanol, methanol) and energy balances in the partial condenser (negligible liquid holdup),

$$V_B - L - V_D = 0 \quad (5)$$

$$V_B \cdot y_B^e - L \cdot x_L^e - V_D \cdot y_D^e = 0 \quad (6)$$

$$V_B \cdot y_B^m - L \cdot x_L^m - V_D \cdot y_D^m = 0 \quad (7)$$

$$V_B \cdot H_B - L \cdot h_L - V_D \cdot H_D - \dot{Q}_C = 0 \quad (8)$$

*Thermodynamic equilibrium ethanol-water*

The mole fraction of ethanol vapour ( $y_B^e, y_D^e$ ), equilibrium temperatures and enthalpies ( $H_{B,D}, h_L$ ) are calculated from empirical correlations (Sacher et al., 2013). These equations depend on the ethanol mole fraction in the liquid phase ( $x_B^e, x_L^e$ ) only.

*Thermodynamic equilibrium relationships for methanol,*

$$y_D^m = K_C^m \cdot x_L^m \quad (9)$$

$$y_B^m = K_B^m \cdot x_B^m \quad (10)$$

$$K_{B,C}^m(x_B^e, x_L^e) = \frac{y_{B,D}^m}{x_{B,L}^m} = \frac{P_m(x_B^e, x_L^e) \cdot \gamma_m(x_B^e, x_L^e)}{P} \quad (11)$$

The activity coefficient for methanol  $\gamma_m$  is estimated using the UNIFAC contribution groups method. Given the assumption of a quasi-binary mixture, the activity coefficient only depends on the ethanol concentration since an infinite dilution of methanol in a mixture of water-ethanol is assumed.

*Heat transfer model*

$$\dot{Q}_B = \dot{Q}_{cal} - U \cdot A \cdot (T_B - T_{amb}) \quad (12)$$

$$\dot{Q}_C = U \cdot A \cdot (T_C - T_{amb}) \quad (13)$$

where  $\dot{Q}_{cal}$  and  $T_{amb}$  are input variables corresponding to the control variable and disturbance of the system, respectively. This model has only one empirical

parameter,  $U \cdot A$ , which can be easily fitted with data normally available in commercial distillation facilities (Pérez-Correa et al., 2013).

*Simulation*

To simulate the model, a reordering of equations is convenient. The distillate molar flow rate is obtained from mass and energy balances in the partial condenser (Eqs. 5, 6 and 8).

$$V_D = \frac{\dot{Q}_C}{(H_B - H_D) + \frac{(y_B^e - y_D^e)}{(x_L^e - y_D^e)} \cdot (H_D - h_L)} - \frac{-\dot{Q}_C}{(h_L - H_D) + \frac{(x_L^e - y_D^e)}{(y_B^e - y_D^e)} \cdot (H_D - H_B)} \quad (14)$$

To calculate the volume of distillate, an empirical correlation which calculates the density of the mixture from ethanol composition is used (Neuburg & Perez-Correa 1994),

$$\rho_L = \frac{x_D^e \cdot PM_E + (1 - x_D^e) \cdot PM_W}{\phi \cdot x_B^e + (1 - x_B^e) \cdot PM_W / \rho_W} \quad (15)$$

$$\phi = f(x_B^e, T_D) \quad (16)$$

Finally to calculate the composition of ethanol in the partial condenser, a rearrangement of the energy balance (Eq. 4) from mass balances in the boiler (Eqs. 1 and 2) is required,

$$(L(x_L^e - x_B^e) - V_B(y_B^e - x_B^e)) \left( \frac{\partial h_B}{\partial x_B} + \frac{\partial h_B}{\partial T_B} \frac{dT_B}{dx_B} \right) = L(h_L - h_B) - V_B(H_B - h_B) + \dot{Q}_B \quad (17)$$

This equation is an implicit function that depends on the value of  $x_L^e$ . This equation was solved in the AMPL code (see appendix) as a constraint in the optimisation problem.

#### 2.4. Optimisation

The multi-objective weighted cost function considers methanol (main objective of this study) and ethanol recoveries in the distillate. The weight defines the priority in the respective distillation. Then, the objective function is,

$$\min_u J(u) = \alpha Rec_{met}(tf) - (1 - \alpha) Rec_{et}(tf) \quad (18)$$

$$0 \leq \alpha \leq 1 \quad (19)$$

where the recoveries are,

$$Rec_k = \frac{\int_{t_0}^{t_f} (D(t) \cdot x_k(t)) dt}{M_0 \cdot x_k(t_0)} \quad (20)$$

$\alpha$  is the adjustable weight,  $Rec_k$  is the recovery of compound  $k$  (met = methanol, et = ethanol),  $D(t)$  is the molar flow rate of distillate,  $x_k$  is the mole fraction of compound  $k$ ,  $M_0$  is the initial mass, and finally,  $t_0$  and  $t_f$  are the initial and final distillation times respectively.

A common practice in the fruit distillates industry is to use the cooling rate in the partial condenser to control the reflux rate. In the alembic used in this study, the reflux rate is given by natural condensation and is not directly controllable. Hence, we manipulated the heating power in the alembic boiler instead. The optimisation process must find an optimal heating power path that minimises methanol recovery and maximises ethanol recovery in the distillate. We applied the simultaneous simulation-optimisation method (direct transcription), using orthogonal collocation on finite elements (Biegler, 2010), which has been shown to be useful in tackling large scale optimisation problems. Only one cut was considered in computing the cost function (Eq. 18), corresponding to the heart/tail cut that was set at 2 h. Additional optimisation constraints are: (i) the heating power varies between 0 and 1100 W; (ii) the minimum distillate flow rate is 2 mL/min.

The optimal control problem in its discrete form can be expressed as:

$$\min_u J(u) \quad (21)$$

subject to

$$\forall i = 1 \dots ne, j = 1 \dots ncp$$

$$x_{i,j} = x_{i-1} + h_i \sum_{j=1}^{ncp} \Omega_j(\tau_j) \cdot \frac{dx}{dt_{i,j}} \quad (22)$$

$$g(x_{i,j}, y_{i,j}, u_{i,j}, \theta) = 0 \quad (23)$$

$$u_L \leq u_{i,j} \leq u_U \quad (24)$$

$$x_L \leq x_{i,j} \leq x_U \quad (25)$$

$$\theta_L \leq \theta \leq \theta_U \quad (26)$$

$$\frac{dx}{dt_{i,j}} = f(x_{i,j}, y_{i,j}, u_{i,j}, \theta) \quad (27)$$

where  $ne$  is the number of finite elements,  $ncp$  is the number of collocation points,  $x$  are differential state variables,  $y$  are algebraic state variables,  $u$  are control variables and  $\theta$  are model parameters. Equation 22 corresponds to the discrete approximation of the differential state variables.  $h_i$  corresponds to the length of the finite element, which is given by the total process time divided by the number of finite elements;  $\tau$  represents the normalised time in the finite element. Finally  $\Omega$  are the interpolation polynomial functions in each finite element. In this study, we used 3 Radau collocation points (Biegler, 2010). The optimisation problem was coded in AMPL and solved using IPOPT (Biegler, 2010). Figure 2 shows the flowchart of the simultaneous simulation-optimisation method, using the orthogonal collocation on finite elements strategy used in this work.

### 3. RESULTS AND DISCUSSION

The optimisation was solved in a Samsung Intel core i7 laptop using the student version of AMPL. Therefore, we considered only 5 finite elements, since the maximum number of variables and constraints allowed were 300. We also tried using the NEOS web interface, but the optimisation problem did not converge with any solver available.

Different values of the weight  $\alpha$  were tried, but the solver only found solutions for  $0.5 \leq \alpha \leq 1$ . For values below 0.5 the cost function took high negative values, resulting in a degenerated optimization problem; hence, some constraints were not met. Table 1 shows the cost function (J), the distillate alcohol grade ( $GA_d$ ), the relative methanol concentration in the distillate ( $Met_d$ ), the ethanol recovery ( $Rec_{et}$ ) and the methanol recovery ( $Rec_{met}$ ) for different values of  $\alpha$ .

Table 1: Optimization results for different values of  $\alpha$ .

$\alpha$	J (-)	$GA_d$ (%)	$Met_d$ (g/L.a.a)	$Rec_{et}$ (%)	$Rec_{met}$ (%)
0.5	-8,1	46.0	1.37	64.8	48.7
0.6	1,8	52.4	1.16	59.8	42.9
0.72	31	78.5	0.75	58.3	41.2
0.8	22	52.4	1.16	59.8	42.9
0.9	31	78.5	0.75	58.3	41.2
1	41	53.5	1.09	58.3	41.0

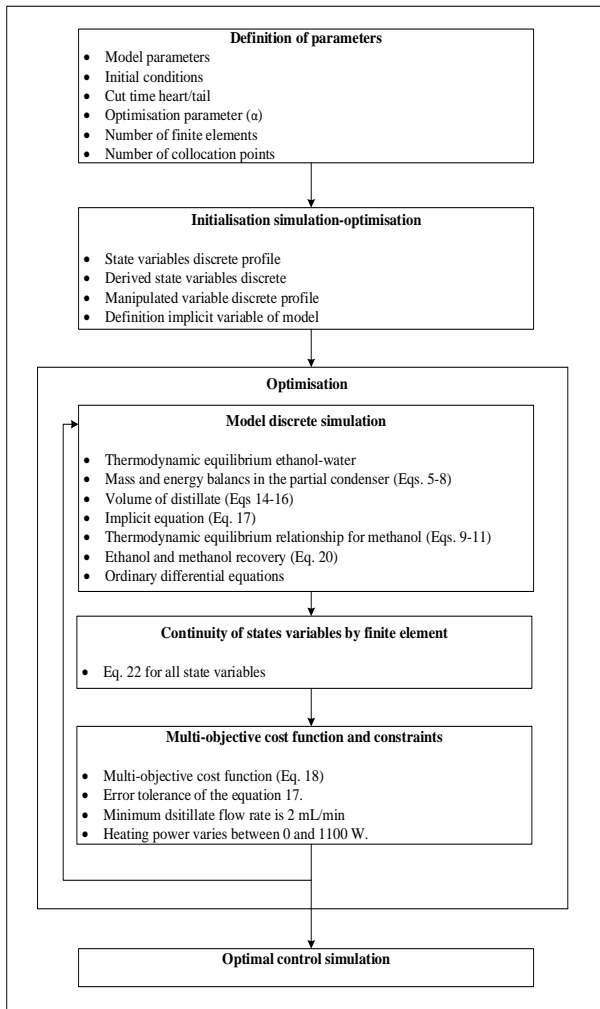


Figure 2: Flowchart of the simultaneous simulation-optimisation method.

It is important to highlight that for values of  $\alpha = 0.75$  and  $0.9$ , a very low relative concentration of methanol in the distillate was obtained. These optimal trajectories reduce the relative concentration of methanol by 50% compared with the initial wine. In previous research (Luna et al., 2014) relative concentration of methanol were reduced by only 29%. In turn, for  $\alpha = 0.5$ , the highest methanol recovery was obtained. We observed that lower methanol relative concentrations were obtained when the distillates were more concentrated in ethanol and less total volume was recovered.

Figure 3 shows optimum trajectories of the heating power for the extreme cases ( $\alpha = 0.5$  and  $\alpha = 0.9$ ). To obtain a distillate with a low concentration of methanol ( $\alpha = 0.9$ ), a high heating power is required at the beginning of the distillation and a low heating power at the end. In turn, if a low heating power is applied at the beginning and a high heating power at the end ( $\alpha = 0.5$ ), a distillate with high methanol concentration will be obtained.

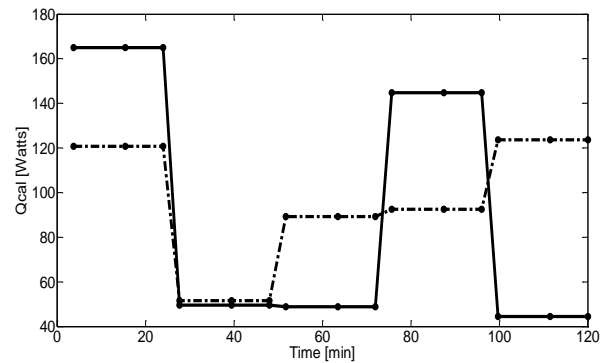


Figure 3: Heating power vs. time. Dashed line:  $\alpha = 0.5$ ; Solid line:  $\alpha = 0.9$ .

#### 4. CONCLUSIONS AND FUTURE WORK

This is the first study that apply a direct transcription method to minimise methanol content in Charentais alembic spirit distillations. Our results indicate that by properly managing the heating power in the boiler, significant reductions in the distillate methanol concentrations can be achieved. We have previously applied several operating strategies, and never achieved such low methanol concentrations as shown in this work. We expect that by testing more robust cost-functions, improving the method for solving the implicit algebraic constraint, and using more finite elements, we will be able to attain smoother heating power trajectories and even lower methanol concentrations.

Future work consider experimental validation, addition of more congeners in the distillation model, develop more complex cost-functions, and including cut times as decision variables.

Results shown here are general, therefore these can be applied to reduce methanol content in any fruit distillate obtained in alembics of any size.

#### APPENDIX

##### AMPL Code

```
# This program minimises the concentration of
methanol (Cmet) and maximises recovery of ethanol
(ETR) for the batch distillation process in a copper
Charentais, with a fixed operating time (tf) and less than
or equal final concentration of 1.5 g methanol/L.a.a. To
achieve this goal, heating power (Qcal) was handled
during distillation with a constant room temperature.
The model was published in its first version in Sacher et
al. 2013.
```

```
# First version: Ricardo Luna 14/01/2015.
```

```
# Process parameters or constants
```

```
param UAb := 0.3196778; # Coefficient and heat
transfer area of the boiler
```

```
param UAc := 0.3333333; # Coefficient and heat
transfer area of the condenser
```

```

param Tamb := 24+273.15; # Room temperature [k]
param MWe := 46.07; # Ethanol Molecular weight
param MWw := 18.0153; # Water Molecular weight
param MWmet := 32.04; # Methanol Molecular weight
param rhow := 1; # Water density [g/mL]
param time := 7200; # Final time [hours]
param alpha := 1; # Weight of function multi-purpose
# Initial conditions for differential variables
param x1_init := 91.789; # Boiler initial moles (Mb)
param x2_init := 0.0437; # Ethanol molar fraction in boiler (xb)
param x3_init := 0; # Distilled initial volume (V)
param x4_init := 0; # Moles of distilled ethanol (M_etd)
param x5_init := 0.000116; # Methanol molar fraction in boiler (xb_met)
param x6_init := 0; # Moles of distilled methanol (M_metd)
param t_init := 0; # Initial time (seg)
# Indexes, number of placement points and number of finite elements
param nfe >= 1 integer;
param ncp >= 1 integer;
let nfe := 5; # Finite element numbers
let ncp := 3; # number of placement points
set fe := 1..nfe; # Finite element index
set cp := 1..ncp; # placement points index
param h{fe} := (time/nfe); # length of a finite element
# Matrix of Radau placement coefficient (implicit Runge-Kutta)
param a{cp,cp};
let a[1,1] := 0.19681547722366;let a[1,2] := 0.39442431473909;
let a[1,3] := 0.37640306270047;let a[2,1] := -0.06553542585020;
let a[2,2] := 0.29207341166523;let a[2,3] := 0.51248582618842;
let a[3,1] := 0.02377097434822;let a[3,2] := -0.04154875212600;
let a[3,3] := 0.11111111111111;
# Discrete profiles of state variables and time
var x1 {fe,cp} >= 0 := x1_init ;
var x2 {fe,cp} >= 0, <=1, := x2_init ;
var x3 {fe,cp} >= 0 := x3_init ;
var x4 {fe,cp} >= 0 := x4_init ;
var x5 {fe,cp} >= 0, <=1, := x5_init ;
var x6 {fe,cp} >= 0 := x6_init ;
var t {fe,cp} >= 0 := t_init ;
# Discrete derivative of state variables and time
var x1dot {fe,cp} ;
var x2dot {fe,cp} ;
var x3dot {fe,cp} ;
var x4dot {fe,cp} ;
var x5dot {fe,cp} ;
var x6dot {fe,cp} ;
var tdot {fe,cp} ;
# Discrete profiles of control variables
var Qcal {fe} >= 0, <= 1000, := 300; # Watt
# Declaration of implicit variable of the model

```

```

var xL {fe,cp} >=0, <=1, := 1.1*x2_init; # Reflux liquid mole fraction L
# Constitutive equations (explicit algebraic)
# Liquid/vapor equilibrium in the boiler
# Steam mole fraction in boiler
var yb {i in fe,j in cp} = -59.6868501+-89.4037240*x2[i,j]+-39.8552042*x2[i,j]^1.5+81.47664393*exp(x2[i,j])-21.7897938*exp(-x2[i,j]);
# Boiler temperature
var Tb {i in fe,j in cp} = 273.15+(-0.02214517+-0.05785120*x2[i,j]^1.5+0.032146591*exp(x2[i,j]))^-1;
# Liquid enthalpy [J/gmol]
var hb {i in fe,j in cp} = (55.678*x2[i,j]+75.425)*Tb[i,j]-15208.44*x2[i,j]-20602.34;
# Gas enthalpy [J/gmol]
var Hb {i in fe,j in cp} = 36172.03-2919.83*yb[i,j]+(31.461-11.976*yb[i,j])*Tb[i,j]+(4.063*10^-4+0.0734*yb[i,j])*Tb[i,j]^2;
# dhb/dxb partial derivative
var dhbdxb {i in fe,j in cp} = 55.678*Tb[i,j]-15208.44;
# dhb/Tb partial derivative
var dhbdTb {i in fe,j in cp} = 55.678*x2[i,j]+75.425;
# dTbdxb partial derivative
var dTbdxb {i in fe,j in cp} = -(1.5*-0.05785120*x2[i,j]^0.5+0.032146591*exp(x2[i,j]))/(-0.02214517+-0.05785120*x2[i,j]^1.5+0.032146591*exp(x2[i,j]))^2;
# Total derivative
var DhbDxb {i in fe,j in cp} = dhbdxb[i,j]+dhbdTb[i,j]*dTbdxb[i,j];
# Liquid/vapor equilibrium in condenser
# Steam mole fraction in boiler
var yD {i in fe,j in cp} = -59.6868501+-89.4037240*xL[i,j]+-39.8552042*xL[i,j]^1.5+81.47664393*exp(xL[i,j])-21.7897938*exp(-xL[i,j]);
# Condenser temperature
var Tc {i in fe,j in cp} = 273.15+(-0.02214517+-0.05785120*(xL[i,j]^(1.5))+0.032146591*exp(xL[i,j]))^-1;
# Liquid enthalpy [J/gmol]
var hL {i in fe,j in cp} = (55.678*xL[i,j]+75.425)*Tc[i,j]-15208.44*xL[i,j]-20602.34;
# Gas enthalpy [J/gmol]
var Hd {i in fe,j in cp} = 36172.03-2919.83*yD[i,j]+(31.461-11.976*yD[i,j])*Tc[i,j]+(4.063*10^-4+0.0734*yD[i,j])*Tc[i,j]^2;
# Stationary mass balances in the condenser
# Molar flow of vapor ascending
var Vb {i in fe,j in cp} = UAc*(Tc[i,j]-Tamb)/((Hb[i,j]-Hd[i,j])+(yb[i,j]-yD[i,j])/(xL[i,j]-yD[i,j]))*(Hd[i,j]-hL[i,j]);
# Molar flow of liquid reflux

```

```

var L {i in fe,j in cp} = -UAc*(Tc[i,j]-Tamb)/((hL[i,j]-
Hd[i,j])+(xL[i,j]-yD[i,j])/(yb[i,j]-yD[i,j]))*(Hd[i,j]-
Hb[i,j]));
# Molar flow of distillate
var D {i in fe,j in cp} = Vb[i,j]-L[i,j];
# Calculation of volumetric flow of distillate
# Molecular weight of the mixture
var MWm {i in fe,j in cp} = MWe*yD[i,j]+(1-
yD[i,j])*MWw;
# Apparent molal volume for mixing [m3/kmol]
var oa {i in fe,j in cp} = 5.1214e-2+6.549e-
3*yD[i,j]+7.406e-5*(Tc[i,j]-273.15);
# Mixture density [g/mL]
var rhom {i in fe,j in cp} =
MWm[i,j]/(oa[i,j]*1000*yD[i,j]+(1-
yD[i,j])*MWw/rhow);
# Distillate volume [mL]
var Vd {i in fe,j in cp} =
D[i,j]*MWm[i,j]*(1/rhom[i,j]);
var tiempo {i in fe,j in cp} = t[i,j]/60;
var Flujo {i in fe,j in cp} = x3[i,j]/tiempo[i,j];
# Implicit algebraic constitutive equation
# Left side of the equation
var g1 {i in fe,j in cp} = (L[i,j]*(xL[i,j]-x2[i,j])-
Vb[i,j]*(yb[i,j]-x2[i,j]))*(DhbDxb[i,j]);
# Right side of the equation
var g2 {i in fe,j in cp} = L[i,j]*(hL[i,j]-hb[i,j])-
Vb[i,j]*(Hb[i,j]-hb[i,j])+(Qcal[i]-UAb*(Tb[i,j]-Tamb));
# Error definition of the equation
var e {i in fe,j in cp} = (g1[i,j]-g2[i,j])^2;
# Calculation of properties and methanol balances
var Kmetanolb {i in fe,j in cp} = -
407274.016707888*x2[i,j]^5+137084.909351167*x2[i,
j]^4-
19714.4353078579*x2[i,j]^3+1757.07481156027*x2[i,
j]^2-125.445056947666*x2[i,j]+7.99009392867259;
var Kmetanolc {i in fe,j in cp} = -
407274.016707888*xL[i,j]^5+137084.909351167*xL[i,
j]^4-
19714.4353078579*xL[i,j]^3+1757.07481156027*xL[i,
j]^2-125.445056947666*xL[i,j]+7.99009392867259;
# Mole fraction of methanol vapor in the boiler
var ybmet {i in fe,j in cp} =
Kmetanolb[i,j]*x5[i,j]*x5_init; # Multiply by x5_init to
return to the original value # Liquid mole fraction in the
gooseneck
var xLmet {i in fe,j in cp} =
(Vb[i,j]*ybmet[i,j])/(L[i,j]+D[i,j]*Kmetanolc[i,j]);
# Vapor mole fraction in the gooseneck
var yDmet {i in fe,j in cp} = xLmet[i,j]*Kmetanolc[i,j];
# Alcoholic strength
var GLb {i in fe,j in cp} = -
38.950116015168470*x2[i,j]^6+231.0609445800233*x
2[i,j]^5-
548.2530421230588*x2[i,j]^4+704.8795688586179*x2
[i,j]^3-
570.0321285089950*x2[i,j]^2+321.3122048471700*x2
[i,j]+0.001131244289002;
var GLd {i in fe,j in cp} = -
38.950116015168470*yD[i,j]^6+231.0609445800233*

```

```

yD[i,j]^5-
548.2530421230588*yD[i,j]^4+704.8795688586179*y
D[i,j]^3-
570.0321285089950*yD[i,j]^2+321.3122048471700*y
D[i,j]+0.001131244289002;
# Methanol concentration [gmetanol/L.a.a]
var Cmet {i in fe,j in cp} =
x6[i,j]*MWmet*1000/((x3[i,j]*(GLd[i,j]/100))+0.0000
0001);
# Ethanol recovery [%]
var Etr {i in fe,j in cp} = x4[i,j]*100/(x1_init*x2_init);
# Methanol recovery [%]
var Mtr {i in fe,j in cp} = x6[i,j]*100/(x1_init*x5_init);
# Differential equations
# Moles balance in the kettle
odex1 {i in fe,j in cp}: x1dot[i,j] = L[i,j]-Vb[i,j];
# Ethanol balance in the kettle
odex2 {i in fe,j in cp}: x2dot[i,j] =
1/x1[i,j]*(L[i,j]*(xL[i,j]-x2[i,j])-Vb[i,j]*(yb[i,j]-
x2[i,j]));
# Distillate volume in time
odex3 {i in fe,j in cp}: x3dot[i,j] = Vd[i,j];
# Moles of distilled ethanol in time
odex4 {i in fe,j in cp}: x4dot[i,j] = D[i,j]*yD[i,j];
# Methanol balance in the kettle
# Normalized equation to x5_init (to avoid tolerance
error)
odex5 {i in fe,j in cp}: x5dot[i,j] =
(1/x1[i,j]*(L[i,j]*(xLmet[i,j]-x5[i,j]*x5_init)-
Vb[i,j]*(ybmet[i,j]-x5[i,j]*x5_init)))/(1/x5_init);
# Moles of distilled methanol in time
odex6 {i in fe,j in cp}: x6dot[i,j] = (D[i,j]*yDmet[i,j]);
# Operating time
odet {i in fe,j in cp}: tdot[i,j] = 1;
# Declaration of real variable (denormalization of x5)
var X5 {i in fe,j in cp} = x5[i,j]*x5_init;
# Continuity of the states through finite element and its
definition in the placement points for finite element
i=2, nfe
# Moles in the kettle
fecolx1 {i in fe diff {1},j in cp}: x1[i,j] = x1[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x1dot[i,k];
# Ethanol fraction in the kettle
fecolx2 {i in fe diff {1},j in cp}: x2[i,j] = x2[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x2dot[i,k];
# Distillate volume
fecolx3 {i in fe diff {1},j in cp}: x3[i,j] = x3[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x3dot[i,k];
# Moles of distilled ethanol
fecolx4 {i in fe diff {1},j in cp}: x4[i,j] = x4[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x4dot[i,k];
# Methanol in the kettle
fecolx5 {i in fe diff {1},j in cp}: x5[i,j] = x5[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x5dot[i,k];
# Moles of distilled methanol
fecolx6 {i in fe diff {1},j in cp}: x6[i,j] = x6[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*x6dot[i,k];
# tiempo
fecolt {i in fe diff {1},j in cp}: t[i,j] = t[i-1,ncp]
+h[i]*sum{k in cp} a[k,j]*tdot[i,k];

```

```

# Continuity of the states through finite element and its
definition in the placement points for the first finite
element
# Moles in the kettle
fecolx10{i in 1..1,j in cp}: x1[i,j] = x1_init
+h[i]*sum{k in cp} a[k,j]*x1dot[i,k];
# Ethanol fraction in the kettle
fecolx20{i in 1..1,j in cp}: x2[i,j] = x2_init
+h[i]*sum{k in cp} a[k,j]*x2dot[i,k];
# Distillate volume
fecolx30{i in 1..1,j in cp}: x3[i,j] = x3_init
+h[i]*sum{k in cp} a[k,j]*x3dot[i,k];
# Moles of distilled ethanol
fecolx40{i in 1..1,j in cp}: x4[i,j] = x4_init
+h[i]*sum{k in cp} a[k,j]*x4dot[i,k];
# Methanol in the kettle
fecolx50{i in 1..1,j in cp}: x5[i,j] = x5_init/x5_init
+h[i]*sum{k in cp} a[k,j]*x5dot[i,k];
# Moles of distilled methanol
fecolx60{i in 1..1,j in cp}: x6[i,j] = x6_init
+h[i]*sum{k in cp} a[k,j]*x6dot[i,k];
# Time
fecolt0{i in 1..1,j in cp}: t[i,j] = t_init
+h[i]*sum{k in cp} a[k,j]*tdot[i,k];
# Objectives
var J1 = alpha*Mtr[nfe,ncp];
var J2 = (1-alpha)*Etr[nfe,ncp];
var Jt = J1+J2;
# Objective Function
Minimize fcosto: alpha*Mtr[nfe,ncp]-(1-
alpha)*Etr[nfe,ncp];
# Constraints
subject to r1 {i in 1..nfe,j in 1..ncp} :
e[i,j]<=0.0000001; # Error tolerance in the implicit
equation
subject to r2 : Etr[nfe,ncp] <=100;
subject to r3 : Mtr[nfe,ncp] <=100;
subject to r4 {i in 1..nfe,j in 1..ncp} : Flujo[i,j]>=2;
subject to r5 : Mtr[nfe,ncp] >=0;
subject to r6 : Etr[nfe,ncp] >=0;

```

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# STUDY AND OPTIMISATION OF A CO<sub>2</sub> SPARGER FOR CARBONATED BEVERAGES AND BEER BY MEANS OF CFD MODELLING

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## ABSTRACT

In the present work, two different geometries of spargers for beverage carbonation were modelled by means of CFD technique. The first geometry presented a radial inlet of liquid food while the second one a tangential one. Calculation allowed to identify the best solution and mathematical results were confirmed by experimental tests with both water and apple juice. CFD resulted a very useful technique for in-silico designing not only for simple parts of plants but also for very complicated ones such as carbon dioxide spargers in which gas and liquid mix together.

Keywords: carbonated beverage, computational fluid dynamic, dense phase carbon dioxide,

## 1. INTRODUCTION

The consumption of soft drinks in their various forms has taken place for many centuries in order to meet the body's fundamental requirement for hydration. The discovery of the means of artificially carbonating water by dissolution of CO<sub>2</sub> under pressure is attributed to Dr Joseph Priestley in the late 1760s, though there were many other workers active in this field at the same time which probably deserve equal credit.

Production of carbonated drinks was traditionally carried out by means of adding concentrated syrup to the bottle and then topping up with carbonated water. A considerable improvement in speed was achieved in 1937, when the Mojonner Brothers Corporation of Chicago introduced a continuous blending/cooling/carbonating system (Steen and Ashurst, 2008).

In general, two basic methods for carbonating a drink are possible: the injection and dispersion of carbon dioxide into the liquid to be carbonated, and the fine spraying of the product into a carbon dioxide atmosphere. However, in-line carbonation methods are being used increasingly. These either sparge carbon dioxide into the liquid or inject the liquid into a gas stream. When the gas is sparged into the liquid this allows small bubbles of gas to be formed which can be easily absorbed by the liquid. The higher the pressure

the smaller the gas bubbles formed at the sparger and the greater the gas bubbles surface area available for the gas to be absorbed by the liquid. In addition, also design of both the spargers and the housing of the sparger play an important role in optimal dissolution of gas into liquid. For this reason the aim of this work was to study and optimize different designs of CO<sub>2</sub> sparger for water or soft drink carbonation by means of CFD modelling.

## 2. MATERIALS AND METHODS

Two sparger geometries were studied and modelled by means of computational fluid dynamics technique.

The modelled sparger geometries were firstly drawn by means of SolidWorks 2013 and then 3D simulated by means of ANSYS 14 software with Boussinesq's approximation that uses a constant density for both fluids but considers gravity acceleration as a function of expansiveness and temperature of fluids. Sparger for carbonation presented a diameter of 64 mm and an height of 203 mm. the two geometries are reported in figure 1.

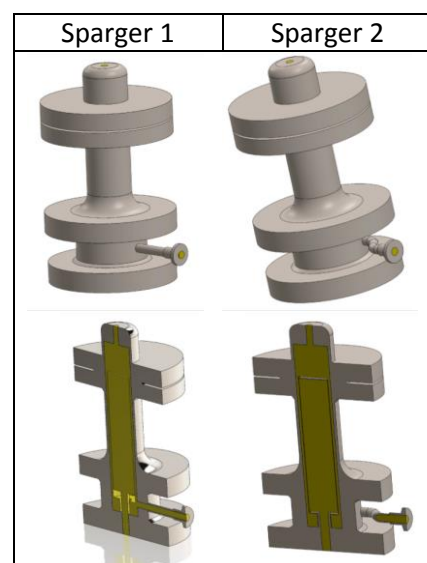


Figure 1: Radial (Sparger 1) and tangential (Sparger 2) sparger geometries



Flow rate of the food liquid was set at 16 L/h while carbon dioxide at 5.6 L/h. The incoming fluid was considered water while carbon dioxide was considered as the gas. Properties of fluids are reported in the following Table 1:

Table 1: Properties of modelled fluids

	CO <sub>2</sub>	Water
Molar mass [kgmol <sup>-1</sup> ]	44.01	18.02
Density [kgm <sup>-3</sup> ]	1.977	997
Dynamic viscosity [mWm <sup>-1</sup> K <sup>-1</sup> ]	0.0149	0.8899
Thermal expansivity [K <sup>-1</sup> ]	0.00366	0.000257

After modelling, water carbonation was experimentally tested on a pilot plant developed by Valfor srl and located in University of Parma laboratories. The pilot plant presented a total volume of about 2 L and a flow rate varying from 10 to 40 L/h. The equipment could work up to 200 bar and presented several water baths to have constant temperature both of carbon dioxide and product to be carbonated.

The carbonation evaluation was performed by means of image analysis: on the pipeline of the pilot three inspection units as reported in Figure 2 made of transparent polycarbonate were added. By means of a camera images of carbonated water were obtained and dimensions of gas bubbles were measured.



Figure 2: Inspection unit for bubble measurement

### 3. RESULTS AND DISCUSSIONS

As shown in Figure 3, tangential connection (Sparger 2) seemed to give a better mixing of gas and liquid at the

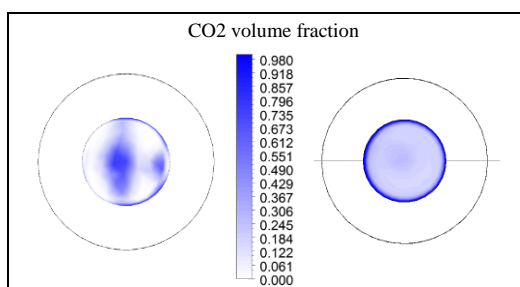


Figure 3: CO<sub>2</sub> volume fraction distribution in the outlet section

On the contrary, in Sparger 1, fluid streamlines were present with less homogenous dispersion at the outlet of the sparger.

In Figure 4, sections of both spargers are reported: as previously observed Sparger 1 presented streamlines with a less homogeneous mixing efficacy.

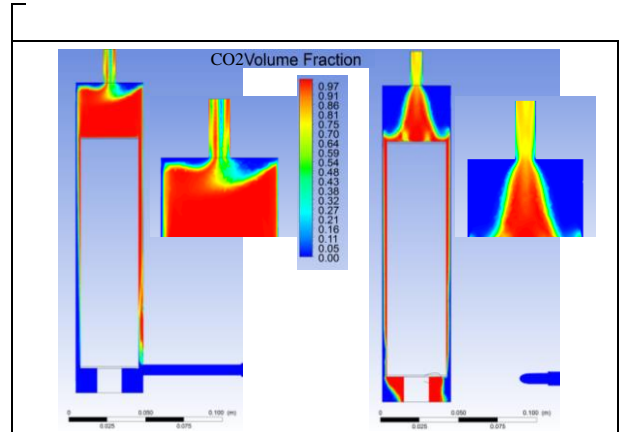


Figure 4: Radial (Sparger 1) and tangential (Sparger 2) sparger geometries results of modelling

As it's possible to see the housing for the sparger was the same while the angle of the inlet flux changed. The gas entered from the basis in the central channel and the liquid food from the lateral connection. Thus, mixing results depended only on fluid flow and relative movements between liquid and CO<sub>2</sub>

In order to quantify mixing efficacy, CO<sub>2</sub> volume on the radius of the outlet of the sparger was calculated and reported in Figure 5

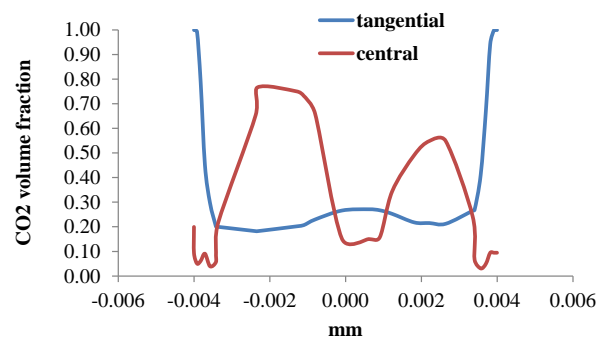


Figure 5: CO<sub>2</sub> volume fraction distribution in the outlet section

Data confirmed what previously observed: Sparger 2 gave a more homogenous distribution but unfortunately gave also a lower mean CO<sub>2</sub> volume fraction with a worse mixing effect. In Sparger 2 a great part of gas exited near the wall of the pipe without an actual dissolution into the liquid.

Picture of inspection element (Figure 6) confirmed the presence of gas bubble near the wall in Sparger 2 that were not mixed with liquid

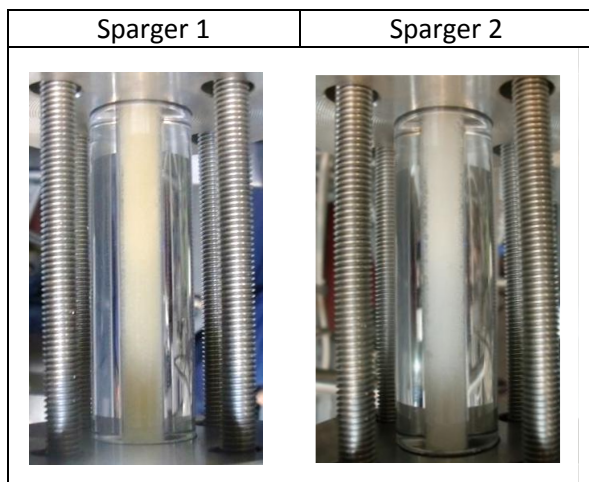


Figure 6: Bubbles inside the inspection unit

Also from experimental tests Sparger 1 gave better results with a better mixing between water and CO<sub>2</sub> and a more stable carbonated water, as consequence. Thus tangential inlet was not the best solution even if the liquid generated itself turbulence inside the housing and prevented fluid streamlines formation. Probably, turbulence inside the Sparger 2 and the spiral flux on the filter limited the spilling of CO<sub>2</sub>.

#### 4. CONCLUSIONS

In conclusion, this work demonstrated the effectiveness of mathematical model to predict fluid mixing also in complex systems and the usefulness of these models for designing and optimizing also gas sparging systems for the beverage industry.

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# ANALYSIS OF THE ENERGETIC AND EXERGETIC EFFICIENCY OF THE ELECTROHYDRODYNAMIC DRYING PROCESS

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## ABSTRACT

Drying is an energy intensive unit operation encountered in many industrial sectors, especially in the food industry. Still over 85% of industrial dryers are convective type. Electrohydrodynamic convective drying (EHD drying) is a novel drying method used to enhance forced convection drying by using electrodes to create an electrostatic field and generate an electric wind. This latter may alter the boundary layer and enhance the heat and mass transfer. In this study, experiments were performed to analyze the drying kinetics during EHD and forced convection (FC) drying experiments. Transient energy and exergy efficiencies expressions were discussed, proposed and computed for each experiment. With airflow of 0.3 m/s in the case of EHD configurations, similar drying rates than FC at 1.0 - 2.0 m/s can be achieved. Moreover, it led to greater energy efficiency (x5) and it was confirmed, using exergy efficiency concept, that EHD better used energy than FC.

Keywords: ElectroHydroDynamic, Drying, Convection, Energy, Exergy

## 1. INTRODUCTION

Drying is an energy intensive unit operation encountered in many industrial sectors, especially in the food industry. It offers many benefits including: extended shelf-life, reduced packaging, storage, handling and transportation costs. Several developed countries have reported that between 12 – 20% of their national industrial energy consumption is due to thermal dehydration operations (Mujumdar, 2014). Still, over 85% of industrial dryers are convective type with hot air or combustion gases as the heat transfer medium (Moses and Norton and Alagusundaram and Tiwari, 2014). Several innovations have been developed to improve product quality and energy efficiency. Forced convective drying (FC drying) uses high air velocity which leads to high energy consumption. One innovative approach to reducing this energy consumption is through ElectroHydroDynamic drying (EHD drying). The main principle behind EHD drying is to generate a corona discharge by one or more electrodes placed in an air stream in order to disrupt the

primary airflow. The ionic air stream significantly alters the boundary layer, and intensifies convective heat transfer between the air and the drying product for low primary airflow velocities. Recent work focus on the influence of operating parameters on energy consumption (Ould Ahmedou and Rouaud and Havet, 2009, Bai and Hu and Li, 2011). Some studies have investigated the EHD drying effect on specific food products (Taghian Dinani and Havet, 2015) and pointed out its interest.

An effective drying process should account for the sustainability issue (Amantea and Fortes and Martins and Ferreira, 2011). Combining energy and exergy analysis is powerful for modeling and optimizing food drying. In a pioneering work, the usefulness of exergy analysis in the thermodynamic assessment of drying processes was demonstrated (Dincer and Sahin, 2004). Based on this approach, an expression for exergetic efficiency was adapted to EHD Drying (Bardy and Hamdi and Havet and Rouaud, 2015). In this work, both energy and exergy analysis are performed on a EHD drying process and the influence of operating parameters is discussed.

## 2. EXPERIMENTAL PROCEDURE

### 2.1. Experimental set-up

The set-up (Figure 1) consisted of a rectangular airflow channel (15 cm x 19 cm x 200 cm) used to perform drying experiments by means of forced convection with and without an electrostatic field (i.e. EHD and FC drying).

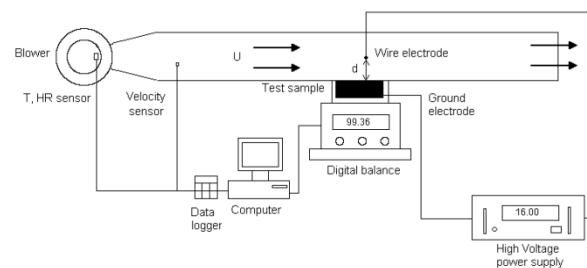


Figure 1: Experimental set-up

The mass loss was measured during drying using a weight scale (Radwag, PS600, Germany). The channel was connected to an air handling unit (ML180, Munters, France) used to control the psychrometric conditions, and the flow rate of the air entering the channel. The averaged temperature and relative humidity at the inlet of the airflow channel were  $T_i=28.9\pm0.5$  °C and  $RH_i=16.8\pm0.7\%$ . A wire-electrode(s) (diameter = 150  $\mu\text{m}$ ) suspended above the test specimen was connected to a high voltage generator (GLHT2260R2, Sefelec, France) in order to produce an electrostatic field. Details on this set-up and the methods are available (Bardy and Hamdi and Havet and Rouaud, 2015). From a preliminary experimental work on convective heat transfer enhancement (Hamdi 2014), a total of three different wire-electrode configurations were used for EHD drying in this study:

- 1 wire-electrode aligned perpendicular to airflow (Perp.),
- 1 wire-electrode aligned parallel to airflow (Paral. 1),
- 2 wire-electrodes arranged parallel to each other and aligned parallel to airflow (Paral. 2).

For each configuration, an applied voltage  $V=16$  kV and an airflow velocity of 0.3 m/s were considered. The evolution of the electric current  $I$  was recorded over the drying experience. The wire-electrodes were placed according to the positions that resulted in the greatest increase in heat transfer (Hamdi 2014). Table 1 shows the exact arrangement of each configuration. For Forced Convection cases (FC drying), the airflow velocity were 1 and 2  $\text{m.s}^{-1}$ .

Table 1: Dimensional placement of wire-electrodes for EHD drying experiments

Electrode arrangement	Perpendicular	Parallel	Parallel
Number of electrodes	1	1	2
Gap between electrode and surface (cm)	6	4	4

For this study, methylcellulose gel was chosen to be used as a test specimen because its thermophysical properties are close to those of meat it and can be controlled to a specific moisture content. Dimensions of test specimens were 15 x 15 x 1 cm and the moisture content was fixed at 80% (i.e. moisture content on dry basis:  $X_{db}=4$  kg w/kg dm). Average test specimen mass was  $231.33 \text{ g} \pm 8.82 \text{ g}$ .

## 2.2. Drying kinetics

The evolution of mass of the product  $m_p$  over time was used to compute the moisture content on dry basis over time as well as the drying kinetics (1) and (2).

$$X_{db}(t) = \frac{m_p(t)}{m_{dc}} - 1 \quad (1)$$

$$\frac{dX_{db}}{dt}(t) = \frac{X_{db}(t_i) - X_{db}(t_{i-1})}{t_i - t_{i-1}} \quad (2)$$

## 3. ENERGY AND EXERGY EFFICIENCIES

As the moisture in the the drying product is assumed to be in a saturated liquid water state at the drying product temperature (i.e. product temperature,  $T_p$ ), the liquid water is then evaporated to a saturated vapor state at the temperature of the air at the inlet of the airflow channel ( $T_i$ ). It is further assumed that  $T_p = T_i$ . Mass, energy and exergy balances for the system (Figure2) were performed by Hamdi (2014).

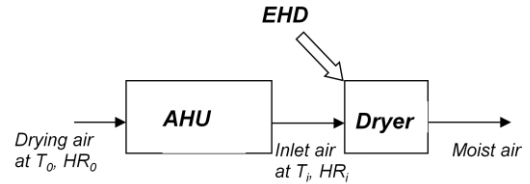


Figure 2: Sketch of Air Handling Unit and Dryer

### 3.1. Energy efficiency

Since the previous assumption, the energy use rate to evaporate moisture from the drying product is trivial (3). When computing energy efficiency, large differences may occur due to the dependence with the type of process used for conditioning the air. To avoid this effect, some author (Kudra 2012) calculated the energy performance by considering the energy used to heat/dry the air. It is possible to calculate this specific energy consumption by the difference between the enthalpy available in the drying air and a reference state (Amantea and Fortes and Martins and Ferreira, 2011). Nevertheless, this approach overestimates the energy efficiency because it completely makes abstraction of technology constraints. We propose in this study to calculate a realistic energy for the AHU based on the minimal energy required to deliver air from the standard conditions ( $T_o, RH_o, \omega_o$ ) to the measured inlet conditions in the dryer ( $T_i, RH_i, \omega_i$ ). We theoretically calculate that air should be dehumidified from the standard conditions ( $T_o, \omega_o$ ) to an intermediate stage ( $T_o, \omega_i$ ) and then heated to the inlet conditions ( $T_i, \omega_i$ ). The energy required to treat the drying air is then broken in two terms (4). This method minimizes the energy consumption in the Air Handling Unit because the dehumidification would require a higher cooling power. Using the energy use rate to evaporate moisture (3), the energy supplied by the AHU (4) and the energy supplied by the EHD system (5) we calculated a transient energetic efficiency (6).

$$\dot{Q}_{vap} = \dot{m}_w L_v \quad (3)$$

$$\dot{Q}_{air} = \dot{m}_{da} |h_{interm} - h_o| + \dot{m}_{da} (h_i - h_{interm}) \quad (4)$$

$$\text{with } h = h_{da} + \omega h_v$$

$$\dot{Q}_{EHD} = VI \quad (5)$$

$$\dot{\eta}_{en}(t) = \frac{\dot{Q}_{vap}}{\dot{Q}_{air} + \dot{Q}_{EHD}} \quad (6)$$

### 3.2. Exergy efficiency

An expression for exergetic efficiency as a function of time (i.e., transient exergetic efficiency) was defined as the ratio of the exergy use rate to evaporate moisture from the drying product ( $\dot{\psi}_{use}$ ) over the exergy supply rate ( $\dot{\psi}_{supply}$ ) as shown in equation (7). The exergy use rate to evaporate moisture from the drying product ( $\dot{\psi}_{use}$ ) is shown in equation (8). The exergy supply rate due to the air (the state difference between the air at the inlet of the drying tunnel and the dead state ( $\dot{\psi}_{air}$ ), is shown in equation (9). It is broken down into three components: thermal, mechanical and chemical. Contrary to the definition of the energy supply rate for air, this definition is shared by the energy community. The exergy supply due to the EHD effect ( $\dot{\psi}_{EHD}$ ) is shown in equation (10).

$$\dot{\eta}_{ex}(t) = \frac{\dot{\psi}_{use}}{\dot{\psi}_{air} + \dot{\psi}_{EHD}} \quad (7)$$

$$\dot{\psi}_{use} = \left(1 - \frac{T_0}{T_p}\right) \dot{Q}_{vap} \quad (8)$$

$$\dot{\psi}_{air} = \dot{m}_{air} \left[ \begin{aligned} & \left( C_{p,a} + \omega_i C_{p,v} \right) \left( T_i - T_0 - T_0 \ln \left( \frac{T_i}{T_0} \right) \right) \\ & + T_0 (R_a + \omega_i R_v) \ln \left( \frac{P_i}{P_0} \right) \\ & + T_0 \left\{ \begin{aligned} & (R_a + \omega_{in} R_v) \ln \left( \frac{1 + 1.6078 \omega_0}{1 + 1.6078 \omega_i} \right) \\ & + 1.6078 \omega_i R_a \ln \left( \frac{\omega_i}{\omega_0} \right) \end{aligned} \right\} \end{aligned} \right] \quad (9)$$

$$\dot{\psi}_{EHD} = VI \quad (10)$$

For each case, the dead state was assumed to be  $T_0 = 20$  °C,  $P_0 = 101.3$  kPa, and  $RH_0 = 50\%$  (standard conditions as stated by ISO 5011). It should be noticed that the overall energetic and exergetic efficiencies, can be determined by integrating equations (3) to (10).

## 4. RESULTS AND DISCUSSION

### 4.1. Drying kinetics

Figure 3 presents the variation of the moisture content  $X_{db}$  of the drying product as a function of time. As expected, the moisture content decreased over time for each experiment. It can be seen that the EHD cases are between the FC drying at 1 m/s and the FC drying at 2 m/s. This latter provides the greatest decrease. For the EHD cases, one wire-electrode parallel to airflow had a greater moisture loss over time than other cases and a moisture loss equivalent to that of FC at 2.0 m/s. It corresponds to the best case also found by Hamdi (2014) who look at the convective heat transfer enhancement. It is noted that the moisture content of all cases would converge if drying continued towards the point of equilibrium moisture content.

It is also interesting to compare the drying kinetics using the evolution of the drying rate (2) according to the moisture content (figure 3). This type of representation is known as Krischer curve. From the initial moisture content to a value of 3.5 g/w/g dm, there was a fluctuation in drying rate with an initial decrease in drying rate followed by an increase. The drying rate then decreased steadily for the remainder of the experiment. FC at 2.0 m/s resulted in the greatest initial fluctuation in drying rate. This initial decrease and then increase was also present for EHD cases although not as pronounced. This initial decrease of the drying rate is attributed to the formation of a thin crust during the gel preparation. The water diffusion across this layer is reduced. The increase is more conventional, it corresponds to a thermal effect. As the product was colder than air, the surface temperature increased to the air wet-bulb temperature and provokes an enhancement of diffusion phenomena. The maximum values of the drying rate correspond to the period when the heat transfer by convection is used to evaporate water at the surface. As drying continues, moisture, entrapped within the product, diffuses towards the surface. This diffusion effect becomes more significant as time progresses and causes the drying rate to decrease. As previously, one can distinguish that the one wire-electrode parallel to airflow is equivalent to that of FC at 2.0 m/s and quite higher than other cases. A deeper analysis of drying phenomena may be performed by measuring shrinkage phenomena and quality attributes. Nevertheless, at this stage, it is important to consider that similar drying were obtained by convection with an airflow velocity at 0.3 m/s assisted by EHD and by forced convection at 2 m/s. The benefit of EHD drying should be discussed from an energetic point of view.

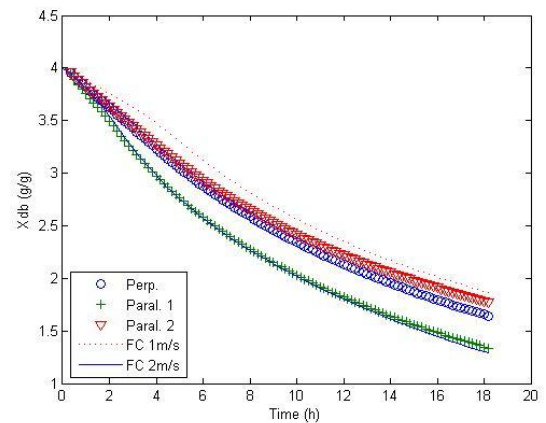


Figure 3: Moisture content as a function of time for each EHD and 2 FC drying cases

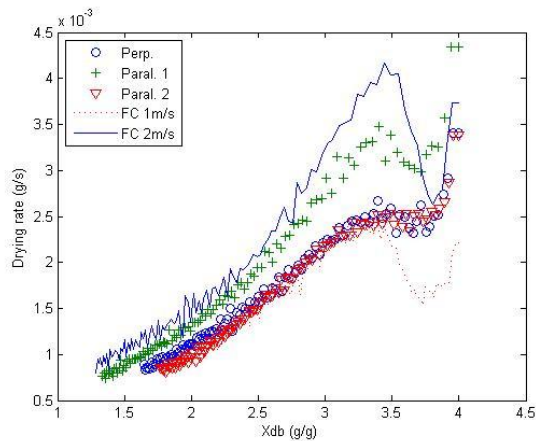


Figure 4: Drying rate as a function of Moisture Content for each EHD and 2 FC drying cases

#### 4.2. Energy and exergy efficiencies

The evolution of the energetic efficiency during the time is depicted on figure 5. It should be firstly mentioned that the maximum energy efficiency is only close to 5%. This value could be considered as extremely low compared to results from literature (Amantea and Fortes and Martins and Ferreira, 2011, Kudra 2012, Hamdi 2014). It is firstly merely due to the definition of the energy supply for air treatment. Considering another definition would increase the energy efficiency but it would not affect the comparison of each case. Moreover, the surface of the specimen was relatively small (15 x 15 cm) compared to the airflow rate and the specific humidity of air did not evolve a lot inside the channel.

It clearly appears that FC cases have very low energy efficiencies compared to EHD case. EHD is 4 to 5 times more efficient. This great difference is due to airflow rate because the velocity is only 0.3 m/s in EHD case whereas it is up to 2 m/s for FC drying. In the first hours, we observe fluctuations due to the evolution of the drying rates but after 5 hours drying, all EHD cases reach similar values of energy efficiency. This values decrease from 3 to 1 % during the drying time.

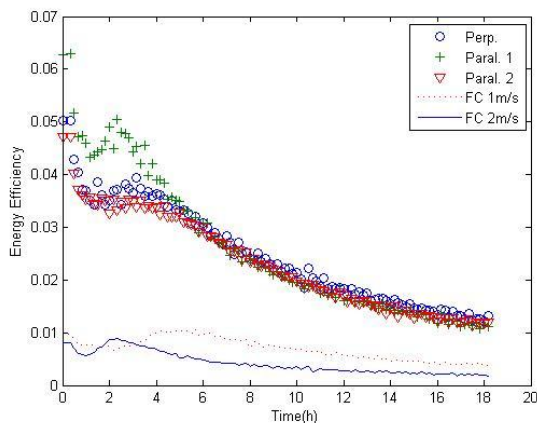


Figure 5: Energetic efficiency as a function of time for each EHD and 2 FC drying cases

The evolution of the exergetic efficiency during the time is depicted on figure 6. It is amazing to see that the values are quite similar to that of energy efficiency. They are in a classical order of magnitude found in literature with values lower than 10% (Dincer and Sahin 2004, Amantea and Fortes and Martins and Ferreira, 2011). The differences in exergetic efficiency between the EHD and FC cases can be attributed to the airflow velocity since the psychrometric inlet conditions of the primary airflow were held constant. The same dead state conditions were assumed for all experimental runs. It is noted that variations in the dead state assumption would lead to variations in computed exergetic efficiencies, but only in scale. For FC cases, exergetic efficiencies are still extremely low and increasing the air velocity reduces the efficiency. It is due to water diffusion that becomes the dominant mode and increasing convection is not relevant at all.

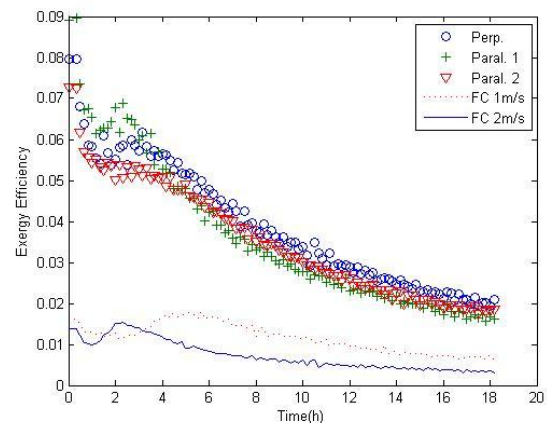


Figure 6: Exergetic efficiency as a function of time for each EHD and FC drying case

#### 5. CONCLUSION

Expressions for the transient energetic and exergetic efficiencies of EHD drying were presented. The energy efficiency considered a minimum energy required to deliver air. The exergy efficiency took into account the power consumed by the EHD effect. The drying experiments indicated, that in the case of the EHD configurations, the same drying rate for FC at 1.0 - 2.0 m/s can be achieved with an airflow of only 0.3 m/s, leading to greater energy and exergy efficiencies. It should also be mentioned that these promising results could possibly be optimized by working with a higher applied voltage. This is a major finding that will also contribute to the design of low energy consumption process for air conditioning.

All EHD drying configurations had significantly higher energetic and exergetic efficiencies than FC drying, especially at the beginning of the drying. The higher exergetic efficiencies were due to the lower airflow velocity associated with EHD drying.

## ACKNOWLEDGMENTS

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# OPTIMIZATION OF HYDRATION PROCESS OF DATE PALM FRUITS FROM EXPERIMENTAL AND NUMERICAL APPROACHES

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## ABSTRACT

This study focuses on the hydration process of dates from experimental and numerical investigations. Dry Tunisian Deglet Nour dates were hydrated at a laboratory scale by using saturated air. A theoretical model based on a finite element scheme was developed to describe mass transfer phenomena that occur during the hydration process. This model considers the real shape of dates with a 2D axisymmetric analysis. Both moisture diffusivity and convective mass transfer coefficient of dates were estimated by using an optimization algorithm based on a least square approach from experimental and numerical average moisture contents. Overall, the experimental moisture contents as a function of hydration time were found in good agreement with the simulated values for various operating conditions. Such a methodology can now be used as a predictive tool to simulate the hydration of dates in order to improve the quality of final product and reducing processing time.

Keywords: Dates hydration, modeling, moisture diffusivity, convective mass transfer coefficient

## 1. INTRODUCTION

In Tunisia, date palm (*Phoenix dactylifera* L.), especially the Deglet Nour Cultivar, constitutes an important food and financial source. The chemical composition, in particular moisture content, of Deglet Nour date palm fruits can vary depending on agronomic practices, climatic conditions as well as ripening stage (El Arem et al. 2010). During last years, there has been an increase in the proportion of dry dates in annual crops. These dry dates are processed in the industry in order to obtain fruits similar to those labeled as Extra category dates. Hydration represents the key unit operation of this process. Therefore, the control of this operation is of paramount importance especially when excessive times of hydration could reduce the shelf stability of dates and induce a waste of energy whereas insufficient hydration durations lead to non acceptable final product quality.

## 2. MATERIAL AND METHODS

### 2.1. Raw material

Deglet Nour dates were harvested in 2014 and stored at 4°C and 65% of relative humidity.

### 2.2. Experimental procedure

Hydration experiments were conducted at laboratory scale approaching industrial conditions by placing dates in a closed environment at atmospheric pressure and where air reaches relative humidity of 100% and inner temperatures between 50 and 65°C. To achieve this, a metallic enclosure was filled with water and heated with a temperature controlled hot plate. Dates were placed in the head space of the enclosure without any contact with the water. The level of heating was set in such a way that temperatures of surrounding air medium, which are recorded by thermocouples placed near the dates, remain in the range of desired temperatures during each experiment. The experimental system designed is presented in Figure 1. Average moisture contents of dates flesh were measured at regular intervals during hydration by a method modified from Singh et al. (2013) which consists of drying 3 g of dates in an oven at 105°C for at least 18h.

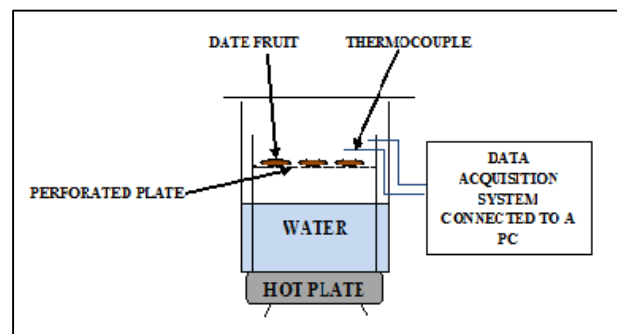


Figure 1: Laboratory Experimental System Designed for Dates Hydration

### 2.3. Modeling of hydration

Prior to modeling, monitoring of the temperatures within the dates flesh for the range of hydration times used for undertaking experiments showed that the



temperature can be considered as homogenous. Therefore a theoretical model based on a finite element scheme was developed to describe only mass transfer phenomena during the hydration operation based on Fick's law according to Eq.(1).

$$\frac{\partial C}{\partial t} = \nabla \cdot (D \nabla C) \quad (1)$$

Where  $C$  and  $D$  are the instantaneous molar concentration ( $mol/m^3$ ) and the diffusivity ( $m^2/s$ ) of moisture within the date flesh. Since that the transport phenomena are symmetric, a 2D axisymmetric domain was considered to represent date flesh as shown in Figure 2.

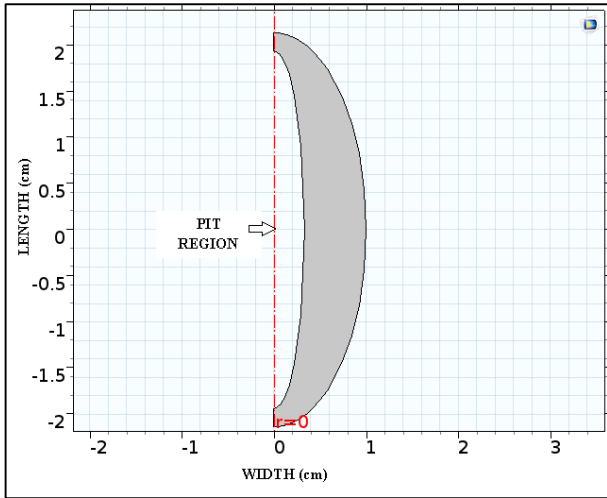


Figure 2: 2D Axisymmetric Domain

The initial and boundary conditions are as follows:

1. Uniform initial moisture molar concentration  $C_0$  which is calculated from Eq.(2).

$$C_0 = \frac{X_0 \rho_0}{M_w} \quad (2)$$

Where  $X_0$ ,  $\rho_0$  and  $M_w$  are the initial average date flesh moisture content on humid basis ( $kg/kg$ ), initial density of date flesh ( $kg/m^3$ ) and the molecular mass of water ( $kg/mol$ ).

$\rho_0$  is determined using Eq.(3).

$$\rho_0 = \frac{m_0}{v} \quad (3)$$

Where  $m_0$  and  $v$  are the initial mass ( $kg$ ) and the volume ( $m^3$ ) of date flesh.

2. Null mass flux at the surface in contact with the date pit.
3. Natural mass convection at the outer surface in contact with the saturated air. The mass flux  $N(mol.m^{-2}.s^{-1})$  of moisture inward date flesh is given by Eq. (4).

$$N = k_c(C_b - C_s) \quad (4)$$

Where  $k_c$ ,  $C_b$  and  $C_s$  are convective mass transfer coefficient ( $m/s$ ), bulk molar water vapor concentration ( $mol/m^3$ ) and molar water vapor concentration in the air adjacent to the outer surface ( $mol/m^3$ ).

$C_b$  and  $C_s$  are determined using respectively Eq.(5) et Eq.(6).

$$C_b = \frac{P_{vs}}{RT} \quad (5)$$

$$C_s = \frac{P_{vs}}{RT} a_w \quad (6)$$

Where  $P_{vs}$ ,  $R$  and  $a_w$  are saturated vapor pressure at the average absolute temperature  $T(K)$  of the air, ideal gas constant ( $8.314 J.mol^{-1}.K^{-1}$ ) and water activity of date flesh. Neglecting sorption hysteresis phenomenon,  $a_w$  was defined using the GAB model in Eq.(7) (Kechaou and Mâalej 1999; Kechaou and Mâalej 2000).

$$X = \frac{X_m C k a_w}{(1 - k a_w)(1 - k a_w + C k a_w)} \quad (7)$$

Where  $X$  is the instantaneous moisture content on dry basis( $kg/kg$ ).  $X_m$ ,  $C$  and  $k$  are the constants of the model.

$k_c$  is determined initially using the correlation for natural mass transfer coefficient around a sphere according to Eq.(8) (Cussler 2009).

$$Sh = \frac{k_c D_{eq}}{D_{w,a}} = 2 + 0.6(Gr)^{\frac{1}{4}}(Sc)^{\frac{1}{3}} \quad (8)$$

Where  $Sh$ ,  $Gr$  and  $Sc$  are Sherwood, Grashof and Schmidt numbers.  $D_{eq}$  is the surface-equivalent sphere diameter of the date ( $m$ ).  $D_{w,a}$  is the diffusivity of the water vapor in air ( $m^2/s$ ) which is estimated using Eq.(9) (Bolz and Tuve 1976).

$$D_{w,a} = -2.775 \times 10^{-6} + 4.479 \times 10^{-8}T + 1.656 \times 10^{-10} T^2 \quad (9)$$

The model was implemented in the software COMSOL®MultiPhysics release 5.1 which solves partial differential equations by using the finite element method. The computational domain is meshed using triangular elements. Date flesh moisture distribution is computed during hydration and the average moisture concentration is calculated as a function of time with the average coupling operator.

Then, by using the optimization module from COMSOL®, both moisture diffusivity and convective mass transfer coefficient at the surface were estimated by minimizing the least-square objective function

calculated from experimental and numerical mean moisture contents. Experimental values were implemented in COMSOL® using equations Eq.(10) and Eq.(11) similar to Eq.(2) and Eq.(3).

$$C_t = \frac{X_t \rho_t}{M_w} \quad (10)$$

Where  $X_t$  and  $\rho_t$  are the instantaneous average date flesh moisture content on humid basis ( $kg/kg$ ) and density of date flesh ( $kg/m^3$ ).  $\rho_t$  is determined using Eq.(11).

$$\rho_t = \frac{m_t}{v} \quad (11)$$

Where  $m_t$  and  $v$  are the instantaneous mass ( $kg$ ) and the volume ( $m^3$ ) of date flesh which is considered constant and equal to the arithmetic mean of initial and hydrated date.

### 3. RESULTS

Although experiments were conducted for several types of Deglet Nour dates, only results for one type of dates which are slightly harder and drier than labeled Extra category dates are presented. The molar moisture concentration distribution is computed at times when average moisture contents were measured experimentally. Figure 3 illustrates molar moisture profile within the flesh of one date after 14640s ( $\approx 4h$ ) of hydration at the average temperature of the air during hydration.

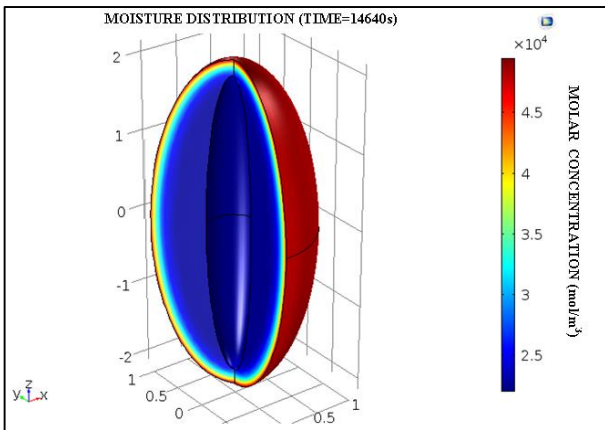


Figure 3: Moisture Concentration Distribution Within Date Flesh After 4h

As mentioned above, the mean water concentration is then computed. The estimation of both moisture diffusivity and mass transfer coefficient by minimizing the least square objective function using Levenberg-Marquardt optimization solver was undertaken as follows:

1. Effective mass diffusivity is estimated by taking initial mass transfer coefficient determined from the Eq.(8).

2. By considering the value of mass diffusivity found from the first run, a second optimization is performed to estimate the mass transfer coefficient.

Table 1 shows the estimated diffusivities and convective mass transfer coefficient for two dates from the considered type.

Table 1: Estimated Moisture Diffusivities and Mass Transfer Coefficients

Date Number	Date 1	Date 2
$D(m^2/s)$	5.54E-11	5.54E-11
$kc (m/s)$	0.00123	0.00126

The values of estimated convective mass transfer coefficient which depend on the dimensions of dates are consistent with the values found using the correlation for natural convection (Eq.8).

Figure 4 and Figure 5 represent experimental and calculated moisture molar concentration data as a function of time using estimated mass diffusivity and convective coefficient.

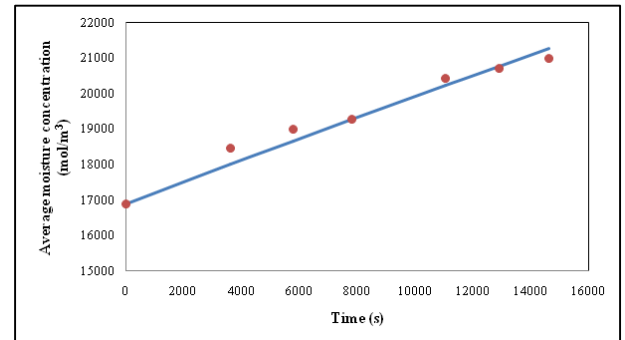


Figure 4: Experimental vs Calculated Average Moisture Concentration Of Date1

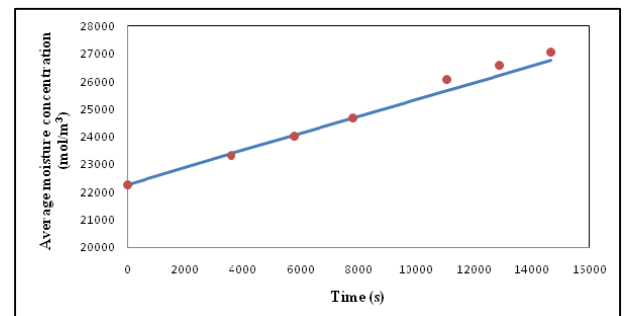


Figure 5: Experimental vs Calculated Average Moisture Concentration Of Date2

It is shown that there is a good agreement between calculated and experimental average moisture molar concentrations.

The root mean square difference (RMS) is also calculated from the Eq. 12 (Kechaou and Maâlej 2000).

$$RMS = \sqrt{\frac{\sum_i \left( \frac{C_{av_i,exp} - C_{av_i,cal}}{C_{av_i,exp}} \right)^2}{n-1}} \cdot 100 \quad (12)$$

Where  $Cav_i$  is the average moisture molar concentration at the  $i$ -th experimental point.

Table 2 confirms the accuracy of the modeling since that root mean square differences (RMS) do not exceed the value of 1.39%.

Table 2: Root Mean Square Differences Between Experimental and Calculated Curves

Date Number	Date 1	Date 2
RMS (%)	1.39	0.97

It is also seen that the phase of decrease in moisture uptake rate is not observed. It could be explained by the relatively short times of hydration (in comparison to maximum processing times in industry).

The estimated moisture diffusivities and convective coefficients shown in Table 1, Figure 4 and Figure 5 indicate that these values do not vary greatly with the initial average moisture concentration. Therefore, this model can be used to predict times necessary to reach desired final moisture contents by taking into account the variability of raw material for this type of dates.

#### 4. CONCLUSION

Hydration is the key unit operation in the industrial thermal process of dates. However there are a scarce data about the capacity of moisture uptake during hydration. In this work, a theoretical model was developed to describe mass transfer phenomena involved in this operation. The model enables to estimate moisture diffusivity and convective mass transfer coefficient using an investigation performed experimentally. This model can be used to simulate hydration of dates in order to predict and optimize this operation. In addition, the model could be used to optimize the maximum time of hydration. For current industrial applications, this operation often lasts during more than 4h. The proposed modeling approach may thus help for the optimization of this operation in order to enhance the mass transfer at the surface and to reduce the energy consumption during hydration.

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# POROVISCOELASTIC MODELING OF PROTEIN HYDROGELS

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## ABSTRACT

Many solid food matrices contain high amounts of solvent, typically water. Hence, the structural behavior depends on its biphasic nature. The time dependent mechanical characterization of foods and hydrocolloid-based solids has typically been analyzed following a viscoelastic approach, omitting the effect of the solvent. However, in solvent rich solids the solvent flows internally as it is compressed, for example, which is typically understood using poroelastic theory. A poroelastic approach allows the determination of parameters, such as the Darcy's diffusivity or the intrinsic permeability, that have a physical meaning. The application of poroelasticity to materials has been traditionally limited due to the complex data analysis. Recently, it has been proposed for polymeric hydrogels a novel experimental methodology, based on relaxation after indentation that greatly simplifies the subsequent analysis. This methodology is applied here for the first time to a complex food-like matrix, to heat induced whey protein hydrogels.

Keywords: poroelasticity, finite elements analysis, hydrogel, whey proteins.

## 1. INTRODUCTION

Solvent rich soft materials, like hydrogels (Chan et al., 2012b) and cells (Trappmann et al., 2012), have recently been characterized using a novel poroelastic relaxation indentation methodology (Hu et al., 2010). The hydrogel, at swelling equilibrium, is indented a small fixed depth using a rigid probe. The relaxation force is measured with time and is subsequently analyzed using correlations obtained from poroelastic finite element analysis. The power of this novel methodology, despite the experimental simplicity, is that several material properties including the shear modulus, diffusion coefficient, average pore size and even the Flory-Huggings interaction parameter, could be obtained by a single experiment (Hu et al., 2011b).

This novel methodology, developed and tested initially for synthetic polymeric systems, has been recently expanded to biomaterials such as gelatin and agar (Strange and Oyen, 2012). We apply it here for the first

time to protein hydrogels, and to whey proteins in particular due to their extensive use in research and commercial applications (Mercadé-Prieto et al., 2008). These initial tests were performed in a typical food lab rheometer with good normal force resolution, where the main limitation was that only one cylindrical indenter was available (Fig. 1). Subsequently it is summarized the mechanical analysis of indentation-relaxation tests with a cylindrical punch.

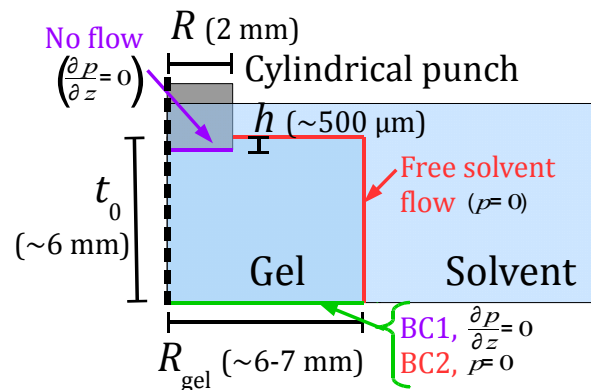


Figure 1. Schematic axisymmetric representation of a swollen protein hydrogel indented with a cylindrical punch. In the finite element analysis it was considered no solvent flow in the area in contact with the indenter (zero solvent flux in the vertical direction,), and free flow (zero pore pressure at the gel surface,  $p = 0$ ) in the non-contact area. For the bottom area, that in contact with the substrate, two boundary conditions are considered: BC1 for no flow, and BC2 for diffusive equilibrium.

## 2. INDENTATION ANALYSIS

### 2.1. Elastic loading with a cylindrical punch

The force at an indentation depth  $h$  using a flat punch indenter in an elastic (E) sample (Fig. 1) is (Hu et al., 2010):

$$F_E = 8 G_E R h \Pi(R/t_0, \nu) \quad (1)$$

where  $G_E$  is the instantaneous shear modulus in a linear elastic material;  $R$  is the radius of the indenter,  $\nu$  is the instantaneous Poisson's ratio, and  $\Pi(R/t_0, \nu)$  is a dimensionless function needed to correct substrate effects, important for ratios of the indenter radius to the sample thickness  $R/t_0 > 0.1$  (Cao et al., 2009). After loading, the indentation depth  $h$  is kept constant, resulting in a constant contact area. Because the indentation step is quick compared to the subsequent relaxation, it is typically treated the hydrogel as incompressible during loading ( $\nu = 0.5$ ).

## 2.2. Poroelastic (PE) relaxation

A purely elastic hydrogel does not show a force decrease with time  $t$  after indentation. A force relaxation in a hydrogel is typically due to the flow of the interstitial fluid through the polymer network – poroelasticity (PE) – and due to the intrinsic viscoelasticity (VE) of the polymer network (Strange et al., 2013). The force relaxation after indentation is first considered for a poroelastic material. Hu et al. (2010) have shown that the normalized relaxation force  $F_{PE}(t)$  is only a function of the normalized time  $\tau_{PE}$ :

$$\frac{F_{PE}(t) - F_{PE}(\infty)}{F_E - F_{PE}(\infty)} = f_{PE}(\tau_{PE}) \quad (2)$$

where  $F_{PE}(\infty)$  is the force in the long-time limit. In a cylindrical punch,  $\tau_{PE} = Dt/R^2$ ; where  $D$  is Darcy's law diffusivity. It is assumed that both the solvent and the solid matrix are incompressible. The function  $f_{PE}(\tau_{PE})$  is found computationally for different indenter shapes and hydrogel configurations, such as different  $R/t_0$  ratios (Hu et al., 2011a). The function  $f_{PE}(\tau_{PE})$  is usually fitted to a series of exponential functions such as:

$$f_{PE}(\tau_{PE}) = \sum_{i=1}^{N_{PE}} A_i e^{(B_i \tau_{PE}^{C_i})} \quad (3)$$

where  $A_i$ ,  $B_i$ , and  $C_i$  are regression constants, and  $N_{PE}$  is an integer typically 2 to 3. The extent of relaxation in PE is determined by the Poisson's ratio of the drained hydrogel,  $\nu_d$  (Hu et al., 2010):

$$\frac{F_{PE}(\infty)}{F_E} = \frac{1}{2(1 - \nu_d)} \quad (4)$$

Poroelastic properties of hydrogels are also typically described using the intrinsic permeability  $k$ :

$$k = \frac{(1 - 2\nu_d)D\eta}{2(1 - \nu_d)G} \quad (5)$$

## 2.3. Viscoelastic (VE) relaxation

The viscoelastic relaxation of a material is usually described empirically using a series of exponentials, known as Prony series, of the shear modulus (Cao et al., 2009):

$$G_{VE}(t) = G_E \left( 1 - \sum_{j=1}^{N_{VE}} g_{VEj} (1 - e^{(-t/\tau_{VEj})}) \right) \quad (6)$$

where  $g_{VEj}$  and  $\tau_{VEj}$  are the relative shear modulus and the relaxation time of the viscoelastic deformation, and  $N_{VE}$  is an integer typically between 1 to 4. The normalized relaxation force with time due to viscoelasticity is therefore

$$\frac{F_{VE}(t)}{F_E} = 1 - \sum_{j=1}^{N_{VE}} g_{VEj} (1 - e^{(-t/\tau_{VEj})}) \quad (7)$$

## 2.4. Poroviscoelastic (PVE) relaxation

Both relaxation methods, PE and VE, can occur simultaneously resulting in a poroviscoelastic (PVE) relaxation. Both phenomena have been suggested that can be decoupled to compute the overall PVE relaxation (Strange et al., 2013), such as:

$$F_{PVE}(t) = \frac{F_{VE}(t)F_{PE}(t)}{F_E} \quad (8)$$

The extent of relaxation in PVE is expected to be:

$$\frac{F_{PVE}(\infty)}{F_E} = \left( 1 - \sum_{j=1}^{N_{VE}} g_{VEj} \right) \left( \frac{1}{2(1 - \nu_d)} \right) \quad (9)$$

In this paper we implement the above relaxation models to swollen protein hydrogels and discuss their validity.

## 3. INDENTATION OF PROTEIN HYDROGELS

### 3.1. Whey protein hydrogels

Heat-induced hydrogels were formed as reported previously (Mercadé-Prieto et al., 2007b), using commercial whey protein isolate powder (BiPro, Davisco, USA). Protein solutions at 15 wt%, with 0.1% sodium azide, were heated inside plastic test tubes for 20 min at 80°C. Gels were stored overnight at 4°C prior being cut in ~6 mm height cylinders, which were then submerged in water solutions at either 0.1, 0.2 or 0.3 M NaCl. The swelling ratio  $SR = (m_{sw}/m_0 - 1)$  was calculated from the gels mass difference after swelling equilibrium was achieved (Mercadé-Prieto et al., 2007b).

### 3.2. Indentation

The swollen hydrogels were placed during testing on a glass petri dish with the corresponding salt solution. Indentation was performed using a rheometer with good normal force resolution, <0.001 N (Malvern Kinexus, UK), with a 2 mm cylindrical indenter. The indenter was placed close to, but not in contact with, the hydrogel surface. Thereafter, the gel was compressed by modifying the gap; the actual indentation depth  $h$  was determined a posteriori during the loading analysis.

Gels were allowed to relax for one hour or until no force change was noticeable.

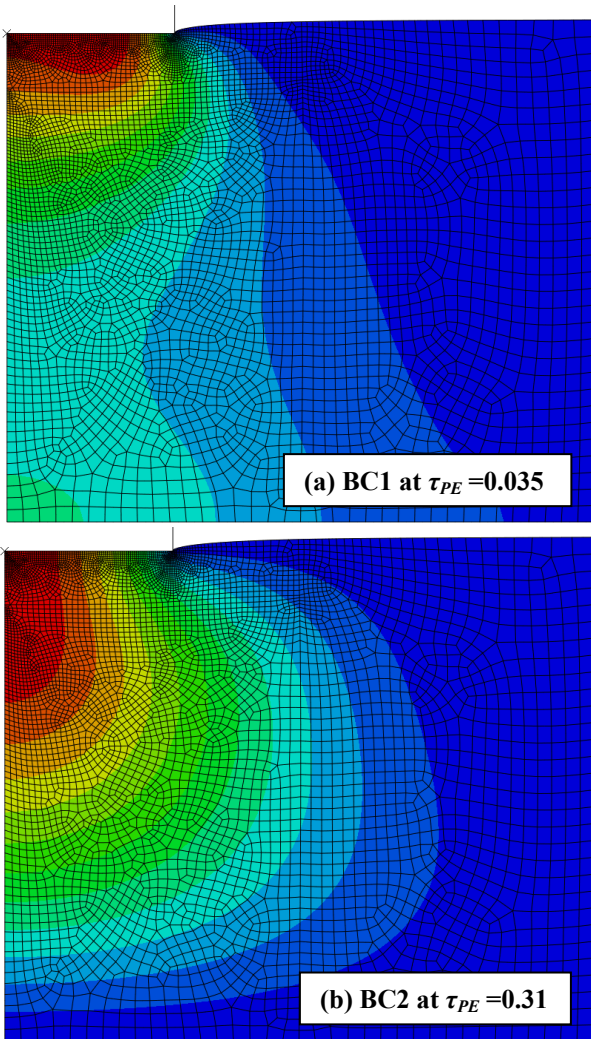


Figure 2. Finite element simulations of the pore pressure inside a swollen hydrogel using  $v_d = 0$ .

#### 4. FINITE ELEMENT ANALYSIS OF INDENTATION RELAXATION

Finite element analysis of the indentation problem shown in Fig. 1 was performed as described by several previous researchers (Hu et al., 2011b; Lin and Hu, 2006; Strange et al., 2013). The main novelty in the present work is that we model small hydrogels using a flat cylindrical indenter, and that we verify PVE eq. 8 using finite elements also for a flat indenter. Although tested gels had some size variations in thickness and diameter, for instance due to the different extent of swelling, the variations were too small to justify using different finite element geometries for each particular gel. For example, the key parameter  $R/t_0$  was  $0.32 \pm 0.02$  (SD) for 0.1 M NaCl gels,  $0.32 \pm 0.01$  at 0.2 M NaCl, and  $0.34 \pm 0.09$  at 0.3 M NaCl, hence an average value of  $R/t_0 = 0.333$  was considered in the simulations. In the same way, the diameter of the simulation hydrogel was chosen as  $R/R_{gel} = 0.286$ . The indentation depth simulated was very small,  $h/t_0 < 0.03$ , although

results were highly insensitive to  $h$  up to large values after normalization.

In previous studies, it was assumed no solvent flow through the bottom of the hydrogel, this boundary condition is termed BC1 here (see Fig. 1). In BC1, the pore pressure does increase with time at the gel bottom, shown in Fig. 2(a). This boundary conditions seemed unlikely in our experiments. Hence, simulations were also performed allowing solvent flow in all directions except the area in contact with the rigid indenter (termed model BC2), Fig. 2(b). PE simulations were fitted to eq. 3 with  $N_{PE} = 2$ ; that equation was then used to calculate  $D$  from normalized experimental relaxations, as reported in the literature. For PVE analysis, using  $N_{VE} = 1$ , three parameters were optimized:  $D$ ,  $g_{VE1}$  and  $\tau_{VE1}$ . The parameter  $D$  was optimized using the simplex search method, function *fminsearch* in Matlab, whereas the last two parameters were optimized within fixed bounds, using the function *fminbnd* in Matlab, in order to assure convergence and a physical meaning ( $0 < g_{VE1} < 0.25$ ;  $0.5 \text{ s} < \tau_{VE1} < 100 \text{ s}$ ). The parameters were optimized one at a time, iteratively, until the mean square error (MSE) did not improve further. All the numerical analysis was performed in Matlab.

#### 5. RELAXATION SIMULATIONS

##### 5.1. Poroelastic simulations

The normalized force for PE given in eq. 2 is very useful to determine the solvent diffusivity  $D$  because the shape of  $f_{PE}(\tau_{PE})$  does not change with the other material parameters, such as  $G_E$  and  $v_d$ , or the extent of indentation  $h$ . However, the shape of  $f_{PE}(\tau_{PE})$  does depend on size of the indenter in proportion to the hydrogel size,  $R/t_0$ , as shown elsewhere for a spherical indenter (Hu et al., 2011a). Figure 3 shows the finite element analysis results for the testing geometry shown in Fig. 1, for the two boundary conditions considered, which is faster compared to a semi-infinite sample, as shown from the results by Hu et al. (2010).

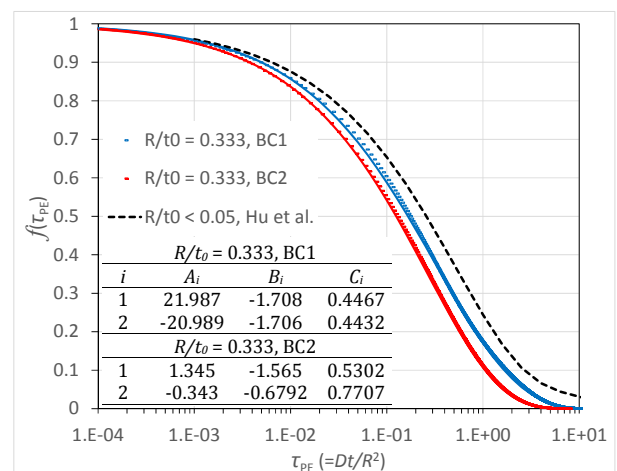


Figure 3. Finite element simulations for the PE relaxation for the geometry shown in Fig. 1, and comparison with the results of Hu et al. (2010) for a

semi-infinite hydrogel. Continuous lines are the best fit exponential regressions eq. 3; inset table shows the regression coefficients for  $N_{PE} = 2$ .

## 5.2. Poroviscoelastic simulations

The fitting of the experimental relaxation curves in PVE was performed in analogy to that of PE, after normalizing the force as  $(F(t) - F(\infty))/(F_E - F(\infty))$ . However, if the shape of PE  $f_{PE}(\tau_{PE})$  only depends on  $D$  for a fixed indenter and sample geometry, in PVE it also depends on the VE parameters  $g_{VEj}$  and  $\tau_{VEj}$ . Here, only one exponential term was needed to model the additional viscoelastic relaxation, i.e.  $N_{VE} = 1$ . In addition, the PVE relaxation profile also depends on the drained Poisson's ratio  $\nu_d$ , as it determines the extent of the PE relaxation, see Fig. 4. The drained Poisson's ratio used for PVE fitting was that calculated from the PE analysis; small  $\nu_d$  modifications due to  $g_{VE1}$  during the overall fitting process were not considered for simplicity. The decoupled eq. 8 can describe the PVE relaxation satisfactorily in the experimental range of  $0.2 > \nu_d > -0.2$  (Table 1); and the final  $F_{PVE}(\infty)/F_E$  using eq. 9 was also well predicted from simulations for the typical parameters considered here, with errors  $<3\%$ .

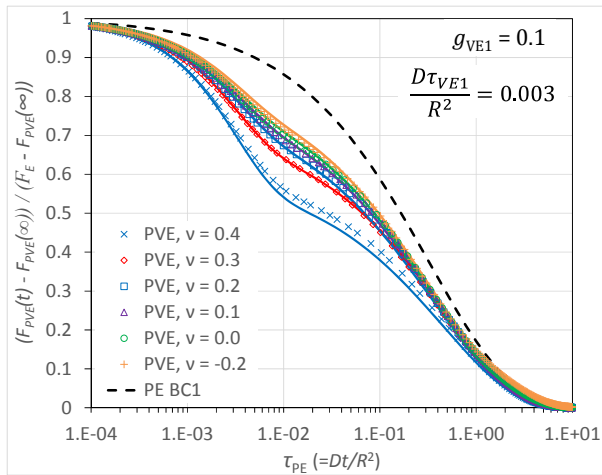


Figure 4. Normalized PVE force at fixed VE conditions and for different drained Poisson's ratio  $\nu_d$ . Points are results from finite element analysis for the BC1 model, lines are calculated using eq. 8. Note that most experimental results were obtained for  $0.2 > \nu_d > -0.2$ , hence the relaxation profile does not change significantly.

## 6. RESULTS

### 6.1. Hydrogels indentation

Heat induced whey protein gels were soaked in water with three different NaCl concentration to achieve different degrees of equilibrium swelling ratios SR (Table 1). The indentation tests include a loading and a relaxation step at a fixed indentation depth  $h$ . From the loading step, the elastic shear modulus  $G_E$  was determined assuming a linear-elastic response, eq. 1. The contact point, required to calculate  $h$  was

determined as shown in Fig. 5. An average corrective value of 1.42 was chosen for  $\Pi(R/t_0, \nu)$  in eq. 1 for all experiments. The average  $G_E$  found, summarized in Table 1, are comparable to those reported previously for unswollen whey protein gels at similar protein concentrations using a different indentation technique (Özkan et al., 2002).

Table 1. Best fit relaxation parameters for whey protein gels at swelling equilibrium with different salt concentrations ( $\pm$  SD). Different superscripts show significant difference ( $p < 0.05$ ): Latin letters for row comparisons, Greek letters for column comparisons of the same propriety.

[NaCl] / M		0.1	0.2	0.3
SR		0.123 <sup>a</sup>	-0.091 <sup>b</sup>	-0.134 <sup>c</sup>
		$\pm 0.017$	$\pm 0.02$	$\pm 0.02$
	# tests	34	19	28
	$G_E$ / kPa	$15^a \pm 3$	$32^b \pm 5$	$36^c \pm 7$
$F(\infty)/F_E$		0.59 <sup>a</sup>	0.50 <sup>b</sup>	0.44 <sup>c</sup>
		$\pm 0.06$	$\pm 0.07$	$\pm 0.08$
	PE $\nu_d$	0.14 <sup>aa</sup>	-0.01 <sup>ba</sup>	-0.2 <sup>ca</sup>
	$\pm 0.09$	$\pm 0.2$	$\pm 0.3$	
PE	$D / 10^{-9} \text{ m}^2 \text{ s}^{-1}$	$8.0^{aa} \pm 3.5$	$6.4^{aba} \pm 3$	$5.8^{ba} \pm 2$
BC1	Av. MSE	$3.7 \times 10^{-4} \text{ aa}$	$4.1 \times 10^{-4} \text{ aa}$	$3.1 \times 10^{-4} \text{ aa}$
PE	$D / 10^{-9} \text{ m}^2 \text{ s}^{-1}$	$5.0^{ab} \pm 3.7$	$4.2^{ab\beta} \pm 2$	$3.9^{b\beta} \pm 1.3$
BC2	Av. MSE	$5.7 \times 10^{-4} \text{ ab}$	$6.7 \times 10^{-4} \text{ aa}$	$4.8 \times 10^{-4} \text{ ab}$
PVE BC1	$D / 10^{-9} \text{ m}^2 \text{ s}^{-1}$	$6.6^{aa} \pm 3$	$5.3^{aba\beta} \pm 3$	$5.0^{ba} \pm 1.9$
	$g_{VE1}$	$0.05^{aa} \pm 0.03$	$0.079^{aa} \pm 0.05$	$0.06^{aa} \pm 0.04$
	$\tau_{VE1} / \text{s}$	$8^{aa} \pm 5$	$7.7^{aa} \pm 4$	$6^{aa} \pm 3$
	$\nu_d$	$0.19^{a\alpha\beta} \pm 0.09$	$0.06^{ba} \pm 0.2$	$-0.13^{ca} \pm 0.3$
	Av. MSE	$1.7 \times 10^{-4} \text{ ay}$	$1.1 \times 10^{-4} \text{ ab}$	$1.5 \times 10^{-4} \text{ ay}$
	PVE BC2	$D / 10^{-9} \text{ m}^2 \text{ s}^{-1}$	$4.0^{ay} \pm 2$	$3.2^{ab\beta} \pm 1.7$
$g_{VE1}$		$0.082^{ab} \pm 0.04$	$0.12^{ba} \pm 0.06$	$0.10^{ab\beta} \pm 0.04$
$\tau_{VE1} / \text{s}$		$13^{a\beta} \pm 6$	$12^{a\beta} \pm 5$	$11^{a\beta} \pm 4$
$\nu_d$		$0.21^{a\beta} \pm 0.1$	$0.10^{ba} \pm 0.2$	$-0.08^{ca} \pm 0.3$
Av. MSE		$1.1 \times 10^{-4} \text{ a}\delta$	$6.5 \times 10^{-5} \text{ b}\beta$	$7.8 \times 10^{-5} \text{ ab}\delta$

### 6.2. Viscoelastic relaxation analysis

The relaxation of food matrices, such as cheese (Lucey et al., 2003) or yoghurt (Puvanenthiran et al., 2002), has been typically modeled as an empirical viscoelastic process (Gallegos and Franco, 1999), where the estimated VE parameters do not have a strong physical meaning. A viscoelastic analysis, using eq. 6, was also applied to the force relaxation of whey protein gels. A value of  $N_{VE} = 3$  was required to fit the relaxation data well (Fig. 6); best fit values are shown in Table 2.

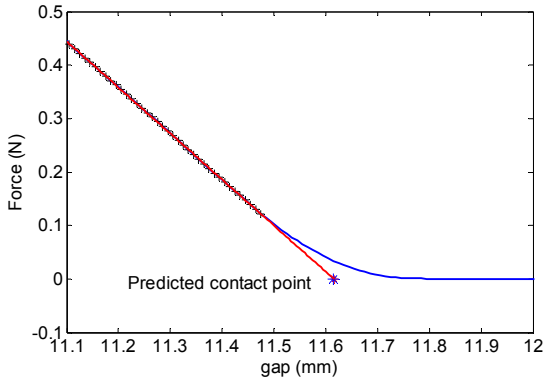


Figure 5. Example of loading step for a hydrogel swollen in 0.3 M NaCl. Red line is the best fit using eq. 1, used to calculate  $G_E$  at about 38 kPa here. The predicted contact point, used to determine  $h$ , is also shown.

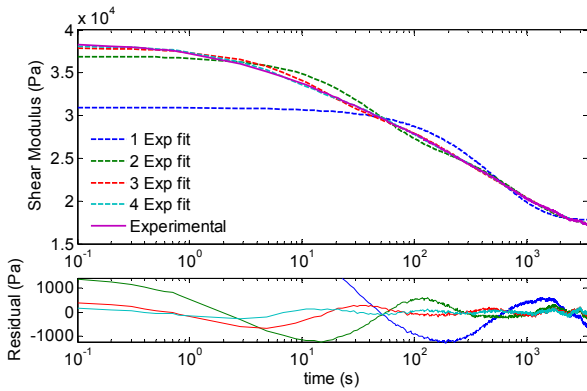


Figure 6. VE analysis with different  $N_{VE}$  values for the indentation shown in Fig. 5.

Table 2. Best fit viscoelastic parameters for  $N_{VE} = 3$ , estimated using nonlinear regressions of eq. 6.

[NaCl] / M	$g_{VE1}$	$\tau_{VE1}$ / s	$g_{VE2}$	$\tau_{VE3}$ / s	$g_{VE3}$	$\tau_{VE3}$ / s	Av. MSE
0.1	$0.12^a$ $\pm 0.03$	$10^a$ $\pm 3$	$0.13^a \pm 107^a$ $\pm 0.02$	$0.17^a$ $\pm 40$	$860^a$ $\pm 0.05$	$860^a$ $\pm 360$	$5 \times 10^5$
0.2	$0.15^b$ $\pm 0.03$	$10^a$ $\pm 3$	$0.15^b \pm 130^{ab}$ $\pm 0.02$	$0.20^a$ $\pm 45$	$1100^{ab}$ $\pm 0.06$	$1100^{ab}$ $\pm 400$	$2 \times 10^5$
0.3	$0.17^b$ $\pm 0.02$	$10^a$ $\pm 2.5$	$0.17^c \pm 125^b$ $\pm 0.01$	$0.25^b$ $\pm 30$	$1000^b$ $\pm 0.07$	$1000^b$ $\pm 220$	$2 \times 10^5$

### 6.3. Poroelastic relaxation analysis

The normalized relaxation force for whey protein hydrogels swollen at different salt concentrations was fitted to the PE  $f_{PE}(\tau_{PE})$  eq. 3 functions to estimate  $D$ . The average  $D$  values for the three swelling conditions tested are given in Table 1, with  $D$  decreasing slightly with higher salt concentrations, although only the values at 0.1 and 0.3 M NaCl are statistically different. The selection of the model BC1 or BC2 is important as statistically different values are estimated, with BC2 resulting obviously in smaller  $D$  values.

The  $D$  values estimated are higher than for the water self-diffusion in a comparable whey protein gel ( $\sim 1.5 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$  at  $20^\circ\text{C}$  (Colsenet et al., 2005) or in pure water. This result is reasonable because the average pore size of the whey protein hydrogels, estimated at  $\sim 20 \text{ nm}$  using the PE data, is much larger than the dimension of the water molecule. Consequently, water is expected to flow more by convection rather than by diffusion (Hu et al., 2011b). The intrinsic permeability  $k$  of the whey protein gels made is calculated at  $(0.8\text{--}2) \times 10^{-16} \text{ m}^2$ , comparable to that reported for agar and acrylamide hydrogels but here for much higher solid concentrations (Oyen, 2014). In fact, the biomaterial more similar to whey protein hydrogels in terms of  $G_E$  and  $k$ , at similar solid concentrations, would be gelatin (Strange and Oyen, 2012).

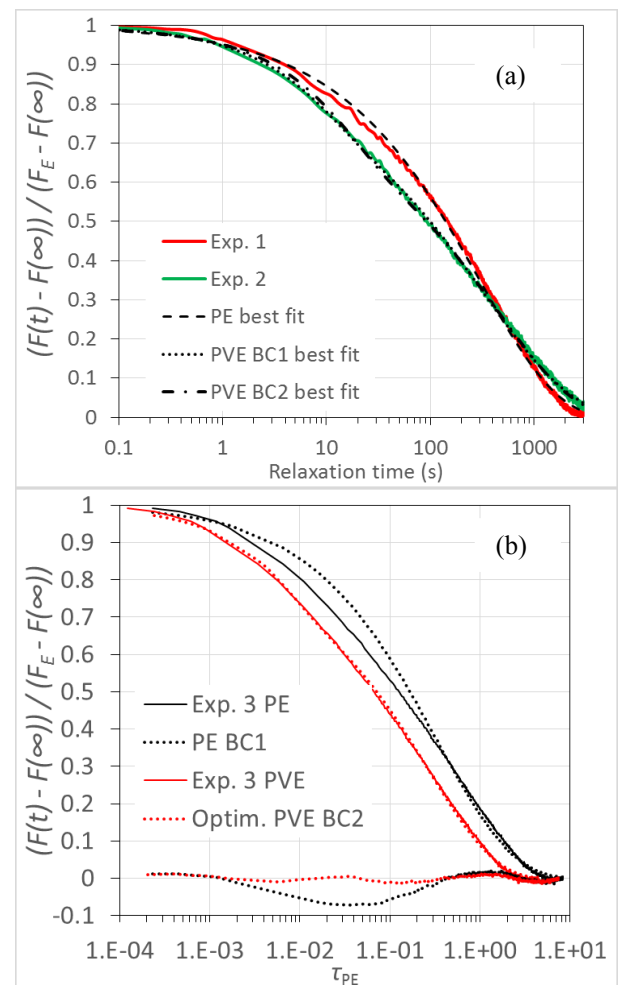


Figure 7. (a) Two representative relaxations of whey protein gels at swelling equilibrium in 0.1 M NaCl. In experiment 1 a PE relaxation fits well (best fit using BC2 model with  $D = 3.6 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$ ), and little improvement is achieved by considering PVE. Experiment 2 represents the more common tests where fitting is substantially improved by considering PVE relaxation (best fit PVE BC1:  $D = 4 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$ ,  $g_{VE1} = 0.074$ ,  $\tau_{VE1} = 10 \text{ s}$ , and using PVE BC2:  $D = 2.4 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$ ,  $g_{VE1} = 0.11$ ,  $\tau_{VE1} = 16 \text{ s}$ ). (b) Normalized relaxation



of a hydrogel in 0.3 M NaCl and the best fit using PE BC 1 (black) and PVE BC2 (red) models,  $D = 9.3 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$  and  $D = 4.9 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$ ,  $g_{VE1} = 0.10$ ,  $\tau_{VE1} = 6 \text{ s}$ ; respectively. Note that as the estimated  $D$  are different, the experimental data (continuous lines) have different  $\tau_{PE}$  values depending on the model considered.

Experiment 1 in Fig. 7(a) shows one of the few examples (<9% of all tests) where PE could represent well the whole relaxation. The majority of the tests showed a poor agreement with the PE models at 10-100 s, see Exp. 2. This is clearly seen in another test shown in Fig. 7(b), where the residuals of the PE fit show a systematic over prediction at  $\tau_{PE} \sim 0.01-0.1$ . Similar relaxation profiles have been reported in the literature, such as for chemically crosslinked polyethylene glycolmethacrylate (PEGMA) (Chan et al., 2012a), and have been explained by considering an additional viscoelastic relaxation, therefore a PVE model.

#### 6.4. Poroviscoelastic relaxation analysis

Two examples of PVE regressions are shown in Fig. 7. The average PVE parameters found are summarized in Table 1. The addition of a viscoelastic relaxation results in lower  $D$  estimates, where again are lower using the BC2 model compared to BC1. On absolute terms, the  $D$  values estimated are similar to the self-diffusivity of free water ( $\sim 2 \times 10^{-9} \text{ m}^2 \text{ s}^{-1}$ ), particularly using the BC2 model. The viscoelastic parameters  $g_{VE1}$  and  $\tau_{VE1}$  calculated were not affected by the swelling degree. We note that the overall  $g_{VE1}$  and  $\tau_{VE1}$  obtained from PVE are comparable to those found using a VE analysis only ( $g_{VE1} \sim 0.15$  and  $\tau_{VE1} = 10 \text{ s}$ , Table 2). The use of model BC2 was hardly justified for PE analysis as the average MSE was indeed worse compared to BC1. However, for PVE analysis, BC2 fits better the experimental data, validating its use for comparison, and suggesting that solvent equilibration could be possible from the bottom of the hydrogels. The quality of the BC2 model fit, as shown by the average MSE values, is comparable to that of a pure VE analysis with  $N_{VE} = 3$  (Table 2), but with half the number of adjustable parameters. The selection of the model did statistically influence the estimated  $g_{VE1}$  and  $\tau_{VE1}$ , which is likely due to the smaller  $D$  estimated with model BC2. We observed a strong correlation between the estimated  $D$  and the VE parameters, not shown. In any case, the values of  $\tau_{VE1}$  at 6-13 s are comparable to those found for PEGMA hydrogels in an analogous PVE analysis, at 6-20 s (Chan et al., 2012a).

## 7. DISCUSSION

It was unexpected that the main difference between the whey protein hydrogels soaked at different salt concentrations, other than  $G_E$  which is directly related to the extent of swelling, was  $F(\infty)/F_E$  and therefore  $v_d$  (Table 1). In addition, it appears that the calculated  $D$  correlate with  $F(\infty)/F_E$ : tests with high  $F(\infty)/F_E$  ratios show comparatively high  $D$  values (Fig. 8), as well as smaller  $g_{VE1}$  and  $\tau_{VE1}$  for PVE fits.

Comparable small and negative  $v_d$  values have been reported in the literature in agar-gelatin composite hydrogels when the hydrogels were more agar rich, with  $0 > v_d > -1$  (Strange and Oyen, 2012); and with  $v_d \sim -0.7$  for pure agar hydrogels calculated with a PVE analysis (Strange et al., 2013). The reasons are unclear, and further tests are required to verify this finding. However, we note that if there were an additional VE relaxation with  $\tau_{VE2} \sim 200 \text{ s}$ , it would be impossible to discern from a PE relaxation in the current experimental setup.

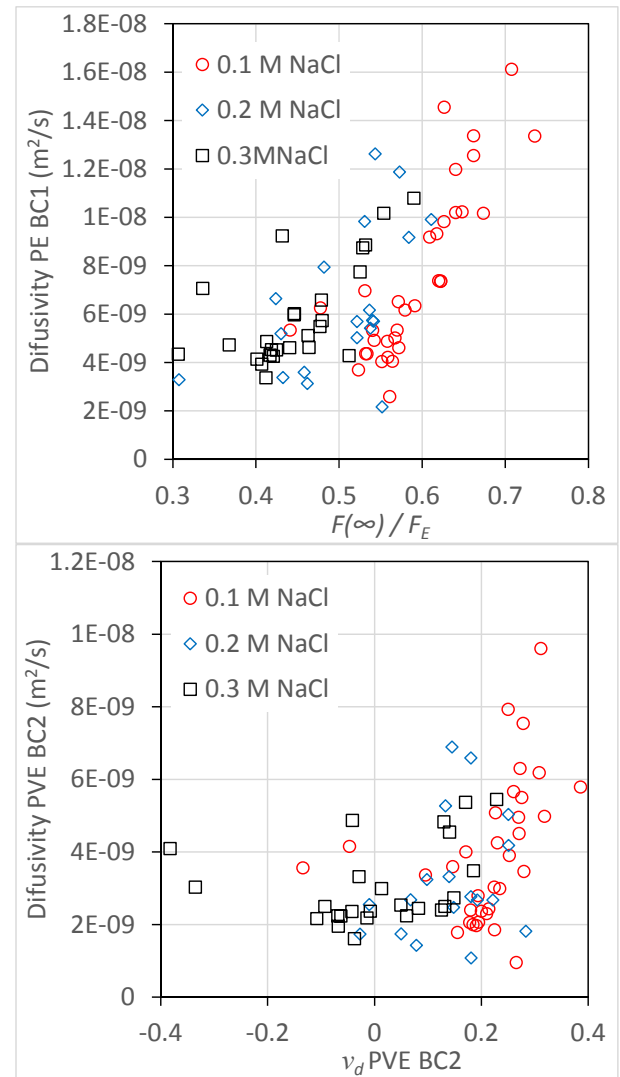


Figure 8. Diffusivities calculated assuming (a) poroelastic relaxation with no solvent flow from the bottom of the gel (BC1), and (b) poroviscoelastic relaxation with solvent flow from the bottom (BC2). Note the different y-axis scale.

One of the advantages of indentation testing to characterize hydrogels is that VE relaxation is independent of the indenter size  $R$ , whereas PE relaxation is strongly dependent on  $R$  (Wang et al., 2014). Hence, it is theoretically possible to perform experiments with different size indenters to elucidate the validity of the  $g_{VE1}$  and  $\tau_{VE1}$  estimated here.

Unfortunately, only one indenter size was currently available in the rheometer used. In order to decouple more both relaxations in time, there are two options: (i) increase the indenter size to  $R > 1$  cm, larger than the current size of the gels  $R_{gel}$ . Keeping the same indenter – hydrogel proportions than used here, the protein hydrogels should be  $R_{gel} > 3$  cm and  $t_0 > 3$  cm, which are highly impractical and expensive to make for routine testing. If the hydrogel dimensions were not so large for  $R > 1$  cm, then the PE relaxation will occur much faster due to the closer boundary surfaces (Hu, Chan, et al., 2011), canceling the time decoupling effect of increasing  $R$ . Option (ii) would be to reduce significantly the indenter size, as in nanoindentation (Oyen, 2015). With very small indenters, the PE relaxation would occur before the VE one; for the conditions used here, it is predicted  $R < 30$   $\mu\text{m}$  to clearly see both relaxations.  $R < 100$   $\mu\text{m}$  should be used to check if there is a VE relaxation with  $\tau_{VE2} \sim 200$  s as discussed above. Different indentation techniques, such as AFM where PE relaxation occurs in the order of seconds or faster (Kalcioğlu et al., 2012), would have to be used for such experiments.

The poroelastic indentation methodology has been further used in polymer hydrogels to obtain reasonable estimates of the Flory-Huggings interaction parameter  $\chi$  (Hu et al., 2011b). It would be very convenient to be able to perform such analysis, despite its many limitations and simplifications, to other more complex food based systems, such as the protein hydrogels considered here. It is known from swelling experiments that  $\chi$  for whey protein gels is  $\sim 0.5$  (ideal or  $\Theta$  conditions) (Mercadé-Prieto et al., 2007a). Yet, according to the figures provided by Hu et al. (2011b), such reasonable  $\chi$  can only be obtained from indentation data for  $0.25 < v_d < 0.3$ , much higher than determined here. Previous studies on whey protein gels highlight that the extent of relaxation  $F(\infty)/F_E$  after indentation is yet poorly understood, as it depends significantly on the gelation conditions for instance (Shim and Mulvaney, 2001). Future work is needed to validate  $v_d$  with a different methodology, to check if a VE relaxation overlaps in time with the PE relaxation, or if a different theoretical framework is required.

## 8. CONCLUSIONS

We have applied the experimental methodology of indentation-relaxation developed to characterize biomaterials to protein hydrogels. A poroelastic (PE) analysis, where the relaxation is solely caused by the internal flow of solvent in the hydrogel, provides reasonably good fits of the experimental data, yet it is significantly improved (as extra regression parameters are used) if a small viscoelastic relaxation at short times is considered. The calculated diffusivity  $D$  and intrinsic permeabilities  $k$  are reasonable and comparable to other hydrogels in the literature. Extensive indentation experiments were conducted with whey protein gels in swelling equilibrium with solutions with different salt

concentrations, thus at different swelling ratios. The estimated  $D$  increased slightly in more swollen gels, as expected, but it was unexpected that the drained Poisson's ratio  $v_d$  decreased at high salt concentrations to values equal to zero or even negative, which requires further investigation. The proposed PE analysis considers for the first time that solvent equilibrium can also occur from the bottom of the hydrogel (BC2 model), reasonable from an experimental point of view, which improves the PVE relaxation fits and yields  $D$  values slightly higher than that of the self-diffusion of free water.

## ACKNOWLEDGEMENTS

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# VIRTUALIZATION STRATEGIES FOR MODELING THE ENERGY TRANSFER IN A FOOD UNDERGOING RF HEATING

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## ABSTRACT

Heating uniformity in radiofrequency (RF) heating of agricultural and food commodities is a challenge to overcome for the development and dissemination of commercial RF heating systems at industrial scale. Different experimental studies have been carried out to test the effectiveness of RF heating in different food processing techniques. Modeling and simulation remains as one of the best tool used to virtualize the challenges of heating non-uniformity and effective tool to analyze and characterize the heating patterns during RF heating. Different strategies used to build-up virtualization of RF heating were briefly discussed in this paper. Dielectric and thermos-physical properties of the product subjected RF heating the key-factors to link the fundamental modeling approach with the data related to the investigated food material. The energy transfer across the food sample due to the RF heating and external convection effects have been discussed. Alternatively, using representative food samples like tylose with known thermo-physical and dielectric properties are also another strategy to be used to describe RF heating system.

Keywords: modelling, RF heating, energy transfer, food heating

## 1. INTRODUCTION

In recent years, the application of radio-frequency (RF) heating in agro-food processing has got wide applications due to the advantage of fast heating. Recently, a lot of interesting studies have been carried out on the application of RF heating in meat processing (Faraq et al., 2008 and 2011; Marra et al., 2007; Uyar et al., 2014 and 2015), in post-harvest treatment and disinfestation of fruits (Birla et al., 2004; Wang et al., 2002, 2003 and 2006a; Alfaifi et al., 2012 and 2015).

In RF heating applications in agricultural and food processing technologies, the heating non-uniformity is mainly dependent on the power absorption capacity of the material subjected to heating. This heating non-uniformity remains a challenge to be solved to widen the application of RF heating commercially in industrial scale.

Different authors have been studied different methods and strategies to improve the heating uniformity during RF heating. For example, Birla et.al. (2008) reported

that placing the sample at the center point between two parallel electrodes improved heating uniformity during RF disinfestation of fruits. Similar strategy was also proposed and used by Tiwari et al. (2011a) and Liu et al. (2013). The use of forced air surface heating, mixing and product movement to improve heating uniformity during RF treatment of walnuts was also promising strategy proposed and applied by Wang et. al. (2005 and 2007). In their recent study, Wang et.al. (2014) also demonstrated a systematic means to evaluate RF heating uniformity by using polyurethane foams in drying of agricultural commodities.

The use of model materials such as tylose with consistent and stable properties to replace more complicated heterogeneous materials is the first step to gain insights about the RF heating patterns during RF assisted heating of food products.

## 2. COMPUTER VIRTUALIZATION OF RF HEATING

Computer virtualization of processes is a powerful tool which can be used as virtual laboratory to get insights of heating patterns and to during RF heating. The mathematical modeling of RF heating systems has been discussed by (Neophytou and Metaxas, 1998; Yang et al., 2003; Chan et al., 2004; Marra et al., 2007 and 2009; Birla et al., 2008; Romano and Marra, 2008; Tiwari et al., 2011; Wang et al., 2012; Uyar et al., 2015; Llave et al., 2015) for different types of products in different heating systems. The heating system in RF heating fall into two categories of energy transfer, which are the heat transfer within the material subjected to RF heating and the electric and magnetic fields.

The mathematical description of heat transfer in the food product which is placed between two electrodes is given by unsteady state heat equation with a generation term derived from electromagnetic equations.

The heat equation is given by:

$$\rho C_p \frac{dT}{dt} = \nabla \cdot k \nabla T + P_{abs} \quad (1)$$

where  $T$  is the temperature within the sample,  $t$  is the time,  $k$  is thermal conductivity,  $\rho$  is the density,  $C_p$  is the heat capacity and  $P_{abs}$  is the power absorbed due to RF.

The power density or power dissipation per unit volume:

$$P_{abs} = 2 \pi f \varepsilon_0 \varepsilon_r'' |\underline{E}|^2 \quad (2)$$

The power density absorbed by lossy dielectric material is proportional to the frequency applied and the dielectric loss factor and proportional to the square of electric field.

The prediction of electromagnetic field distribution in RF heating is important to determine the power absorption density within the food product. Hence Maxwell's equations which relate the electric charge density,  $\rho_c$ ; electric field,  $\underline{E}$ ; electric displacement,  $\underline{D}$ ; current,  $\underline{J}$ ; magnetic field intensity,  $\underline{H}$  and magnetic flux density,  $\underline{B}$ .

$$\nabla \times \underline{H} = \underline{J} + \frac{\partial \underline{D}}{\partial t} \quad (3)$$

$$\nabla \times \underline{E} = -\frac{\partial \underline{B}}{\partial t} \quad (4)$$

$$\nabla \cdot \underline{D} = \rho_c \quad (5)$$

$$\nabla \cdot \underline{H} = 0 \quad (6)$$

Based on the assumption of fields to be time-harmonic (Neophytou and Metaxas, 1998), combination of Maxwell's equation could be reduced to the following wave equation in frequency domain (Marra et al., 2009).

$$\nabla \times \frac{1}{\mu_r} \nabla \times \underline{E} - \omega^2 \mu_0 \varepsilon_0 \varepsilon_r \underline{E} = 0 \quad (7)$$

When a quasi-static approach is assumed, equation (5) could be reduced to the following:

$$\nabla \cdot [(\sigma + j \omega \varepsilon) \nabla V] = 0 \quad (8)$$

where the electric potential,  $V$  is related to the electric field by:

$$\underline{E} = -\nabla V \quad (9)$$

The solution of the both heat equation and electromagnetic field equation are subjected assigning appropriate to initial and boundary conditions to come up with consistent and valid mathematical simulation of the process. The common initial and boundary conditions (BCs) used in solving heat equation are, uniform initial temperature of product and convection heat transfer on the boundaries of the product subjected to RF heating respectively. For the case of electromagnetic field distribution, an electric potential value or distribution is applied to the upper electrode while the bottom electrode is at ground condition. The RF applicator walls are also electrically insulated.

Figure 1, where typical configuration of parallel electrode RF system is shown, allow to better focus on the whole domain representation and to visualize where the previously discussed BCs are applied.

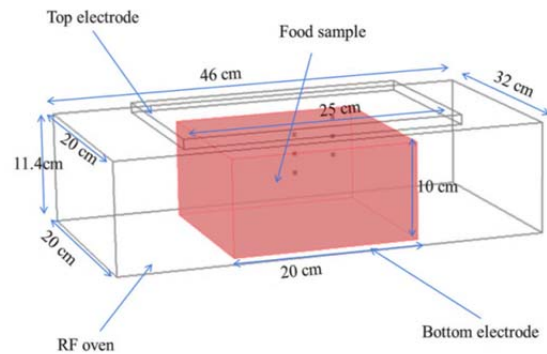


Figure 1: Typical configuration of parallel electrode RF system (adapted from Uyar et al., 2015)

### 3. VIRTUALIZATION STRATEGIES IN RF HEATING

Modeling of RF heating involves characterization of possible parameters and consideration of best configurations in the RF heating system. Different virtualization strategies have been proposed by different researchers since the first 3D FEM based mathematical model of RF heating by Neophytou and Metaxas, (1998). These strategies involve a number of issues which arise when processes assisted by RF are considered. Before modeling RF heating of foods, a number of questions require proper answers:

- In order to describe the electric field displacement, it is necessary to consider the whole set of Maxwell's equations?
- When the so called quasi-static approach is allowed?
- It is possible to use constant values of dielectric (and thermos-physical) properties? Or it is always recommended to use empirical values/expressions as functions of the local temperature?
- In the last case of point c), it is sufficient to consider the dielectric and thermos-physical properties as functions of the temperature, or also their dependence on moisture content (and, in some cases, in ionic concentration)?
- What is the importance (and the role) of mass transfer? In other words, it is always sufficient to consider the set of equations previously introduced or also mass transfer equation(s) must be considered?
- If also mass transfer has to be considered, what mass component is the more relevant? Moisture content? Ionic content? Fraction of free water with respect to bounded water?

- g) If also moisture transfer has to be considered, what is the more appropriate mass transfer approach? The porous media approach (Datta, 2007)? The kinetic like approach (Marra et al., 2010)?
- h) What happens when phase change has to be considered?
- i) What about boundary conditions? When external convective heat transfer should be taken into account by solving the momentum transfer and the heat and mass transfer in the environment surrounding the food sample?
- j) What about validation?

The answers to the questions listed above are not unique and the strategy of virtualization must be driven by the objective(s) of the research. Some examples are discussed below.

Under the hypothesis of quasi-static approach, Marra et al. (2007) solved coupled heat transfer and electromagnetic equations for a cylindrical shaped meat batters and demonstrated that quasi-static hypothesis to be successful in lab scale RF system. Different strategies have been applied by different researchers to find out how to improve the heating uniformity of the RF heating. Moving or rotating in water surrounding system has been proposed and used in modeling/experimental study about RF heating of high water content commodities like fruits (Birla et al., 2008) and for low moisture foods, use of polyetherimide (PEI) blocks, hot air assistance, intermittent stirring, electrode configuration and use of plastic sheets surrounding the product to enhance heating (Jiao et al, 2015; Alfaifi et al., 2014).

Virtualization of the processes also includes the assumptions and specific considerations about numerical parameters and also about process parameters involved in it. When a rough solution is sufficient for a preliminary analysis, one may choose to work with a coarse mesh, in order to deal with a numerical procedure less demanding in terms of computing power. The same strategy is not acceptable when the goal of virtualization is to have a model which is independent by the mesh grid. Another example, the latent heat peak during thawing of foods needs special consideration in order to simplify simulation complexities and time. Uyar et al. (2015) used apparent specific heat method in virtualization of lean beef meat undergoing thawing assisted by RF heating. The following section is about this topic.

As mentioned above, the virtualization strategy should include also a motivated decision on the set of BCs to be considered. In case of relevant convective contributions from the external environment toward the food product, the model should take into account (and, thus, should include) also the heat and mass transfer in the surrounding environment, and with it the momentum transfer too.

### 3.1 An example: meat thawing assisted by RF

In food processing industries tempering and thawing are the two main processes involved in heating frozen food products to required temperature for further processing or to make it ready for consumption. Tempering is considered as pre-thawing process, since it is used to increase the temperature of the product from frozen state to a temperature point where it is usable without causing harm to subsequent processing operations. Thawing is increasing frozen product temperature to melted or unfrozen state. RF assisted tempering and thawing have been experimentally studied and models were also developed to virtualize the heating patterns as well as to allocate the hot and cold spots during the process (Frag et al., 2008 and 2011; Llave et al., 2015; Uyar et al., 2015).

As shown in figure 2, where the temperature distribution (in °C) in a sample of beef meat during RF thawing is reported, the thawing process cannot be followed as a simple problem of monitoring the cold spot (which is located in the heart of the sample) but also in terms of a problem of overheating (as it happens in the areas colored in brown-red). Thus, here the virtualization strategy cannot ignore complex mechanisms happening at boundaries (at corners especially). These areas undergo phase change with subsequent severe changes in thermos-physical and dielectric properties.

Determination of RF power density ( $P_{abs}$ ) distribution in the sample during the process is a key-step as well. Figure 3 shows the power density (in  $W/m^3$ ) distribution (at side slice) in lean beef thawed by RF assisted heating at early stage of the process. In this figure it is possible to appreciate the great variability of power density values. At same time, it is possible to appreciate that in the strategy of virtualization – in this case – is to have a preliminary rough solution, being the mesh grid quite coarse.

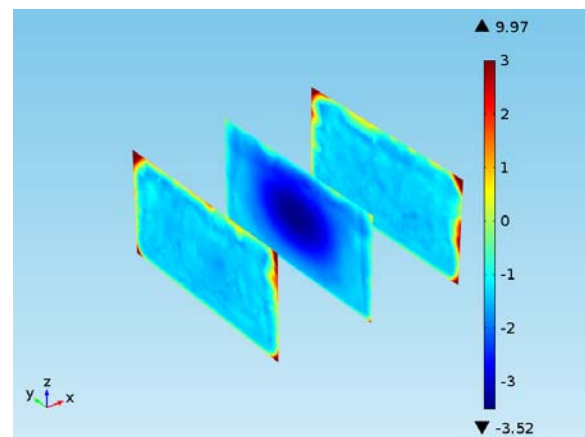


Figure 2: Temperature distribution [°C] in the sample during RF thawing of beef meat.

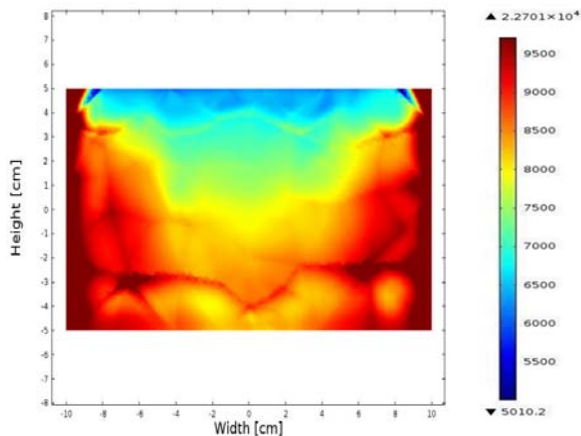


Figure 3: Power density [ $\text{W}/\text{m}^3$ ] distribution (at side slice) in lean beef thawed by RF assisted heating, after 1 minute of processing.

#### 4. CONCLUSIONS

Virtualization of complex processes, such as food heating assisted by radio-frequencies, should be based on a strategy which includes a number of steps related to what is simulated (definition of the process), where it is simulated (definition of a domain), how it is simulated (definition of the equations' set and of the solution procedure). A number of essential questions as been proposed and example of virtualization strategy have been discussed, with a focus on RF thawing of foods.

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# REACTION ENGINEERING METHODOLOGY AS AN EFFECTIVE APPROACH TO MODEL DRYING, BAKING AND WATER VAPOR SORPTION PROCESSES

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## ABSTRACT

An effective model of simultaneous heat and mass transfer processes is useful to assist in process design, optimization of existing processes and monitoring product quality. Previously, the lumped reaction engineering approach (L-REA) has been shown to model the global drying rate of several heat and mass transfer processes very well. Here, the REA framework is implemented to model the local evaporation/condensation rate in drying of non-food materials, baking and water vapor sorption. The REA is combined with a set of equations of conservation of heat and mass transfer to yield the S-REA (spatial reaction engineering approach) to model these processes. For modeling each process, the activation energy is generated from one accurate run and evaluated according to environmental temperature and humidity. The relative activation energy implemented in the L-REA is implemented in the S-REA by applying local variables. The results indicate that the S-REA is accurate to describe baking, drying and water vapor sorption process, which shows the applicability of the REA to model the local evaporation/condensation rate of these processes. The S-REA is readily implemented to assist in process design, evaluation of existing processes and maintenance of product quality. In near future, by coupling with solid mechanics, the REA may be employed to predict material deformation and shape change during heat and mass transfer processes by coupling with solid mechanics. The development of the REA to describe migration of volatiles inside materials undergoing heat and mass transfer processes is also underway.

Keywords: reaction engineering approach, drying, baking, water vapor sorption, model

## 1. INTRODUCTION

Several heat and mass transfer processes including drying, baking and water vapor sorption are commonly applied in industry. For process design of new equipment and evaluation of current processes, a reliable mathematical model is useful. The model may also be used to predict product quality during the processes. The reaction engineering approach (REA), a

semi-empirical approach, was proposed in 1996 and initially used to model thin layer drying of food materials (Chen and Lin, 2005; Lin and Chen, 2006; 2007). The model accurately captures the major physics of drying. The relative activation energy, a material characteristic, basically expresses the difficulty to remove the moisture from the materials. The removal of both free and bound moisture is captured by the relative activation energy. The REA has advantages of simplicity, accuracy and effectiveness with respect to generating the drying parameters. Only one accurate drying run is required to generate the parameters (Chen, 2008). The REA in its lumped format, later called as the lumped reaction engineering approach (L-REA), has accurately modeled several challenging heat and mass transfer processes. In this study, the REA is combined with a set of equations of conservation of heat and mass transfer to yield the S-REA to model drying of non-food materials, baking and water vapor sorption. This study aims to develop S-REA and implement it in the study of several heat and mass transfer processes. Its accuracy is examined by benchmarking model results against high quality experimental data reported in the literature.

## 2. MATHEMATICAL MODELING

For drying, the applicability of the S-REA to describe the drying processes was benchmarked towards the experimental data of drying of honeycomb (Dhall et al, 2012) while for baking and water vapor sorption, the validity of the S-REA is tested towards Zhang and Chen (2014) and Li et al (2014), respectively. The S-REA consists of a set of equations of microscopic heat and mass balance in which the REA is used to describe the local evaporation/condensation rate. For drying of honeycomb (Dhall et al, 2012), the mass balance of liquid water in Lagrangian coordinate, which allows expansion and contraction of the matrix, is written as (Putranto and Chen, 2013a,b):

$$\frac{\partial(C_s X)}{\partial t} = \frac{1}{r} \frac{\partial}{\partial r} \left( D_w r \frac{\partial(C_s X)}{\partial r} \right) - \dot{I} \quad (1)$$

where  $D_w$  is the local surface diffusivity ( $m^2 s^{-1}$ ),  $X$  is the concentration of liquid water ( $kg H_2O kg dry solids^{-1}$ ),  $C_s$  is the solids concentration ( $kg dry solids m^{-3}$ )

which can change if the structure changes,  $\dot{I}$  is the evaporation or condensation rate (kg H<sub>2</sub>O m<sup>-3</sup> s<sup>-1</sup>) and is positive when evaporation occurs locally.

The mass balance of water in the vapor phase (water vapor) in spherical coordinate system is expressed as (Putranto and Chen, 2013a,b):

$$\frac{\partial C_v}{\partial t} = \frac{1}{r} \frac{\partial}{\partial r} (D_v r \frac{\partial C_v}{\partial r}) + \dot{I} \quad (2)$$

where  $D_v$  is the local effective vapor diffusivity (m<sup>2</sup> s<sup>-1</sup>) and  $C_v$  is the concentration of water vapor (kg H<sub>2</sub>O m<sup>-3</sup>).

The heat balance is represented by the following equation (Putranto and Chen, 2013<sup>a,b</sup>):

$$\rho C_p \frac{\partial T}{\partial t} = \frac{1}{r} \frac{\partial}{\partial r} (k r \frac{\partial T}{\partial r}) - \dot{I} \Delta H_v \quad (3)$$

where  $T$  is the sample temperature (K),  $k$  is the local thermal conductivity of sample (W m<sup>-1</sup> K<sup>-1</sup>),  $\Delta H_v$  is the water vaporization heat (J kg<sup>-1</sup>),  $C_p$  is the local specific heat of sample (W m<sup>-1</sup> K<sup>-1</sup>) and  $\rho$  is the local sample density (kg m<sup>-3</sup>).

The initial and boundary conditions for equations (1) to (3) are:

$$t=0, X=X_o, C_v=C_{v_o}, T=T_o \quad (4)$$

$$\begin{aligned} \frac{dX}{dr} &= 0, \quad \frac{dC_v}{dr} = 0, \quad \frac{dT}{dr} = 0 \\ r=0, & \\ r=R, & \end{aligned} \quad (5)$$

$$-C_s D_w \frac{dX}{dr} = h_m \varepsilon_w \left( \frac{C_{v,s}}{\varepsilon} - \rho_{v,b} \right) \quad (7)$$

$$-D_v \frac{dC_v}{dr} = h_m \varepsilon_v \left( \frac{C_{v,s}}{\varepsilon} - \rho_{v,b} \right) \quad (8)$$

$$k \frac{dT}{dr} = h(T_b - T) \quad (9)$$

where  $R$  is the sample radius (m),  $h$  is the heat transfer coefficient (W.m<sup>-2</sup>.K<sup>-1</sup>),  $\varepsilon_w$  and  $\varepsilon_v$  are fraction of surface area influenced by liquid water and water vapor respectively.

$\dot{I}$  is the local evaporation/condensation rate within the solid structure described as (Putranto and Chen, 2013a,b):

$$\dot{I} = h_{m,in} A_{in} (C_{v,s} - C_v) \quad (9)$$

where  $h_{m,in}$  is the internal mass transfer coefficient (m.s<sup>-1</sup>),  $A_{in}$  is the total internal surface area available for phase change (m<sup>2</sup>.m<sup>-3</sup>),  $C_{v,s}$  is the internal-solid-surface water vapor concentration (kg.m<sup>-3</sup>). Although the void space inside the samples may be large enough, the incorporation of the advection term in the S-REA is not necessary since its term has been captured by the local evaporation/condensation term as represented by equation (9).

By implementing the REA, the internal-surface water vapor concentration can be written as (Putranto and Chen, 2013a,b):

$$C_{v,s} = \exp\left(\frac{-\Delta E_v}{RT}\right) C_{v,sat} \quad (10)$$

where  $C_{v,sat}$  is the internal-saturated water vapor concentration (kg m<sup>-3</sup>).

Therefore, the local evaporation rate can be expressed as (Putranto and Chen, 2013<sup>a,b</sup>):

$$\dot{I} = h_{m,in} A_{in} \left[ \exp\left(\frac{-\Delta E_v}{RT}\right) C_{v,sat} - C_v \right] \quad (11)$$

The relative activation energy of honeycombs is generated from one accurate drying run at temperature of 103 °C. The relationship between the relative activation energy and average moisture content is assumed to be represented by simple mathematical equation obtained by least square method using Microsoft Excel®. The relative activation energy was found to be suitably fitted as:

$$\begin{aligned} \frac{\Delta E_v}{\Delta E_{v,b}} &= -138685(X - X_b)^5 + 50405(X - X_b)^4 \\ &- 6953.5(X - X_b)^3 + 466.12(X - X_b)^2 \\ &- 17.616(X - X_b) + 1.0053 \end{aligned} \quad (12)$$

For baking and water vapor sorption, the S-REA, similar to the ones written in equations (1) to (3) are setup and implemented. This is then combined with appropriate relative activation energies of dough during baking (Zhang and Chen, 2014) and silica gel during water vapor sorption (Li et al, 2007). The relative activation energy of dough can be written as:

$$\begin{aligned} \frac{\Delta E_v}{\Delta E_{v,b}} &= -55.754(\bar{X} - X_b)^5 + 113.95(\bar{X} - X_b)^4 \\ &- 96.768(\bar{X} - X_b)^3 + 40.609(\bar{X} - X_b)^2 \\ &- 8.385(\bar{X} - X_b) + 1.006 \end{aligned} \quad (13)$$

While the relative activation energy of silica gel can be expressed as:

$$\begin{aligned} \frac{\Delta E_v}{\Delta E_{v,b}} &= -4.19(\bar{X} - X_b)^4 - 15.66(\bar{X} - X_b)^3 \\ &+ 3.192(\bar{X} - X_b)^2 - 10.063(\bar{X} - X_b) + 1 \end{aligned} \quad (14)$$

In order to yield the spatial profiles of moisture content, concentration of water vapor and temperature, equations (1) to (3) are solved simultaneously in conjunction with the initial and boundary conditions (shown in equations (4) to (8) and the appropriate relative activation energy.

### 3. RESULTS AND DISCUSSION

For convective drying of honeycombs, as suggested by Dhall et al (2012), the liquid diffusivity of 10<sup>-10</sup> m<sup>2</sup>.s<sup>-1</sup> is used. Interestingly, as shown in Figure 1, the liquid

diffusivity of lower than  $1 \times 10^{-10} \text{ m}^2 \text{ s}^{-1}$  does not result in any noticeable differences in the moisture content and temperature profiles. This may indicate that the modeling is independent on surface diffusivity and hence the surface diffusion term may be neglected in the modeling. Without incorporating the surface diffusion term, the results of modeling of drying of honeycombs at temperature of 103 and 137 °C are shown in Figures 2 and 3. A good agreement towards the experimental data is shown ( $R^2$  of higher than 0.998 for moisture content and  $R^2$  of higher than 0.993 for temperature). Benchmarks against the modeling implemented by Dhall et al (2012) shows that the S-REA yields closer agreement to the experimental data. The S-REA is accurate to model both moisture content and temperature profiles at drying air temperature of 103 and 137 °C.

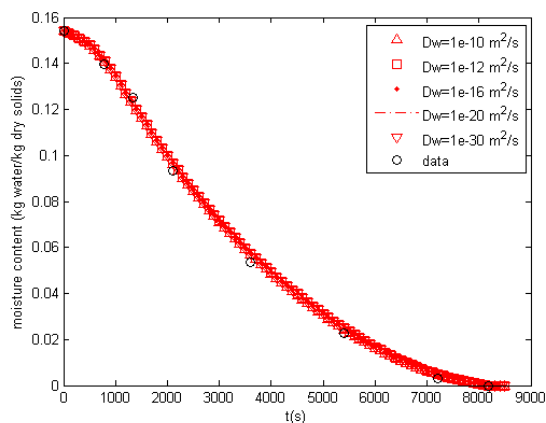


Figure 1. Sensitivity of effective liquid diffusivity towards profiles of moisture content during drying of honeycombs at temperature of 137 °C (Putranto and Chen, 2015)

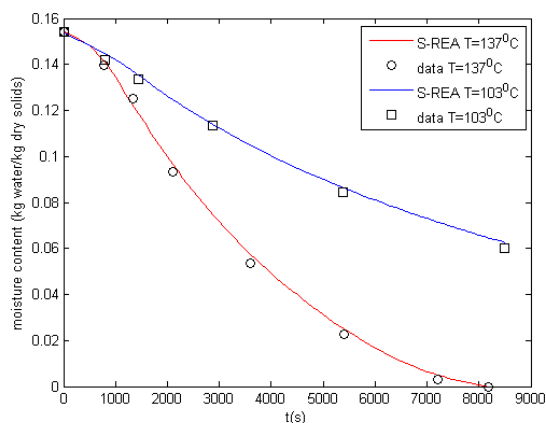


Figure 2. Profiles of moisture content during drying of honeycombs at temperature of 103 and 137 °C (Putranto and Chen, 2015)

For baking, the moisture content profiles during baking of 0.5 and 1 g bread are shown in Figure 4. A good agreement towards the experimental data is shown ( $R^2$  of 0.992 and 0.988 for 0.5 and 1 g bread, respectively). Figure 5 shows the spatial profiles of moisture content

during baking of 1 g of bread at baking temperature of 235 °C. The profiles at the outer part of the samples are lower than those at the inner part which indicate that the moisture migrate outwards during baking. The profiles decrease during baking time since the moisture in the samples depletes as baking progresses. The S-REA models well the baking process which could be due to the applicability of the REA to represent the local evaporation/condensation rate during baking. The relative activation energy seems to be sufficiently accurate to capture the local evaporation/condensation inside the samples coupled with starch gelatinization, protein denaturation and browning.

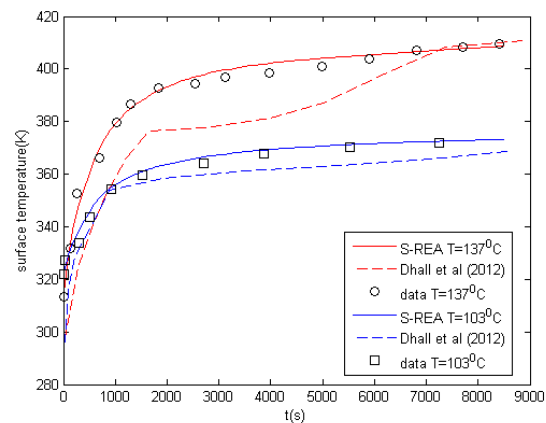


Figure 3. Profiles of surface temperature during drying of honeycombs at temperature of 103 and 137 °C (Putranto and Chen, 2015).

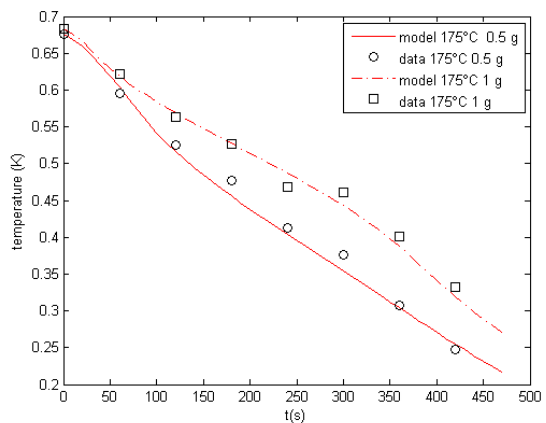


Figure 4. Profiles of average moisture content during baking of bread at baking temperature of 175 °C (Putranto and Chen, 2015)

For water vapor sorption, similar to the convective drying of honeycombs, the varied values of surface diffusivity do not give any noticeable differences in the profiles. Without using the surface diffusion term, the profiles of average moisture content are shown in Figure 6 while the profiles of spatial one are indicated in Figure 7. The S-REA models accurately the water vapor sorption which may be because the REA describes well the local condensation rate. The applicability of the REA indicates that the relative

activation energy used to model the global wetting rate (refer to equation (16)) (Putranto and Chen, 2014) is also applicable to model the local wetting rate of water vapor sorption.

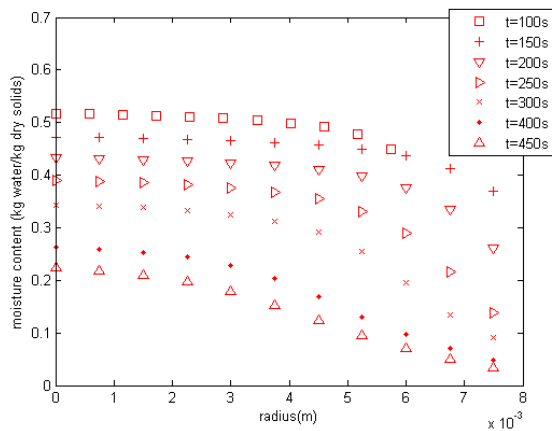


Figure 5. Spatial profiles of moisture content during bread baking of 1 g of bread at baking temperature of 235 °C (Putranto and Chen, 2015)

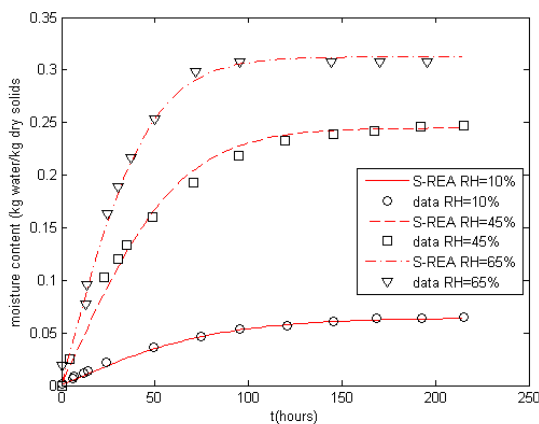


Figure 6. Profiles of moisture content during water vapor sorption at various relative humidity (Putranto and Chen, 2015)

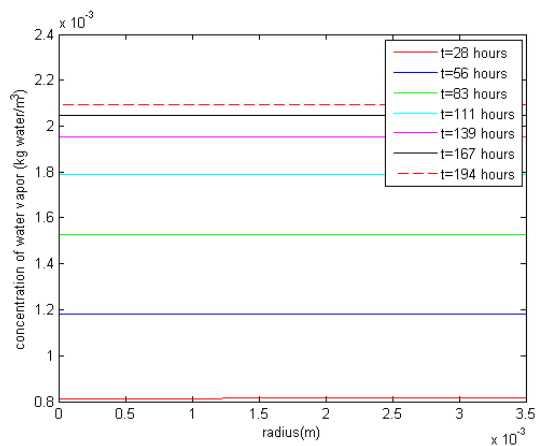


Figure 7. Spatial profiles of concentration of water vapor inside silica gel during water vapor sorption at relative humidity of 10% (Putranto and Chen, 2015)

#### 4. CONCLUSIONS

In this study, the applicability of the S-REA to describe several simultaneous heat and mass transfer processes is demonstrated. The S-REA is accurate to model the convective drying, baking and water vapor sorption. This accuracy indicates that the REA is applicable to model the local evaporation/condensation rates during the processes. The combination of relative and equilibrium activation energy results in a unique relationship of activation energy which can describe the local structural changes during the simultaneous heat and mass transfer processes, as affected by the environmental conditions.

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Xiao Dong Chen was born in Beijing in 1965. BE in Engineering Mechanics from Tsinghua University (1987). PhD in Chemical and Process Engineering from Canterbury University in NZ (1991). MSc in Mathematics from University of New South Wales in Australia (2014). After working for Fonterra NZ for 2.5 years on food particle technology, he took up a lectureship at The University of Auckland, NZ. He was made a Personal Chair of Chemical Engineering in 2001 at The University of Auckland. In 2006, he was appointed to the Chair of Biotechnology and Professor of Chemical Engineering at Monash University, Melbourne, Australia. He has published over 440 refereed journal articles and 220 conference papers (over 30 were Keynote or Plenary Lectures), 3 books, 20 book chapters, and over 50 reports on industrial consulting projects. From early 2010 to the end of 2012, he held a 1000-Talent National Chair Professor of Chemical Engineering and the Executive Headship of Chemical and Biochemical Engineering at Xiamen University. He is an Elected Fellow of Royal Society of New Zealand since 2001 and an Elected Fellow of Australian Academy of Technological Sciences and Engineering since 2007, and is a Fellow of IChemE (UK) from 2001.

His main areas of research are in bio-inspired chemical engineering, food engineering and biotechnology, separation processes including drying and membrane processing, heat transfer and mass transfer including fouling, cleaning and drying, safety processes including spontaneous combustion of solid materials, powder technology and powder functionality including food and pharmaceutical particulates.

In January 2013, he has begun his role as the University Distinguished Professor and Head of School at Soochow University (Suzhou, PR China). He has established a new School of Chemical and Environmental Engineering there. He has remained as Adjunct Professors at Xiamen University (China) and Monash University (Australia).



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